

Civilian Radioactive Waste Management System

Transportation, Aging and Disposal Canister System Performance Specification

Revision 0

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REVISION HISTORY

Revision	<u>Change</u>
A	Initial Issue
В	For requirement number (5) and (6) in Section 3.1.1 changed " or less then 5 years out-of-reactor" to "and no less than 5 years out-of-reactor"
0	Initial issue of Final TAD Performance Specification. Incorporated comments on the Preliminary TAD Performance Specification, Rev. B

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Attachment A Seismic Data for Yucca Mountain Geologic Repository Operations Area

GLOSSARY_____

Attachment B Postclosure Criticality Loading Curves

Attachment C TAD Canister Lifting Feature

Transportation, Aging and Disposal Canister

Attachment D Aging Overpack Details

4.0

Attachment E Supplemental Soils Report

ACRONYMS

ALARA as low as is reasonably achievable

BWR boiling water reactor

CFR Code of Federal Regulation CSNF commercial spent nuclear fuel

DCRA disposal control rod assembly DOE U.S. Department of Energy

GROA geologic repository operations area

HLW high-level radioactive waste

HVAC heating, ventilation and air-conditioning

ICRP International Commission on Radiological Protection

ISFSI independent spent fuel storage installation

ITS important to safety

MTU metric tons of uranium

NRC U.S. Nuclear Regulatory Commission

NWPA Nuclear Waste Policy Act

OCRWM Office of Civilian Radioactive Waste Management

PWR pressurized water reactor

SNF spent nuclear fuel

SSC structures, systems and components

STC shielded transfer cask

TAD transportation, aging and disposal TEDE total effective dose equivalent TWPS TAD waste package spacer

USL upper subcritical limit

YMP Yucca Mountain Project

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ABBREVIATIONS

° C degrees Centigrade ° F degrees Fahrenheit

BTU International Table British thermal unit BTU/hr-ft² British thermal unit per hour-square foot

Bq becquerel

cm centimeter

cm² square centimeter

dpm disintegrations per minute

ft feet

ft/s feet per second

g acceleration due to gravity g/cm² grams per square centimeter

GWd gigawatt-day

h or hr hour

in. inches

 k_{eff} effective neutron multiplication factor

kg kilogram km kilometer km/hr kilometer/hour kPa kilopascal kW kilowatt

kW/m² kilowatt per square meter

lb pound(s) (weight; unless otherwise specified)

lb/ft² pounds per square foot lb/in² pounds per square inch

lb/in²/sec pounds per square inch per second

m meter

m/s meter per second m² square meter(s)

mho Conductance in mho being the reciprocal of resistance in ohms

mm millimeter
MPa megapascal
mph miles per hour

Transportation, Aging and Disposal Canister System Performance Specification

milli roentgen equivalent man mrem

metric tons MT

рН potential of hydrogen parts per million ppm psi, lb/in² pounds per square inch

s or sec second

short ton (2,000 lb weight) ton

pressure that causes the Hg column to rise 1 millimeter torr

year yr

1.0 Introduction

1.1 Purpose

This document provides specifications for selected system components of the Transportation, Aging and Disposal (TAD) canister-based system. A list of system specified components and ancillary components are included in Section 1.2.

The TAD canister, in conjunction with specialized overpacks will accomplish a number of functions in the management and disposal of spent nuclear fuel. Some of these functions will be accomplished at purchaser sites where commercial spent nuclear fuel (CSNF) is stored, and some will be performed within the Office of Civilian Radioactive Waste Management (OCRWM) transportation and disposal system. This document contains only those requirements unique to applications within Department of Energy's (DOE's) system. DOE recognizes that TAD canisters may have to perform similar functions at purchaser sites. Requirements to meet reactor functions, such as on-site dry storage, handling, and loading for transportation, are expected to be similar to commercially available canister-based systems.

This document is intended to be referenced in the license application for the Monitored Geologic Repository (MGR). As such, the requirements cited herein are needed for TAD system use in OCRWM's disposal system. This document contains specifications for the TAD canister, transportation overpack and aging overpack. The remaining components and equipment that are unique to the OCRWM system or for similar purchaser applications will be supplied by others.

1.2 Transportation, Aging and Disposal (TAD) System Description

A TAD system consists of a canister, together with other equipment, that allows for management of commercial spent nuclear fuel.

1.2.1 TAD canister

The TAD canister is loaded with commercial spent nuclear fuel (CSNF) and sealed at purchaser sites (e.g., reactors) or the repository. The loaded TAD canister may be used for storage for a period of time at purchaser sites; for this purpose it must be approved contents for a storage system certified under title 10 CFR part 72. The loaded TAD canister may be delivered to DOE for transportation to the geologic repository operations area (GROA), for which it would be listed as approved contents for packaging, including the transportation overpack, certified under title 10 CFR part 71. At the GROA, a loaded TAD canister may also be handled using a shielded transfer cask or aged in an aging overpack; and shall be disposed of in a waste package. All three of these functions will be covered by the repository license granted under title 10 CFR part 63.

1.2.2 Transportation Overpack

The transportation overpack is an overpack certified under title 10 CFR part 71 as a packaging component used to enclose TAD canisters for transportation. The transportation overpack: protects the TAD canister during normal conditions of transport and design basis accidents; dissipates decay heat from the contained CSNF; and, protects workers and the public from radiation.

1.2.3 <u>Transportation Skid</u>

The transportation skid is the means of handling assembled transportation packages at various sites and during inter-modal transfers.

1.2.4 Ancillary Equipment

Ancillary equipment is any general or site specific equipment, not specifically described within this document, required to operate and handle TAD system components in accordance with their certificates of compliance and other regulatory or operational requirements. Ancillary equipment to be used at the repository will be provided by others. Any ancillary equipment needed for use at purchaser sites is expected to be similar to commercially available equipment in common usage.

1.2.5 Shielded Transfer Cask

The shielded transfer cask (STC) is used to transport a loaded TAD canister among the various surface facilities at the GROA prior to loading into an aging overpack or waste package. The STC protects the TAD canister from damage, protects workers from radiation and allows for proper heat dissipation. The STC for use at the repository will be provided by others. STC to be used at purchaser sites are expected to be similar to commercially available equipment commonly used.

1.2.6 Aging Overpack

Aging overpacks are used to safely contain a loaded TAD canister on the aging pad until repository emplacement thermal limits are met. The aging overpack protects the TAD canisters from damage, dissipates decay heat and protects workers from radiation.

1.2.7 Site Transporter

The site transporter is a vehicle to be used for transporting loaded and unloaded STCs and aging overpacks at the GROA. The transporter will also provide support for STCs and aging overpacks during loading and unloading operations. The site transporter will be provided by others. A site transporter is expected to be required to perform analogous functions at purchaser sites. Any site transporter that is part of a site specific independent spent fuel storage installation (ISFSI) system is expected to be similar to commercially available equipment in common usage.

1.2.8 Waste Package

The waste package is the disposal container that the TAD canister will be sealed inside prior to final emplacement in the drift.

1.2.9 Storage Overpack

The storage overpack provides functions analogous to the aging overpack at purchaser sites. Storage overpacks which are part of a purchaser site specific ISFSI will be designed to meet the requirements of title 10 CFR part 72. Storage overpacks used at purchaser sites as part of a site specific ISFSI are expected to be similar to commercially available equipment in common usage.

1.3 Definitions

Accident- An undesirable event; especially one that could potentially do damage or harm to a cask or its contents.

Approved Contents- Used in the context of this performance specification, the term "approved contents" means one of the following:

<u>Transportation Overpack</u>: The contents of Type B packaging as defined NRC Regulatory Guide 7.9 *Standard Format and Content of Part 71 Applications for Approval of Packages for Radioactive Material* and listed in section 5b "Contents of Packaging" of Certificates of Compliance issued under 10 CFR part 71.

Storage Overpack: The materials to be stored as defined in NRC Regulatory Guide 3.61 Standard Format and Content for a Topical Safety Analysis Report for a Spent Fuel Dry Storage Cask and listed in Section 6 "Approved Contents" of Certificates of Compliance issued under 10 CFR part 72.

Normal- A term used to define expected radioactive wastes, operations and/or processes.

Off-normal- A term used to define any combination of radioactive waste, operations or processes that are not expected during normal activities; usually associated with damaged or failed materials, equipment or processes.

Purchaser- Any person, other than a Federal agency, who is licensed by the Nuclear Regulatory Commission to use a utilization or production facility under the authority of sections 103 or 104 of the Atomic Energy Act of 1954 (42 U.S.C. 2133, 2134) or who has title to spent nuclear fuel or high-level radioactive waste and who has executed a contract for disposal of spent nuclear fuel and/or high-level radioactive waste with DOE.

1.4 Safety Classification of the Components

Safety classification of the components in this specification has not been assigned. However; the TAD canister, the transportation overpack, and the aging overpack covered by this specification are expected to be Important to Safety (ITS).

1.5 Limitations

No portion of this specification shall be interpreted such that it suggests, implies or intimates that the vendor is responsible for showing compliance with 10 CFR part 63, *Disposal of High-Level Radioactive Wastes in a Geologic Repository at Yucca Mountain, Nevada*. That responsibility remains the sole purview of the Department of Energy.

Those conditions unique to the operations at the GROA are included in this performance specification.

2.0 APPLICABLE DOCUMENTS/REFERENCES

2.1 Regulations

10 CFR part 19- 2006 Energy: *Notices, Instructions and Reports to Workers: Inspection and Investigations.*

10 CFR part 20- 2006 Energy: Standards for Protection Against Radiation.

10 CFR part 21- 2006 Energy: Reporting of Defects and Noncompliance.

10 CFR part 26- 2006 Energy: Fitness for Duty Programs.

10 CFR part 50- 2006 Energy: *Domestic Licensing of Production and Utilization Facilities*.

10 CFR part 63- 2006 Energy: Disposal of High-Level Radioactive Wastes in a Geologic Repository at Yucca Mountain, Nevada.

10 CFR part 71- 2006 Energy: *Packaging and Transportation of Radioactive Material*.

10 CFR part 72- 2006 Energy: Licensing Requirements for the Independent Storage of Spent Nuclear Fuel, High-Level Radioactive Waste and Reactor-Related Greater than Class C Waste.

10 CFR part 73- 2006 Energy: Physical Protection of Plants and Materials.

10 CFR part 74- 2006 Energy: *Material Control and Accounting of Special Nuclear Material*.

10 CFR part 140- 2006 Energy: Financial Protection Requirements and Indemnity Agreements.

10 CFR part 835- 2006 Energy: Occupational Radiation Protection.

10 CFR part 961- 2006 Energy: Standard Contract for Disposal of Spent Nuclear Fuel and/or High-Level Radioactive Waste.

40 CFR part 261- 2006 Protection of Environment: Identification and Listing of Hazardous Waste.

49 CFR part 173- 2006 Transportation: Shippers--General Requirements for Shipments and Packagings.

66FR 55732- Disposal of High-Level Radioactive Wastes in a Proposed Geologic Repository at Yucca Mountain, NV, Final Rule. 10 CFR parts 2, 19, 20, 21, 30, 40, 51, 60, 61, 63, 70, 72, 73 and 75.

Nuclear Waste Policy Act of 1982. 42 U.S.C. 10101 et seq.

Resource Conservation and Recovery Act of 1976. 42 U.S.C. 6901 et seq.

2.2 DOE Documents

DOE O 450.1-Change 2; 2005; *Environmental Protection Program;* Washington, D.C.: U.S. Department of Energy.

DOE-STD-1090-2004. 2004. *Hoisting and Rigging (Formerly Hoisting and Rigging Manual)*. Washington, D.C.: U.S. Department of Energy.

DOE O 435.1. 1999. *Radioactive Waste Management*. Washington, D.C.: U.S. Department of Energy.

2.3 NRC Documents

NUREG-1567, Standard Review Plan for Spent Fuel Dry Storage Facilities

NUREG-1536, Standard Review Plan for Dry Cask Storage Systems

NUREG-1617, Standard Review Plan for Transportation Packages for Spent Nuclear Fuel

NUREG-0612, Control of Heavy Loads at Nuclear Power Plants

NUREG-0800, Standard Review Plan for the Review of Safety Analysis Reports for Nuclear Power Plants

NUREG/CR-4461, Tornado Climatology of the Contiguous United States

NUREG-1804, Yucca Mountain Review Plan

Regulatory Guide 1.23, Rev. 0, 1972; *Onsite Meteorological Programs;* Washington, D.C.: U.S. Atomic Energy Commission.

Regulatory Guide 1.76, Rev. 0, 1974; *Design Basis Tornado for Nuclear Power Plants;* Washington, D.C.: U.S. Atomic Energy Commission.

NRC Regulatory Guide 7.9 Standard Format and Content of Part 71 Applications for Approval of Packages for Radioactive Material

NRC Regulatory Guide 3.61 Standard Format and Content for a Topical Safety Analysis Report for a Spent Fuel Dry Storage Cask

SFPO-ISG-11, Revision 3, Cladding Considerations for the Transportation and Storage of Spent Fuel

SFPO-ISG-18, The Design/Qualification of Final Closure Welds on Austenitic Stainless Steel Canisters as Confinement Boundary for Spent Fuel Storage and Containment Boundary for Spent Fuel Transportation; NRC Interim Staff Guidance

2.4 Codes and Standards

AAR (Association of American Railroads) 1993. *Manual of Standards and Recommended Practices, Section C – Part II, Specifications for Design, Fabrication and Construction of Freight Cars M-1001, Volumes I and II Standards*. Washington, D.C.: Association of American Railroads. TIC: 10188.

AAR 2004. *Manual of Standards and Recommended Practices*. Washington, D.C.: Association of American Railroads. TIC: 256289.

AASHTO (American Association of State Highway and Transportation Officials) 2004. *A Policy on Geometric Design of Highways and Streets.* 5th Edition. Washington, D.C.: American Association of State Highway and Transportation Officials. TIC: 257443.

ANSI/ANS-57.7-1988. American National Standard Design Criteria for an Independent Spent Fuel Storage Installation (Water Pool Type). Revision of ANSI/ANS 57.7-1981. La Grange Park, Illinois: American Nuclear Society. TIC: 238870.

ANSI N14.5-97. 1998. American National Standard for Radioactive Materials - Leakage Tests on Packages for Shipment. New York, New York: American National Standards Institute. TIC: 247029.

ANSI/ANS-57.9. 1992. Design Criteria for an Independent Spent Fuel Storage Installation (Dry Type). La Grange Park, Illinois: American Nuclear Society. TIC: 3043.

ASCE 7-98. 2000. *Minimum Design Loads for Buildings and Other Structures*. Revision of ANSI/ASCE 7-95. Reston, Virginia: American Society of Civil Engineers. TIC: 247427.

ASME (American Society of Mechanical Engineers) 2004. 2004 ASME Boiler and Pressure Vessel Code. 2004 Edition. New York, New York: American Society of Mechanical Engineers. TIC: 256479.

ASTM A-276-06. 2006. Standard *Specification for Stainless Steel Bars and Shapes*. West Conshohocken, PA: ASTM International. TIC: 258258

ASTM A887-89. 2004 Standard Specification for Borated Stainless Steel Plate, Sheet, and Strip for Nuclear Application; Conshohocken, PA 19428: ASTM International. TIC: 258746

ASTM B 932-04. 2004. Standard Specification for Low-Carbon Nickel-Chromium-Molybdenum-Gadolinium Alloy Plate, Sheet and Strip. West Conshohocken, Pennsylvania: American Society for Testing and Materials. TIC: 255846.

ISO 11611984/Cor.1:1990(E). 1990. Series 1 Freight Containers - Corner Fittings - Specification (including Technical Corrigendum 1), 4th Edition. Geneva, Switzerland: International Organization for Standardization. TIC: 258256; 258247.

SEI/ASCE 7-02. 2003. *Minimum Design Loads for Buildings and Other Structures*. Reston, Virginia: American Society of Civil Engineers. TIC: 255517.

IEEE/ASTM SI 10-1997. 1997. Standard for Use of the International System of Units (SI): The Modern Metric System. New York, New York: Institute of Electrical and Electronics Engineers.

2.5 Other References

Transportation, Aging and Disposal Canister System Performance Specification Requirements Rationale; DOC ID: WMO-TADCS-RR-000001 Washington, D.C.: U.S. Department of Energy.

3.0 Performance Requirements

For the purposes of this specification, the following English unit designations and conventions are intended:

lb. = pound force not pound mass

ton = short ton (2,000 lb.)

3.1 TAD Canister

When necessary, the following TAD canister-based system components shall work in conjunction with the TAD canister to meet objectives of this performance specification:

- Transportation Overpack (Section 3.2)
- Aging Overpack (Section 3.3)
- Ancillary Equipment (Not Included in this Specification)
- Shielded Transfer Cask (Not Included in this Specification)
- Site Transporter (Not Included in this Specification)

3.1.1 General

This section applies to the TAD canister, which will be part of a Nuclear Regulatory Commission (NRC) certified system, approved for confining CSNF during storage, transportation, aging and disposal. The TAD canister includes a canister shell, lid(s) and components (e.g., basket for holding fuel assemblies, thermal shunts and neutron absorbers, etc.) needed to perform its functions.

- (1) The TAD canister shall be a right circular cylinder with a diameter of 66.5 in. $\left(\frac{+\ 0.0\ \text{in.}}{-\ 0.5\ \text{in.}}\right)$. The TAD canister height shall not be less than 186.0
 - in. and not greater than 212.0 in. including the lifting feature shown in Attachment C considering all relevant factors (e.g., tolerance stack-up, thermal expansion, internal pressure).
 - a. For a TAD canister with a height less than the maximum, a TAD waste package spacer (TWPS) meeting requirements in Section 3.1.1(17-20) shall be included. If required, the TWPS shall have a diameter of

66.5 in.
$$\left(\frac{+\ 0.0\ \text{in.}}{-\ 0.5\ \text{in.}}\right)$$
 and length such that the combined height of the TWPS

and TAD canister shall be 212.0 in.
$$\left(\frac{+\ 0.0\ in.}{-\ 0.5\ in.}\right)$$
 considering all relevant

factors (e.g., tolerance stack-up, thermal expansion, internal pressure).

- b. If required, the TWPS shall be placed in a waste package prior to loading of the TAD canister for disposal. The TWPS function is to restrict axial motion of the TAD canister within the waste package after emplacement.
- (2) The TAD canister loaded weight shall be consistent with the height determined in accordance with 3.1.1(1). The combined weight of the loaded TAD canister and TWPS shall not exceed 54.25 tons.

- (3) The capacity of the TAD canister shall be either 21 pressurized water reactor (PWR) spent fuel assemblies or 44 boiling water reactor (BWR) spent fuel assemblies.
- (4) The loaded and closed TAD canister shall be capable of being reopened while submerged in a borated or unborated pool.
- (5) A TAD canister for PWR assemblies shall be limited to accepting CSNF with characteristics less than 5% initial enrichment, less than 80 GWd/MTU burn up and no less than 5 years out-of-reactor cooling time. 1,3
- (6) A TAD canister for BWR assemblies shall be limited to accepting CSNF with characteristics less than 5% initial enrichment, less than 75 GWd/MTU burnup and no less than 5 years out-of-reactor cooling time.^{2,3}
- (7) A TAD canister shall be capable of being loaded with CSNF from one or more facilities that are licensed by the NRC and hold one or more contracts with the DOE for disposal of CSNF.³
- (8) All external edges of the TAD canister shall have a minimum radius of curvature of 0.25 in.
- (9) To the extent practicable, projections or protuberances from reasonably smooth adjacent surfaces shall be avoided or smoothly blended into the adjacent smooth surfaces.
- (10) The TAD canister shall be designed to store vendor defined design basis CSNF at a purchaser site in accordance with 10 CFR part 72 in either a horizontal or vertical orientation.
- (11) A TAD canister shall be designed to transport vendor defined design basis CSNF to the GROA in a horizontal configuration.
- (12) A TAD canister shall be designed to dispose of vendor defined design basis CSNF in a waste package in a horizontal configuration.
- (13) A TAD canister shall be designed to be handled at the GROA loaded with vendor defined design basis CSNF in a vertical configuration.
- (14) A TAD canister shall be designed to age vendor defined design basis CSNF in a vertical configuration.

¹ These characteristics represent bounding PWR characteristics used in the repository design basis and provide enveloping conditions for repository shielding, thermal and dose consequence analysis.

² These characteristics represent bounding BWR characteristics used in the repository design basis and provide enveloping conditions for the repository shielding, thermal and dose consequence analysis.

³ TAD canister design basis SNF (i.e., approved contents) chosen by the vendor shall be any assembly subset with characteristics bounded by the limits defined by 3.1.1(5) or 3.1.1(6).

- (15) At the time of delivery to the repository, a loaded TAD canister shall have a remaining service lifetime for aging of 50 years without maintenance.⁴
- (16) The service lifetime environmental conditions shall be site appropriate for the period of deployment at reactors. Yucca Mountain environmental conditions apply for repository aging service.
- (17) TWPS shall be constructed of materials specified in 3.1.8 (1).
- (18) TWPS shall be a right circular cylinder, either solid or hollow with sides and ends formed from plates at least 2 inches thick.
- (19) The TWPS shall have an average mass density equal to or greater than that of the loaded TAD canister.⁵
- (20) The TWPS shall include four (4) threaded holes in its top for the purpose of attaching temporary rigging meeting requirements of NUREG-0612, *Control of Heavy Loads at Nuclear Power Plants* to be used when inserting the TWPS into an otherwise empty waste package.

3.1.2 Structural

- (1) For each of the following design basis seismic events and configurations, the TAD canister shall meet the performance specifications. Seismic return vertical and horizontal accelerations are detailed in Attachment A.
 - a. Following a 2,000-year seismic return period event, a TAD canister shall maintain a maximum leakage rate of 1.5×10⁻¹² fraction of canister free volume per second⁶ (normal), maximum cladding temperature of 752° F (normal) and remain within design codes while in the configurations described below.
 - While suspended by a crane inside an ASTM A-36 cylindrical steel cavity with an inner diameter of 72.5 inches with 12 inch thick wall.
 - While contained in a vendor defined transportation overpack (with impact limiters) described in Section 3.2 of this performance specification.
 - While contained in a vendor defined transportation overpack (without impact limiters) described in Section 3.2 of this performance specification that is constrained in an upright position. A constrained transportation overpack is one properly secured into GROA transfer trolley and restrained from tip-over in a seismic event.
 - While contained in a vendor defined aging overpack as described in Section 3.3 of this performance specification.

⁴ Prior to delivery to the repository, a loaded TAD canister may have been stored at a reactor site for up to 60 years.

⁵ The average mass density is determined by dividing the total mass of the TAD canister/TWPS by the volume of a right circular cylinder with same diameter and height.

⁶ This leakage rate meets the leak-tight criterion of ANS/ANSI-N14.5, American National Standard for Radioactive Materials - Leakage Tests on Packages for Shipment.

- b. Following a 10,000-year seismic return period event, a TAD canister shall maintain a maximum leakage rate of 1.5×10^{-12} fraction of canister free volume per second⁶ (normal), cladding temperature limit of 1,058° F (offnormal) and remain within design codes while in the configurations described below.
 - While suspended by a crane inside an ASTM A-36 cylindrical steel cavity with an inner diameter of 72.5 inches with 12 inch thick wall.
 - While contained in a vendor defined transportation overpack (with impact limiters) described in Section 3.2 of this performance specification.
 - While contained in a vendor defined transportation overpack (without impact limiters) described in Section 3.2 of this performance specification that is constrained in an upright position. A constrained transportation overpack is one properly secured into GROA transfer trolley and restrained from tip-over in a seismic event.
 - While contained in a vendor defined aging overpack as described in Section 3.3 of this performance specification.
- c. Following a seismic event characterized by horizontal and vertical peak ground accelerations of 96.52 ft/s 2 (3g) a TAD canister shall maintain a maximum leakage rate of 1.5×10^{-12} fraction of canister free volume per second 6 (normal) while in the configurations described below. For this initiating event, canister design codes may be exceeded (i.e., vendor may rely on capacity in excess of code allowances).
 - A TAD canister in a vendor defined transportation cask described in Section 3.2 that drops 10 feet onto an unyielding surface in the most damaging orientation. The transportation cask configuration shall be with or without impact limiters.
 - While contained in a vendor defined transportation overpack (without impact limiters) described in Section 3.2 of this performance specification that is constrained in an upright position. A constrained transportation overpack is one properly secured into GROA transfer trolley and restrained from tip-over in a seismic event.
 - While contained in a vendor defined aging overpack as described in Section 3.3 of this performance specification.
- (2) A TAD canister in a vendor defined aging overpack shall maintain a maximum leakage rate of 1.5×10^{-12} fraction of canister free volume per second⁶ (normal) and cladding temperature limits (see inset) during and following exposure to the environmental conditions listed below.

For a - e, the cladding temperature limits are 752° F and 1,058° F for "normal" and "off-normal" limits, respectively.

a. These environmental conditions are not cumulative but occur independently:

- Outdoor average daily temperature range of 2° F to 116° F with insolation as specified in 10 CFR part 71 (normal)
- An extreme wind gust of 120 mph for 3-sec (normal)
- Maximum tornado wind speed of 189 mph with a corresponding pressure drop of 0.81 lb/in² and a rate of pressure drop of 0.30 lb/in²/sec (off-normal). The spectrum of missiles from the maximum tornado is provided in Table 3.1-1 (off-normal):

Table 3.1-1 Spectrum of Missiles

Missile	Mass (lb)	Dimensions (ft)	Hor. Vel. (ft/s)
Wood Plank	114.6	$0.301\times0.948\times12$	190.2
6" Schedule 40 pipe	286.6	$0.551D \times 15.02$	32.8
1 in. steel rod	8.8	$0.0833D \times 3$	26.3
Utility Pole	1,124	$1.125D \times 35.04$	85.3
12" Schedule 40 pipe	749.6	1.05D × 15.02	23.0

b. Annual precipitation of 20 inches/year (normal). The spectrum of rainfall is provided in Table 3.1-2 (normal):

Table 3.1-2 Spectrum of Rainfall

Table 5.1-2 Spectrum of Kannan			
Parameter and Frequency	Nominal Estimate	Upper Bound 90% Confidence Interval*	
Maximum 24-hr precipitation (50-year return period)	2.79 in./day	3.30 in./day	
Maximum 24-hr precipitation (100-year return period)	3.23 in./day	3.84 in./day	
Maximum 24-hr precipitation (500-year return period)	4.37 in./day	5.25 in./day	
Precipitation 1-hr intensity (50-year return period)	1.35 in./hr	1.72 in./hr	
Precipitation 1-hr intensity (100-year return period)	1.68 in./hr	2.15 in./hr	

^{*}Use the values for upper bound 90% confidence interval.

- c. Maximum daily snowfall of 6.0 in. (normal)
- d. Maximum monthly snowfall of 6.6 in. (normal)
- e. A lightning strike with a peak current of 250 kiloamps over a period of 260 microseconds and continuous current of 2 kiloamps for 2 seconds (off-normal).
- (3) A TAD canister in a transportation overpack (with impact limiters) shall maintain a maximum leakage rate of 1.5×10^{-12} fraction of canister free volume per second⁶ (off-normal) and cladding temperature limits (see inset) during and following exposure to the environmental conditions listed below.

For a - e, the cladding temperature limits are 752° F and 1,058° F for "normal" and "off-normal" limits, respectively.

- a. These environmental conditions are not cumulative but occur independently:
 - Outdoor average daily temperature range of 2° F to 116° F with insolation as specified in 10 CFR part 71 (normal)
 - An extreme wind gust of 120 mph for 3-sec (normal)
 - Maximum tornado wind speed of 189 mph with a corresponding pressure drop of 0.81 lb/in² and a rate of pressure drop of 0.30 lb/in²/sec (off-normal). The spectrum of missiles from the maximum tornado is provided in Table 3.1-3 (off-normal):

Table 3.1-3 Spectrum of Missiles

		L	
Missile	Mass (lb)	Dimensions (ft)	Hor. Vel. (ft/s)
Wood Plank	114.6	$0.301\times0.948\times12$	190.2
6" Schedule 40 pipe	286.6	$0.551D \times 15.02$	32.8
1 in. steel rod	8.8	$0.0833D \times 3$	26.3
Utility Pole	1,124	$1.125D \times 35.04$	85.3
12" Schedule 40 pipe	749.6	$1.05D \times 15.02$	23.0

b. Annual precipitation of 20 inches/year (normal). The spectrum of rainfall is provided in Table 3.1-2 (normal):

Table 3.1-4 Spectrum of Rainfall

=			
Parameter and Frequency	Nominal Estimate	Upper Bound 90% Confidence Interval*	
Maximum 24-hr precipitation (50-year return period)	2.79 in./day	3.30 in./day	
Maximum 24-hr precipitation (100-year return period)	3.23 in./day	3.84 in./day	
Maximum 24-hr precipitation (500-year return period)	4.37 in./day	5.25 in./day	
Precipitation 1-hr intensity (50-year return period)	1.35 in./hr	1.72 in./hr	
Precipitation 1-hr intensity (100-year return period)	1.68 in./hr	2.15 in./hr	

- *Use the values for upper bound 90% confidence interval.
 - c. Maximum daily snowfall of 6.0 in. (normal)
 - d. Maximum monthly snowfall of 6.6 in. (normal)
 - e. A lightning strike with a peak current of 250 kiloamps over a period of 260 microseconds and continuous current of 2 kiloamps for 2 seconds (off-normal).

(4) The TAD canister shall have a flat bottom.

3.1.3 Thermal

- (1) Except as noted in 3.1.3 (2), CSNF cladding temperature in TAD canisters shall not exceed 752° F during normal operations. Normal operations include storage at purchaser sites, transportation from purchasers to the GROA and handling at the GROA (e.g., aging, storage, onsite transfer, etc).
- (2) CSNF cladding temperature shall not exceed 1,058° F during draining, drying and backfill operations following TAD canister loading.
- (3) The maximum leakage rate of a TAD canister shall be 9.3×10^{-10} fraction of canister free volume per second (off-normal) after a fully-engulfing fire characterized by an average flame temperature of 1,720 °F and lasting 30 minutes. During this event the TAD canister is in either a closed vendor defined transportation overpack (with or without impact limiters) or an open vendor defined transportation overpack without impact limiters. For this event, canister design codes may be exceeded (i.e., vendor may rely on capacity in excess of code allowances).
- (4) TAD canister cooling features and mechanisms shall be passive.
- (5) To ensure adequate thermal performance of the TAD canister when emplaced in the waste package, the peak cladding temperature shall be less than 662° F for each set of conditions in Table 3.1-3.

Table 3.1-3 Thermal Conditions for Cladding Temperature Determination

Temperature Betermination			
Thermal Output	Canister Surface Temperature		
(kW)	Boundary Conditions (°F)		
11.8	525		
18	450		
25	358		

3.1.4 Dose and Shielding

- (1) For GROA operations, the combined neutron and gamma integrated average dose rate over the top surface of a loaded TAD canister shall not exceed 800 mrem/hr on contact.
- (2) For GROA operations, the combined contact neutron and gamma maximum dose rate at any point on the top surface of the TAD canister shall not exceed 1,000 mrem/hr.
- (3) The TAD canister shall be designed such that contamination on an accessible external surface shall be removable to:
 - a. $1,000 \text{ dpm}/100 \text{ cm}^2$ beta-gamma with a wipe efficiency of 0.1.

b. $20 \text{ dpm}/100 \text{ cm}^2$ - alpha with a wipe efficiency of 0.1

3.1.5 Criticality

- (1) No specific requirements beyond those of 10 CFR Part 71, Subpart E, Paragraph 55(b).
- (2) Postclosure Criticality control shall be maintained by employing either the items in (a) or the analysis in (b), as follows:
 - a. Include the following features in the TAD canister internals:
 - 1. Neutron absorber plates or tubes made from borated stainless steel produced by powder metallurgy and meeting ASTM A887-89, *Standard Specification for Borated Stainless Steel Plate, Sheet, and Strip for Nuclear Application*, Grade "A" alloys.
 - 2. Minimum thickness of neutron absorber plates shall be 0.433 inches. Maximum and nominal thickness may be based on structural requirements. Multiple plates may be used if corrosion assumptions (250 nm/year) are taken into for all surfaces such that 6 mm remains after 10,000 years.
 - 3. The neutron absorber plate shall have a boron content of 1.1 wt % to 1.2 wt %, a range that falls within the specification for 304B4 UNS S30464 as described in ASTM A887-89, *Standard Specification for Borated Stainless Steel Plate, Sheet, and Strip for Nuclear Application*.
 - 4. Neutron absorber plates or tubes shall extend along the full length of the active fuel region inclusive of any axial shifting of the assemblies within the TAD canister.
 - 5. Neutron absorber plates or tubes must cover all four longitudinal sides of each fuel assembly.
 - 6. TAD canister designs for PWR fuel assemblies shall accommodate assemblies loaded with a disposal control rod assembly (DCRA⁷). A DCRA is intended for acceptance of PWR CSNF with characteristics outside limits set in the postclosure criticality loading curves. Current postclosure criticality loading curves are shown in Attachment B of this performance specification. Updated postclosure criticality loading curves that represent a PWR TAD canister with features described in items 1 through 5 of this subsection may be provided at a later date.
 - b. Perform analyses of TAD canister-based systems to ensure the maximum calculated effective neutron multiplication factor $(k_{eff})^8$ for a TAD canister containing the most reactive CSNF for which the design is approved shall

OCRA is similar to control rod assemblies, reactivity control assemblies, reactivity control cluster assemblies or burnable poison rod assemblies placed in fuel assemblies during irradiation in reactors. A primary difference is extra thick zircaloy cladding, absorber materials that extend beyond the active fuel length and spiders that hold rods have thick zircaloy or titanium locking mechanism(s).

⁸ The maximum k_{eff} for a configuration is the value at the upper limit of a two-sided 95% confidence interval.

 $\underline{\text{not}}$ exceed the critical limit 9 for four postclosure archetypical proxy configurations. 10,11

3.1.6 Containment

- (1) The TAD canister design shall meet either of the requirements below.
 - a. The qualification of the TAD canister final closure welds shall meet SFPO-ISG-18, Design/Qualification of Final Closure Welds on Austenitic Stainless Steel Canisters as Confinement Boundary for Spent Fuel Storage and Containment Boundary for Spent Fuel Transportation, for assuring no credible leakage for containment and confinement.
 - b. The TAD canister shall be designed to facilitate helium leak testing of closure features using methods that can demonstrate the defined leak-tight requirements have been met. Leak testing shall be performed in accordance with ANSI N14.5-97, *American National Standard for Radioactive Materials Leakage Tests on Packages for Shipment*.
- (2) Helium shall be the only gas used for final backfill operations.
- (3) TAD canister shell and lid shall be designed and fabricated in accordance with *ASME Boiler and Pressure Vessel Code*, Section III, Division 1, Subsection NB (for Class 1 Components). Vendor shall identify applicable exceptions, clarifications, interpretations, and code cases.
- (4) In accordance with industry standards and regulatory guidance, the TAD canister shall be designed to facilitate the following:
 - a. Draining and drying to remove water vapor and oxidizing material shall be carried out in accordance with NUREG-1536, *Standard Review Plan for Dry Cask Storage Systems Final Report, USNRC*, January 1997.
 - b. Filling with helium to atmospheric pressure or greater as required to meet leak test procedural requirements.
 - c. Sampling of the gas space to verify helium purity.
 - d. Limiting maximum allowable oxidizing gas concentration within the loaded and sealed TAD canister to 0.20% of the free volume in the TAD canister at atmospheric pressure.

The critical limit is the value of k_{eff} at which a configuration is considered potentially critical including biases and uncertainties (BSC 2004, Section 6.3.1).

The Criticality Input to Canister Based System Performance Specification for Disposal (BSC 2006, Section 3.1) provides a set of considerations for determining the proxy configurations based upon analyses of different, but similar, waste package designs. A list of the four proxy configuration cases are:

a. Nominal case, basket assembly degraded, CSNF intact.

b. Seismic case-I, basket assembly intact, CSNF degraded.

c. Seismic case-II, basket assembly degraded, CSNF degraded.

d. Igneous intrusion case, basket assembly degraded, CSNF degraded, waste package and TAD structural deformation.

A system performance assessment is a comprehensive analysis estimating dose incurred by reasonably maximally exposed individual, including associated uncertainties, as a result of repository releases caused by all significant features, events, processes, and sequences of events and processes, weighted by their probability of occurrence (YMP 2003, Appendix B).

(5) A loaded TAD canister shall maintain a leakage rate of 1.5×10^{-12} fraction of canister free volume per second⁶ (normal) and cladding temperature below 752° F (normal) following a 12 inch vertical flat-bottom drop. The impacted surface is a solid carbon steel plate, simply supported as shown in Figure 3.1-1. The material conforms to ASTM A36/A36M, Standard Specification for Carbon Structural Steel. Centerline of the TAD canister may be offset from centerline of the plate by as much as three (3) inches.

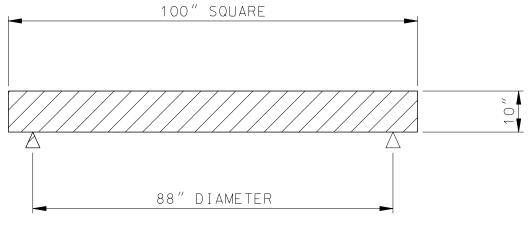


Figure 3.1-1

3.1.7 Operations

- (1) The TAD canister lid shall be designed for handling under water with the TAD canister in a vertical orientation.
- (2) The TAD canister body and lid shall have features to center and seat the lid during submerged installation. The maximum off-center value is ½ in.
- (3) A feature for lifting a vertically oriented, loaded TAD canister from the lid shall be provided. The lifting feature may be integral with the lid or mechanically attached. The lifting feature shall be in place and ready for service prior to transport to the repository. A sketch of the lifting feature that shall be used is shown in Attachment C.
- (4) An open, empty and vertically oriented TAD canister shall have integral lifting feature(s) provided to allow lifting by an overhead handling system.
- (5) The TAD canister shall be designed with features such that draining, drying and backfill operations take advantage of "as low as reasonably achievable" (ALARA) principles.

3.1.8 Materials

(1) Required Materials- Except for thermal shunts and criticality control materials, the TAD canister and structural internals (i.e., basket) shall be constructed of a Type 300-series stainless steel (UNS S3XXXX, such as

UNS S31603, which may also be designated as type 316L) as listed in ASTM A-276-06, *Standard Specification for Stainless Steel Bars and Shapes*.

(2) The TAD canister and its basket materials shall be designed to be compatible with either borated or unborated repository pool water as defined in Table 3.1-4.

Table 3.1-4.	Repository Pool Water Specifications			
Average annual pool	<90° F (Pool water ter	<90° F (Pool water temperature may exceed		
water temperature	110° F for no more that	110° F for no more than 5% of the time		
	during June, July, Aug	gust, and September.)		
	Unborated Pool	Unborated Pool Borated Pool		
Average annual pool water conductivity	<3 μ-mho/cm	<3 μ-mho/cm		
Pool water chloride concentration	<0.5 ppm	<0.5 ppm		
Pool water pH	5.3 to 7.5	4.5 to 9.0		
Pool water boron concentration	-	2000 to 2500 ppm		

(3) Prohibited or Restricted Materials

- a. The TAD canister shall not have organic, hydrocarbon-based materials of construction.
- b. All metal surfaces shall meet surface cleanliness classification C requirement defined in ASME NQA-1-2000 Edition, Subpart 2.1 *Quality Assurance Requirements for Cleaning of Fluid Systems and Associated Components for Nuclear Power Plants*.
- c. The TAD canister shall not be constructed of pyrophoric materials.
- d. The TAD canister, including the steel matrix, gaskets, seals, adhesives and solder, shall not be constructed with materials that would be regulated as hazardous wastes under the Resource Conservation and Recovery Act (RCRA) and prohibited from land disposal under RCRA if declared to be waste.

(4) Markings

- a. The TAD canister shall be capable of being marked on the lid and body with an identical unique identifier prior to delivery for loading.
- b. The unique identifier space shall be of suitable length and height to contain nine (9) alphanumeric and two (2) special characters (e.g., -, /, "space", etc.) to be specified by the DOE.
- c. Alphanumeric characters shall have a minimum height of 6 in.

- d. The markings shall remain legible without intervention or maintenance during/after any of the following events:
 - The entire service life defined in Section 3.1.1.
 - Normal operations to include loading, closure, storage, transportation, aging and disposal.
 - Dose, heat and irradiation associated with the vendor defined design basis PWR or BWR, as applicable.

3.2 Transportation Overpack

3.2.1 General

- (1) The transportation overpack cavity shall accommodate a TAD canister formed as a right-circular cylinder with a length including the lifting feature as specified by the vendor in accordance with 3.1.1(1) and a diameter of 66.5 in.; and Attachment C.
- (2) The transportation overpack shall function with a vendor defined TAD canister that meets the requirements of Section 3.1.
- (3) The loaded transportation overpack (without impact limiters) shall be designed to be lifted in a vertical orientation by an overhead crane.
- (4) The loaded transportation overpack (without impact limiters) shall be able to stand upright when set down upon a flat horizontal surface without requiring the use of auxiliary supports.
- (5) The size and weight of the loaded transportation overpack shall be limited to the characteristics provided in Table 3.2-1.

Table 3.2-1 Transportation Overpack Characteristics

Characteristic	Value
Maximum cask length without impact limiters (in.)	230
Maximum cask length with impact limiters (in.)	333
Maximum cask diameter without impact limiters (in.)	98
Maximum cask lid diameter (in.)	84
Maximum distance across upper trunnions (in.)	108
Maximum diameter of impact limiters (in.)	126
Maximum weight of fully loaded overpack without impact limiters (lb.)	250,000
Maximum weight of fully loaded overpack, impact limiters and transportation skid (lb.)	360,000

(6) Lifting attachments and appurtenances on transportation overpacks, overpack lids and impact limiters shall be designed, documented and

fabricated in accordance with NUREG-0612 Control of Heavy Loads at Nuclear Power Plants.

3.2.2 Structural

A loaded TAD canister contained within a transportation overpack assembled with any other components included in the packaging, as defined in 10 CFR part 71, shall meet the requirements for a Type B cask as specified in 10 CFR part 71, as evidenced by a valid Certificate of Compliance.

3.2.3 Thermal

- (1) During normal operations, the CSNF cladding temperature in the TAD canister shall not exceed 752° F. Normal operations include transportation from purchaser sites to the GROA.
- (2) Transportation overpacks cooling features and mechanisms shall be passive.

3.2.4 <u>Dose and Shielding</u>

- (1) The transportation overpack impact limiters shall include design and handling features that use standardized tools and features that simplify removal operations. Standard tools are those that can be found in industrial tool catalogs.
- (2) Supplemental shielding shall not be required in vacant trunnion locations to meet dose requirements for transporting the TAD canister with vendor defined contents.
- (3) Transportation overpack shall be designed such that contamination on accessible external surfaces shall be removable to:
 - a. $1,000 \text{ dpm}/100 \text{ cm}^2$ beta-gamma with a wipe efficiency of 0.1.
 - b. 20 dpm/100 cm² alpha with a wipe efficiency of 0.1.

3.2.5 <u>Criticality</u>

No specific requirements beyond those of 10 CFR part 71.

3.2.6 Containment

The loaded transportation overpack shall have a tamper indicating device (TID) that meets requirements of 10 CFR part 73 *Physical Protection of Plants and Materials*.

3.2.7 Operations

 Normal operational procedures shall <u>not</u> require submergence of transportation overpack into CSNF pool at repository or loading site. Transportation overpacks may be submerged in pool in unusual or offnormal circumstances.

- (2) Transportation overpack shall have closures that can be bolted and unbolted using standard tools. Standard tools are those that can be found in industrial tool catalogs.
- (3) The transportation overpack shall have trunnions that meet the following requirements.
 - a. There shall be two (2) upper (lifting) trunnions with the centerline located between 8 and 24 inches from the top of the vendor defined transportation overpack.
 - b. There shall be two (2) lower (rotation) trunnions with the centerline located less than 36 inches from the bottom of the vendor defined transportation overpack.
 - c. The centerline of each trunnion set shall be outside the area of the spent fuel region to provide maximum ALARA benefits.
- (4) The transportation overpack shall have upper lifting trunnions with dual seats.
 - a. The smaller seat (lifting yoke interface) shall have a diameter of 6.75 ±0.25 inches and an axial width of no less than 2.5 inches.
 - b. The diameter of the end caps shall not exceed 8.75 inches.
- (5) Transportation skid shall be designed to permit the loaded transportation overpack, without impact limiters, to be upended by rotation about its lower trunnions and removed from the transportation skid in a vertical orientation via overhead crane.
- (6) The lower turning trunnions shall be pocket trunnions and recessed into the cask body.
- (7) The upper trunnions shall:
 - a. Be mechanically fastened to the cask body.
 - b. Incorporate features for installation and removal that maximize ALARA principles. Repository goal is to limit total dose for installing or removing the trunnions to less then 40 millirem per pair.
- (8) The upper trunnions shall be removed and stowed during transport.
- (9) The transportation overpack lid shall have a lifting ring that is:
 - a. Identical to that of the TAD canister as shown in Attachment C.
 - b. Is removable from the transportation overpack lid.
 - c. Capable of handling the unencumbered transportation overpack lid.

- (10) The transportation skid to be used with the TAD canister-based system shall have the following characteristics:
 - a. Secures the transportation overpack during normal conditions of transport in accordance with requirements of 10 CFR part 71.45.
 - b. Secures to the railcar in accordance with requirements of AAR Interchange Rule 88, A.15.c.3. (AAR Field Manual 2006)
 - c. Design shall facilitate lifting of the loaded package in its transportation configuration, including the skid and impact limiters, and transfer of the package from one conveyance to another.
 - d. The footprint of the transportation skid shall not exceed 124 inches wide by 360 inches long.
 - e. Vendor skid design shall be compatible with all variations of their TAD canister-based system in a transportation configuration (e.g., PWR and BWR variants).
 - f. Shall be designed to permit the loaded vendor defined transportation overpack, without impact limiters, to be upended by rotation about its lower trunnions and removed in a vertical orientation via overhead crane.
 - g. Skid shall be designed such that the bottom of loaded vendor defined transportation overpack (in a vertical orientation) shall not be required to be lifted more than 12'-3" above grade elevation (top of rail). The conveyance deck height will not be greater than 54" above grade elevation.

3.2.8 Materials

Materials selections shall be as necessary to meet requirements of 10 CFR part 71 and other requirements of this specification.

3.3 Aging Overpack

3.3.1 General

- (1) The aging overpack cavity shall accommodate a TAD canister formed as a right-circular cylinder with a length including the lifting feature as specified by the vendor in accordance with 3.1.1(1) and a diameter of 66.5 in.; and Attachment C.
- (2) The aging overpack shall function with a TAD canister that has a loaded weight consistent with vendor specified dimensions in accordance with 3.1.1(1, 2).
- (3) The combined size and weight of the loaded TAD canister-based system in an aging overpack shall be limited to ensure handling at the GROA. The limits are provided in Table 3.3-1.

Table 3.3-1 Combined Size and Weight		
Limits		
Maximum overpack diameter	144 in.	
Maximum overpack lid diameter	84 in.	
Maximum overpack lid thickness	18 in.	
Maximum overpack length	264 in.	
Maximum overpack weight (loaded)	250 tons	

- (4) The aging overpack shall meet the operational requirements detailed in sketch presented in Attachment D.
- (5) The aging overpack shall be designed to be moved in a vertical orientation.
- (6) The aging overpack lid shall have a lifting ring that is:
 - a. Identical to that of the TAD canister as shown in Attachment C.
 - b. Capable of handling the unencumbered aging overpack lid.
- (7) The designed maintainable service lifetime of the aging overpack shall be a minimum of 100 years.

3.3.2 Structural

- (1) For each design basis seismic events defined below, the TAD canister in an aging configuration shall meet the following performance specifications. Seismic return vertical and horizontal accelerations are detailed in Attachment A.
 - a. Following a 2,000-year seismic return period event:
 - TAD canister in an aging overpack, shall maintain a maximum leakage rate of 1.5×10^{-12} fraction of canister free volume per second⁶ (normal)
 - Maintain a maximum cladding temperature of 752° F (normal)
 - Canister design codes shall not be exceeded.
 - The aging overpack shall remain upright and free standing.
 - b. Following a 10,000-year seismic return period event:
 - TAD canister in an aging overpack, shall maintain a maximum leakage rate of 1.5×10^{-12} fraction of canister free volume per second⁶ (normal)
 - Maintain a maximum cladding temperature of 1,058° F (off-normal)
 - Canister design codes shall not be exceeded.
 - The aging overpack shall remain upright and free standing.
 - c. Following a seismic event characterized by horizontal and vertical peak ground accelerations of 96.52 ft/s² (3g):
 - TAD canister in an aging overpack, shall maintain a maximum leakage rate of 1.5×10^{-12} fraction of canister free volume per second⁶ (normal)

- <u>Canister design codes may be exceeded</u> (i.e., vendor may rely on capacity in excess of code allowances).
- The aging overpack shall remain upright and free standing during and following the event.
- (2) During GROA operations, aging overpack shall be designed to maintain a maximum TAD canister leakage rate of 1.5×10⁻¹² fraction of free volume per second⁶ (normal) and cladding temperature limits (see inset) during and following exposure to the environmental conditions listed below.

For 2a - 2e, the cladding temperature limits are 752° F and 1,058° F for "normal" and "off-normal" limits, respectively.

- a. These environmental conditions are not cumulative but occur independently:
 - Outdoor average daily temperature range of 2° F to 116° F with insolation as specified in 10 CFR part 71 (normal)
 - An extreme wind gust of 120 mph for 3-sec (normal)
 - Maximum tornado wind speed of 189 mph with a corresponding pressure drop of 0.81 lb/in² and a rate of pressure drop of 0.30 lb/in²/sec (off-normal). The spectrum of missiles from the maximum tornado is provided in Table 3.3-2 (off-normal).

Table 3.3-2 Spectrum of Missiles

Missile	Mass (lb)	Dimensions (ft)	Hor. Vel. (ft/s)
Wood Plank	114.6	$0.301\times0.948\times12$	190.2
6" Schedule 40 pipe	286.6	$0.551D \times 15.02$	32.8
1 in. steel rod	8.8	$0.0833D \times 3$	26.3
Utility Pole	1,124	$1.125D \times 35.04$	85.3
12" Schedule 40 pipe	749.6	$1.05D \times 15.02$	23.0

b. Annual precipitation of 20 inches/year (normal). The spectrum of rainfall is provided in Table 3.3-3 (normal):

Table 3.3-3 Spectrum of Rainfall

Parameter and Frequency	Nominal Estimate	Upper Bound 90% Confidence Interval*
Maximum 24-hr precipitation (50-year return period)	2.79 in./day	3.30 in./day
Maximum 24-hr precipitation (100-year return period)	3.23 in./day	3.84 in./day
Maximum 24-hr precipitation (500-year return period)	4.37 in./day	5.25 in./day
Precipitation 1-hr intensity (50-year return period)	1.35 in./hr	1.72 in./hr
Precipitation 1-hr intensity (100-year return period)	1.68 in./hr	2.15 in./hr

^{*}Use the values for upper bound 90% confidence interval.

- c. Maximum daily snowfall of 6.0 in. (normal)
- d. Maximum monthly snowfall of 6.6 in. (normal)
- e. A lightning strike with a peak current of 250 kiloamps over a period of 260 microseconds and a continuing current of 2 kiloamps for 2 seconds (off-normal).
- (3) Following an impact (with resultant fire) from an F-15 military aircraft into an aging overpack, the TAD canister shall maintain a maximum leak rate of 9.3×10^{-10} fraction of canister free volume per second (off-normal) and maximum cladding temperature 1,058° F (off-normal). The analysis shall assume the following:
 - a. The crash speed is 500 ft/sec.
 - b. Impact orientation analyzed shall be that which results in maximum damage.
 - c. 12,000 lbs of JP-8 fuel.
 - d. F-15 airframe.
 - e. Two engine components of 3,740 lbs. and dimensions of 46.5 inches D \times 191 inches each spaced 96 inches apart.
 - f. One (1) M61A1 20-mm cannon mounted internally just off center of axis.
 - g. 1,000 lbs of inert armaments (i.e., dummy bombs) located between the engines.
- (4) The TAD canister in an aging overpack shall be designed to a maximum leakage rate of 1.5×10^{-12} fraction of canister free volume per second⁶ (normal) and maximum cladding temperature of $1,058^{\circ}$ F (off-normal) following 4 in. of volcanic ash accumulation. The aging overpack may be

- on a site transporter. The ash fall loads are estimated at 21 lb/ft 2 with a thermal conductivity of 0.11 BTU/hr-ft- $^\circ$ F.
- (5) The aging overpack shall retain the TAD canister following a drop and/or tip-over event.
- (6) The aging overpack top shall have one (1) lift feature in each quadrant to allow for lifting using temporary rigging and portable crane. The lifting features shall be of sufficient size to allow any two (2) to upright and lift a loaded aging overpack.
- (7) For analysis purposes, the aging pad shall be assumed to have the following characteristics:
 - a. 5,000 PSI concrete with a minimum thickness of three feet and a maximum thickness of seven feet.
 - b. Concrete surface is a light broom finish.
 - c. Reinforcing steel shall be #11 on 12 in. centers, each direction, top and bottom, standard cover top and bottom.
 - d. Soil data is in Attachment E.

3.3.3 Thermal

- (1) Aging overpack cooling features and mechanisms shall be passive.
- (2) A loaded aging overpack shall be capable of withstanding a fully engulfing fire without the TAD canister exceeding a leakage rate of 9.3×10^{-10} fraction of canister free volume per second (off-normal) and maximum fuel cladding temperature of 1,058° F (off-normal) under the conditions below.
 - a. The resulting fire described in section 3.3.2 (3) (aircraft impact) of this performance specification.
 - b. The fire described in 10 CFR 71.73.c (4) *Hypothetical Accident Condition* requirements as modified below.
 - 1. The 30-minute period shall be replaced by a period to be determined by calculation of a pool spill fire formed by 100 gallons of diesel fuel.
 - 2. Additionally, a surrogate fully engulfing fire of duration twice the duration of the pool fire which starts simultaneously with the pool fire and with a steady-state heat release rate of 10 MW shall be used to model the burning rate of all other solid and liquid combustible materials. For this purpose, assume the heat transfer conditions specified in 10 CFR 71.73.c (4). Temperature conditions from this fire shall be consistent with a totally engulfing black body emitting from the 10 MW requirement.

c. A loaded aging overpack shall withstand a deflagration blast wave, fuel tank projectiles and incident thermal radiation resulting from the worst case engulfing fire¹² determined in the previous fire protection requirement without the TAD canister exceeding a leakage rate of 9.3×10⁻¹⁰ fraction of canister free volume per second (off-normal) and maximum fuel cladding temperature of 1,058° F (off-normal).

3.3.4 Dose and Shielding

When the loaded aging overpack is on the aging pad with its vertical axis in its normal orientation, the combined neutron and gamma contact dose rate on any accessible exterior surface (excluding the underside of the aging overpack) shall not exceed 40 mrem per hour at any location. This is inclusive of air circulation ducts, penetrations and other potential streaming paths on the overpack surface.

3.3.5 Criticality

No criticality requirements beyond those detailed in Section 3.1.5 of this performance specification.

3.3.6 Containment

The aging overpack shall be designed such that following a 3-ft vertical drop or tip over from a 3-ft high site transporter, the TAD canister maximum leak rate is 9.3×10^{-10} fraction of canister free volume per second (off-normal) under applicable repository environmental conditions. The impacted surface characteristics are as follows:

- 5,000 PSI concrete with a minimum thickness of three feet and a maximum (1) thickness of seven feet with a broom finish.
- Reinforcing steel shall be #11 on 12 in. centers, each direction, top and (2) bottom, standard cover top and bottom.
- (3) Soil data is in Attachment E.

3.3.7 **Operations**

- The aging overpack shall be designed to receive, age, and discharge a (1) loaded TAD canister in a vertical orientation.
- (2) The loaded aging overpack shall be transportable on site in a vertical orientation.
- (3) The loaded aging overpack shall be designed to remain in its transport orientation when set down on a flat horizontal surface without use of auxiliary supports.

¹² For this analysis, assume the total quantity of fuel shall vaporize into an efficient fuel-air mixture producing an explosive event. Effects of heat generation, fuel tank projectiles and blast wave propagation shall be considered.

- (4) The aging overpack shall have a vendor designed fixture(s) such that the loaded aging overpack can be handled via an overhead crane.
- (5) The loaded aging overpack shall be designed to be moved to the aging pad via site transporter using a pair of lift beams (e.g., forklift). A sketch showing the interface is shown in Attachment D.
- (6) The aging overpack shall be capable of being transported by air pallet.

3.3.8 Materials

No material requirements, prohibitions, or restrictions have been identified for the aging overpack.

4.0 GLOSSARY

The following section incorporates the definitions and descriptions of major "terms of art" used throughout this document.

Aging- Safely placing commercial CSNF in a site-specific overpack on an aging pad for a long period of time (years) for radioactive decay. Radioactive decay results in a cooler waste form to ensure thermal limits can be met. Safely aging CSNF is an integral part of GROA operations to ensure material has significantly decayed to meet licensed thermal limitations.

Burnup- A measure of nuclear reactor fuel consumption expressed either as the percentage of fuel atoms that have undergone fission or as the amount of energy produced per initial unit weight of fuel.

Canister- The structure surrounding the waste form that facilitates handling, storage, aging and/or transportation.

- 1. The canister may provide structural support for intact CSNF, loose rods, non-fuel components and confinement of radionuclides.
- 2. Canistered waste shall be placed in waste packages prior to emplacement.

Cladding- The metallic outer sheath of a fuel rod generally made of a zirconium alloy. It is intended to isolate the fuel from the external environment.

Design Bases- That information that identifies the specific functions to be performed by a structure, system, or component of a facility and the specific values or ranges of values chosen for controlling parameters as reference bounds for design. These values may be constraints derived from generally accepted "state-of-the-art" practices for achieving functional goals or requirements derived from analysis (based on calculation or experiments) of the effects of a postulated event under which a structure, system, or component must meet its functional goals. The values for controlling parameters for external events include:

- 1. Estimates of severe natural events to be used for deriving design bases that will be based on consideration of historical data on the associated parameters, physical data, or analysis of upper limits of the physical processes involved; and.
- 2. Estimates of severe external human-induced events to be used for deriving design bases, which will be based on analysis of human activity in the region, taking into account the site characteristics and the risks associated with the event.

Event Sequence- A series of actions and/or occurrences within the natural and engineered components of a GROA that could potentially lead to exposure of individuals to radiation. An event sequence includes one or more initiating events and associated combinations of repository system component failures, including those produced by the action or inaction of operating personnel. Those event sequences that are expected to occur one or more times before permanent closure of the geologic repository operations area are referred to as Category 1 event sequences. Other event sequences that have at least one chance in 10,000 of occurring before permanent closure are referred to as Category 2 event sequences.

Fuel assembly- A number of fuel rods held together by plates and separated by spacers used in a reactor. This assembly is sometimes called a fuel bundle or fuel element.

Geologic Repository Operations Area (GROA)- A high-level radioactive waste facility that is part of a geologic repository, including both surface and subsurface areas, where wet handling activities are conducted.

Hypothetical Accident Conditions- The sequential conditions and tests defined in 10 CFR part 71 subpart E (Package Approval Standards) and subpart F (Package, Special Form and LSA-III Tests) that a package (or array of packages) must be evaluated against.

High-Level Radioactive Waste (HLW)- (1) The highly radioactive material resulting from the reprocessing of spent nuclear fuel, including liquid waste produced directly in reprocessing and any solid material derived from such liquid waste that contains fission products in sufficient concentrations; (2) Irradiated reactor fuel; and (3) Other highly radioactive material that the Commission, consistent with existing law, determines by rule requires permanent isolation.

Important to Safety- In reference to structures, systems and components, means those engineered features of the GROA whose function is:

(1) To provide reasonable assurance that high-level waste can be received, handled, packaged, stored, emplaced, and retrieved without exceeding the requirements of §63.111(b)(1) for Category 1 event sequences; or

(2) To prevent or mitigate Category 2 event sequences that could result in radiological exposures exceeding the values specified at §63.111(b)(2) to any individual located on or beyond any point on the boundary of the site.

Important to Waste Isolation- With reference to design of the engineered barrier system and characterization of natural barriers, means those engineered and natural barriers whose function is to provide a reasonable expectation that high-level waste can be disposed of without exceeding the requirements of 10 CFR 63.113(b) and (c).

Neutron Absorber- A material (e.g., boron) that absorbs neutrons used in nuclear reactors, transportation overpacks and waste packages to control neutron multiplication.

Normal Conditions of Transport- The conditions and tests defined in 10 CFR part 71 subpart E (Package Approval Standards) and subpart F (Package, Special Form and LSA-III Tests) that all packages must be evaluated against.

Postclosure- The period of time after closure of the geologic repository.

Preclosure- The period of time before and during closure of the GROA disposal system.

Site¹- An area surrounding the GROA for which the DOE exercises authority over its use in accordance with the provisions of 10 CFR part 63.

Site²- The owner controlled area defined for a utility under 10 CFR part 50.

Site Transporter- A self-powered vehicle designed to haul the TAD canister and contents while within either a shielded transfer cask or aging overpack between GROA surface facilities.

Shielded Transfer Cask (STC)- A cask that meets applicable requirements for safe transfer of a TAD canister and its contents between various surface facilities.

Spent Nuclear Fuel (SNF)- Fuel withdrawn from a nuclear reactor following irradiation, the constituent elements of which have not been separated by reprocessing.

Storage– For the purposes of this specification, the placement, by a licensee of spent nuclear fuel in independent spent fuel storage installations (ISFSI) certified under title 10 CFR part 72.

TAD System- The set of components consisting of one or more TAD canisters, transportation overpacks, transportation skids, ancillary equipments, shielded transfer casks, aging overpacks and site transporters used to facilitate handling of CSNF.

Total Effective Dose Equivalent- For purposes of assessing doses to workers, the sum of the deep-dose equivalent (for external exposures) and committed effective dose equivalent (for internal exposures).

Transportation Overpack- The assembly of components of the packaging intended to retain the radioactive material during transport.

Trunnion- Cylindrical protuberance for supporting and/or lifting located on the outside of a container or cask (e.g., waste package, aging overpack, etc.)

Waste package- The waste form and any containers, shielding, packing and other absorbent materials immediately surrounding an individual waste container.

Attachment A Seismic Data for Yucca Mountain Geologic Repository Operations Area

Summary of Seismic Data for Yucca Mountain Surface Facilities

Table 1. Peak Ground Motions Associated with Ground Motion Categories

Part A: Horizontal Ground Accelerations

Ground Motion Category	Return Period ^b (years)	Horizontal Peak Ground Acceleration ^a (PGA) (g) Surface ^{c, d} Subsurface		DTN
DBGM-1	1,000	0.37	0.13	MO0411SDSDE103.003 MO0405SDSTPNTB.001
DBGM-2	2,000	0.58	0.19	MO0411SDSTMHIS.006 MO0407SDARS104.001
BDBGM	10,000	1.19	0.43	MO0411WHBDE104.003 MO0306SDSAVDTH.000

Part B: Vertical Ground Accelerations

Ground Motion	Return Period b		eak Ground n ^a (PGA) (g)	DTN
Category	(years)	Surface c, d	Subsurface	
DBGM-1	1,000	0.28	0.12	MO0411SDSDE103.003, MO0405SDSTPNTB.001
DBGM-2	2,000	0.52	0.23	MO0411SDSTMHIS.006, MO0407SDARS104.001
BDBGM	10,000	1.49	0.62	MO0411WHBDE104.003, MO0306SDSAVDTH.000

NOTES:

- a) The PGA value is the spectral acceleration at a frequency of 100 Hz (period = 0.01 second) at 5% damping.
- b) A return period of 1,000 years equals a mean annual probability of exceedance (MAPE) of $1.0 \times 10-3$; similarly, a return period of 2,000 years equals a MAPE of $5.0 \times 10-4$ and a return period of 10,000 years equals a MAPE of $1.0 \times 10-4$.
- c) Surface values were defined for the Geotechnical Data for a Potential Wet handling Building and for Ground Motion Analyses for the Yucca Mountain Site Characterization Project (BSC 2002, Figure 1) based on profiles for 35 ft (11 m) and 110 ft (34 m) of alluvium (soil).
- d) PGA values for surface facilities are computed at Point D/E. Location of computation points for surface facilities and emplacement level is shown in Development of Earthquake Ground Motion Input for Preclosure Seismic Design and Postclosure Performance Assessment of a Geologic Repository at Yucca Mountain, NV (BSC 2004, Figure 1).

BDBGM = beyond design basis ground motion; DBGM = design basis ground motion; DTN = document tracking number; g = acceleration due to gravity.

Table 2. Spectral Ground Motions Associated with Ground Motion Categories

Part A: Horizontal Spectral Accelerations

Ground Motion	Return Period		Horizontal accelerations a	DTN		
Category	(years)	Range	Surface b, c	Subsurface	DIN	
DBGM-1	1,000	$S_{A(1-2.5)}$	0.43	0.17	MO0411SDSDE103.003,	
DBOM-1		$S_{A(5-10)}$	0.80	0.25	MO0405SDSTPNTB.001	
DBGM-2	2,000	S _{A(1-2.5)}	0.67	0.24	MO0411SDSTMHIS.006,	
DBGW 2		$S_{A(5-10)}$	1.22	0.37	MO0407SDARS104.001	
BDBGM	10,000	$S_{A(1-2.5)}$	1.58	0.55	MO0411WHBDE104.003,	
	10,000	S _{A(5-10)}	2.52	0.83	MO0306SDSAVDTH.000	

Part B: Vertical Accelerations

Ground Motion	Return Period		rage Vertical S Accelerations ^a	DTN	
Category	(years)	Range	Surface b, c	Subsurface	
DBGM-1	1,000	S _{A(1-2.5)}	0.20	0.13	MO0411SDSDE103.003,
DBGWI-1	1,000	S _{A(5-10)}	0.53	0.23	MO0405SDSTPNTB.001
DBGM-2	2,000	S _{A(1-2.5)}	0.34	0.22	MO0411SDSTMHIS.006,
DBGW 2		S _{A(5-10)}	0.90	0.41	MO0407SDARS104.001
BDBGM	10.000	S _{A(1-2.5)}	0.86	0.58	MO0411WHBDE104.003,
	10,000	S _{A(5-10)}	2.47	1.03	MO0306SDSAVDTH.000

NOTES:

BDBGM = beyond design basis ground motion; DBGM = design basis ground motion; DTN = document tracking number; g = acceleration due to gravity; SA = spectral acceleration; $S_{A(X-Y)} =$ average spectral acceleration for a range, computed as the average of spectral accelerations at frequencies of X and Y.

^a Spectral accelerations are defined as: $S_{A(1-2.5)} = [(SA_1 + SA_{2.5}) / 2]$ and $S_{A(5-10)} = [(SA_5 + SA_{10}) / 2]$, where $SA_{1,0} = SA_{2.5}$, $SA_{1,0} = SA_{1,0}$, and $SA_{1,0} = SA_{1,0} = SA_{1,0}$, and $SA_{1,0} = SA_{1,0} = SA_{1,0$

b Surface values are defined for the *Geotechnical Data for a Potential Wet handling Building and for Ground Motion Analyses for the Yucca Mountain Site Characterization Project* (BSC 2002, Figure 1) based on profiles for 35 ft (11 m) and 110 ft (34 m) of alluvium (soil).

^c Acceleration values for surface facilities are computed at Point D/E. Location of computation points for surface facilities and emplacement level is shown in *Development of Earthquake Ground Motion Input for Preclosure Seismic Design and Postclosure Performance Assessment of a Geologic Repository at Yucca Mountain, NV* (BSC 2004, Figure 1).

Table 3. Maximum Horizontal Spectral Accelerations at Surface for 2,000-Year Return Period Seismic Event

	Spectral Acceleration At Different Damping Levels (g)								
Period (sec)	0.5%	1.0%	2.0%	3.0%	5.0%	7.0%	10%	15%	20%
0.010	0.5802	0.5802	0.5802	0.5802	0.5802	0.5802	0.5802	0.5802	0.5802
0.011	0.5973	0.5973	0.5973	0.5973	0.5973	0.5973	0.5973	0.5973	0.5973
0.012	0.6194	0.6194	0.6194	0.6194	0.6194	0.6194	0.6194	0.6194	0.6194
0.014	0.8274	0.7628	0.6982	0.6604	0.6470	0.6426	0.6343	0.6249	0.6182
0.017	1.0026	0.8910	0.7795	0.7142	0.6808	0.6685	0.6522	0.6337	0.6205
0.020	1.2153	1.0497	0.8842	0.7873	0.7302	0.7083	0.6826	0.6533	0.6325
0.025	1.4818	1.2525	1.0233	0.8892	0.8031	0.7545	0.7145	0.6689	0.6365
0.034	1.8822	1.5604	1.2386	1.0504	0.9194	0.8239	0.7606	0.6886	0.6375
0.050	2.4206	1.9810	1.5414	1.2842	1.0897	0.9551	0.8638	0.7600	0.6864
0.100	2.9209	2.3812	1.8414	1.5256	1.2512	1.0728	0.9464	0.8027	0.7007
0.110	2.9076	2.3719	1.8362	1.5229	1.2453	1.0662	0.9383	0.7929	0.6898
0.123	2.8864	2.3574	1.8284	1.5189	1.2380	1.0582	0.9287	0.7815	0.6771
0.142	2.8545	2.3356	1.8167	1.5132	1.2292	1.0491	0.9180	0.7690	0.6632
0.167	2.7904	2.2892	1.7881	1.4950	1.2106	1.0318	0.9001	0.7504	0.6442
0.201	2.6829	2.2092	1.7354	1.4583	1.1779	1.0031	0.8725	0.7239	0.6185
0.248	2.5572	2.1160	1.6749	1.4168	1.1422	0.9726	0.8436	0.6969	0.5929
0.335	2.2450	1.8732	1.5014	1.2839	1.0341	0.8816	0.7623	0.6268	0.5306
0.498	1.7176	1.4514	1.1852	1.0295	0.8309	0.7107	0.6131	0.5021	0.4233
1.000	0.7299	0.6332	0.5365	0.4799	0.3918	0.3384	0.2918	0.2389	0.2014
1.123	0.6204	0.5407	0.4611	0.4144	0.3394	0.2937	0.2535	0.2077	0.1753
1.262	0.5282	0.4626	0.3970	0.3586	0.2946	0.2556	0.2208	0.1812	0.1530
1.417	0.4495	0.3956	0.3416	0.3101	0.2557	0.2224	0.1923	0.1580	0.1337
1.668	0.3575	0.3168	0.2761	0.2523	0.2092	0.1827	0.1583	0.1305	0.1108
2.009	0.2734	0.2442	0.2151	0.1980	0.1654	0.1452	0.1261	0.1044	0.0890
2.477	0.2012	0.1814	0.1616	0.1500	0.1265	0.1118	0.0975	0.0813	0.0697
3.351	0.1215	0.1110	0.1006	0.0945	0.0810	0.0724	0.0636	0.0537	0.0467
4.978	0.0614	0.0572	0.0530	0.0505	0.0445	0.0405	0.0362	0.0312	0.0277
10.000	0.0126	0.0122	0.0119	0.0116	0.0110	0.0105	0.0098	0.0089	0.0083

Source: MO0411SDSTMHIS.006. Seismic Design Spectra and Time Histories for the Surface Facilities Area (Point D/E) at 5E-4 Annual Exceedance Frequency.

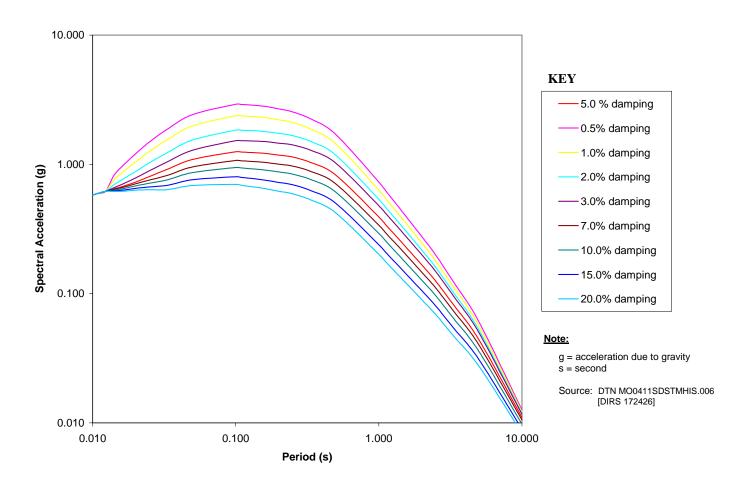


Figure 1 Maximum Horizontal Spectra at Surface for Multiple Damping Levels for 2,000-Year Return Period Seismic Event

Table 4. Vertical Spectral Accelerations at Surface for 2,000-Year Return Period Seismic Event

	Spectral Acceleration At Different Damping Levels (g)								
Period (sec)	0.5%	1.0%	2.0%	3.0%	5.0%	7.0%	10%	15%	20%
0.010	0.5188	0.5188	0.5188	0.5188	0.5188	0.5188	0.5188	0.5188	0.5188
0.011	0.5413	0.5413	0.5413	0.5413	0.5413	0.5413	0.5413	0.5413	0.5413
0.012	0.5709	0.5709	0.5709	0.5709	0.5709	0.5709	0.5709	0.5709	0.5709
0.014	0.8412	0.7613	0.6814	0.6346	0.6086	0.5902	0.5745	0.5568	0.5442
0.017	1.0315	0.9036	0.7756	0.7008	0.6557	0.6224	0.5971	0.5684	0.5480
0.020	1.2579	1.0739	0.8900	0.7824	0.7141	0.6633	0.6265	0.5846	0.5549
0.025	1.5055	1.2604	1.0152	0.8718	0.7768	0.7065	0.6568	0.6002	0.5600
0.034	1.8666	1.5353	1.2039	1.0101	0.8745	0.7758	0.7069	0.6286	0.5730
0.050	2.2647	1.8411	1.4176	1.1698	0.9842	0.8529	0.7614	0.6573	0.5834
0.100	2.4411	1.9793	1.5176	1.2474	1.0166	0.8619	0.7520	0.6270	0.5384
0.110	2.4119	1.9572	1.5026	1.2366	1.0050	0.8506	0.7407	0.6156	0.5269
0.123	2.3722	1.9276	1.4830	1.2229	0.9907	0.8372	0.7273	0.6024	0.5138
0.142	2.3149	1.8847	1.4545	1.2029	0.9715	0.8199	0.7107	0.5866	0.4985
0.167	2.0742	1.6936	1.3129	1.0903	0.8779	0.7403	0.6404	0.5267	0.4461
0.201	1.8236	1.4948	1.1659	0.9735	0.7819	0.6590	0.5689	0.4664	0.3937
0.248	1.5396	1.2684	0.9972	0.8386	0.6723	0.5667	0.4884	0.3994	0.3363
0.335	1.1633	0.9666	0.7699	0.6549	0.5246	0.4429	0.3811	0.3110	0.2612
0.498	0.7810	0.6574	0.5338	0.4615	0.3705	0.3139	0.2699	0.2200	0.1845
1.000	0.3751	0.3243	0.2736	0.2438	0.1980	0.1694	0.1459	0.1192	0.1003
1.123	0.3315	0.2880	0.2445	0.2191	0.1784	0.1530	0.1319	0.1079	0.0908
1.262	0.2929	0.2557	0.2185	0.1968	0.1608	0.1382	0.1192	0.0977	0.0824
1.417	0.2585	0.2268	0.1951	0.1766	0.1448	0.1248	0.1077	0.0884	0.0747
1.668	0.2166	0.1914	0.1662	0.1515	0.1249	0.1080	0.0935	0.0769	0.0652
2.009	0.1748	0.1558	0.1367	0.1255	0.1043	0.0907	0.0787	0.0650	0.0553
2.477	0.1374	0.1236	0.1098	0.1017	0.0853	0.0747	0.0650	0.0540	0.0462
3.351	0.0904	0.0825	0.0745	0.0699	0.0596	0.0528	0.0463	0.0389	0.0336
4.978	0.0504	0.0469	0.0434	0.0413	0.0362	0.0327	0.0291	0.0249	0.0219
10.000	0.0126	0.0123	0.0119	0.0117	0.0110	0.0105	0.0097	0.0087	0.0080

Source: MO0411SDSTMHIS.006. Seismic Design Spectra and Time Histories for the Surface Facilities Area (Point D/E) at 5E-4 Annual Exceedance Frequency.

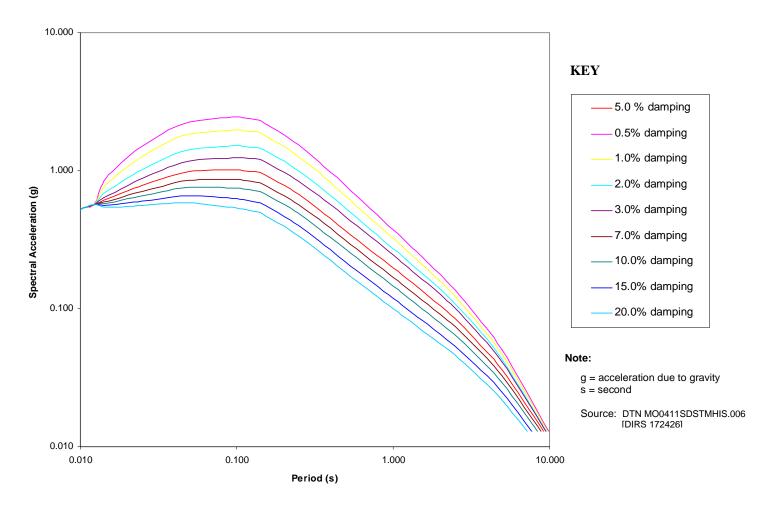


Figure 2. Vertical Spectra at Surface for Multiple Damping Levels for 2,000-Year Return Period Seismic Event

Table 5. Maximum Horizontal Spectral Accelerations at Surface for 10,000-Year Return Period Seismic Event

	Spectral Accelera	ation At Different	Damping Levels (g)						
Period (sec)	0.5%	1.0%	2.0%	3.0%	5.0%	7.0%	10%	15%	20%
0.010	1.1926	1.1926	1.1926	1.1926	1.1926	1.1926	1.1926	1.1926	1.1926
0.011	1.2296	1.2296	1.2296	1.2296	1.2296	1.2296	1.2296	1.2296	1.2296
0.012	1.2775	1.2775	1.2775	1.2775	1.2775	1.2775	1.2775	1.2775	1.2775
0.014	1.7104	1.5768	1.4432	1.3651	1.3374	1.3284	1.3112	1.2917	1.2779
0.017	2.0765	1.8454	1.6144	1.4792	1.4100	1.3845	1.3508	1.3124	1.2852
0.020	2.4923	2.1528	1.8133	1.6147	1.4975	1.4526	1.3998	1.3397	1.2971
0.025	3.0403	2.5699	2.0996	1.8244	1.6478	1.5482	1.4659	1.3724	1.3060
0.034	3.9040	3.2366	2.5692	2.1788	1.9070	1.7089	1.5776	1.4283	1.3224
0.050	5.0495	4.1325	3.2154	2.6790	2.2732	1.9925	1.8020	1.5855	1.4319
0.100	5.9642	4.8620	3.7599	3.1151	2.5548	2.1905	1.9324	1.6389	1.4307
0.110	5.9413	4.8467	3.7521	3.1118	2.5446	2.1786	1.9173	1.6203	1.4095
0.123	5.9030	4.8211	3.7392	3.1063	2.5318	2.1641	1.8993	1.5983	1.3848
0.142	5.8441	4.7818	3.7195	3.0980	2.5166	2.1478	1.8795	1.5744	1.3579
0.167	5.7601	4.7256	3.6912	3.0860	2.4990	2.1299	1.8580	1.5490	1.3297
0.201	5.6363	4.6411	3.6459	3.0638	2.4746	2.1074	1.8329	1.5209	1.2995
0.248	5.2724	4.3629	3.4533	2.9213	2.3550	2.0054	1.7393	1.4369	1.2224
0.335	4.7596	3.9713	3.1831	2.7220	2.1924	1.8690	1.6162	1.3288	1.1249
0.498	4.0273	3.4031	2.7790	2.4139	1.9482	1.6663	1.4374	1.1772	0.9926
1.000	1.9942	1.7299	1.4656	1.3110	1.0704	0.9244	0.7973	0.6528	0.5502
1.123	1.7558	1.5303	1.3048	1.1729	0.9605	0.8313	0.7174	0.5879	0.4960
1.262	1.4946	1.3089	1.1232	1.0146	0.8336	0.7233	0.6247	0.5126	0.4331
1.417	1.2211	1.0745	0.9280	0.8423	0.6946	0.6041	0.5223	0.4293	0.3633
1.668	0.9655	0.8556	0.7456	0.6813	0.5650	0.4934	0.4274	0.3524	0.2991
2.009	0.7473	0.6676	0.5878	0.5412	0.4521	0.3968	0.3447	0.2855	0.2434
2.477	0.5417	0.4884	0.4350	0.4038	0.3406	0.3010	0.2625	0.2188	0.1878
3.351	0.3401	0.3109	0.2816	0.2645	0.2268	0.2027	0.1782	0.1504	0.1307
4.978	0.1813	0.1689	0.1565	0.1492	0.1315	0.1198	0.1069	0.0923	0.0819
10.000	0.0395	0.0384	0.0372	0.0365	0.0345	0.0330	0.0307	0.0280	0.0261

Source: DTN MO0411WHBDE104.003. Seismic Design Spectra and Time Histories for the Surface Facilities Area (Point D/E) at 10-4 Annual Exceedance Frequency.

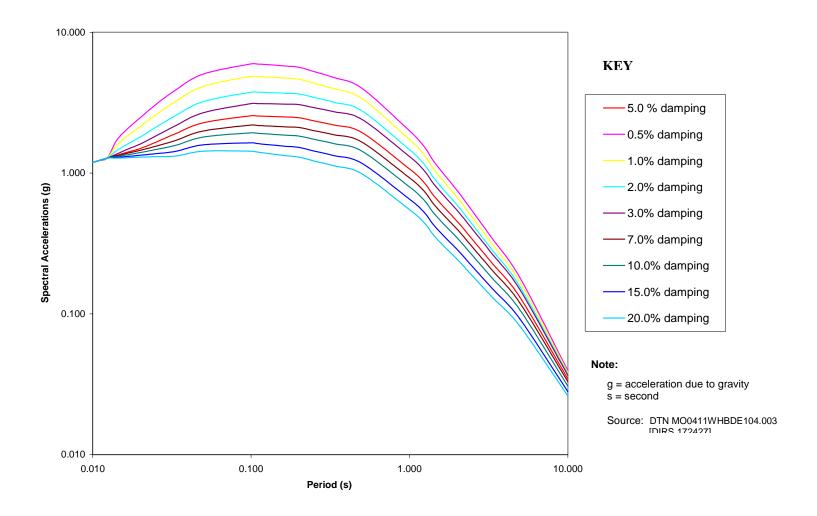


Figure 3. Maximum Horizontal Spectra at Surface for Multiple Damping Levels for 10,000-Year Return Period Seismic Event

Table 6. Vertical Spectral Accelerations at Surface for 10,000-Year Return Period Seismic Event

	Spectral Acceleration At Different Damping Levels (g)								
Period (sec)	0.5%	1.0%	2.0%	3.0%	5.0%	7.0%	10%	15%	20%
0.010	1.4932	1.4932	1.4932	1.4932	1.4932	1.4932	1.4932	1.4932	1.4932
0.011	1.5703	1.5703	1.5703	1.5703	1.5703	1.5703	1.5703	1.5703	1.5703
0.012	1.6723	1.6723	1.6723	1.6723	1.6723	1.6723	1.6723	1.6723	1.6723
0.014	2.4807	2.2450	2.0093	1.8714	1.7947	1.7403	1.6943	1.6420	1.6048
0.017	3.0555	2.6765	2.2975	2.0758	1.9423	1.8437	1.7688	1.6837	1.6233
0.020	3.6683	3.1319	2.5955	2.2817	2.0825	1.9343	1.8269	1.7049	1.6183
0.025	4.3386	3.6321	2.9256	2.5124	2.2386	2.0360	1.8926	1.7296	1.6140
0.034	5.2596	4.3260	3.3924	2.8463	2.4641	2.1860	1.9919	1.7712	1.6146
0.050	6.3125	5.1319	3.9512	3.2606	2.7433	2.3774	2.1222	1.8321	1.6263
0.100	6.9021	5.5965	4.2908	3.5271	2.8744	2.4369	2.1262	1.7729	1.5223
0.110	6.7469	5.4750	4.2031	3.4591	2.8113	2.3795	2.0718	1.7221	1.4740
0.123	6.3388	5.1507	3.9626	3.2676	2.6472	2.2369	1.9434	1.6097	1.3729
0.142	5.8683	4.7778	3.6873	3.0495	2.4628	2.0785	1.8017	1.4870	1.2638
0.167	5.3486	4.3671	3.3856	2.8115	2.2638	1.9090	1.6512	1.3583	1.1504
0.201	4.7952	3.9305	3.0657	2.5598	2.0560	1.7327	1.4958	1.2265	1.0354
0.248	3.9885	3.2859	2.5833	2.1723	1.7416	1.4681	1.2653	1.0347	0.8711
0.335	3.0229	2.5118	2.0007	1.7018	1.3632	1.1509	0.9904	0.8080	0.6786
0.498	1.9764	1.6637	1.3509	1.1680	0.9376	0.7943	0.6831	0.5567	0.4669
1.000	1.0074	0.8710	0.7346	0.6548	0.5317	0.4549	0.3919	0.3202	0.2694
1.123	0.9048	0.7861	0.6673	0.5979	0.4869	0.4175	0.3599	0.2944	0.2479
1.262	0.8123	0.7092	0.6060	0.5457	0.4459	0.3832	0.3306	0.2708	0.2284
1.417	0.7291	0.6397	0.5503	0.4980	0.4084	0.3519	0.3039	0.2494	0.2107
1.668	0.6263	0.5534	0.4805	0.4379	0.3611	0.3124	0.2703	0.2224	0.1884
2.009	0.5213	0.4644	0.4076	0.3743	0.3110	0.2704	0.2345	0.1938	0.1648
2.477	0.4150	0.3733	0.3315	0.3071	0.2576	0.2255	0.1963	0.1631	0.1395
3.351	0.2786	0.2542	0.2297	0.2154	0.1837	0.1627	0.1426	0.1198	0.1036
4.978	0.1454	0.1353	0.1252	0.1192	0.1045	0.0945	0.0839	0.0719	0.0633
10.000	0.0339	0.0329	0.0320	0.0314	0.0295	0.0282	0.0259	0.0234	0.0216

Source: MO0411WHBDE104.003. Seismic Design Spectra and Time Histories for the Surface Facilities Area (Point D/E) at 10-4 Annual Exceedance Frequency.

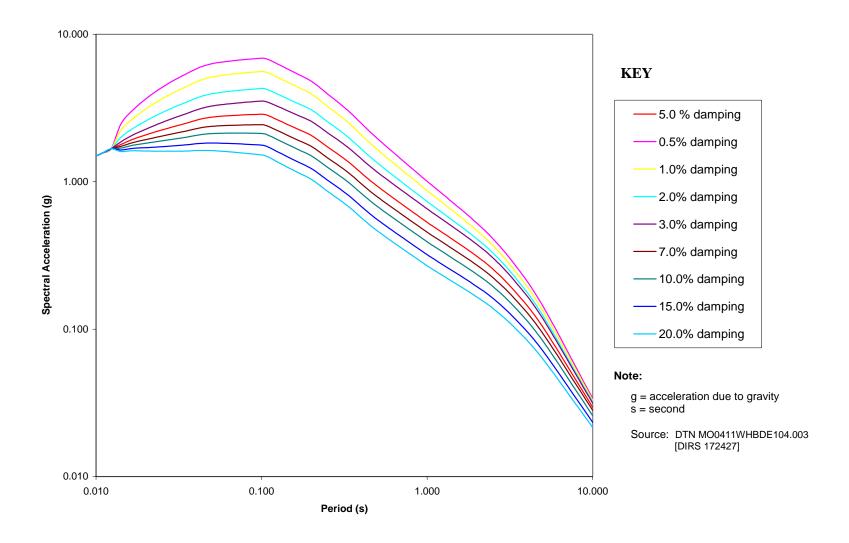
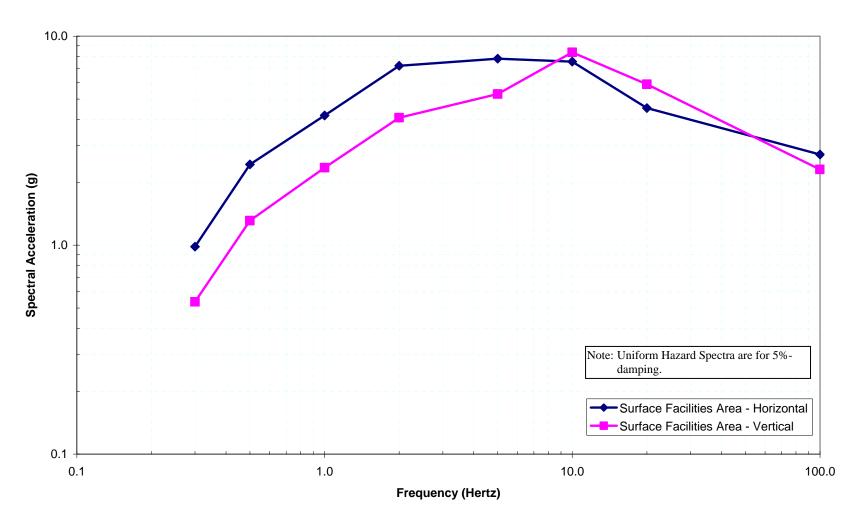


Figure 4. Vertical Spectra at Surface for Multiple Damping Levels for 10,000-Year Return Period Seismic Event

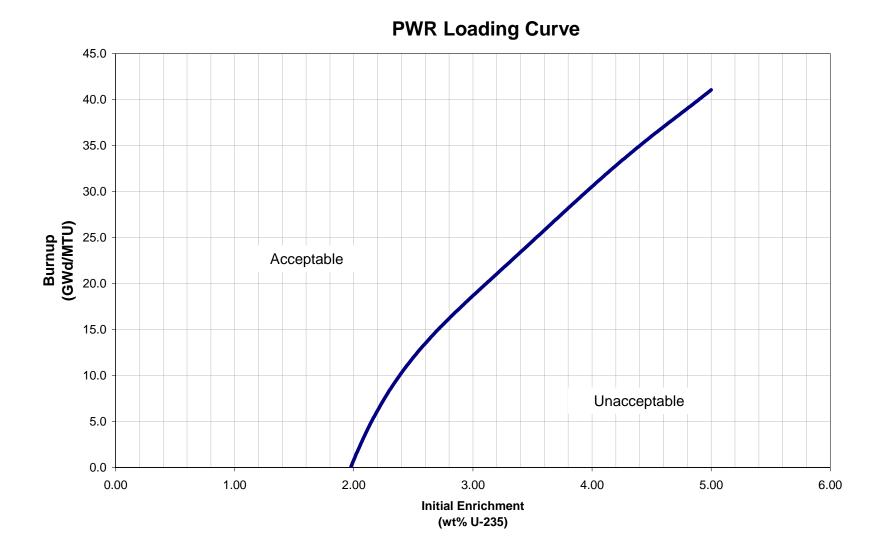
Uniform Hazard Spectra - Surface Facilities Area 3 g Peak Ground Acceleration Earthquake

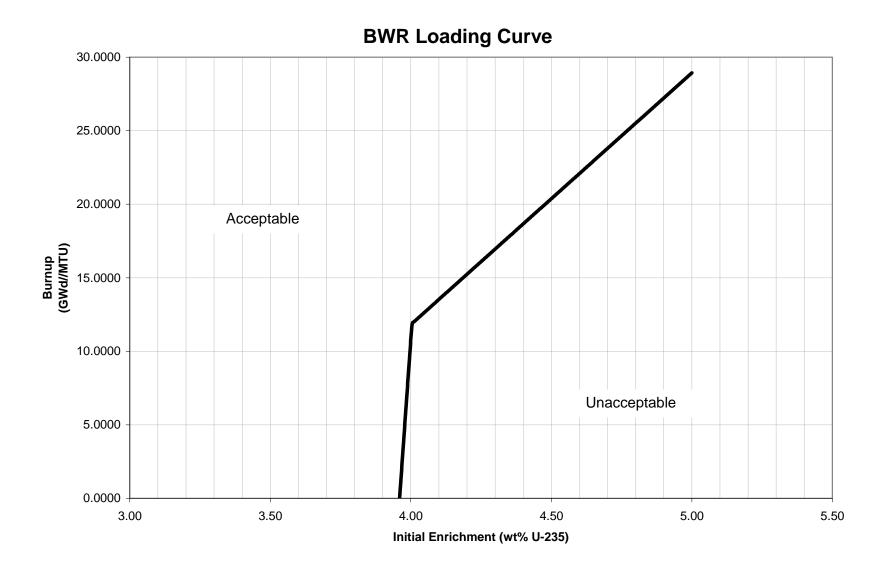


Attachment B Current Postclosure Criticality Loading Curves

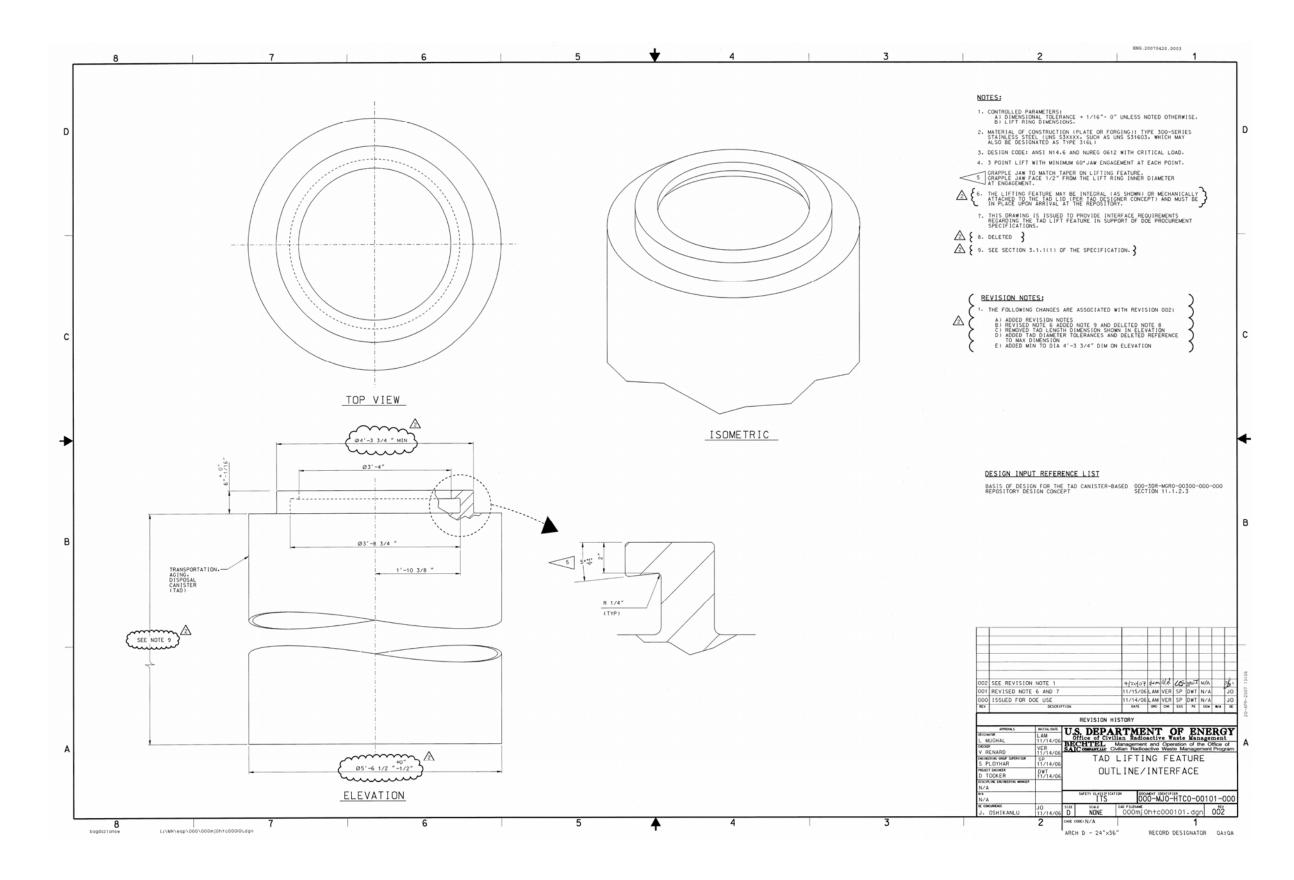
The Department of Energy is currently developing finalized PWR and BWR postclosure criticality loading curves. The following PWR and BWR loading curves represent the currently defined TAD configuration and materials baseline as detailed in Section 3.1.5 of this Performance Specification.

WMO-TADCS-000001 Rev. 0 DOE/RW-0585

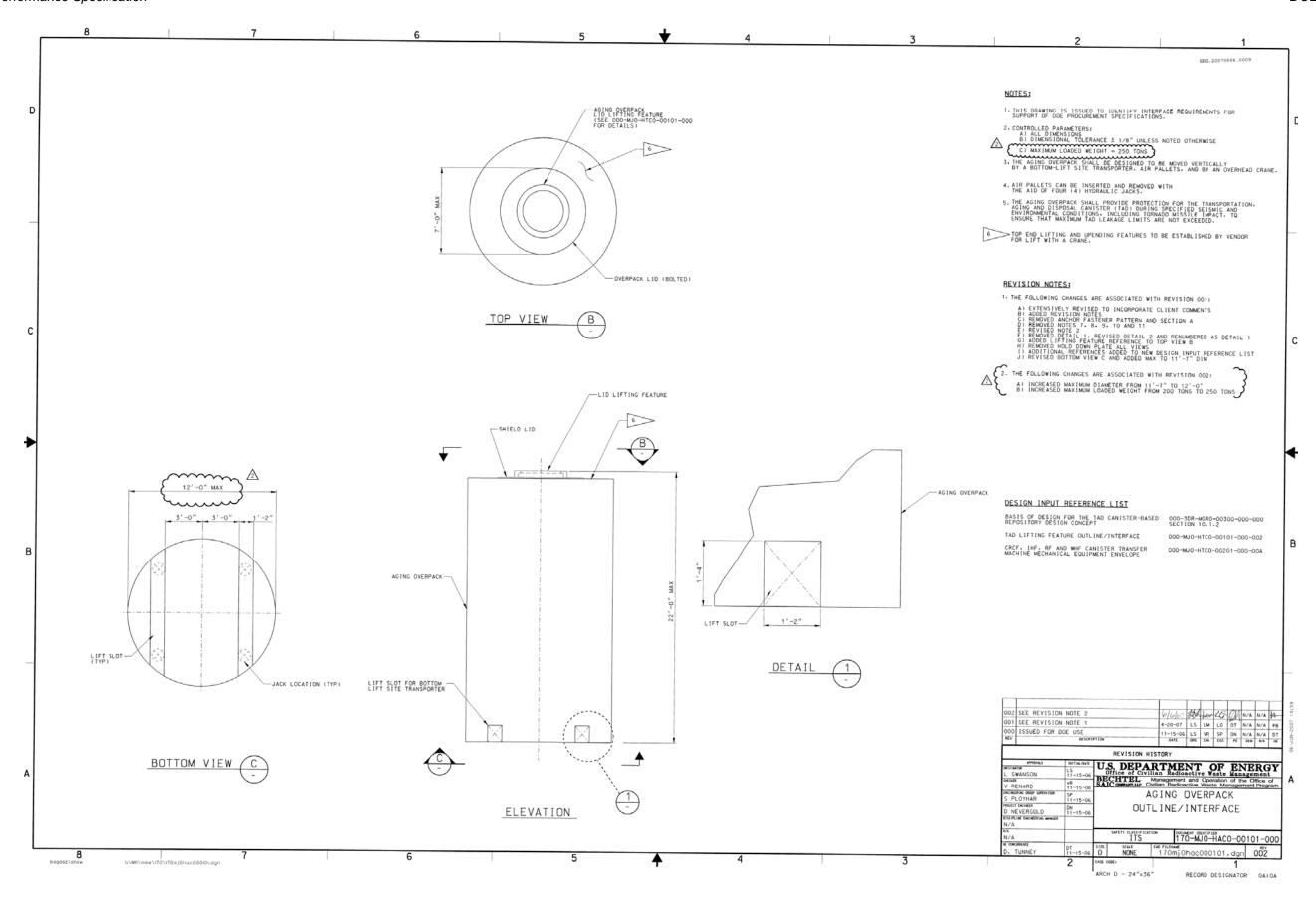




Attachment C TAD Canister Lifting Feature



Attachment D Aging Overpack Details



Attachment E Supplemental Soils Report

BSC

Design Calculation or Analysis Cover Sheet

1. QA: QA

Complete only applicable items.

2. Page 1

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		Attachm	ents				Pages
	ppendix A-Seismic Wave Velocity						57
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				İ			R. E. Pernisi 11/4/04
00B	Complete Revision; revised format to conform to EG-	255	C-26	James T. Camero	Nan Deng		Farhang Ostadan
	DSK-3003; revised contents of 6.4.2.3 (formerly 10.2.3)			8/15/06	8/15/06		8/16/06
	and 7.1.11 (formerly 11.11) and replaced figure 7-1 (formerly 11-1) to resolve CR5565 concerns. Also						Raj S. Rajagopal 8/21/06
	addresses issues raised in CR8288 regarding consistency					ĺ	
	with Ground Motion Report (Section 6.4.2). Note: the layout of the facilities is currently under revision. Even						
İ	though the exact number and configuration of facilities						
	will change, the design philosophy will continue to use multiple facilities of different sizes located within the						
	North Portal area similar to that shown on Figure 1-1.						
	The calculation will be updated when the layout for the facilities is revised.						
	lacinites is revised.						
00C	Complete Revision; replaced Figs. 1-1 and 6-7; modified	265	C-26	Janes T/Cameron	Nan Deng	Richard Kotas	Farhang Ostadan
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ACRONYMS AND ABBREVIATIONS

ACC Accession Number

ACI American Concrete Institute

ASCE American Society of Civil Engineers

ASTM American Society for Testing and Materials

BSC Bechtel SAIC Company

c cohesion

C_c coefficient of curvature

CCCF Central Control Center Facility

CF fines content

CPT Cone penetrometer test

CRCF Canister Receipt and Closure Facility

CRWMS Civilian Radioactive Waste Management System

C₁₁ coefficient of uniformity

D₁₀ grain diameter (in mm) corresponding to 10% passing, by weight (or mass)
D₃₀ grain diameter (in mm) corresponding to 30% passing, by weight (or mass)
grain diameter (in mm) corresponding to 60% passing, by weight (or mass)

DIRS Document Input Reference System

DOE U.S. Department of Energy DTN Data Tracking Number

E Young's modulus or secant Young's modulus

elev. elevation

EMWB Equipment Maintenance/Warehouse Building

EPRI Electrical Power Research Institute

Eq. equation

ESF Exploratory Studies Facility

Fpm feet per minute fps feet per second

ft foot, feet (unit of measurement)

ft/s feet per second ft² feet squared ft³ feet cubed

G shear modulus

G_{max} small-strain (maximum) shear modulus

GP poorly-graded gravels or gravel-sand mixtures, little or no fines

GSF Ground Surface Facility

GW well-graded gravels or gravel-sand mixtures, little or no fines

HEMF Heavy Equipment Maintenance Facility

ACRONYMS AND ABBREVIATIONS (CONTINUED)

ICC International Code Council

ID identification

IHF Initial Handling Facility

in. inch, inches

K_A coefficient of active earth pressureK_P coefficient of passive earth pressure

kip 1,000 pounds (kilopound) kips/ft² kips per square foot kips/ft³ kips per cubic foot

K_o coefficient of at-rest soil pressure

Kcf kips per cubic foot ksf kips per square foot

lb/ft² pounds per square foot lb/ft³ pounds per cubic foot

lb pounds (usually pounds-force)

LL liquid limit

mm millimeter

M&O Management and Operating Contractor

MWV Midway Valley

N SPT penetration resistance (blow count)

N₆₀ SPT penetration resistance corrected to 60% efficiency

NNWSI Nevada Nuclear Waste Site Investigation

NRC National Regulatory Commission

NRG North Ramp Geotechnical NRSF North Ramp Surface Facilities

p page

pcf pounds per cubic foot PI plasticity index

pp pages

psf pounds per square foot psi pounds per square inch

"Q" "quality"

QA quality assurance Qal Quaternary alluvium

ACRONYMS AND ABBREVIATIONS (CONTINUED)

RCSC roller compacted soil cement

RCTS Resonant Column & Torsional Shear

Rev. revision REV. revision

RF Repository Facility
RF Receipt Facility

SASW spectral analysis of surface waves

SFS Surface Facility System

SM silty sands, sand-silt mixtures

SN Scientific Notebook

SP poorly-graded sand or gravelly sands, little or no fines

SPT Standard Penetration Test

SW well-graded sand or gravelly sands, little or no fines

tcf tons (American) per cubic foot
TIC Technical Information Center
Tmbtl pre-Rainier Mesa Tuff bedded tuff

Tmr Rainer Mesa Tuff of the Timber Mountain Group

tons/ft³ tons (American) per cubic foot

Tpbt5 pre-Tuff unit "x" bedded tuffs (also known as post-Tiva Canyon Tuff bedded

tuffs)

Tpcpll Tiva Canyon Tuff: crystal-poor member, lower lithophysal zone
Tpcpln Tiva Canyon Tuff: crystal-poor member, lower nonlithophysal zone
Tpcpun Tiva Canyon Tuff: crystal-poor member, middle nonlithophysal zone
Tpcpun Tiva Canyon Tuff: crystal-poor member, upper lithophysal zone
Tpcpun Tiva Canyon Tuff: crystal-poor member, upper nonlithophysal zone

Tpcrn Tiva Canyon Tuff: crystal-rich member, nonlithophysal zone, but used in BSC

(2002) to mean the Tpcr member

Tpki Tuff unit "x"

tsf tons (American) per square foot

UF Utility Facility

USBR U.S. Bureau of Reclamation USN U.S. Department of the Navy

USS United States Steel

UTA University of Texas, Austin

V_p compression-wave seismic velocity

V_s shear-wave seismic velocity

ACRONYMS AND ABBREVIATIONS (CONTINUED)

WHB waste handling surface facilities formally designated as WHB or Waste Handling

Building

WHF Wet Handling Facility

WNNRF Warehouse and Non-Nuclear Receipt Facility

YMP Yucca Mountain Project

GLOSSARY

This glossary presents definitions for geologic and geotechnical terms as used in this report. Other definitions may be used in other disciplines or in other contexts.

bedded tuff—a rock unit composed of volcanic ejecta that was deposited in layers and that exhibits distinct planes of weakness (bedding planes) parallel to layering; deposited either by water or by compositional sorting by air fall.

coefficient of uniformity—the ratio of D_{60} to D_{10} , where D_n is the sieve opening that would allow n percent of the soil particles (on a dry mass basis) to pass. In practice, D_n is determined by interpolation of the results of a particle-size distribution test.

coefficient of vertical subgrade reaction, k (mass per length squared per time squared, e.g., pound-force/ft³ or kN/m³)—the ratio of the vertical pressure acting at the foundation/subgrade interface at a point to the settlement at the same point.

compression-wave velocity–velocity of the compression (P) wave from a seismic energy source.

density, ρ (mass per length cubed, e.g., pound-mass/ft³ or kg/m³)—the total mass (solids plus liquid plus gas) per total volume. Synonyms: bulk density, total bulk density, moist density, total density, wet density.

density of solid particles, ρ_s (mass per length cubed, e.g., pound-mass/ft³ or kg/m³)—the mass of solid particles divided by the volume of solid particles.

dry density, ρ_d (mass per length cubed, e.g., pound-mass/ft³ or kg/m³)—the mass of solid particles per the total volume of soil or rock.

embedment—the depth at which the base of a foundation is situated below the ground surface.

engineered fill—a fill placed by man that meets several criteria, including: (1) the fill is designed to meet established criteria (e.g., bearing capacity, settlement) for a particular purpose (building, embankment, etc.); (2) criteria are established on drawings and in a written specification for the material placed in the fill; (3) the fill is placed in accordance with drawings and written specifications; (4) the fill placement operations are observed by a geotechnical engineer (usually a geotechnical technician working under the geotechnical engineer's supervision); (5) the material being placed in the fill is sufficiently tested to establish its geotechnical characteristics(6) the degree of compaction of the fill is verified by either (a) in-situ density tests and compaction tests if relative compaction or relative density is specified, or (b) documenting adherence to a method specification, depending on which acceptance criteria is stipulated in the construction contract documents; (7) all fill material and all compacted fill that do not meet the contract requirements is either removed and replaced or reworked in an appropriate manner; (8) the geotechnical engineer prepares detailed written daily reports stating the geotechnical engineer's observations for the day, which are distributed on a daily basis; and (9) the geotechnical engineer writes and files a report at the conclusion of earthwork construction summarizing the geotechnical engineer's observations and testing made during construction and

providing his opinion that the fill was or was not constructed in accordance with the specifications and is suited or not for its intended use.

fines content—the percent of a materials' particles, on a dry weight basis, that pass through a U.S. Standard No. 200 sieve.

kip—a unit of force (weight) equal to one thousand pounds-force (1000 lbf).

lithophysae—hollow, bubble-like structures composed of concentric shells formed by the concentration of gasses during cooling of portions of a volcanic flow deposit.

lithophysal—containing lithophysae.

low-amplitude shear modulus—see shear modulus, low-amplitude.

moist density-synonym of density.

non-engineered fill—an artificial (man-made) fill that does not meet the definition of engineered fill.

nonwelded tuff—a volcanic rock consisting of fragments that were deposited with insufficient heat to have become fused.

overburden pressure–at point A at depth, d, $\sigma_v = \int_0^d \gamma \, dz$ where γ is unit weight and z is depth

below the point on the ground surface directly above Point A. Note: For this report, groundwater is not a consideration, so effective overburden pressure is taken to be the same as total overburden pressure.

percent core recovery—in a given cored interval, the ratio of the length of core recovered to the length of the interval, expressed as a percentage.

Poisson's ratio, υ —in Hooke's Law for isotropic materials, for a material subjected to a stress in some direction, the ratio of the strain in the transverse direction to the strain in the direction of stress application.

relative compaction—the ratio, expressed as a percentage, of the dry unit weight of a soil mass to the reference maximum dry unit weight of the material as determined by a test, such as ASTM D 1557, Standard Test Methods for Laboratory Compaction Characteristics of Soil Using Modified Effort (56,000ft-lbf/ft³ (2,700kN-m/m³)).

relative density—the ratio of (1) the difference between the void ratio of a cohesionless soil in the loosest state and its actual void ratio, to (2) the difference between the void ratios in the loosest and in the densest states.

shear modulus—the stiffness factor for a material under shear stress, expressed by the relationship of the applied shear force to the change in position produced by this force, calculated as the product of the total mass density (total unit weight divided by gravity) and the square of the shear wave velocity. Symbol: G.

shear modulus, low-amplitude—shear modulus determined as the ratio of the shearing stress divided by the shearing strain at low strain values (< 0.001%). Symbol: G. Synonym: small-strain shear modulus.

shear-wave velocity–velocity of the shear (S) wave from a seismic energy source.

shear-wave velocity, low-amplitude -the velocity of a seismic body wave propagating with a shearing motion that oscillates particles at right angles to the direction of propagation measured at low strain values (< 0.001%). Synonym: small-strain shear-wave velocity.

small-strain shear modulus—synonym of low-amplitude shear modulus

small-strain shear-wave velocity—synonym of low-amplitude shear-wave velocity.

total density-synonym of density.

total unit weight-synonym of unit weight.

unit weight, γ (mass per length squared per time squared, e.g., pound-force/ft³ or kN/m³)—the total weight (solids plus liquid plus gas) per total volume. This parameter is also referred to as "moist unit weight," "wet unit weight," or "total unit weight."

unit weight, dry, γ_d (mass per length squared per time squared, e.g., pound-force/ft³ or kN/m³)—the total weight of solid particles per total volume.

unit weight, total—synonym of unit weight.

vitric tuff—an indurated deposit of volcanic ash composed mainly glassy fragments blown out of a volcano during a volcanic eruption.

water content—the ratio of the mass of water contained in the pore spaces of soil or rock material, to the solid mass of particles in that material, expressed as a percentage. Also referred to as gravimetric water content. Note that adsorbed water is not considered part of the water in the pore spaces but as water bound to the solid particles—synonym of moisture content.

welded tuff—a rock consisting of volcanic fragments that has been indurated by the heat retained by particles and the enveloping gases.

wet density-synonym of density.

1 PURPOSE

1.1 PURPOSE

This report is written as a companion report to *Soils Report for North Portal Area, Yucca Mountain Project*, Document Identifier 100-00C-WRP0-00100-000, dated October 2002 (BSC 2002b). The primary purpose of the current report is to adopt, clarify, and summarize the findings and recommendations of BSC (2002a) and BSC (2002b) into design charts and tables to be used for the preliminary design of waste handling surface facilities (formally designated as WHB, or waste handling building) to be constructed near the North Portal of the Exploratory Studies Facility (ESF) at the Yucca Mountain Project Site (YMP). The surface facilities include all associated surface structures for the nuclear waste handling and storage facility. This report also recommends additional investigation and testing for the final design of the proposed facilities. These recommendations have been developed for use in design of the potential waste handling facilities to a level suitable to support License Application.

Subsequent to the issuance of Revision 00A of this calculation a ground motion report for the site was written (BSC 2004a) more thoroughly addressing dynamic properties and other seismic considerations. This current calculation revision includes consideration of the BSC 2004a report regarding the dynamic soil properties, including shear and compression wave velocities and material degradation relationships.

1.2 SCOPE

The scope of this report is to provide simplified charts and recommendations of geotechnical parameters to be used for preliminary design and analysis of the surface facilities. Where pertinent, the recommendations provided in BSC (2000b) are used. The current report summarizes the pertinent field and laboratory investigations, the results of material property tests, and provides engineering design parameters including allowable bearing capacity, settlement, lateral earth pressures on retaining walls, and slope evaluation based on site-specific subsurface soil information. Additional recommendations provided include pavement design parameters, percolation rates, and frost penetration. Construction considerations and additional investigations and testing are also discussed.

1.3 PROJECT DESCRIPTION

The configuration of the nuclear waste handling surface facilities area has changed over much iteration from a single building encompassing all aspects of the waste handling process to the configuration used herein, which consists of several major process facilities. The largest potential facility of this layout is the Canister Receipt and Closure Facility (CRCF 1) (as shown in Drawing 100-C00-MGR0-00501-000-00A). Other major structures include the Canister Receipt and Closure Facility (CRCF 2 & 3); Wet Handling Facility (WHF); Initial Handling Facility (IHF); Receipt Facility (RF); Central Control Center Facility (CCCF); Warehouse and Non-Nuclear Receipt Facility (WNNRF) and the heavy equipment maintenance facility (HEMF). The southeast portion of the site area contains an evaporation pond and a stormwater/retention pond. Several smaller facilities (administration, fire rescue, medical, storage, etc.) are located in

the southern portion of the site. The currently planned layout is shown in Figure 1-1. The nuclear handling surface facilities are typical constructed with heavy reinforced concrete walls, floor and roof slabs, and heavy structural steel framing systems. Foundation pressures are expected to be on the order of 3 to 5 ksf (static) and 10 ksf (dynamic) under the planned structures

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Supplemental Soils Report

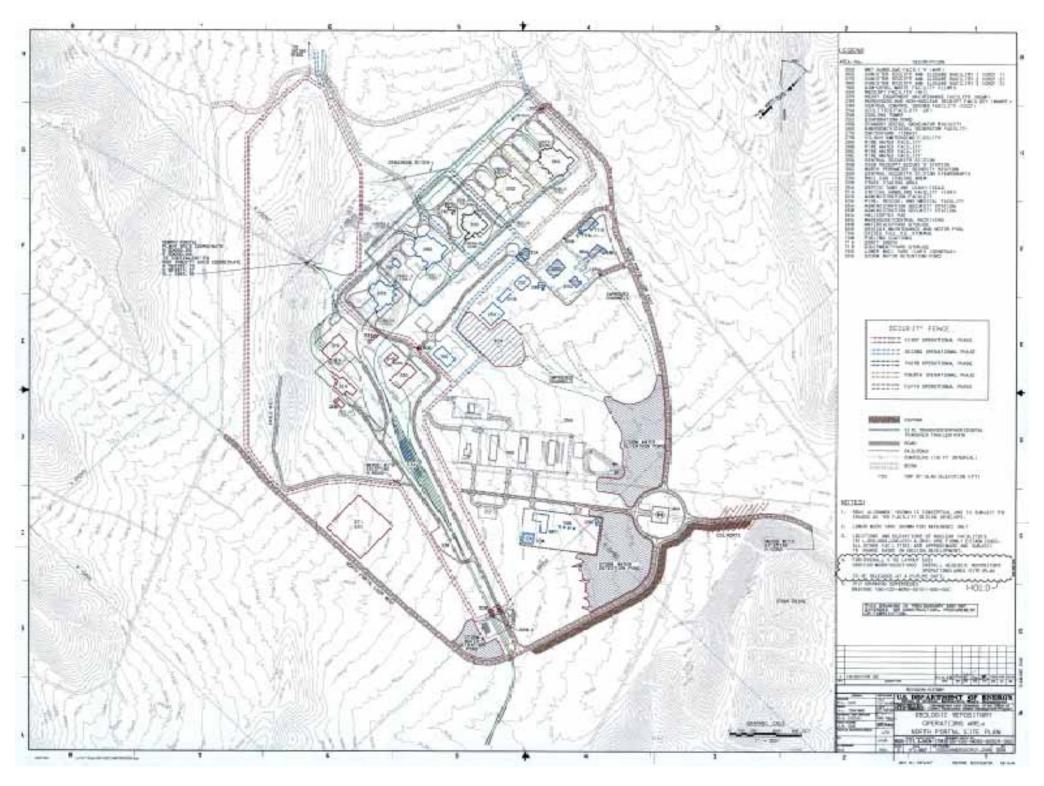


Figure 1-1. Building Layout Sketch. (Ref. Drawing: 100-C00-MGR0-00501-000-00A)

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1.4 LIMITATIONS

Limitations stated in Section 1.3 of BSC (2002b) apply to this report and are briefly summarized below (refer to BSC 2002b for full descriptions):

- 1. These recommendations are intended to provide geotechnical input for the surface facilities to support License Application.
- 2. When the final building configuration and borrow source are defined the recommendations should be reviewed to evaluate whether any changes or additional confirmatory borings or field tests are needed (These items are addressed in Section 7.3 of this report.).
- 3. The bases for the recommendations are limited to the borings, field tests, and laboratory tests performed in the vicinity of the site to date. Although not likely, unanticipated subsurface conditions may be present. The recommendations provided in this report are based on no major deviations occurring from what was observed in the studies to date.
- 4. The recommended bearing capacities and lateral earth pressures are for near horizontal ground conditions (i.e., less than or equal to a 3% slope). However, modifications to the recommendations can be made on a case-by-case basis for any specific conditions that vary appreciably from the near horizontal ground condition.
- 5. Any person using this report for bidding purposes should perform independent investigations, as they deem necessary to satisfy themselves that the surface and subsurface conditions are suitably accurate to determine construction procedures and methods.

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2.2 DESIGN INPUTS

The input data used or considered in the calculation herein are primarily adopted from the following references (for the surface facilities area):

- Geotechnical Data for Potential Waste Handling Building and for Ground Motion Analyses for the Yucca Mountain Site Characterization Project, BSC (2002a)
- Soils Report for North Portal Area, Yucca Mountain Project, BSC (2002b)
- Ground Motion Input Report, BSC (2004a)

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2.2.4 Drawings

BSC 2007. *Geological Repository Operations Area North Portal Site Plan.* 100-C00-MGR0-00501-000-00A (DC #503740). Las Vegas, Nevada: Bechtel SAIC Company.

2.3 DESIGN CONSTRAINTS

None.

2.4 DESIGN OUTPUTS

This calculation will be used as input for other calculations. Summaries of material properties and design parameters derived from this calculation are provided in Tables 2-1 and 2-2.

Table 2-1. Recommended Material Parameters

Design Parameter	Layer			
	Engineered	Roller	Alluvium	Bedrock
	Fill	Compacted		
		Cement ^a		
Moist Density, γ (pcf)	127 pcf	130-140 pcf	114–117 pcf	100 pcf
Specific Gravity, G _s	2.5		2.5	Not Applicable
Shear Strength Parameters	$ \phi = 42^{\circ} $ $ c = 0 $	$\phi = 0$ c = 400 psi (unconf. comp.)	$ \phi = 39^{\circ} $ $ c = 0 $	Not Applicable
At-Rest Earth Pressure Coefficient, K _o	0.33	Not Applicable	0.37	Not Applicable
Active Earth Pressure Coefficient, K _A	0.20	Not Applicable	0.23	Not Applicable
Passive Earth Pressure Coefficient, K _P	5.0	Not Applicable	4.4	Not Applicable
Static Elastic Modulus, E (ksi)	14–28	Not Available	30–75	Not Applicable
Poisson's Ratio, υ	0.3-0.4	0.3	0.23-0.44	0.3
Shear Wave Velocity, Vs (fps)	630–1,500	2,000–3,000	Figure 6-21, Figure 6-23, and Base Case - Figure 6-25	Figure 6-26 and Figure 6-28
Compression Wave Velocity, Vp (fps)	1,500–3,700	3,700–5,600	Figure 6-22 and Figure 6-24	Figure 6-27 and Figure 6-29
Low-Strain Shear Modulus, G (ksi)	10–60	100–270	40–200	150–1,000
Low-Strain Elastic Modulus, E (ksi)	30–170	260–700	100–500	400–2,500
Shear Modulus Reduction, G/G _{max}	Figure 6-32	Figure 6-34	Figure 6-30– upper figure	Figure 6-31– upper figure
Material Damping Ratio, D%	Figure 6-33	Figure 6-35	Figure 6-30– lower figure	Figure 6-31– lower figure
Resistivity (ohm-m)	To Be	To Be	To Be	Not Applicable
	Determined	Determined	Determined	
CBR	20-60	Not Applicable	20–60	Not Applicable
Soil Profile Type	S_D	S _C (very dense	S_{C}	S _B (rock) to
(ICC 2000)	(stiff soil)	soil and soft rock) to S _B (rock)	(very dense soil and soft rock)	S _A (hard rock)

^a additional testing required for verification

Table 2-2. Summary of Recommended Surface Facilities Foundation Design Parameters

Design Parameter	mmended Surface Facilities Foundation Design Parameters Results / Recommendations		
Soil Material Properties	Table 2-1		
Foundation Pressure	Settlement controls design		
Foundation Flessure	Square and Continuous footings: Figure 7-2 and Figure 7-3		
Estimated Settlements	i ü ü		
Estimated Settlements	Square and strip footings Figure 7-4 through Figure 7-6		
	rigure 7-4 unrough rigure 7-6		
	Mat foundation (450' × 500')		
	Load (ksf) Center (in) Corner (in) Differential		
	(in)		
	3 0.2–0.4 negligible 0.4		
	5 0.5-1.6 < 0.1 1.5		
	7 1.3–3.0 < 0.1 3.0		
Lateral Pressures	Yielding walls		
Editoral Fressares	Static and seismic pressures: Figure 7-7		
	Surcharge loads: Figure 7-8 and Figure 7-9		
	Suremarge roads.		
	Non-yielding walls		
	Static and seismic pressures: Figure 7-10		
	Compactor-induced pressures: Figure 7-11 thru		
	Figure 7-15		
Lateral Load Resistance	Friction Coefficient, tan ϕ for alluvium:		
	0.81		
	for engineered fill: 0.90		
	Passive resistance: 515 pcf equivalent fluid		
Temporary Shoring	For braced excavation		
	Lateral pressure: 17H psf		
Temporary Slopes	1.5H:1V		
Permanent Slopes	2H:1V		
Modulus of Subgrade	<u>Alluvium</u> <u>Engineered Fill</u>		
Reaction	Horizontal: 104-120 kcf (60-70 pci) 60-96 kcf (35-55		
(static loading; ranges	pci)		
may be doubled for short-			
term loading)	Vertical:		
	1ft ×1ft footing 1000 kcf (580 pci) 600 kcf (350 pci)		
	Large mats 155-520 kcf (90-300 pci) 75-250 kcf (45-		
	145 pci)		
Percolation Rates	5×10^{-5} to 5×10^{-4} fpm		
Depth of Frost Penetration	10 inches: see Figure 7-16		
Minimum Footing Depth	2 feet		
winning rooting Depth	2 ICCt		

3 ASSUMPTIONS

This calculation is a compilation of available geotechnical information for use in preliminary design of waste handling surface facilities. It is written to adopt, clarify, and summarize findings and recommendations of BSC (2002a) and BSC (2002b) into design charts and tables. The same assumptions as listed in Section 5 of BSC (2002b) have been used in this calculation. There were no assumptions requiring verification in BSC, 2002b.

3.1 ASSUMPTIONS REQUIRING VERIFICATION

There are no assumptions used in this calculation requiring verification.

3.2 ASSUMPTIONS NOT REQUIRING VERIFICATION

There are no additional assumptions (to those listed in BSC, 2002b) used in this calculation.

4 METHODOLOGY

4.1 QUALITY ASSURANCE

This calculation was prepared in accordance with procedures EG-PRO-3DP-G04B-00037, Calculation and Analyses (ACC: ENG 20070122.0010). The Basis of Design for the TAD Canister-Based Repository Design Concept (ACC: ENG 20061023.0002) classifies the nuclear waste handling surface facilities as Important to Safety. Hence, the approved version of this document is designed as QA:QA.

4.2 USE OF SOFTWARE

Excel 2003 and Word 2003, which are part of this Microsoft Office 2003 suite of programs, were used in this report. Microsoft Office 2003 as used in this calculation is classified as Level 2 software usage as defined in IT-PRO-0011 *Software Management* (ACC;DOC 20061221.0003) and is listed on the *Repository Project Management Automation Plan* (ACC: ENG.20060703.0001).

Mathcad version 13 was utilized in this calculation. Mathcad was operated on a PC system running the Window 2003 operating system. Mathcad as used in this calculation is considered as Level 2 software usage as defined in IT-PRO-0011, *Software Management* (ACC: DOC.20061221.0003). Mathcad version 13 is listed on the *Repository Project Management Automation Plan.* (ACC: ENG 20060703.0001).

4.3 CALCULATION APPROACH

This calculation reviews existing analyses, reports, drawings, and other documents to determine relevant aspects that have the potential to contribute to and enhance the evaluation of soil

materials present at the site. Analytical methods of relevant engineering concepts with arithmetic computation and logic are used.

4.4 DESIGN CRITERIA

The criteria itemized in Section 4.2 of BSC (2002b) are, in general, applicable for this calculation. The current project design criteria are contained in BSC (2006b). Applicable criteria are briefly summarized below.

- 1. The final building grades will be above the probable maximum flood level (BSC 2006b, Section 6.1.9).
- 2. The nominal grades within pad areas shall be as required to provide proper drainage (BSC, 2006b, Section 4.2.1.7).
- 3. The pad configuration will prevent ponding of water (BSC 2006b, Section 4.2.1.6).
- 4. Site drainage will direct natural surface runoff around surface facilities (BSC 2006b, Section 4.2.1.6).
- 5. Fill side slopes will be no greater than 2 horizontal to 1 vertical (BSC 2006b, Section 4.2.1.7).
- 6. A minimum surcharge pressure of 300 psf shall be applied for the design of all subsurface walls (BSC, 2006b, Section 4.2.11.3.5).
- 7. The layout will locate the surface facilities near the North Portal of the repository.

Refer to Sections 4.2 of BSC (2002b) and BSC (2006b) for more thorough descriptions.

5 LIST OF ATTACHMENTS

5.1 APPENDICES

Analyses performed for use in the study herein are documented in the following attached appendices:

Appendix A: Seismic Wave Velocity

Appendix B: Bearing Capacity and Settlement

Appendix C: Lateral Earth Pressures and Resistance to Lateral Loads

6 BODY OF CALCULATION

6.1 SITE DESCRIPTION

6.1.1 Location

The YMP site is situated in the southwestern part of U.S. Department of Energy (DOE) Nevada Test Site (NTS), and on parts of adjacent Nellis Air Force Range and U.S. Bureau of Land Management (BLM) lands (See Section 1.2.1 of CRWMS M&O 1999). The site of the potential surface facilities is totally within Area 25 of the NTS. The surface facilities site extends east from the North Portal Pad, which is the fill pad that was constructed for the Exploratory Studies Facility (ESF). A small portion of the site in the northwest corner lies within engineered fill. The site is approximately 27 miles west-northwest of Mercury, Nevada (Figure 6-1) and is located in Nye County, Nevada approximately 100 miles northwest of the city of Las Vegas.

The approximate northing and easting coordinate ranges of the proposed site are N764,000 to N767,000 and E570,000 to E573,000, respectively (Nevada State Plane). The latitudinal and longitudinal coordinates are 36° 50' and 116° 26.5', respectively.

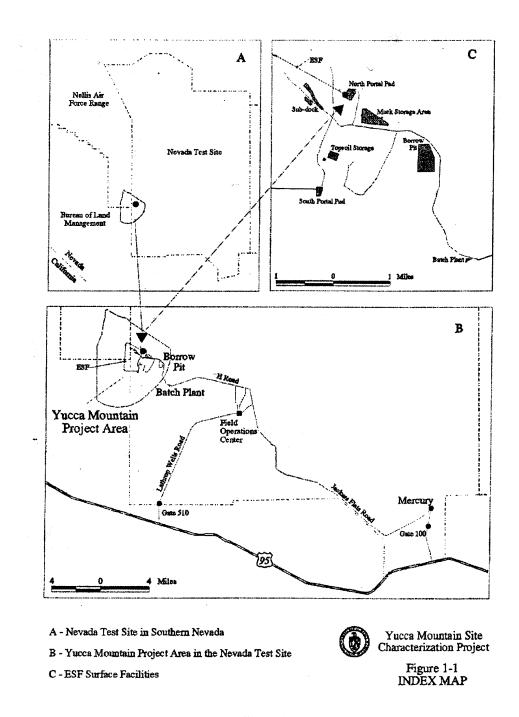


Figure 6-1. Site Vicinity Map

(Figure 1-1 from CRWMS M&O 1999).

6.1.2 Summary of Site Geology

The surface facilities site lies on the western edge of the central portion of the Midway Valley at the eastern toe of Exile Hill. Yucca Mountain lies about 2 miles west of the surface facilities site. Figure 6-2 shows the general geologic features in the vicinity of the site, with the surface facilities area indicated near the center of this figure.

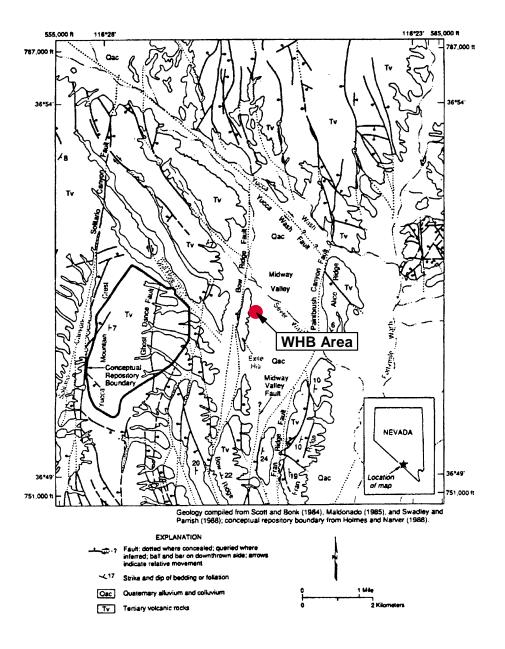


Figure 6-2. Generalized Map of the Midway Valley area

(Fig. 1-1 from Gibson et al. 1992).

The generalized geology of the site consists of alluvial and colluvial deposits overlying tuffitic bedrock. Volcanic rocks of Miocene age dominate the area. Small, intermittent flood-type drainage deposits cross the site area from west to east. The alluvial and colluvial deposits, which originated from Yucca Mountain on the west, vary from about 60 to 120 feet thick under the current building layout and deepen to several hundred feet in the center of the Midway Valley. Thorough descriptions of the geologic settings in the area can be found in Section 2 of CRWMS M&O (1999) and Section 6.6 of BSC (2002a) and their corresponding references.

6.1.3 Existing Conditions and Surface Features

The existing surface conditions and features are succinctly summarized in the following paragraphs, which were excerpted from Section 1.2.1 of CRWMS M&O (1999):

"The ground surface elevation in the vicinity of the WHB [surface facilities] site ranges from about 3,000 feet in the lower reaches of Forty Mile Wash, southeast of the site, to over 6,000 feet in the closer areas of Timber Mountain Caldera, about 4 miles to the north.

The crest of Yucca Mountain averages roughly 4,900 feet in elevation. Relief near the site of the WHB [surface facilities] site is approximately 250 feet, from roughly 3,850 feet elevation at the crest of Exile Hill, immediately west of the site, to roughly 3,600 feet elevation at the center of Midway Valley, east of the site.

The North Portal Pad is located along the western margin of Midway Valley, at the eastern base of Exile Hill. It is an area of approximately 800 to 1,200 feet by 600 to 700 feet of man-made fill sloping roughly 2 degrees to the east, and is situated at approximately 3,670 to 3,683 feet elevation. Muck piles along the eastern side of the North Portal Pad rise to approximately 3,700 feet elevation. The eastern part of the surface facilities footprint is in the area of the present muck piles.

Beneath fill placed for the North Portal Pad is a variable thickness of colluvial and alluvial material overlying Tertiary volcanic bedrock units. The North Portal Pad is the surface at which the ESF tunnel portal was constructed. The pad supports the muck-handling facilities for the tunnel excavation, as well as offices, shops and rail equipment supporting the boring of the ESF tunnel, and facilities for engineering and scientific testing in the ESF."

6.1.4 Subsurface Conditions

This section provides a general description of some of the subsurface conditions at the surface facilities area. The descriptions of the subsurface conditions are based on information obtained from existing boreholes in the area. Refer to BSC (2002a) Section 6.6.2 and BSC (2002b) Section 6 for more detail. Figure 6-4 and Figure 6-5 show existing geologic cross-sections near the site. The cross-sections are taken from Figures 225, cross-section A-A' and 226, cross-section B-B' of BSC (2002a) and span in the NW-SE and NE-SW directions, respectively. The locations of these cross-sections and the layout of the proposed facilities are shown in Figure 6-7. Although these cross-sections do not span through the area of the current layout of the proposed

facilities, they present a general summary of the expected subsurface conditions. It should be noted that these cross-sections were based on data tracking numbers GS020383114233.003 and MO0008GSC00286.000. GS020383114233.003 has been superseded by GS030783114233.001 to account for bedrock depth corrections. The revisions in GS030783114233.001 are not reflected in Figure 6-4 and Figure 6-5. However, the differences are relatively minor and will not affect the recommendations of this calculation. A sketch of the stratigraphy beneath a typical surface facility is shown in Figure 6-6.

6.1.4.1 Existing Fill

Non-engineered fill, varying in thickness from 5 to 22.4 feet (refer to Tables 4 and 5 of BSC (2002a) for fill contact depths), covers the surface of the western edge of the proposed structures at the site. The existing fill it is planned to be removed prior to the construction of the surface facilities (see BSC 2002b, Section 6.1) and be replaced by an engineered fill. Section 3.7 of CRWMS M&O (1999) provides more information about the existing fill. It is understood that the fill consists of tunnel muck material from the exile hill, and from borrow areas of Fran Ridge and Forty-mile Wash. Note that Section 5, Assumption 10 of BSC (2002b) states that 28 feet of existing fill was initially logged in one of the borings at the surface facilities area (UE-25 RF#20) and may have been misidentified during field exploration. For that location, the existing fill may, instead, have been only 9 feet thick.

6.1.4.2 Alluvium

Beneath the existing fill there is a layer of alluvial material, consisting of interbedded calcitecemented (caliche) and non-cemented poorly sorted, coarse-grained gravel with sand and some fines, cobbles, and boulders (refer to Tables 4 and 5 of BSC 2002a, for alluvium contact depths). Available information indicates that the thickness of the alluvium is likely to vary considerably at some locations due to irregular erosion. Furthermore, cemented and un-cemented soil layers appear randomly within this soil unit. Based on previous borings performed in the area, the alluvium ranges in thickness from a few feet to approximately 120 feet under the proposed site with the thickness increasing to the east of Exile Hill. As seen in Figure 6-7, a large portion of the site lies outside the North Portal area where no subsurface investigations were performed. Thus, the actual alluvium thickness is not certain. (Note that Section 5, Assumption 9 of BSC 2002b states that alluvium logged in borehole UE-25 RF#21 between 70 and 115 feet may have been misidentified and may, in fact, be bedrock.)

6.1.4.3 Bedrock

As Section 6.3 of BSC (2002b) asserts, there are non-welded and welded tuffs from the units of Timber Mountain and Paintbrush groups underlying the surface deposits of fill and alluvium.

The non-welded units include the following:

- Pre-Rainier Mesa Tuff bedded tuffs (Tmbt1) of the Timber Mountain Group
- Tuff unit "x" (Tpki) of the Paintbrush Group
- Pre-Tuff unit "x" bedded tuffs (Tpbt5) of the Paintbrush Group

Beneath the non-welded units is the densely welded Tiva Canyon Tuff consisting of the following:

- Younger crystal-rich member (Tpcr)
- Older crystal-poor member (Tpcp)

Both of the Tiva Canyon Tuff members are further divided into zones. Refer to Section 6.6.2 and Attachments I and II of BSC (2002a) for a detailed geologic description of the bedrock. Figure 6-3 shows elevation contours for the top-of-bedrock (Figure 232 of BSC 2002a).

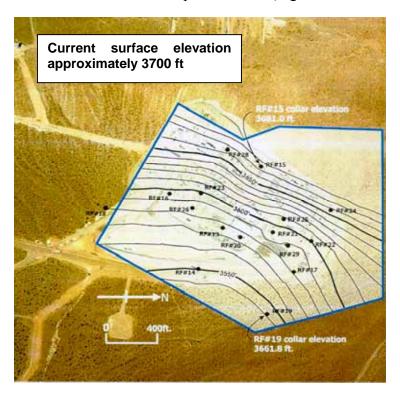


Figure 6-3. Elevation Contours for Top-of-Bedrock Encountered in Boreholes
(Figure 232 of BSC 2002a)

6.1.4.4 Groundwater

Groundwater data relevant to the area is summarized in Section 6.6.3 of BSC (2002a). The groundwater table is located at a typical depth of 1,270 feet below the present ground surface, and is over 1,000 feet below the top of bedrock in the North Portal Area. Hence, groundwater does not affect the geotechnical calculations presented in this study.

6.1.4.5 Proposed Engineered Fill

It is assumed that the existing fill will be removed and that the surface facilities will be founded on the native alluvium soil. Any required engineered fill will likely be obtained from the alluvial sand and gravel deposits at the Fran Ridge Borrow Area, which is located approximately 1.5 miles southeast of the surface facilities. However, due the large design lateral and vertical accelerations, alternative measures are being considered to lock the structures to the ground by a more positive means, such as roller-compacted soil cement (RCSC). Section 6.1.4.6 below discusses estimated properties of RCSC for design evaluation purposes.

6.1.4.6 Roller-Compacted Soil Cement

A literature review was performed to define typical soil properties for use in evaluating potential benefits of using roller-compacted soil cement to replace the tunnel muck that currently underlies the surface facilities site (see BSC 2004b). Papers regarding properties of roller-compacted concrete as well as deep soil mix technologies were reviewed. It is anticipated that a soil-concrete mixture could provide the desired soil response properties for seismic design of the structures and simultaneously provide a high quality control in the field. The report resulting from the literature review is provided in BSC (2004b).

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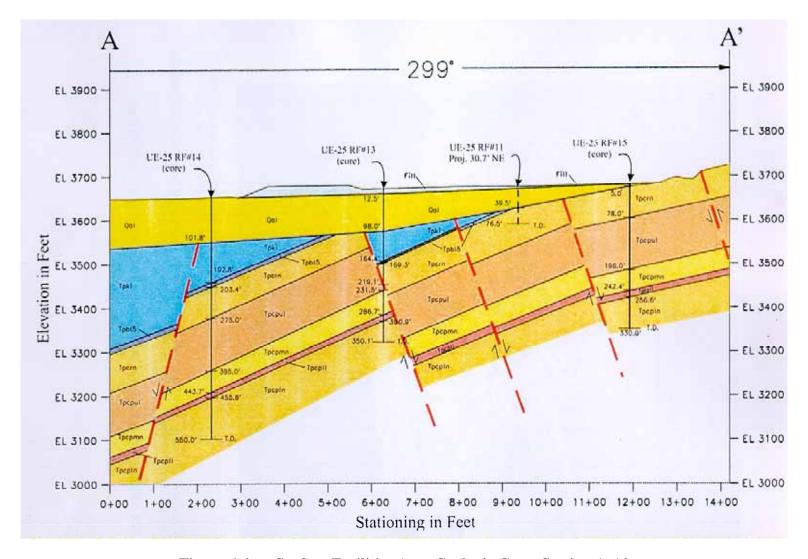


Figure 6-4. Surface Facilities Area Geologic Cross Section A-A' (Figure 225 of BSC 2002a and Assumption 6 of BSC 2002a, DTN:MO0008GSC00286.000–see Figure 6-7 for the location of the cross-section)

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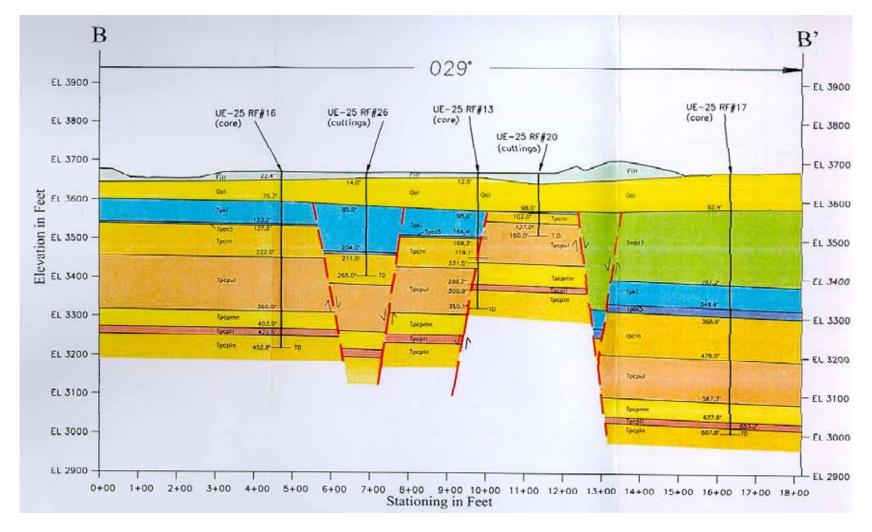


Figure 6-5. Surface Facilities Area Geologic Cross Section B-B' (Figure 226 of BSC 2002a, DTN:MO0008GSC00286.000–see Figure 6-7 for the location of the cross-section)

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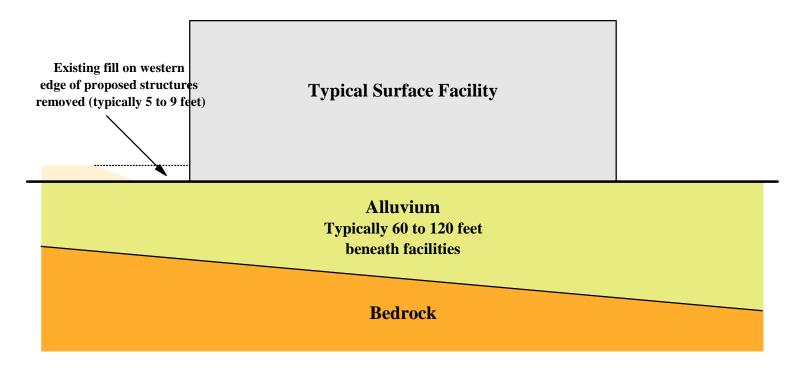


Figure 6-6. Sketch of Stratigraphy Underlying Typical Surface Facility (not to scale).

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6.2 FIELD EXPLORATION AND TESTING

The following sections summarize the soil investigations and field tests performed in the surface facilities area. Soil investigations in the surface facilities area have been conducted since the mid-1980s. The most recent data obtained at the site (as presented in the BSC 2002a and BSC 2002b references) is primarily relied upon as the direct input for the analyses contained in this report. The subsurface investigations for BSC 2002a and BSC2002b were performed within 2000 and 2001. Other data acquired from previous explorations are used as corroborative information.

6.2.1 Field Exploration

6.2.1.1 Borings

Within the surface facilities area, 15 total boreholes (UE-25 RF#14 to RF#26, RF#28, and RF#29) were drilled in 2000 using core hole and mud rotary drilling techniques. Depths of the borings ranged from approximately 100 to 670 feet below top of bedrock (Table 4, Bechtel 2002a). A previous boring (UE-25 RF#13) was cored in 1998 to a depth of approximately 350 feet (Table 5, Bechtel 2002a).

A boring designated as NRG#1 was drilled at the top of the nearby Exile Hill in 1992 (McKeown 1992). Studies performed between 1984 and 1985 (Neal 1985, and Neal 1986) produced 4 borings located within the surface facilities area (UE-25 RF#3, RF#3b, RF#9, and RF#11). The location of all borings drilled in the surface facilities area is provided in Table 6-1. The locations of the borings are shown in Figure 6-7. As seen in the figure, the coverage of the borings is restricted mostly in the northeastern portion of the most recent proposed building layout.

6.2.1.2 Test Pits and Trenches

Investigations performed from 2000 to 2001 included four test pits (TP-WHB-1 to -4) excavated in the surface facilities area. The test pits were each excavated to a depth of approximately 20 feet into the alluvial material. No fill was encountered in the test pits. A total of 22 samples of the alluvium were obtained from the four test pits for laboratory testing. The test pit locations are shown in Figure 6-7.

Previous investigations in the surface facilities area during the 1980s and 1990s included numerous excavations of shallow test pits (designated as NNWSI, SFS, NRSF, GSF and MWV-P) and trenches (MWV-T). Documentation of these test pits is provided in Holmes & Narver 1983, Ho et al. 1986, McKeown 1992, and Map ID SA95-9-15 of DOE 1995. Table 6-2 provides a summary of all known test pits and trenches excavated in the surface facilities area. All of the test pits were performed in the alluvial material.

Four disturbed samples of material to be potentially used as engineered fill were obtained from the existing borrow area (Fran Ridge Borrow Area) at widely spaced locations. The location of Fran Ridge is shown in Figure 1-1. Figure 6-8 (taken from Figure 213 of BSC 2002a) shows the

sampling locations taken from the Fran Ridge Borrow Area. These samples were combined into a composite sample and taken to offsite laboratory facilities for testing.

Table 6-1. Boring information in surface facilities area.

Date	Identification	Coordinates (Nevada State Plane), ft		Total Depth (ft)	Source
		Northing	Easting	(11)	
March 1984 -	UE-25 RF#3	765,575	571,100	301	Neal (1985) &
July 1985	UE-25 RF#3B	765,695	571,066	111	Neal (1986)
	UE-25 RF#9	765,945	570,643	105	
	UE-25 RF#11	765,622	570,435	77	
November 1992	UE-25 NRG#1	765,359	569,803	150.1	McKeown (1992)
October 1998	UE-25 RF#13	765,500	570,720	350.1	BSC (2002a)
June -	UE-25 RF#14	765,309	571,065	550	
November 2000	UE-25 RF#15	765,774	570,225	330	
2000	UE-25 RF#16	765,056	570,473	452.8	
-	UE-25 RF#17	766,076	571,042	667.8	
	UE-25 RF#18	764,522	570,627	493.6	
	UE-25 RF#19	765,880	571,384	645.2	
	UE-25 RF#20	765,637	570,797	160	
	UE-25 RF#21	765,899	570,739	192.2	
	UE-25 RF#22	766,206	570,793	540.6	
	UE-25 RF#23	765,311	570,465	159.1	
-	UE-25 RF#24	766,344	570,542	268	
	UE-25 RF#25	765,968	570,626	159	
	UE-25 RF#26	765,248	570,580	264.9	
	UE-25 RF#28	765,510	570,105	99.8	
	UE-25 RF#29	766,018	570,836	430	

- NRG–North Ramp Geotechnical
 RF–Repository Facility

Table 6-2. Test Pit and Trench Information in surface facilities area.

Date	Identification	Coordinates (Nevada State Plane), ft		Source
		Northing	Easting	
May 1983	NNWSI 2	764,850	570,941	Holmes & Narver (1983)
May 1984	SFS-3	764,850	570,941	Ho et al (1986)
Spring 1992	NRSF-TP-1	765,193	569,828	McKeown (1992) &
	NRSF-TP-2	765,313	569,892	DOE (1995)
	NRSF-TP-3	765,359	569,946	
	NRSF-TP-4	765,383	569,998	
	NRSF-TP-5	765,430	569,977	
	NRSF-TP-6	765,510	570,002	
	NRSF-TP-7	765,463	570,093	
	NRSF-TP-8	765,506	570,101	
	NRSF-TP-9	765,571	570,029	
	NRSF-TP-10	765,669	570,015	
	NRSF-TP-11	765,638	570,206	
	NRSF-TP-12	765,641	570,035	
	NRSF-TP-13	765,798	570,140	
	NRSF-TP-14	765,700	570,244	
	NRSF-TP-15	765,837	570,228	
	NRSF-TP-16	765,790	570,344	
	NRSF-TP-17	765,916	570,277	
	NRSF-TP-18	765,860	570,382	
	NRSF-TP-19	765,621	570,511	
	NRSF-TP-20	765,541	570,436	
	NRSF-TP-21	765,599	570,346	
	NRSF-TP-22	765,521	570,313	
	NRSF-TP-23	765,462	570,390	
	NRSF-TP-24	765,218	570,255	
	NRSF-TP-25	765,113	570,360	
	NRSF-TP-26	765,016	570,036	
	NRSF-TP-27	765,256	570,246	
	NRSF-TP-27a	765,259	570,330	
	NRSF-TP-28	765,093	570,256	
	NRSF-TP-29	765,107	570,201	
	NRSF-TP-30	765,127	570,156	
	NRSF-TP-31	764,987	570,135	
	NRSF-TP-32	765,084	569,969	

- 1. NNWSI-Nevada Nuclear Waste Site Investigation

- 2. SFS-Surface Facility System
 3. NRSF-North Ramp Surface Facilities
 4. SFS-3 was deepened from pre-existing NNWSI 2

Table 6-2. Test Pit and Trench Information in surface facilities area (continued)

Date	Identification	Coordinates (Nevada State Plane), ft		Source
		Northing	Easting	
September	GSF-TP-1	765,966	570,884	Bureau of Reclamation (1992)
1992	GSF-TP-2	765,539	571,110	
	GSF-TP-3	765,040	571,110	
	GSF-TP-4	764,519	571,040	
	GSF-TP-5	764,000	570,935	
1992	MWV-P1	765,405	570,849	DOE (1995)
	MWV-P2	765,259	571,652	
	MWV-P3	764,148	570,845	
	MWV-P9	762,931	572,751	
	MWV-P28	765,178	571,005	
	MWV-P29	765,147	570,387	
	MWV-P30	765,149	570,599	7
	MWV-P31	765,150	570,717	7
	MWV-P32	765,189	571,029	
	MWV-P32a	765,144	571,028	7
1992	MWV-T5A	765,212	570,501	DOE (1995)
	MWV-T6	765,173	569,987	
	MWV-T7	765,482	570,059	
July 2000	TP-WHB-1	766,304	570,772	BSC (2002a)
	TP-WHB-2	765,595	571,106	
	TP-WHB-3	765,306	571,161	
	TP-WHB-4	765,950	571,453	

Notes:

- GSF–Ground Surface Facility
 MWV–Midway Valley
 WHB–Waste Handling Building



Figure 6-7. Locations of Soil Exploration in the surface facilities area.

Cross-Sections shown in Figure 6-4, Figure 6-5, and Figure 7-1.

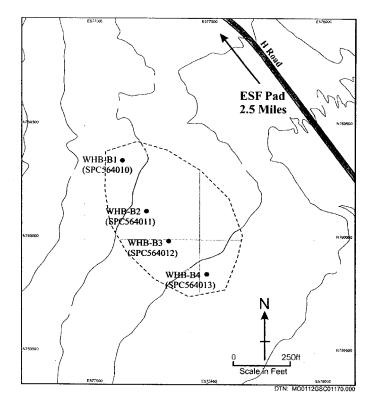


Figure 6-8. Location of Fran Ridge Borrow Samples

(DTN: MO0112GSC01170.000)

6.2.2 Field Tests

6.2.2.1 In-Situ Density Testing

Six 6-foot diameter ring density tests, and sixteen 20-inch diameter sand cone density tests were performed on the alluvial material within test pit excavations in the Fran Ridge borrow area (TP-WHB-1 through TP-WHB-4) from depths of 4 to 20 feet. Caliper and gamma-gamma wireline surveys were also performed in some of the borings primarily to determine the density of the subsurface materials. This is discussed in Section 6.2.2.4. Table 6-3 lists the standards used for the testing.

Test	Standard
Ring density test	• USBR 7221-89, Procedure for Determining Unit Weight of Soils In-Place by the Water Replacement Method in a Test Pit
Sand cone density test	• USBR 7205-89, Procedure for Determining Unit Weight of Soils In-Place by the Sand-Cone Method
Gamma-gamma wireline survey	• PA-PRO-0312), Rev. 0, ICN 0, The Preparation, Planning, and Field Verification of Surface-Based Geophysical Logging Operations (this information is considered historical, therefore, only shown as reference)

Table 6-3. Test Standards Used for In-Situ Density Testing.

Results of the in-situ density tests are shown in Table 6 of BSC (2002a) and discussed in the material properties (Section 6.4) of this report. The materials from these tests were sealed and taken to an offsite geotechnical laboratory for further soil property and classification testing (See Section 6.4, Material Properties section of this report).

In-place density tests were also conducted for materials from several test pits and borings performed in the mid-1980s to early 1990s (see Table 6-1 and Table 6-2). Methods used to measure the densities included water replacement tests (McKeown 1992), and sand cone and nuclear densometer tests (Ho et al. 1986). Data from these tests are compiled and used as corroborative information in the analyses contained herein.

6.2.2.2 Standard Penetration Tests

Standard Penetration Test (SPT) blowcounts were obtained at 5-foot intervals up to 100 feet in depth in RF#13 using a Modified California (MC) sampler (140-pound hammer with a drop of 30 inches). A review of the literature also revealed that SPT blowcounts were performed in TP-NNWSI2 (up to 5 feet depth) in May 1983 (Holmes & Narver 1983).

6.2.2.3 Seismic Velocity Surveys at Surface Facilities Area

Several seismic velocity surveys were conducted in the surface facilities area in order to determine the dynamic characteristics of the subsurface materials. The following 3 methods were used:

- 1. Downhole (DH)
 - 22 total surveys extending down to approximately 640 feet in depth
- 2. Suspension logging (DH)
 - 16 receiver-to-receiver surveys extending down to approximately 650 feet in depth
 - 16 source-to-receiver surveys extending down to approximately 650 feet in depth
- 3. Spectral-analysis-of-surface waves (SASW)
 - 40 survey lines extending down to approximately 500 feet in depth

The results and comparisons of the surveys are documented in BSC (2002a) and summarized in Section 6.4.2.1 of this report. Table 6-4 shows a list of the references containing the procedures used to conduct the seismic surveys.

 Method
 Procedure

 Downhole
 • Redpath Geophysics: SN-M&O-SCI-030-V1 (Wong 2002b)

 • GEOVision: SN-M&O-SCI-025-V1 (Luebbers 2002c)

 Suspension
 • SN-M&O-SCI-024-V1 (Luebbers 2002a)

 • SN-M&O-SCI-024-V2 (Luebbers 2002b)

 SASW
 • SN-M&O-SCI-022-V1 (Wong 2002c)

 • SN-M&O-SCI-040-V1 (Wong 2002a)

Table 6-4. References of Seismic Survey Procedures.

Table 6-5 (Table 31 from BSC 2002a) describes and compares the different seismic velocity surveying methods. Table 6-6 lists the borings in which the seismic velocity surveys were performed in the surface facilities area. The locations of the borings in which downhole and suspension seismic surveys were performed are shown in Figure 6-7. Figure 6-9 (Figure 43 of BSC 2002a) shows the locations of the SASW lines at the surface facilities site.

Table 6-5. Comparison of downhole seismic, suspension seismic and SASW methods (Table 31 of BSC 2002a)

technique and interpreter were all apparently constant, suggesting that variation of this magnitude can occur over short distances due to geologic variability.

Table 31. Comparison of Downhole Seismic, Suspension Seismic and SASW Methods

Characteristic	Suspension Seismic	Downhole Seismic	SASW
Energy source	Built-in solenoid hammer	Hammer on plank	Hammer at close source- receiver spacings; sledgehammer, dropped weight, bulldozer or vibroseis at longer spacings
Type of wave generated	P and S	P and S	Rayleigh or other surface wave
Ability to reverse polarity	Yes	Yes	No
Primary direction of wave motion	Upward, vertical	Downward, near vertical but becoming more inclined at shallow depth	Horizontal
Wave frequency, Hz	S wave 500 - 1,000	S wave 20 - 40	5 – 500 or more
	P wave 1,000 – 3,000	P wave 50 – 200	
Boreholes required	One	One	None
Borehole requirements	Liquid-filled; uncased generally preferred; plastic casing is acceptable	Dry preferred; casing optional	Not applicable
Maximum effective depth, ft	1,600	300 to 700	Up to 500
Resolution	Resolution constant with depth	Resolution decreasing with depth	Resolution decreasing with depth
Borehole drift survey	Not required	Not required	Not applicable
Space limitations	Can be performed wherever a borehole can be drilled	Can be performed wherever a borehole can be drilled	Line length is about 2 times the depth surveyed, so on-site and off-site constraints may limit survey depth
Type of wave interpreted	P and S _H	P and S _H	R, converted to S using theory and assumed Poisson's ratio
Interval velocity	Yes	Only with geophones at multiple depths	No
Average velocity	Yes, by accumulation of individual travel times	Yes	Yes

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Table 6-6. Seismic Velocity Survey Summary

Borehole ID	Downhole seismic			Suspension seismic (source-to-receiver and receiver-to-receiver)		SASW ^[6]	
	Reynolds and Associates	URS ^[2]	Redpath Geophysics [3]	GEOvision Inc.		Luebbers M. J.	University of Texas at Austin ^[5]
UE-25 RF#3	×						
UE-25 RF#3B	×						
UE-25 RF#9	×						
UE-25 RF#10	×						
UE-25 RF#13		×	×	×	×		SASW-1
UE-25 RF#14			×			×	
UE-25 RF#15			×			×	SASW-10+37
UE-25 RF#16 ^[7]			×			×	SASW-29
UE-25 RF#17				×		×	SASW-34+36 ^[8]
UE-25 RF#18 ^[7]			×			×	
UE-25 RF#19			×			×	
UE-25 RF#20 ^[7]			×			×	
UE-25 RF#21 ^[7]			×			×	SASW-2
UE-25 RF#22 ^[7]			×			×	SASW-23
UE-25 RF#23			×			×	SASW-32+35, SASW-33
UE-25 RF#24 ^[7]			×			×	SASW-4
UE-25 RF#25			×			×	
UE-25 RF#26			×			×	
UE-25 RF#28 ^[7]			×			×	SASW-8
UE-25 RF#29			×			×	

- [1] 1985 surveys
- [2] December 1998 survey
- [3] October through December 2000 surveys
- [4] September through December 2000 surveys
- [5] July through August 2000 surveys
- [6] A total of 40 SASW surveys were performed in the proposed surface facilities area. Five of these surveys were combined with other adjacent surveys resulting in 35 dispersion curves. A total of 35 shear-wave velocity profiles were developed. 11 of these profiles were performed on top of existing boreholes (BSC 2002a). Refer to Figure 6-9 for all SASW line locations.

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- [7] Caliper and gamma-gamma wireline surveys were performed in these boreholes
- [8] 2 velocity profiles measured at SASW line survey



Figure 6-9. Locations of SASW lines at the surface facilities site (Figure 43 of BSC 2002a)

6.2.2.4 Borehole Wireline

Caliper and gamma-gamma wireline surveys were performed in 7 boreholes (RF#16, #18, #20, #21, #22, #24, and #28.). Caliper measurements were performed in order to assess the extent of erosion of the borehole walls by the drilling fluid and its potential effects on the suspension seismic results. The main purpose of performing the gamma-gamma measurements was to evaluate the density of the subsurface materials.

The process established in PA-PRO-0312 (this information is considered historical, therefore, only shown as reference), Yucca Mountain Site Characterization Project Field Verification of Geophysical Operation, and AP-SIII.6Q, Geophysical Logging Programs for Surface-Based Testing Program Boreholes, were followed for both the caliper and gamma-gamma wireline surveys.

6.3 LABORATORY TESTING

This section discusses laboratory testing conducted on samples taken during 2000 to 2001 from the borings and test pits performed at the surface facilities area.

6.3.1 Static Testing

Documentation of all static laboratory testing is found in Sections 6.2.9 and 6.5.2 of BSC (2002a) for the alluvial and borrow pit materials, respectively. A summary of the static laboratory test results is presented in Section 6.4 of this report.

6.3.1.1 Alluvium

The following static tests were performed on each of 22 samples of alluvial material obtained from test pits TP-WHB-1 through –4 (see Figure 6-8). All tests were conducted at a geotechnical laboratory located in Denver, Colorado. The tests conducted are listed in Table 6-7 along with the testing standards used.

Table 6-7. Laboratory Tests and Standards Conducted on Alluvium.

Test	Standard
Atterberg Limits	 USBR 5350-89, Procedure for Determining the Liquid Limit of Soils by the One-Point Method USBR 5360-89, Procedure for Determining the Plastic Limit and Plasticity Index of Soils.
Maximum	, , ,
Maximum and Minimum Index	For particles passing the 3-inch sieve:
Unit Weights	• USBR 5525-89, Procedure for Determining the Minimum Index Unit Weight of Cohesionless Soil
	• USBR 5530-89, Procedure for Determining the Maximum Index Unit Weight of Cohesionless Soils.
Particle-Size Distribution	• USBR 5325-89, Procedure for Performing Gradation Analysis of Gravel Size Fraction of Soils
	• USBR 5330-89, Procedure for Performing Gradation Analysis of Fines and Sand Size Fraction of Soils, Including Hydrometer Analysis
	• USBR 5335-89, Procedure for Performing Gradation Analysis of Soils Without Hydrometer–Wet Sieve.
Specific Gravity	For particles passing the 4.75 mm (No. 4) sieve:
	• USBR 5320-89, Procedure for Determining Specific Gravity of Soils (volume method)
	For particles retained on the 4.75 mm (No. 4) sieve:
	• USBR 5320-89, Procedure for Determining Specific Gravity of Soils (suspension method).
Unified Soil Classification System	• USBR 5000-86, Procedure for Determining Unified Soil Classification (Laboratory Method).
Water Content	• USBR 5300-89, Procedure for Determining Moisture Content of Soil and Rock by the Oven Method.

6.3.1.2 Engineered Fill

Disturbed samples of the borrow material were taken from 4 locations (WHB-B1 to WHB-B4) in the Fran Bridge Borrow Area and then combined into one bulk sample. Testing was conducted at laboratory facilities in Denver, Colorado and Santa Ana, California (URS Greiner Woodward Clyde). The tests conducted are listed in Table 6-8 below along with the testing standards used (where provided).

Table 6-8. Laboratory Tests and Standards Conducted on Engineered Fill Material.

Test	Standard
Atterberg Limits	• USBR 5350-89, Procedure for Determining the Liquid Limit of Soils by the One-Point Method.
Compaction Test	• ASTM D 1557, Standard Test Method for Laboratory Compaction Characteristics of Soil Using Modified Effort (56,000 ft-lbf/ft ³ (2,700 kN-m/m ³)).
Maximum and	For particles passing the 3-inch sieve:
Minimum Index Unit Weights	• USBR 5525-89, Procedure for Determining the Minimum Index Unit Weight of Cohesionless Soils
	• USBR 5530-89, Procedure for Determining the Maximum Index Unit Weight of Cohesionless Soils.
Particle-Size Distribution	• USBR 5325-89, Procedure for Performing Gradation Analysis of Gravel Size Fraction of Soils
	• USBR 5335-89, Procedure for Performing Gradation Analysis of Soils Without Hydrometer—Wet Sieve.
	• ASTM C 136, Standard Test Method for Sieve Analysis of Fine and Coarse Aggregates, for 3 conditions: (1) as received; (2) after scalping on the ½-inch sieve and prior to compaction; and (3) after the compaction test on the ½-inch minus material.
Specific Gravity	For particles passing the 4.75 mm (No. 4) sieve:
	• USBR 5320-89, Procedure for Determining Specific Gravity of Soils (volume method).
	Denver, Colorado laboratory for particles retained on the 4.75 mm (No. 4) sieve:
	• USBR 5320-89, Procedure for Determining Specific Gravity of Soils (suspension method).
Triaxial Test	• Four triaxial tests performed on reconstituted specimens under isotropically consolidated, drained conditions.
Unified Soil Classification System	• USBR 5000-86, Procedure for Determining Unified Soil Classification (Laboratory Method).

6.3.2 Dynamic Testing

Dynamic properties of the alluvium, bedrock (tuff), and engineered fill were evaluated using combined resonant column and torsional shear (RCTS) tests. The laboratory dynamic testing was performed in the Geotechnical Engineering Center at the University of Texas at Austin. Testing procedures are presented in Section 6.2.10.1 of BSC (2002a) and SN-M&O SCI-033-V1 (Wong 2002d).

Dynamic properties, including the shear modulus and material damping relative to shearing strain, were determined from the laboratory tests on samples of alluvium, bedrock, and engineered fill. A summary of the results of the dynamic testing is presented in Section 6.4.2 of this report. Table 6-9 lists the testing standard and reference used for the dynamic tests.

Table 6-9. Standard and Reference Used for Dynamic Testing.

Test	Standard and Reference
Resonant column and torsional shear (RCTS)	 PA-PRO-0310, Laboratory Dynamic Rock/Soil Testing (this information is considered historical, therefore, only shown as reference). SN-M&O-SCI-033-V1 (Wong 2002d)

6.3.2.1 Alluvium

One combined alluvial sample was collected from boreholes RF#14 to #17. The specimen was reconstituted in the laboratory due to sampling disturbance, using the standard under-compaction method of Ladd (1978).

Additionally, dynamic testing was also performed on a soil sample taken from borehole RF#13 in 1999. A summary of the test results from this sample is provided in CRWMS M&O (1999, Appendix Q).

6.3.2.2 Engineered Fill

Ten reconstituted specimens taken from the Fran Ridge borrow area were tested. Four of the samples were tested in 2 stages to investigate the dynamic property effects of increasing the water content of the granular fill after placement.

6.3.2.3 Bedrock (**Tuff**)

Eighteen undisturbed specimens taken from boreholes RF#14 to #17 were tested. During testing, the specimens were divided into three groups based on their dry unit weight, γ_d :

- Group 1: γ_d from 133 pcf to 147 pcf
- Group 2: γ_d from 117 pcf to 132 pcf
- Group 3: γ_d from 78 pcf to 94 pcf

6.4 MATERIAL PROPERTIES

This section presents a summary and discussion of the results of both static and dynamic laboratory tests on the soil units at the site. All information presented in the following sections is based on data presented in BSC (2002a) and BSC (2002b). A summary of recommended material properties for design is presented in Table 2-1.

6.4.1 Static Soil Properties

6.4.1.1 Alluvium

Results of the in-situ density tests and laboratory tests conducted on the alluvial material from TP-WHB-1 to TP-WHB-4 are shown in Tables 6 and 13 of BSC (2002a), respectively. A plot of the gradation test results reported in BSC (2002a) is provided in Figure 6-10.

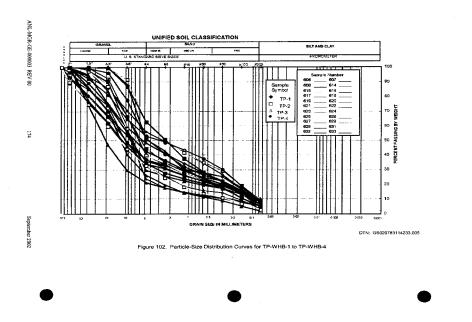


Figure 6-10. Particle-size distribution curves for Alluvium for TP-WHB-1 to TP-WHB-4

The following sections describe the results of testing on 22 samples obtained at depths ranging from 4 to 20 feet. There were no alluvium samples obtained for depths greater than 20 feet.

6.4.1.1.1 General Characteristics

The alluvium material is generally medium dense to dense, and varies between a well-graded gravel (GW), well-graded gravel with silt (GW-GM), poorly graded sand with silt (SP-SM), and well-graded sand with silt (SW-SM). Intermittent layers of calcite-cemented material (caliche) are present in the alluvium (BSC 2002b, Section 6.2). However, these areas were conservatively not considered in this report. Table 6-10 provides a summary of average soil properties determined from the laboratory testing.

Table 6-10. Results from tests performed on alluvial samples at surface facilities area (DTN: GS020483114233.004).

Test	Results
Particle size distribution	$59 \pm 12\%$ (gravel & cobbles)
	$34 \pm 11\%$ (sand)
	$7 \pm 2.5\%$ (fines)
Plasticity	Non-plastic
Average Density	117 pcf maximum index (passing 3-inch sieve)
	91 pcf minimum index (passing 3-inch sieve)
	107 pcf dry in-place
	68 % relative
Average minimum index density	91 pcf (passing 3-inch sieve)
Average specific gravity and absorption	2.37 apparent
(passing 3-inch sieve)	2.25 bulk (saturated surface dry)
	2.16 bulk (oven dry)
	4.0% absorption
Average specific gravity and absorption	2.47 apparent
(retained on No. 4 and passing 3-inch sieve)	2.26 bulk (saturated surface dry)
	2.12 bulk (oven dry
	9.4% absorption
Average specific gravity	2.52
(passing No. 4)	
Average water content	7.1 % (passing No. 4 sieve)
	4.9 % (retained on No. 4 sieve)

A comparison of the data from Table 6-10 with soil data from earlier geotechnical investigations (1980's and early 1990's) shows good corroboration of the soils properties. The specific gravity of the alluvium at the site is less than typically encountered for sand and gravel soils. See, for instance, USN 1986 (pp. 7.1-23), which uses a specific gravity of 2.65 for granular soils in their tables of typical values.

6.4.1.1.2 Density

From the 22 samples taken within the alluvium from the recent field tests, in-place dry density, and minimum and maximum index density tests were performed (see Table 6-10). An average relative density of 68% was determined from these tests.

Density testing included 6 ring density and 16 sand cone tests taken up to 20 feet in depth into the alluvium, and gamma-gamma surveys extending up to a 480 depth through the alluvium and into bedrock. Based on the field tests, Sections 8.2.1 and I.2.1 of BSC (2002b) recommends the average moist unit weight of the alluvium to be approximately 114 pcf in the upper 8 feet and 117 pcf below 8 feet. Moisture contents vary between about 5 and 7 percent. The data from the gamma-gamma surveys are the only known density measurements at lower depths of the alluvium and are generally lower in value by approximately 25 to 30%. However, Section 8.2.1 of BSC (2002b) indicates that these results may correspond to the bedrock material rather than the alluvium.

Densities from earlier soil investigations were measured by water replacement tests (McKeown 1992), laboratory tests on drive tube samples (Neal 1986), and sand cone and nuclear tests (Ho et al. 1986). A comparison of the data obtained from these measurements to recent field tests show good agreement. Hence, a conservative moist unit weight of 114 to 117 pcf for the alluvium is recommended.

6.4.1.1.3 Shear Strength

Because undisturbed alluvial samples were not obtained in prior geotechnical investigations, correlations from several sources between the relative density and friction angle are used to estimate the strength of the alluvium. Table I-17 of BSC (2002b) presents a summary of the friction angles obtained from the various correlations used. The mean, and mean plus/minus one standard deviation values of relative density are used for the calculation. Based on the correlations between relative density and friction angle, an effective friction angle of 39 degrees, corresponding to halfway between the mean minus one standard deviation and the mean values of the relative density, is recommended for the alluvium for a pressure of 1 atmosphere.

Sections 8.2.2, I.2.2.1, I.2.2.2, I.2.2.3, I.2.2.4, and I.2.2.5 of BSC (2002b) recommend different strength envelopes to be used for different types of analyses (i.e., passive pressures, bearing capacity, and slope stability). A linear failure envelope with no cohesion (c = 0) and producing an equivalent effective friction angle, ϕ_{eff} , of 39 degrees is considered to adequately characterize the alluvial material and is recommended for preliminary design.

SPT data (discussed in Section 6.2.2.2) from borehole RF#13 and TP-NNWSI2 corroborate the conservative shear strength friction angle selected for the alluvium, revealing blow counts on the order of 100 to 300 blows/foot. This also holds true for correlations between shear wave velocity and shear strength, as the measured shear wave velocities at the site correlate to unrealistically high shear strength values.

6.4.1.1.4 Earth Pressure Coefficients

Earth pressure coefficients are calculated for at-rest, active and passive conditions to be used for analyses of lateral earth pressures (Section 7.1.5). Appendix C documents the derivation of these coefficients. The results are shown in Table 2-1.

6.4.1.1.5 Young's Modulus

Static Young's Modulus, E, for the alluvium can be calculated using the results of the elastic settlement analyses contained in Appendix C. For expected vertical loads of 3 and 5 ksf, the elastic settlements computed are 0.4 and 1.6 inches, respectively. Using a maximum alluvium thickness of 120 ft, the average strains induced in the alluvium from the 3 and 5 ksf vertical loading are determined to be 0.03% and 0.11%, respectively. Static Young's modulus can then be determined using:

$$E = \frac{\sigma}{\varepsilon}, (Eq. 1)$$

where E is the Young's modulus, ε is the axial strain, and σ is the vertical stress. E is determined to be 30 to 75 ksi.

Sections 8.2.3 and I.2.3 of BSC (2002b) recommend the following equation to be used for a strain range of 0.1 to 0.5% and for a stress range for 0 to 6 ksf:

$$E = 777.37(\epsilon)^{-0.6505} \sigma^{0.5}$$
, where (Eq. 12 and I-66 of BSC 2002b) (Eq. 2)

 ϵ is the axial strain in percent and σ is the vertical overburden stress. Using an average overburden stress in the alluvium of 11 ksf (σ = 0.117 kcf × 60 from the alluvium weight plus 4 ksf from vertical loading) and an axial strain of 0.1% yields E to be 80 ksi. For design, use a static Young's Modulus of 30 to 75 ksi for static loading conditions (Table 2-1).

6.4.1.1.6 Resistivity

Electrical resistivity of the soil will be required for design of grounding and evaluation of corrosion potential. Field measurements will be required for the alluvium and any engineered fill that is placed. It is expected that the main source of engineered fill will be alluvium and, therefore, the resistivity of these two materials will be similar. Measurements made at eight locations on the alluvial surface prior to building the construction-support pad at the North Portal (U.S. Bureau of Reclamation, 1992, and Bureau of Reclamation, 1993) provide a typical range of values for these materials. The results indicated resistivities measuring between 60 and 540 ohm-meters.

6.4.1.2 Engineered Fill

It is anticipated that engineered fill will be obtained from alluvial soils, possibly processed to some extent. The Fran Ridge Borrow Area is an example of such material. The information

presented in this section are provided for corroborative purposes only. Actual design values will be obtained after a source pit is identified.

Results of the static tests conducted in a laboratory located in Denver, CO, on the fill obtained from the Fran Ridge Borrow Area are presented in Table 6-11 (Table 27 and Figure 214 of BSC 2002a). Results of static strength tests conducted in Santa Ana, CA are presented in Table 28 and Figures 215 through 217 of BSC (2002a). The following sections summarize the results of the laboratory testing on disturbed samples obtained at widely spaced locations of the Fran Ridge Borrow Area.

6.4.1.2.1 General Characteristics

The borrow material is classified as a poorly graded sand to gravel (SP/GP), and, after compaction, a poorly graded sand with silt and gravel (SP-SM). Table 6-11 below presents soil properties determined for engineered fill from laboratory testing.

Table 6-11. Results from tests performed in Denver, CO on a composite sample of Fran Ridge Borrow materials (Table 27 of BSC 2002a, DTN: MO0206EBSFRBLT.018).

Test	Results
Particle size distribution	48% (gravel)
	49% (sand)
	3% (fines)
Plasticity	Non-plastic
Average maximum index density	112.4 pcf (passing 3-inch sieve)
Average minimum index density	94 pcf (passing 3-inch sieve)
Average specific gravity and absorption	2.39 apparent
(passing 3-inch sieve)	2.24 bulk (saturated surface dry)
	2.13 bulk (oven dry)
	5.3% absorption
Average specific gravity and absorption	2.45 apparent
(retained on No. 4 and passing 3-inch sieve)	2.24 bulk (saturated surface dry)
	2.10 bulk (oven dry
	6.9% absorption
Average specific gravity	2.52
(passing No. 4)	

6.4.1.2.2 Total Unit Weight

The results of the compaction test on a composite sample of the Fran Ridge Borrow material indicate a maximum dry unit weight of 114.5 pcf for an optimum water content of 11 percent. Based on the results from the compaction test and on the standard practice presented in ASTM D

4718 for the correction of unit weight and water content for soils containing oversized particles, the moist unit weight for the engineered fill is computed to be 127 pcf [114.5 pcf \times (1+0.11)].

6.4.1.2.3 Shear Strength

A set of four drained triaxial compression tests performed on the composite sample of the Fran Ridge material are used to obtain the shear strength of the engineered fill material. The material was compacted to an average dry density of 110 pcf and water content of 12.5%. The results of these tests are shown in Table 28 and Figures 216 and 217 of BSC (2002a). Figure 6-11 below (Figure 217 of BSC 2002a) shows the results of the triaxial tests.

Sections 8.1.2, I.1.2.3, I.1.2.4, I.1.2.5, and I.1.2.6 of BSC (2002b) recommend various strength envelopes to be used for different types of analyses (general purpose, passive pressures, bearing capacity, slope stability). However, a linear failure envelope with no cohesion (c=0) and producing an equivalent effective friction angle, ϕ_{eff} , of 42 degrees is considered to adequately characterize the alluvial material.

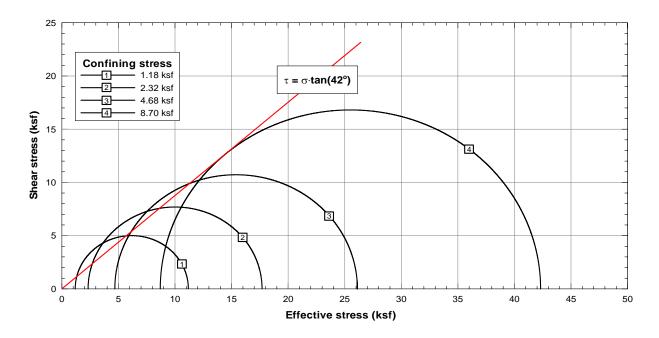


Figure 6-11. Strength envelopes fitted to triaxial tests on engineered fill

(DTN: MO0203EBSCTCTS.016)

6.4.1.2.4 Earth Pressures Coefficients

Earth pressure coefficients are calculated in Appendix C. Table 2-1 presents the results.

6.4.1.2.5 Young's Modulus

Typical Young's Modulus values for a dense sand and gravel material are recommended to be 14–28 ksi (Bowles 1996).

A secant Young's modulus was calculated in Sections 8.1.3 and I.1.3 of BSC (2002b) from the drained triaxial test results. Equations 6 and I-19 of BSC (2002b) are recommended for use in computing the Young's modulus:

$$E = 911.19(\sigma')^{0.4541}$$
, where (Eq. 6 and I-19 of BSC 2002b) (4)

 σ' is the initial isotropic consolidation stress prior to loading. The above equation corresponds to a strain of 0.25%. For an overburden stress of 5.5 ksf, Young's Modulus is estimated to be approximately 14 ksi.

For design, use a static Young's Modulus of 14 to 28 ksi (Table 2-1).

6.4.1.3 Bedrock

6.4.1.3.1 Moist Unit Weight

Density measurements were obtained from the gamma-gamma wireline surveys and dynamic laboratory testing for the bedrock material. The results are discussed in Section 6.4.2.3.

6.4.1.3.2 Shear Strength

Since, the structures will be underlain by a significant amount of alluvium over bedrock, estimation of shear strength of the bedrock is not required for purposes of these analyses and is conservatively ignored. This information can be derived from other project sources if needed.

6.4.2 Dynamic Soil Properties

Dynamic soil properties, including seismic wave velocity, Poisson's ratio, and strain dependent parameters of shear modulus degradation and material damping ratio were developed for use in the future dynamic analyses of the structures and foundations at the surface facilities site. The following sections are a summary of the data compiled and reported in BSC (2002a).

6.4.2.1 Shear and Compression Wave Velocity

Statistical values of shear (V_s) and compression (V_p) wave velocities for each soil layer present at the site (existing fill, alluvium and bedrock) are provided in tabular form in Sections 6.2.5.3, 6.2.6.4, and 6.2.7.3 in BSC (2002a), for the downhole, suspension, and SASW seismic surveys. For this report, the available data was compiled and divided by soil unit (where known), using geologic information provided in the boring logs of the surface facilities area. Representative velocity values of the soil materials are summarized in Table 2-1. Section 6.2.2.3 discusses the data used in the analysis. A detailed discussion of this analysis is provided in Appendix A of this report. This analysis determined a linear fit of shear wave velocity versus depth in the fill and average shear wave velocity for rock types grouped into two main divisions:

- 1. Those occurring lithostratigraphically between Tmbt1 through Tpcpul,
- 2. and those occurring lithostratigraphically between Tpcpmn through Tpcpln.

Shear and compression wave velocities determined from dynamic laboratory tests on the alluvium and bedrock materials discussed in Sections 6.2.10.2, and 6.2.10.3 of BSC (2002a) were not considered to be as accurate as measurements in the field of the seismic wave velocities and thus were not used in the analysis herein. Results of dynamic laboratory tests (RCTS) performed on the engineered fill were considered since no geophysical surveys could be performed on this material. A discussion of the seismic wave velocity values for the roller compacted cement is provided in Section 6.4.3.

Sections 6.2.5, 6.2.6, and 6.2.7, BSC (2002a) provide numerous figures comparing the results of the seismic wave velocity surveys in the surface facilities area. These figures should be referred to for details on individual surveys and specific comparisons between survey methods. Figure 6-12 through Figure 6-17 are taken from Figures 22, 30, 31, 91, 34, and 23 of BSC (2002a), respectively, and show statistical analyses of the shear- and compression-wave velocities measured in the surface facilities area by the three surveying methods. A discussion and comparison of the surveying methods and results of individual velocity profiles are provided in Sections 6.2.5 to 6.2.7, Section 6.7, and Attachments V through IX of BSC (2002a).

Additional analyses of the available shear wave velocity and compression wave velocity data were performed in BSC (2004a) after the initial issue of the current calculation. The 2004 analysis subdivided the analyses based on data measurements taken west (on the upthrown side) and east (on the downthrown side) of the Exile Hill Fault Splay (see Figure 6-18). Additional culling of data was also made based on evaluation of the quality of each data source. This created modified distributions of the available field measurement data for the various subsurface materials. As demonstrated in Figure 6-18, the majority of the available measurements were performed on the upthrown side (west) of the Exile Hill Fault Splay.

Due to its more detailed data analysis, the results from BSC (2004a) are recommended for design. However, the results of the previous analyses are presented for informational purposes.

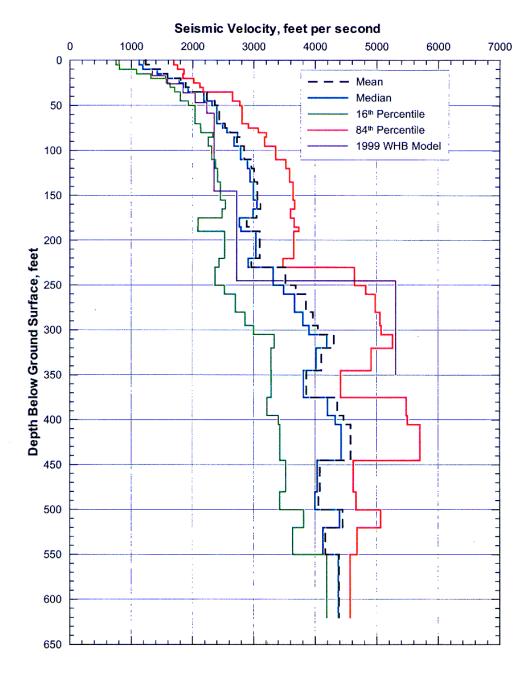


Figure 6-12. Statistical analyses of shear-wave velocities from downhole measurements in the surface facilities area.

(Figure 22 of BSC 2002a)

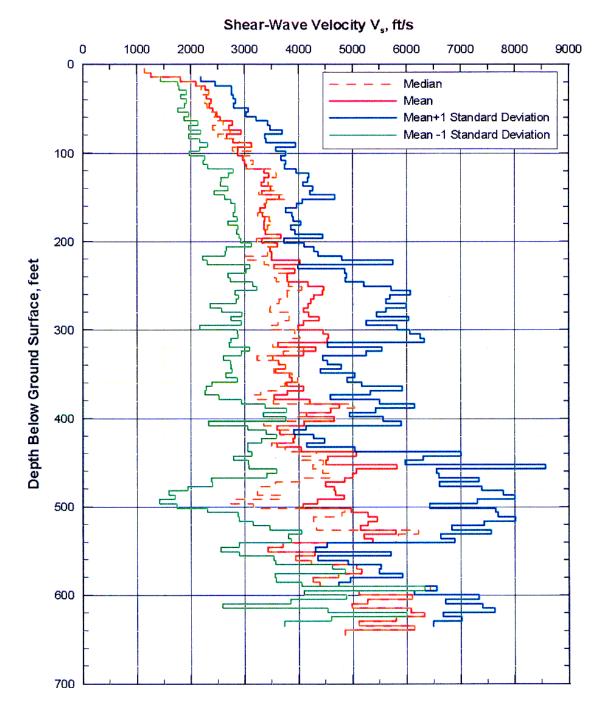


Figure 6-13. Shear wave velocity by depth interval from receiver to receiver interval suspension surveys in surface facilities area.

(Figure 30 of BSC 2002a, DTN: MO02045FTDSUSP.001)

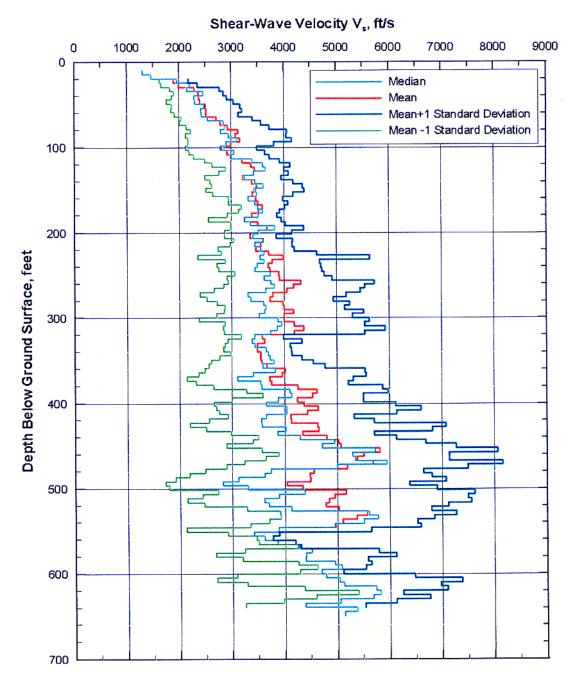


Figure 6-14. Shear wave velocity by depth interval from source to receiver interval suspension surveys in surface facilities area.

(Figure 31 of BSC 2002a, DTN: MO02045FTDSUSP.001)

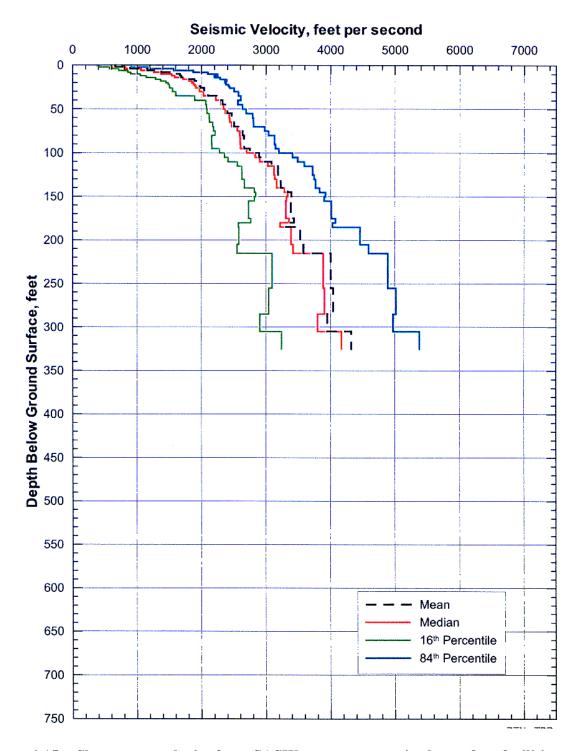


Figure 6-15. Shear wave velocity from SASW measurements in the surface facilities area. (Figure 91 of BSC 2002a)

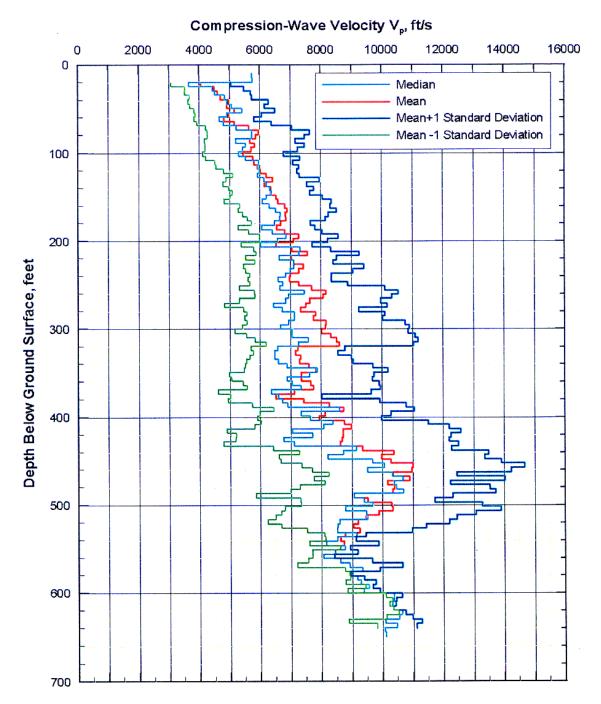


Figure 6-16. Compression wave velocity by depth interval from source to receiver interval suspension surveys in surface facilities area.

(Figure 34 of BSC 2002a, DTN: MO02045FTDSUSP.001)

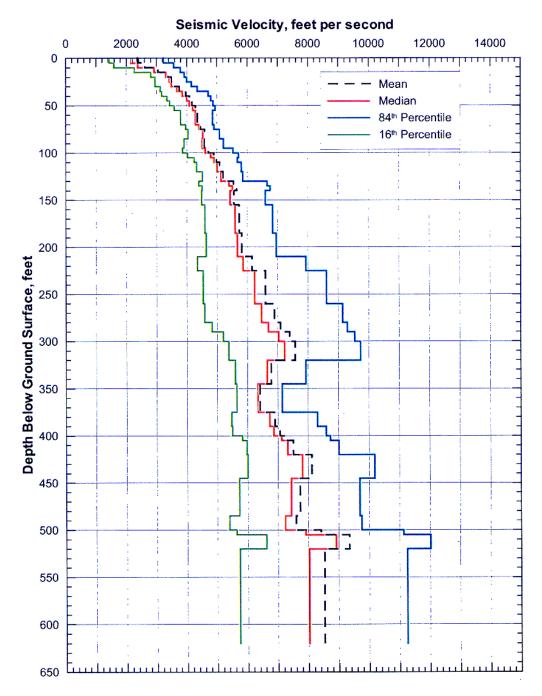
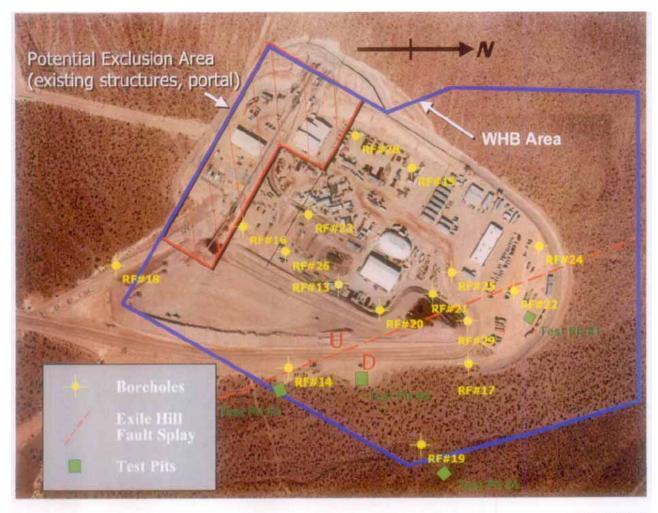


Figure 6-17. Compression-wave velocities from downhole measurements in the surface facilities area.

(Figure 23 of BSC 2002a)



NOT TO SCALE

Source: BSC 2002a [DIRS 157829], Figures 2 and 3, YMP Photograph #BN8811_50

NOTE: "U" and "D" signify the upthrown and downthrown sides, respectively, of the Exile Hill fault splay.

Figure 6-18. WHB showing location and upthrown and downthrown sides of Exile Hill Fault Splay

(Figure 6.2-89 from BSC 2004a)

6.4.2.1.1 Alluvium

Figure 6-19 and Figure 6-20 show compilation plots of the shear and compression wave velocities measured by the various seismic survey methods for the alluvium layer. The data discussed in Section 6.4.2.1 is used in the figures. The seismic velocity data from the downhole and SASW (DTN: MO0110SASWWHBS.000) surveys are plotted at the mid-depth of the velocity intervals. As Figure 6-9 indicates, the SASW survey lines were conducted at various locations in the surface facilities area. Not all of the surveys corresponded directly with a known boring. Hence, for shear-wave velocity profiles provided in BSC 2002a (Figures 54, 55, 57, 61,

63, 76, 82, 85, and 87), only a select number of SASW surveys corresponding to the lithology of known borings were used.

Data obtained from suspension logging surveys (source-to-receiver method) shown in Tables VII-2 and VII-3 from BSC (2002a) are also plotted in Figure 6-19. The data contains average seismic wave velocities calculated for each boring drilled in the alluvium.

Average seismic velocities at various depth intervals (5-15 feet, 15-30 feet, 30-60 feet, and 60-100 feet) were calculated from the downhole and SASW survey data and fitted to a linear best-fit relationship for the shear wave velocity shown in Figure 6-19. A detailed discussion is provided in Appendix A of this report.

The shear wave and compression wave velocities were also determined in BSC (2004a) for the alluvium separately for each side of the Exile Hill Fault Splay. The shear wave and compression wave velocities on the upthrown side of the Exile Hill Fault Splay are presented in Figure 6-21 and Figure 6-22, respectively. The shear wave and compression wave velocities determined for alluvium on the downthrown side of the Exile Hill Fault Splay are presented in Figure 6-23 and Figure 6-24, respectively.

Figure 6-25 presents a comparison of the shear wave velocity relationships estimated in Appendix A (simple averaging technique) to the more rigorous analysis performed in BSC 2004a. The two methods provide similar results.

Recommended seismic wave velocity values for alluvium are provided in Table 2-1.

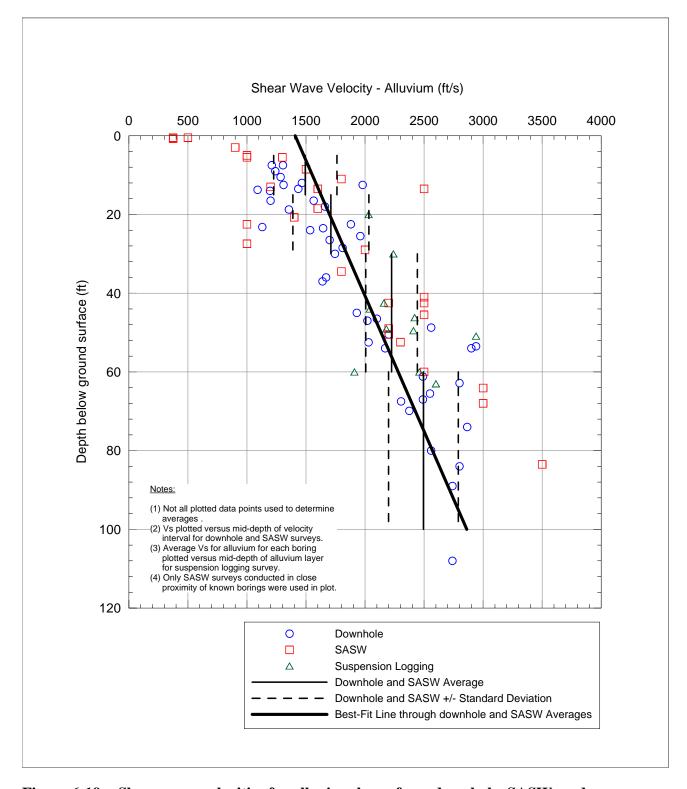


Figure 6-19. Shear-wave velocities for alluvium layer from downhole, SASW, and suspension surveys.

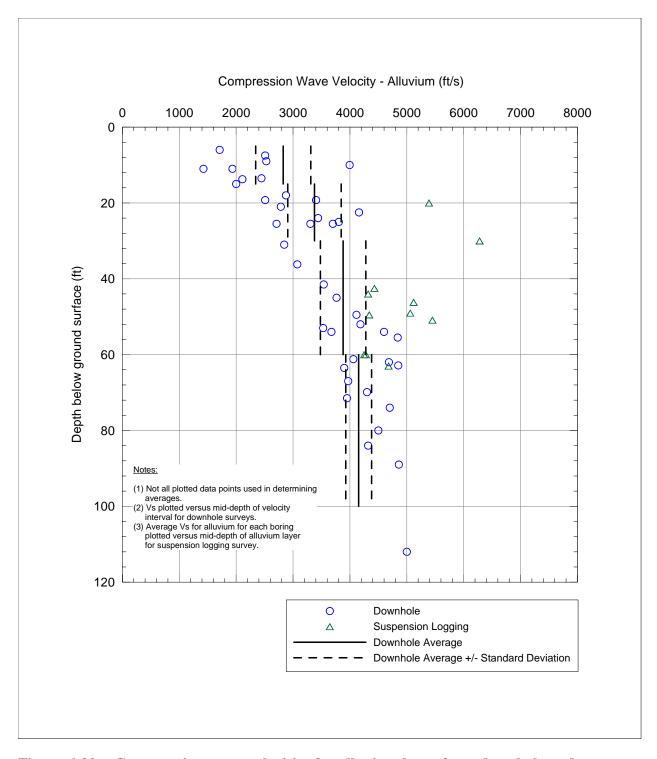
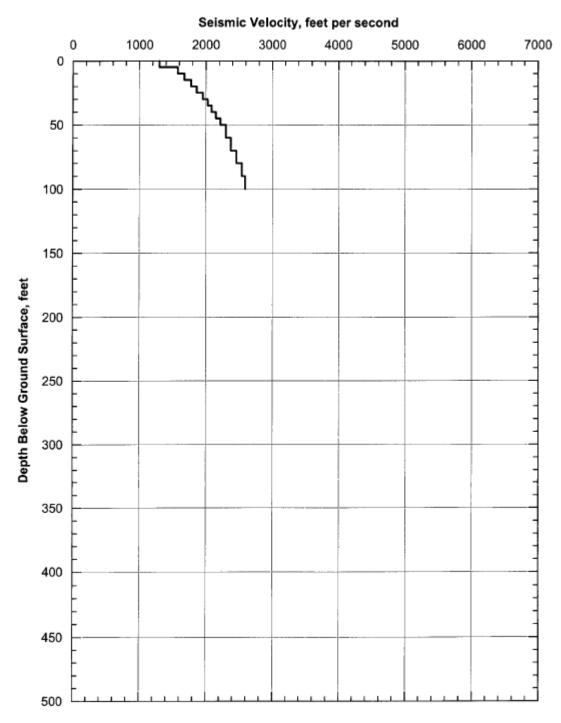


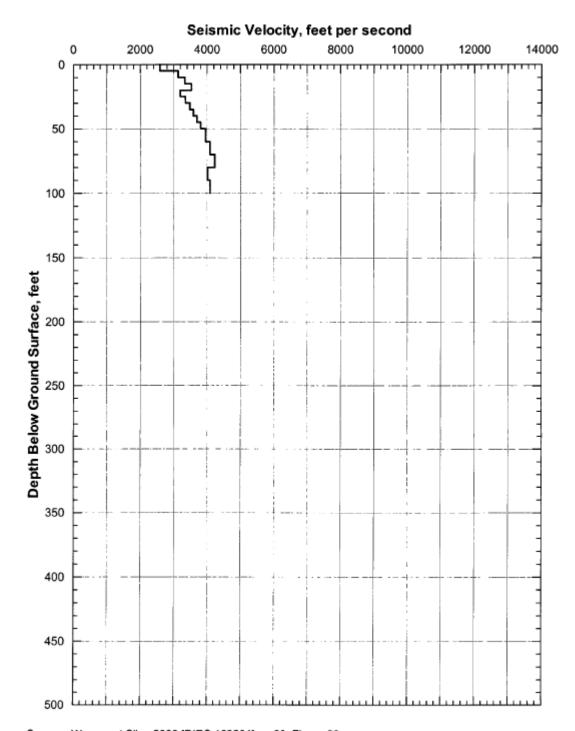
Figure 6-20. Compression-wave velocities for alluvium layer from downhole and suspension surveys.



Source: Wong and Silva 2003 [DIRS 163201], p. 88, Figure 27

Figure 6-21. Base Case shear wave velocity profile for alluvium in the surface facilities area—upthrown side

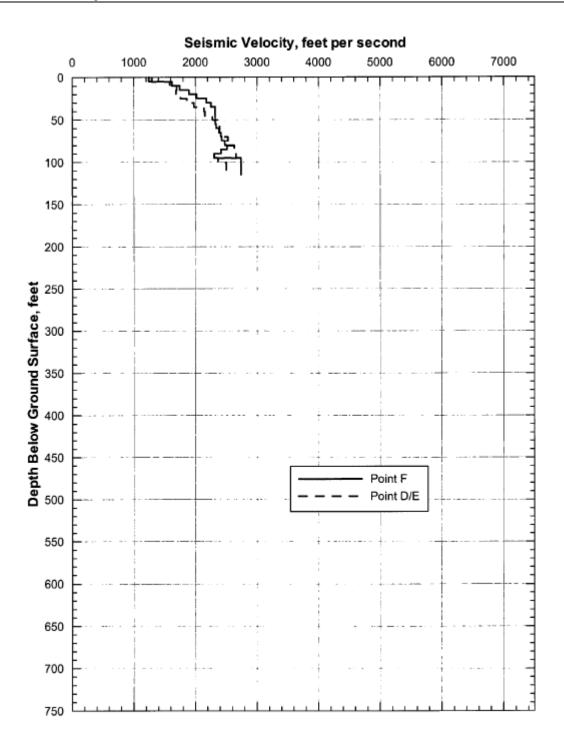
(Figure 6.2-121 from BSC 2004a)



Source: Wong and Silva 2003 [DIRS 163201], p. 90, Figure 29

Figure 6-22. Base Case compression wave velocity profile for alluvium in the surface facilities area—upthrown side

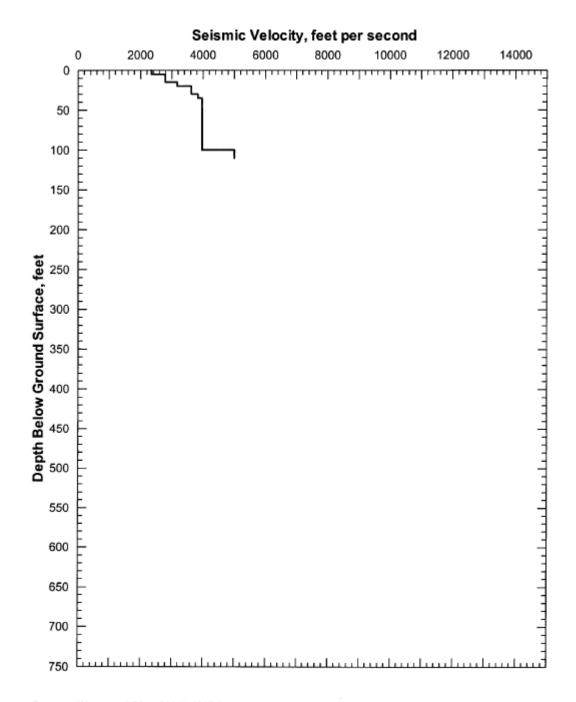
(Figure 6.2-122 from BSC 2004a)



Source: Wong and Silva 2004b [DIRS 170444], page 102

Figure 6-23. Base Case shear wave velocity profile for alluvium in the surface facilities area—downthrown side

(Figure 6.3-176 from BSC 2004a)



Source: Wong and Silva 2004b [DIRS 170444], page 108

Figure 6-24. Base Case compression wave velocity profile for alluvium in the surface facilities area—downthrown side

(Figure 6.3-179 from BSC 2004a)

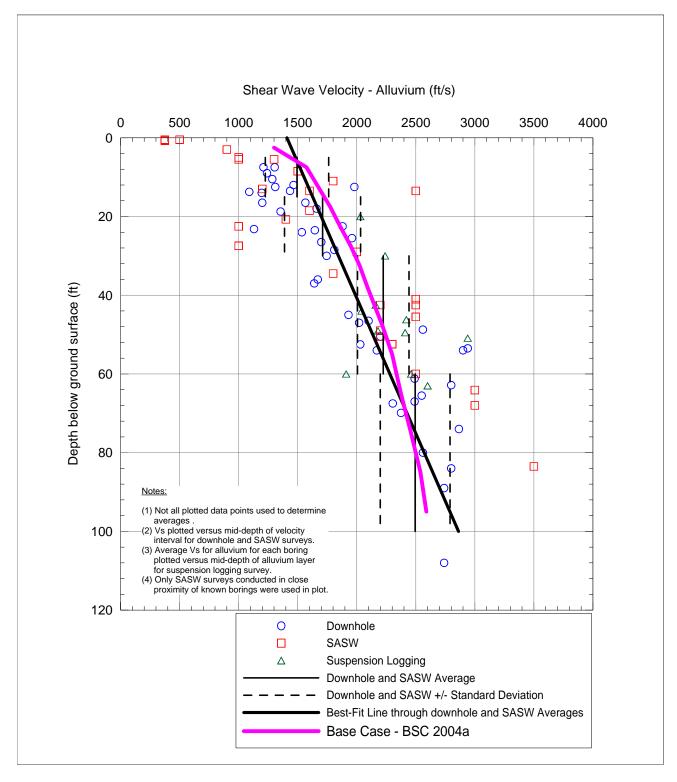


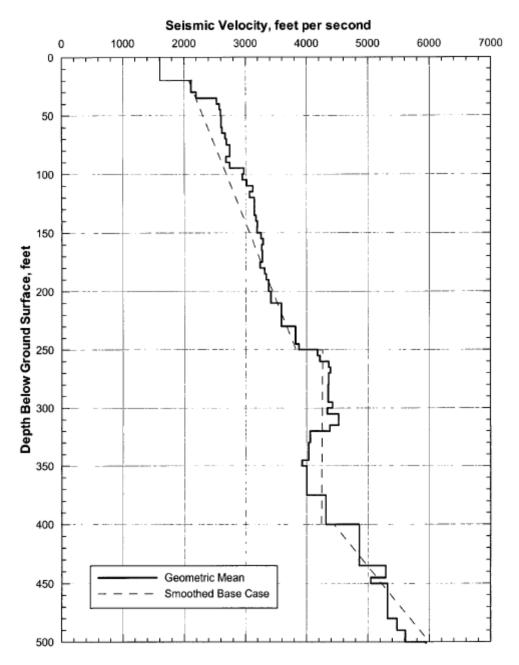
Figure 6-25. Comparison of simple averaging (Appendix A) and Base Case (2004a downthrown side) shear wave velocity profiles for alluvium in the surface facilities area

6.4.2.1.2 Bedrock

Shear and compression wave velocity averages were also computed for the bedrock (tuff) units identified in Section 6.1.4.3 using a methodology similar to that performed for the alluvium. This was performed for the downhole, suspension logging, and SASW line surveys. A detailed discussion is provided in Appendix A of this report.

The shear wave velocities determined in BSC (2004a) for tuff on the upthrown and downthrown sides of the Exile Hill Fault Splay are presented in Figure 6-26 and Figure 6-28, respectively. The compression wave velocities determined in BSC (2004a) for tuff on the upthrown and downthrown sides of the Exile Hill Fault Splay are presented in Figure 6-27 and Figure 6-29, respectively.

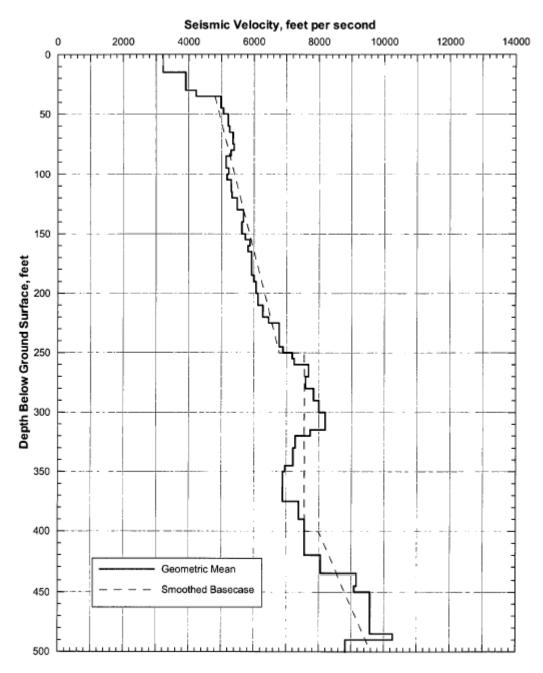
Recommended seismic wave velocity values for the bedrock are provided in Table 2-1.



Source: Wong and Silva 2003 [DIRS 163201], p. 91, Figure 30

Figure 6-26. Base Case shear wave velocity profile for tuff in the surface facilities areaupthrown block

(Figure 6.2-119 from BSC 2004a)



Source: Wong and Silva 2003 [DIRS 163201], p. 93, Figure 32

Figure 6-27. Base Case compression wave velocity profile for tuff in the surface facilities area–upthrown block

(Figure 6.2-120 from BSC 2004a)

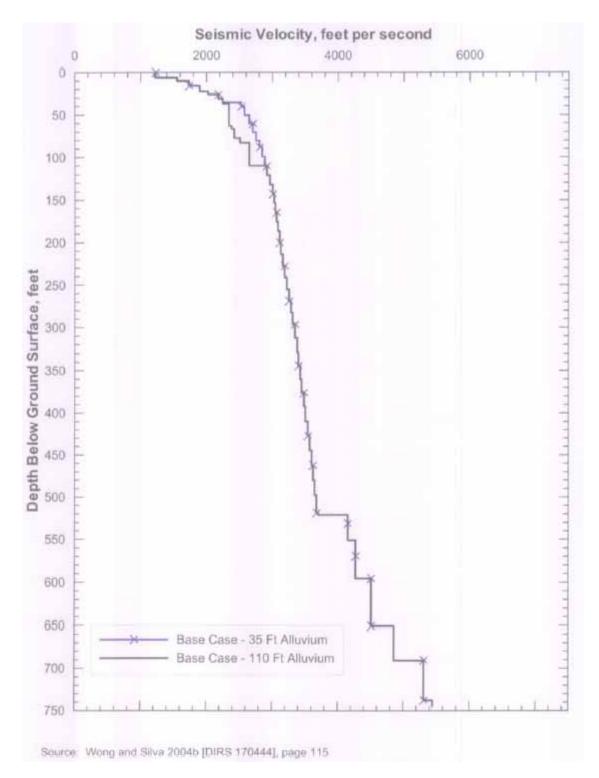


Figure 6-28. Base Case shear wave velocity profile for alluvium and tuff in the surface facilities area—downthrown block

(Figure 6.3-177 from BSC 2004a)

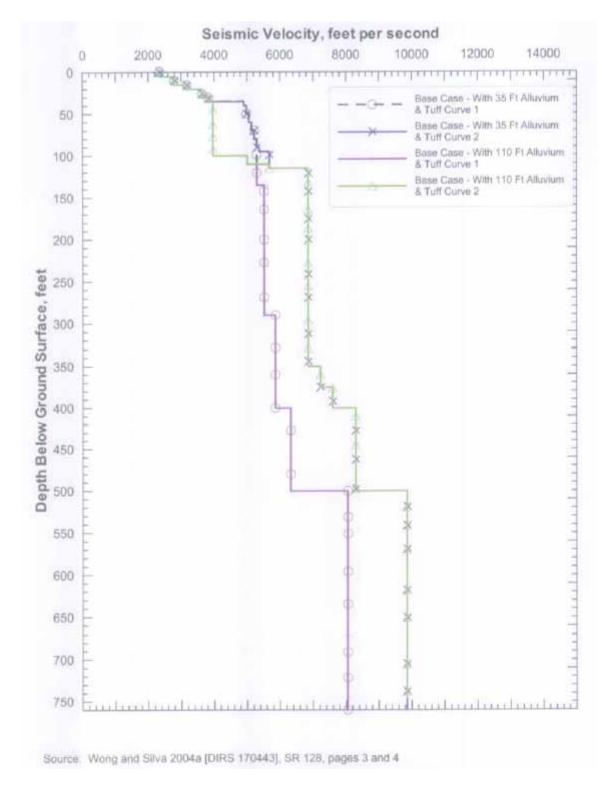


Figure 6-29. Base Case compression wave velocity profiles for alluvium and tuff in the surface facilities area—downthrown block

(Figure 6.3-180 from BSC 2004a)

6.4.2.1.3 Engineered Fill

No geophysical surveys could be performed on the proposed engineered fill. Hence, shear wave velocities measured from the dynamic laboratory testing (RCTS) were considered (Attachment XVII of BSC 2002a). Ten reconstituted specimens were tested for dynamic response characteristics by low-amplitude RC tests with confinement pressures ranging from 2 to 64 psi. The average total unit weight and percent of Modified Proctor of the specimens were 116 pcf and 93%, respectively. Four of these samples were tested in 2 stages to investigate the impact on the dynamic properties of increasing the water content of the granular fill after placement. For an estimated mean total stress of 8 psi, the average shear wave velocity measured from the dynamic testing was 700 ± 70 ft/s. This appears low for dense sand. As an upper bound limit, 1500 ft/s is recommended for the shear wave velocity.

From Section 6.4.2.2, using the range of Poisson's ratio for the engineered fill and equation (5), the range of compression wave velocity was estimated to be 1500 to 3700 ft/s.

6.4.2.1.4 Roller Compacted Soil Cement

Typical seismic velocities for roller-compacted soil cements are discussed in Section 6.4.3.

6.4.2.2 Poisson's Ratio

From the estimated average shear and compression wave velocities, representative Poisson's ratios for the alluvium and bedrock were determined using the following relationship:

$$\upsilon = \frac{2V_s^2 - V_p^2}{2V_s^2 - 2V_p^2},$$
 (Eq. 5)

where $V_p = compression$ wave velocity

 V_s = shear wave velocity v = Poisson's ratio

For the engineered fill, Bowles (1996) recommends a range of 0.3 to 0.4 for a dense cohesionless sand. The recommended values are summarized in Table 2-1.

6.4.2.3 Total Density

Table 12 of BSC (2002a), provides a statistical summary of density measurements by lithostratigraphic unit from the borehole wireline geophysical surveys (made in boreholes RF#16, #18, #20, #21, #22, #24, and #28). Figure 101 of BSC (2002a), shows the total densities measured versus depth. This information is summarized in Table 6-12 below.

able 6-12.	Mean values of soil density from borehole geophysical surveys				
	(adopted from Table 12 of BSC 2002a).				
	Unit	Mean	Density		

Unit	Mean (pcf)	Density
Existing Fill	115	
Alluvium, Qal	116 ⁽¹⁾	
Bedrock, Tmbt1	110	
Bedrock, Tpki	98	
Bedrock, Tpbt5	112	
Bedrock, Tpcrn	117	
Bedrock, Tpcpun	132	
Bedrock, Tpcpul	130	
Bedrock, Tpcpmn	145	
Bedrock, Tpcpll	136	
Bedrock, Tpcpln	132	

⁽¹⁾ Assumption 4 of Section 5 in BSC (2002a) was not considered in the values.

Densities of the bedrock were also measured in the dynamic laboratory tests. BSC (2002a) compares the mean values from the in-situ tests and dynamic laboratory tests, as well as values obtained from previous borehole samples in the area (Table 34 of BSC 2002a). Some variability exists between the different methods of measurement. Since the number of measurements obtained from dynamic tests was too small to provide reliable numbers compared to the in-situ tests, it was not considered. In accordance with Section I.3.1 of BSC (2002b), it is recommended that the lowest density value obtained for bedrock (approximately 100 pcf from the Tpki rock unit) be used for design as this provides the most conservative value for bearing capacity calculations.

6.4.2.4 Shear Moduli

Although dynamic shear moduli, G, values for the alluvium and bedrock were determined from the dynamic soil laboratory testing, the laboratory tests are primary performed to measure the modulus reduction and damping ratio curves. Factors that may not be representative in the laboratory samples such as aging and cementation may affect G measurements. Hence, the dynamic soil shear modulus for the alluvium and bedrock is calculated using the theory of elasticity. The following equation is used:

$$G_{\text{MAY}} = \rho V_s^2, \tag{Eq. 6}$$

where ρ = mass density, which is the unit weight of the soil divided by the acceleration of gravity

 V_s = average shear wave velocity estimated in Section 6.4.2.1

The upper and lower bound shear-wave velocity values are used to determine the maximum shear modulus as shown in Table 2-1

6.4.2.5 Modulus Degradation and Material Damping

6.4.2.5.1 Alluvium

Figure 6-30 (Figure 6.2-147 of BSC 2004a) presents the effects of the shear strain on the normalized shear modulus (G/G_{max}) and material damping ratio (D_{min}). These figures were derived from dynamic laboratory testing performed on reconstituted samples. The following testing–related factors, in order of largest influence, were considered in developing the design curves (BSC 2004a, Section 6.2.4.3):

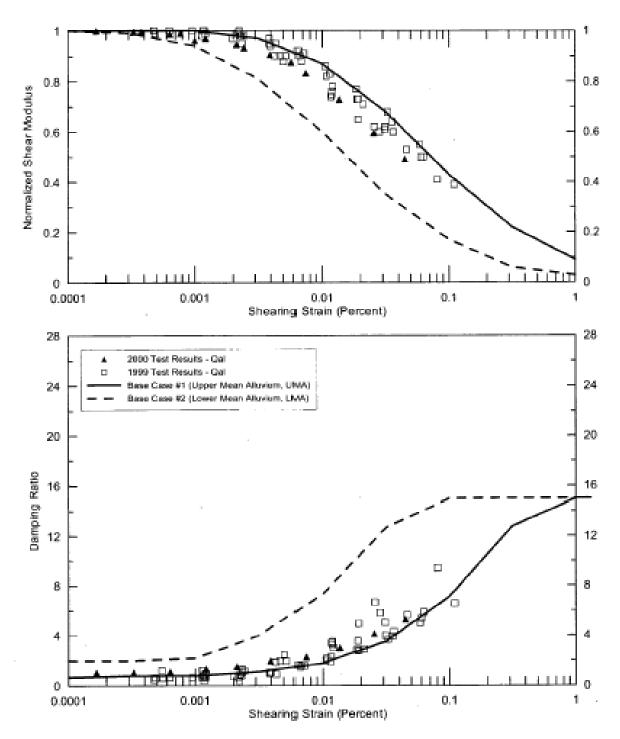
- 1. Destruction of cementation
- 2. Decrease in coefficient of uniformity
- 3. Variation in confining pressure in the field
- 4. Variation in density in the field
- 5. Increase in mean particle size
- 6. The test boundary size

Research available in the geotechnical literature provide guidance on the difference in dynamic behavior between reconstituted, scalped samples and field conditions relative to these factors. Two sets of mean curves are recommended to adequately incorporate uncertainty in the dynamic response of the alluvium as a result of the limited test data available.

The lower mean average (LMA, dashed) lines represent the case where natural cementation in the field breaks down under ground motion producing strains and follows the relationship presented in Figure 7.A-3 of EPRI (1993) for the middle of the range indicated for gravels.

The upper mean average (UMA, solid) lines were developed as an envelope of the data above 0.01% strain and a general fit to the data at small strains (<0.01%) and generally fit the relationships for cohesionless soils between depths of 250 ft and 500 ft as presented in Figure 7.A-18 of EPRI (1993).

Both the UMA and LMA curves were developed in a subjective manner. The UMA curves acknowledge the lack of experience in the geotechnical community with this type of soil and the limited test data. The six factors listed were considered in developing both sets of curves, but more directly with the LMA curves.



DTN: MO0403SDIAWHBC.003 [DIRS 170434] (CURVES), MO0203DHRSSWHB.001 [DIRS 158082] (DATA), MO9905LABDYNRS.000 [DIRS 103792] (DATA)

Figure 6-30. Normalized shear modulus and damping ratio for alluvium

(Figure 6.2-147 of BSC 2004a)

6.4.2.5.2 Bedrock

As stated in Section 6.3.2.3 of this report, dynamic tests were performed on a total of 18 tuff specimens divided into three groups based on their dry unit weight:

- Group 1: γ_d from 133 pcf to 147 pcf
- Group 2: γ_d from 117 pcf to 132 pcf
- Group 3: γ_d from 78 pcf to 94 pcf

Based on these tests it was determined that there were no correlations between the shear modulus and damping ratio degradation relationships and lithostratigraphic unit, degree of welding, or dry unit weight, and therefore, all dynamic test results on tuff could be grouped together (BSC 2004a).

However, test specimen size, amount of fractures, voids, and planes of weakness, do play a role in the strain behavior of the tuff. In consideration of this, similar to the alluvium, two sets of degradation relationship curves were also developed for the tuff–for upper mean tuff (UMT) and lower mean tuff (LMT). Figure 6-31 shows the shear modulus reduction and damping ratio curves for the tuff bedrock (taken from Figure 6.2-139 of BSC 2004a).

The UMT shear modulus reduction curve was developed considering the generalized shape of cohesionless soil curves from EPRI (1993, Figure 7.A-18) fitted through the most linear laboratory test data. For the damping ratio curve the corresponding EPRI (1993, Figure 7.A-19) curves were used, constrained to have 1) a small-stain damping ratio of 0.5% for consistency with the site attenuation (kappa of 0.0186 sec) used in the PSHA, and 2) a maximum of 15% in accordance with guidance from NUREG-0800, Section 3.7.2 (NRC 1989).

The LMT modulus reduction curve was developed considering the generalized shape of cohesionless soil curves from EPRI (1993, Figure 7.A-18) fitted through the middle of the laboratory test data, then adjusted downward by a factor of 4 based on the ratio of G_{max} in the field (based on Vs) to that in the laboratory to account for heterogeneity and fracturing in the field. The resulting curve corresponds to the 21 ft to 50 ft curve in EPRI (1993, Figure 7.A-18). For material damping the corresponding curve from Figure 7.A-19 of EPRI (1993) was used. As for the UMT curves, a small-strain material damping of 0.5% was used to constrain the curves at small strains.

Note that for both the UMT and LMT relationships developed; all values for strains exceeding 0.1% are extrapolated based solely on the shape of the EPRI (1993) curves.

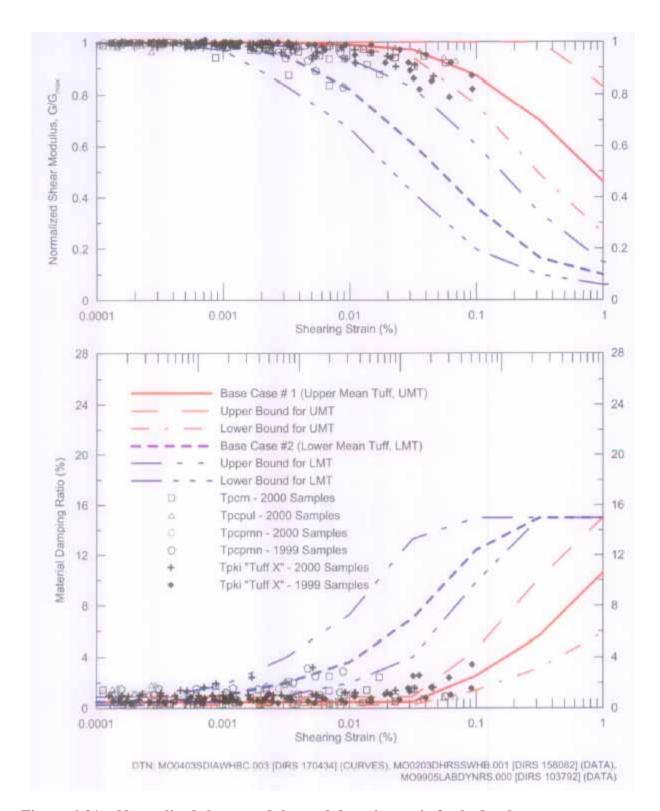


Figure 6-31. Normalized shear modulus and damping ratio for bedrock

(Figure 6.2-139 of BSC 2004a)

6.4.2.5.3 Engineered Fill

Figure 6-32 and Figure 6-33 show the composite shear modulus reduction and damping ratio curves for this material (taken from Figures 221 and 222 from BSC 2002a, respectively).

 G_{max} increases with increasing dry unit weight of the compacted material, and decreases with increasing water content for denser specimens. The soil behavior curve is similar to that for sandy material proposed by Seed et al. (1986). However, it exhibits a higher minimum damping ratio. In general, the values of G_{max} evaluated using the resonant column and torsional shear devices agreed within 10 percent. The values of D_{min} evaluated using the resonant column and torsional shear devices also agreed within 10 percent.

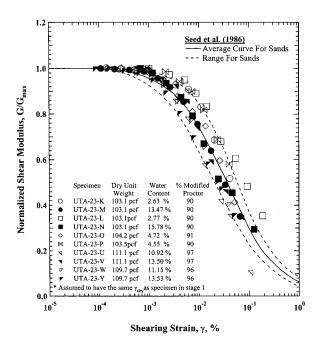


Figure 6-32. Normalized shear modulus for engineered fill from Fran Ridge Borrow Area (Figure 221 of BSC 2002a, DTN: MO0203DHRSSWHB.001)

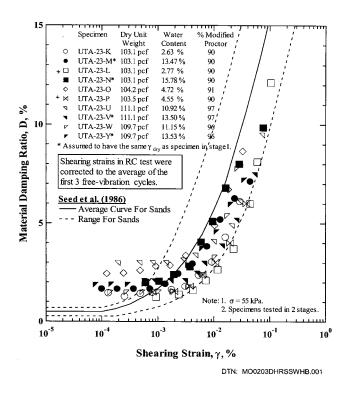


Figure 6-33. Material damping ratio for engineered fill from Fran Ridge Borrow Area (Figure 222 of BSC 2002a, DTN: MO0203DHRSSWHB.001)

6.4.3 Roller Compacted Soil Cement

6.4.3.1 Recommended Properties

The following recommendations are based on the review of properties of roller compacted soil cement (RCSC) and deep soil mixes that are summarized in Section 9 of BSC (2004b). These parameters are provided as a first estimate for dynamic evaluation of roller-compacted soil-cement should RCSC be considered for use at the site:

• Percent cement: 4% to12% by weight

• Unit weight: 130 pcf to 140 pcf

Poisson's ratio: 0.30
Shear-wave velocity, V_s:

Lower bound: 2000 ft/sAverage: 2500 ft/sUpper bound: 3000 ft/s

• Shear modulus at low strain:

Lower bound: 100 ksiAverage: 180 ksiUpper bound: 270 ksi

No information was found in the literature regarding shear modulus reduction curves specific to RCSC. However, the following presents a limited collection of shear modulus reduction curves for cement treated soils identified in the literature:

- 1. Dupas and Pecker (1979)–From cyclic triaxial tests performed on soil-cement samples. Soil was fine to medium grain sand with 5% cement by weight and compacted to 100% of the maximum density as determined by ASTM D 558. Curing time 180 days.
- 2. Wang (1986)—From triaxial and simple shear tests on artificially cemented sand. The material was a mixture of Monterey #0 and #20 sand with 2% cement and 74% relative density.
- 3. Kohata et al. (1997)–From cyclic triaxial tests performed on soil-cement samples cured for 28 days. Soil was fine-grained. Cement percentage unknown.
- 4. Sato et al. (1995)—From dynamic triaxial tests performed on sand-cement samples. Sand was fine to medium grained with100% less than 0.84 mm, D60 = 0.35 mm, D30 = 0.31 mm, and uniformity coefficient = 1.59. Cement percentage unknown.
- 5. McGinn and O'Rourke (2003)–From pressuremeter tests performed on stiff clays treated with 12% to 15% cement by weight using the deep soil mixing method.

Figure 6-34 presents the normalized shear modulus data from these sources. The EPRI (1993) curve for sand for depths of 0-20 ft is included for comparison.

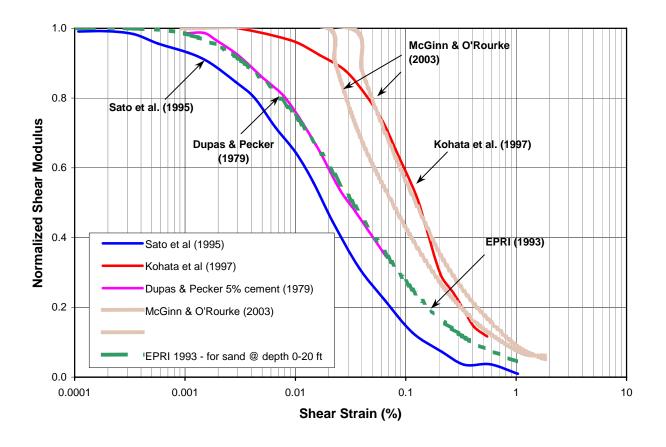


Figure 6-34. Normalized shear modulus reduction curves for cement treated soils

Note that the curves by Kohata et al. (1997), and McGinn and O'Rourke (2003) were computed from fine-grained soils treated with cement, while all other curves were obtained from sand-cement mixtures.

No information was found in the literature regarding damping ratio reduction curves specific to RCSC. Only Dupas and Pecker (1979), Wang (1986) and Kohata et al. (1997) present damping ratio degradation data for soil cement mixes. Figure 6-35 presents the damping ratio degradation data from these sources. The EPRI (1993) curve for sand for depths of 0-20 ft is included for comparison.

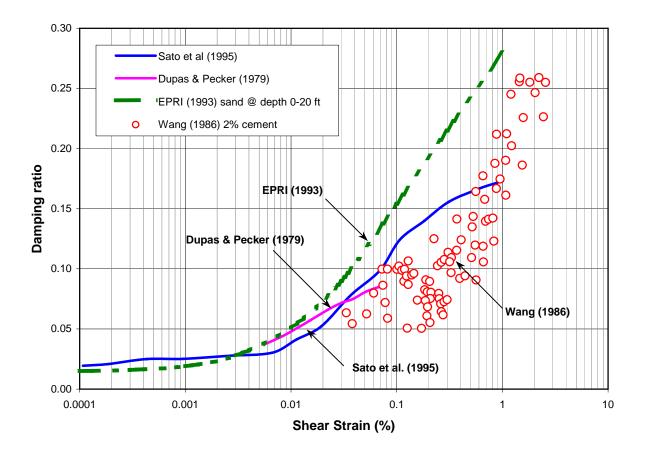


Figure 6-35. Damping ratio degradation curves for cement treated soils

Note: the curves presented in Figure 6-35 were obtained from sand-cement mixtures.

6.4.3.2 Limitations of Use

Since very little data are available on non-linear properties they should be reviewed by the peer review to develop recommendations for application for design evaluation. Prior to final design a comprehensive laboratory and field testing program should be created to determine all of the above listed properties. In addition, non-linear soil-structure interaction analyses should be performed to optimize the depth and extent of soil treatment. Depending on the extent of the treatment areas the above inputs may be used either in the free-field or in SSI models.

7 RESULTS AND CONCLUSIONS

7.1 Engineering Design Parameters

Analyses outputs obtained from the calculations prepared for this report are reasonable compared to the input parameters used. The results are considered suitable for use in preliminary design.

Figures and tables containing supporting design parameters are located at the end of this section. Table 2-2 shows key results of the analyses contained herein.

7.1.1 Material Properties

Table 2-1 summarizes the recommended static and dynamic soil properties discussed in Sections 6.4 and 7 for design.

7.1.2 Foundation Pressures

The recommended foundation pressures of the soil at the surface facilities area were determined for various conditions using conventional geotechnical bearing capacity theory. Design charts are provided for allowable foundation pressures for different footing geometries resting on alluvium. Design settlements of 1-inch, or less, and $\frac{1}{2}$ -inch, or less, were used in the analyses. The strength parameters used for the alluvium are discussed in Section 6.4. A minimum factor of safety of 3.0 against bearing capacity failure was used (i.e., $q_{\text{allowable}} = q_{\text{ultimate}} / 3.0$).

Figure 7-2 and Figure 7-3 show the recommended bearing pressures on square and strip footings for 2-foot and 6-foot embedment depths with widths ranging from 2 to 30 feet, for 1-inch and ½-inch design settlement, respectively. Figure 7-4 and Figure 7-5 show the variation of immediate settlement with bearing pressure for square and strip footings of 5-foot, 10-foot, and 20-foot widths, for embedment depths of 2 feet and 6 feet, respectively. Figure 7-6 shows the variation of long-term settlement with foundation width. Note that in Figure 7-6, strip and square footings provide nearly identical solutions. Details of the foundation analyses are documented in Appendix B.

7.1.3 Settlement

7.1.3.1 Short-Term Settlement

Settlement of foundations is a function of the footing size, average footing load, the depth of footing embedment, and characteristics of the soil material type. Two methods were considered to estimate the settlements for the surface facilities area: (1) Burland and Burbidge (Terzaghi et al., 1996), which uses an average N₆₀ blow count value (correlated from a relationship to the relative density), and (2) Schmertmann (Terzaghi et al., 1996), which uses Young's modulus (correlated from measured shear-wave velocities). Immediate settlements induced under different foundation pressures are presented in Figure 7-4 and Figure 7-5 for a variety of conditions. A detailed description of the analyses is provided in Appendix B.

7.1.3.2 Long-Term Settlement

Over time, some additional settlement will occur due to long-term, secondary settlement effects. This settlement is in addition to that estimated in Section 7.1.3.1. The long-term or secondary settlements for the surface facilities area were computed based on the method developed by Burland and Burbidge (Terzaghi et al., 1996). The settlement was determined to be less than ½

inch. Long-term settlements are presented for different footing widths (square and strip footings) and different depths of foundation embedment in Figure 7-6. A detailed description of the analysis is provided in Appendix B.

7.1.3.3 Elastic Settlement

Elastic settlements were computed for a large mat foundation based on a uniform vertical stress distribution, representative average shear wave velocities (see Section 6.4.2.1.1), and Young's modulus (derived from modulus degradation curves). The dimensions of $(450^{\circ} \times 500^{\circ})$ were used in the analysis. Settlements under the corner and center of the mat were determined for loads of 3, 5, and 7 ksf. A detailed description of the analysis is provided in Appendix B. Results of the analysis are shown in Table 2-2.

7.1.3.4 Differential Settlement

In accordance with Peck, et al. (1974) Chapter 14, differential settlement between adjacent footings can be ³/₄ of the maximum estimated value.

The following are allowable angular distortions, δ/L (allowable differential settlement over a given distance), for buildings (Fig. 5.59 of Fang 1991):

δ/L	Building type
1/500	Buildings where cracking is not permissible; Rigid circular mat or ring footing for tall and slender rigid structures

where δ = allowable differential settlement and L = spacing distance

7.1.3.5 Seismically-Induced Settlement

Seismically-induced settlement is not considered to be a significant issue due to the dry and dense nature of the soils encountered at the YMP site. In addition, cementation of the native alluvium will also reduce the potential for dynamic settlement.

7.1.4 Coefficient of Subgrade Reaction and Equivalent Soil Springs

All shallow footings and mat foundations will be supported by the alluvium. For the design of large footings and mats it is typical to represent the soil with equivalent springs. The vertical coefficient of subgrade reaction for the alluvium is estimated based on Terzaghi (1955). For dry dense sand, the recommended value for a one-foot by one-foot plate, $k_{\rm s1}$, is 600 -2000 kcf (tons/cubic foot). For the dense gravelly alluvium present at the site, it is recommended that a best estimate value of 1000 kcf (580 pci) be used. For the anticipated dense engineered fill, it is recommended that a best estimate value of 600 kcf (350 pci) be used.

These values must be reduced for large loaded sizes in accordance with the following relationship:

$$k_s = k_{s1} \left(\frac{B+1}{2B}\right)^2 \tag{Eq. 8}$$

where B is the least footing dimension and k_s is the coefficient of subgrade reaction for the footing or mat. For large footings or mats k_s will approach $\frac{k_{s1}}{4}$. Therefore, for preliminary design, it is recommended to use 155 to 520 kcf (90–300 pci) for alluvium and 75 to 250 kcf (45–145 pci) for engineered fill. It is common practice to double the static load for dynamic load cases.

Figure 9 of USN (1986) was used to estimate the horizontal coefficient of subgrade reaction by correlations with relative density. For the very dense alluvium material, the values were estimated to be 104–120 kcf (60–70 pci). For the dense engineered fill, the values were estimated to be 60–96 kcf (35–55 pci). Results are summarized in Table 2-1.

7.1.5 Lateral Earth Pressures

Currently a 55-foot below-grade wall is planned for construction of a pool for the wet-process building. Lateral earth pressures were determined to estimate the loads that will act on subgrade walls. Both static and seismic conditions for yielding and non-yielding walls were considered in the analyses, including effects from compaction-induced earth pressures and static surcharge loads. Live loads and dynamic surcharge loads were not considered in the analyses. No factor of safety was applied to the calculated earth pressures. The calculations were performed using the measured properties of the alluvium (see Section 6.4.1). A coefficient of horizontal acceleration of 1g was used in the seismic analysis. The results from the seismic analysis may be scaled by any selected peak ground acceleration value.

A schematic summary of the results for yielding and non-yielding walls is shown in Figure 7-7 through Figure 7-15. A detailed description of the analyses is provided in Appendix C.

7.1.5.1 Lateral Earth Pressures for Temporary Shoring

For a braced and shored excavation the lateral pressures can be estimated using a uniform pressure of 17H psf, where H is the height of the wall. Details of the supporting analysis are provided in Appendix C.

7.1.5.2 Surcharge and Compaction Loads

Surcharge loading due to nearby point, line, uniform surcharge, strip, and footing loads are presented in Figure 7-8 and Figure 7-9. These relationships are based on those presented in USN (1986).

Compaction stresses imposed on the wall as a result of compaction are addressed in Appendix C. The calculated compaction stresses due to various compaction devices are presented Figure 7-11 through Figure 7-15. In accordance to Section 4.2.2.3.5 of BSC (2005), a minimum surcharge load of 300 psf shall be used (see Section 2.2.1).

7.1.6 Resistance to Lateral Loads

Resistance to lateral loads acting on footings, mats, and subgrade walls can be developed from passive resistance of the soil and from friction acting between the structural base and the subgrade soils.

Passive resistance can be determined assuming an equivalent fluid unit weight of 515 pcf acting on the sides of the foundations. When applying passive resistance for external footings or building walls, the effective depth of embedment should be reduced by one foot. Supporting calculations are provided in Appendix C.

The ultimate coefficients of sliding friction for mat and footing foundations underlain by alluvium and engineered fill are estimated to be 0.81 and 0.90, respectively (see Appendix C).

7.1.7 Slope Considerations

Figure 7-1 is a cross-section through the site and the lower end of Exile Hill (located on the far right hand side of the figure), which is located west of the planned surface facilities. Note that the column of geologic labels on left-hand-side of Figure 7-1 has been shifted upward and do not label the corresponding geologic strata. However, that information is not critical to the intent of this figure, as only the right-hand-side of figure is needed to illustrate the relatively gentle slope of Exile Hill directly adjacent to the surface facilities to be located east of RF#28 at the toe of the slope. As the cross-section illustrates, the surface facilities site is on relatively level ground. Exile Hill immediately west of the surface facilities site slopes at about 2.5H:1V (horizontal: vertical) in its upper portion and flattens to about 6H:1V near its base adjacent to the surface facilities site. The steeper, upper portions of Exile Hill, west of the surface facilities, are composed of rock at the surface. The alluvium and colluvium constitute the flatter lower portion. Due the flatness of the adjacent alluvial/colluvial portions and the presence of rock in the upper portions, slope stability of Exile Hill is not anticipated to be a significant concern for Additional reconnaissance of the slope as recommended in the surface facilities site. Section 7.3.1 will determine if a detailed stability analysis of Exile Hill is necessary.

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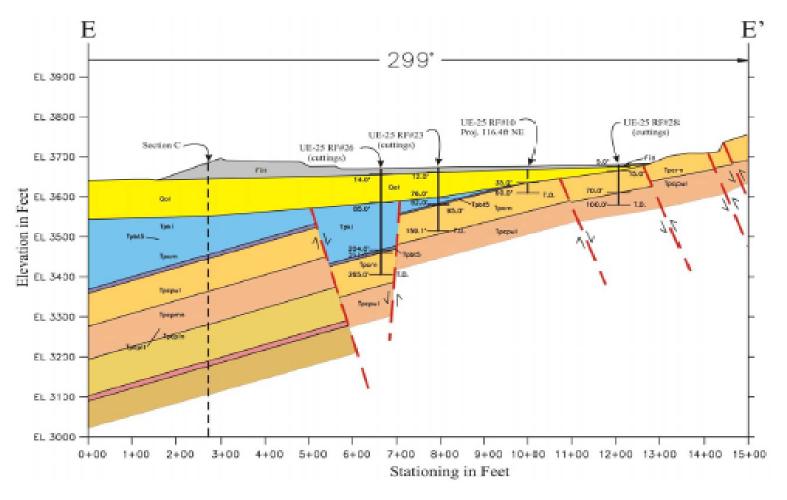


Figure 7-1. WHB Area Geologic Cross Section E-E', looking South

(see Figure 6-7 for location of cross section)

(Excerpted from BSC 2002a. Geotechnical Data for a Potential Waste Handling and for Ground Motion Analyses for the Yucca Mountain Site Characterization Project)

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Temporary cuts in the alluvium should be no steeper than 1.5H:1V. Permanent fill slopes should be no steeper than 2H:1V. Permanent cut and fill slopes should be provided with erosion protection by placement of at least 3 inches of coarse aggregate shouldering material.

7.1.8 Pavements

The designs of all pavement sections including the gravel construction phase pavements and any heavy transport routes should be developed using an approved pavement analysis method.

CBR was not directly measured on the site materials. Preliminary designs may be based on a CBR of 20 percent for the alluvium and engineered fill. This is the most conservative value recommended for the gravelly soils in Table 1, page 7.2-39, of USN (1986).

7.1.9 Percolation Rates

Percolation rates may be required for design of septic drain fields. Although no direct infiltration tests or other measurements have been made, an estimate of the permeability of the alluvial soils can be made based on the fines percentage (see Section 6.4.1.1.1) and the relationship presented in Figure 8-5 of USN (1986) for the effect of fines on permeability. Based on this figure, the permeability of the alluvium can be estimated to be between 5×10^{-5} fpm and 5×10^{-4} fpm. These numbers can be refined by performing field percolation tests (ASTM D 5126) or laboratory constant-head permeability tests on reconstituted samples (ASTM D 2434).

7.1.10 2000 International Building Code (IBC) Soil Type

Using the averaged shear wave velocities developed in Appendix A and reported in Table 2-1, a Soil Profile Type from the 2000 International Building Code (IBC) was selected to characterize the dynamic soil properties of the surface facilities area (ICC 2000, Table 1615.1.1).

Table 2-1 summarizes the soil profile type determined for the soil units at the site.

7.1.11 Frost Penetration

Figure 7-16 (Figure III-1 of BSC 2002b) below shows the potential frost penetration for the western United States. Based on this map, the potential for frost penetration at the YMP site is approximately 10 inches. Use 10 inches for design purposes.

7.1.12 Liquefaction Potential

As discussed in Section 6.1.4.4, groundwater is located 1270 feet below the ground surface. Therefore, there is no potential for liquefaction to occur beneath the planned structures at the site.

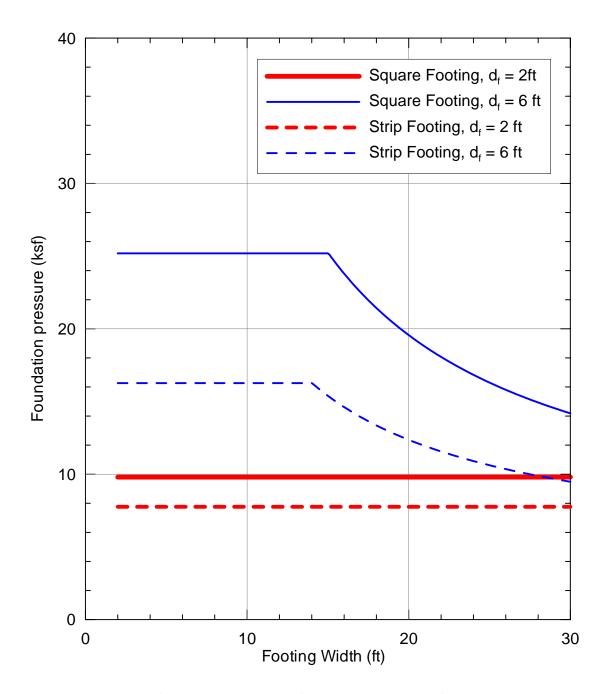


Figure 7-2. Allowable foundation pressure for square and strip footings on alluvium vs. foundation width and foundation embedment (1-inch design settlement).

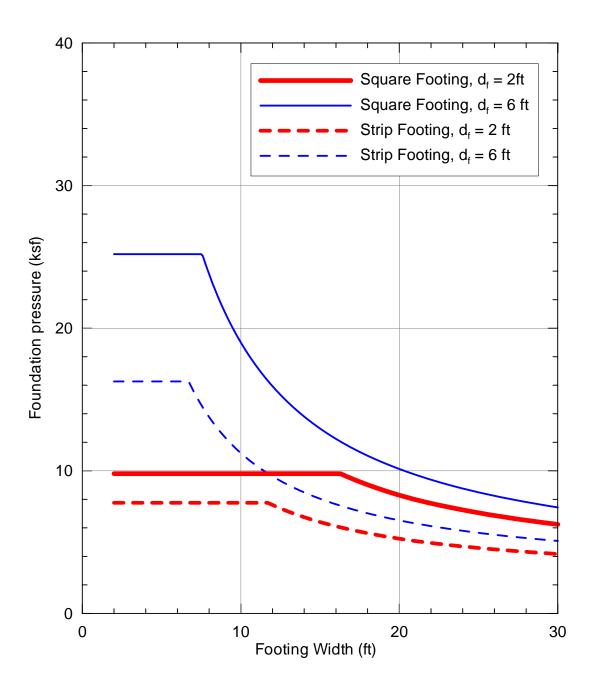


Figure 7-3. Allowable foundation pressure for square and strip footings on alluvium vs. foundation width and foundation embedment (½-inch design settlement).

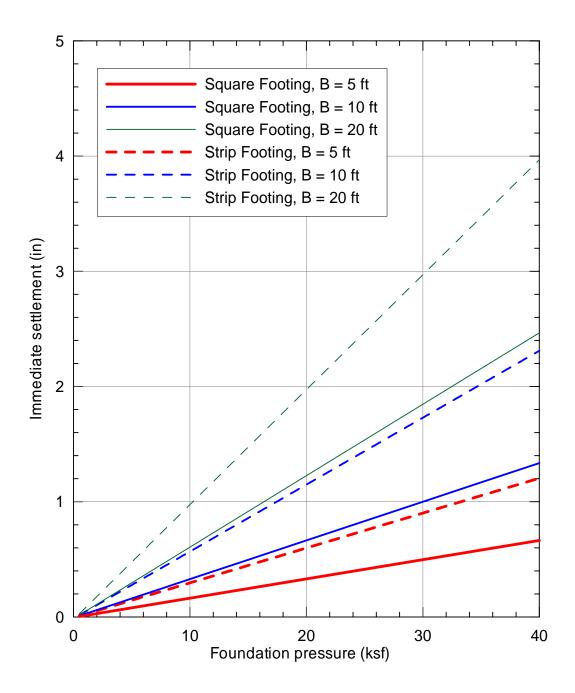


Figure 7-4. Immediate settlements for different widths of square and strip footings on alluvium vs. foundation pressure $(d_f=2\ ft)$

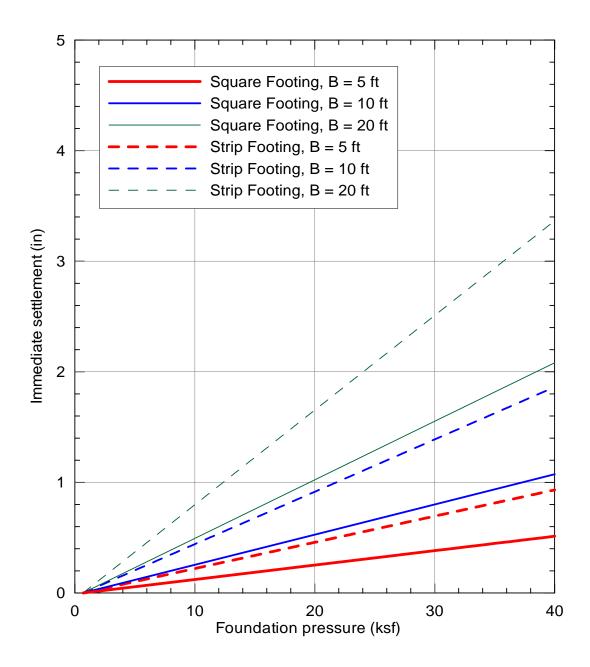


Figure 7-5. Immediate settlements for different widths of square and strip footings on alluvium vs. foundation pressure ($d_f = 6 \text{ ft}$).

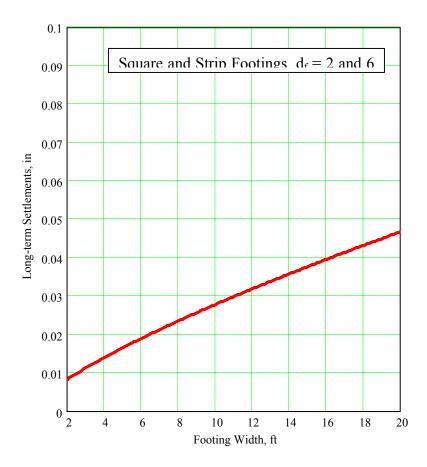
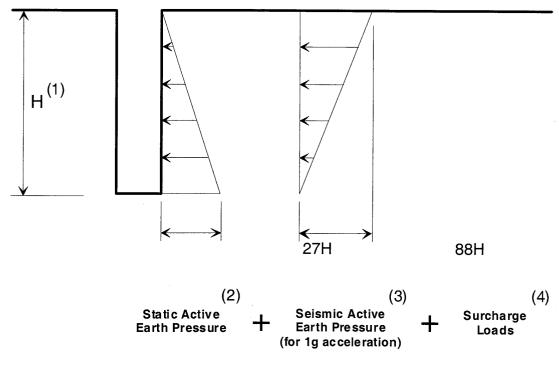


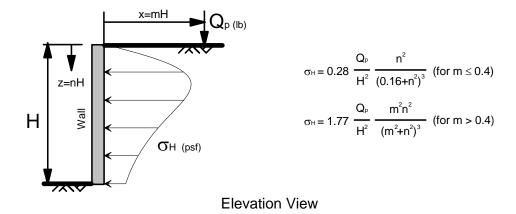
Figure 7-6. Long-term settlements for square and strip footings and different depths of foundation embedment.

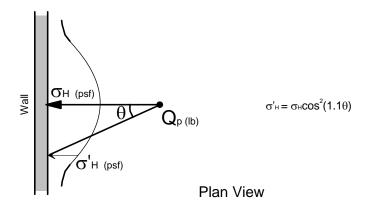


Notes:

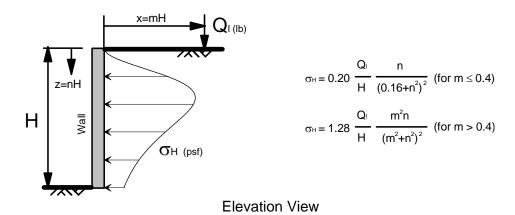
- (1) Height of wall, H, is presented in feet.
- (2) Static active earth pressure for alluvium: $K_A = 0.23$, $\gamma = 117$ pcf.
- (3) Seismic active earth pressure for alluvium based on Seed and Whitman (1970) simplified method where $K_h = 1g$ (to be scaled by actual peak ground acceleration, PGA).
 - (4) Surcharge loads are shown in next figure.
 - (5) Pressures are presented in psf.

Figure 7-7. Lateral earth pressures for yielding walls



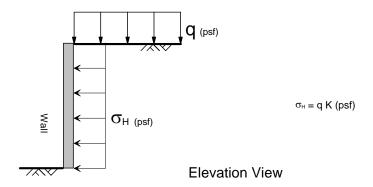


Lateral Pressure due to Point Load

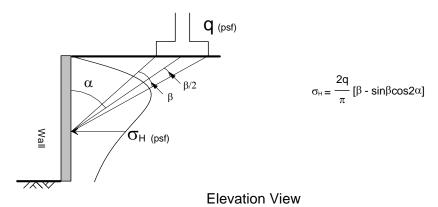


Lateral Pressure due to Line Load

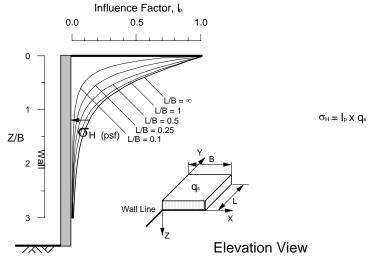
Figure 7-8. Surcharge loading for yielding walls (not drawn to scale, USN 1986)



Lateral Pressure due to Uniform Surcharge



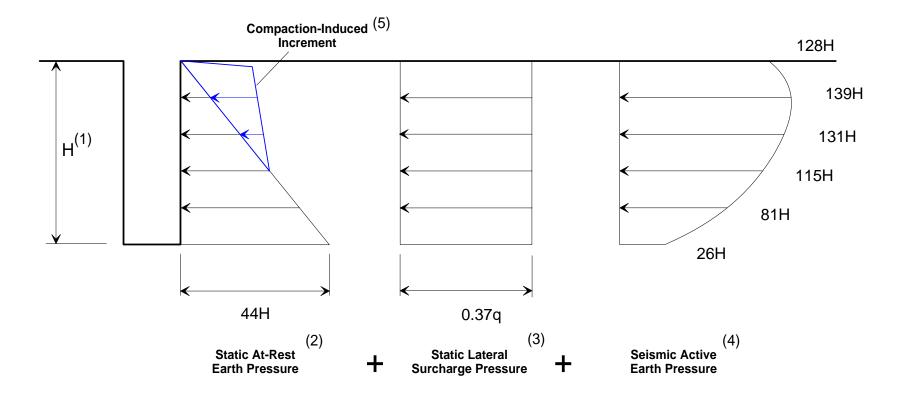
Lateral Pressure due to Strip Load



Lateral Pressure due to Footing

Figure 7-9. Surcharge loading for yielding walls, continued (not drawn to scale, USN 1986)

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Notes:

- (1) Height of wall, H, is presented in feet.
- (2) Static at-rest earth pressures for alluvium: $K_0 = 0.37$, $\gamma = 117$ pcf.
- (3) Static lateral surcharge pressure based on K_oq, where q is surcharge to be determined.
- (4) Seismic active earth pressure based on methods from ASCE 4-98 (2000), where $k_h = 1g$ (to be scaled by actual peak ground acceleration, PGA);

Does not include dynamic contribution due to surcharge load

- (5) Compaction-induced pressure increments for specific compaction equipment provided in the next following figures.
 - (6) Pressures are presented in psf.

Figure 7-10. Lateral earth pressures for non-yielding walls.

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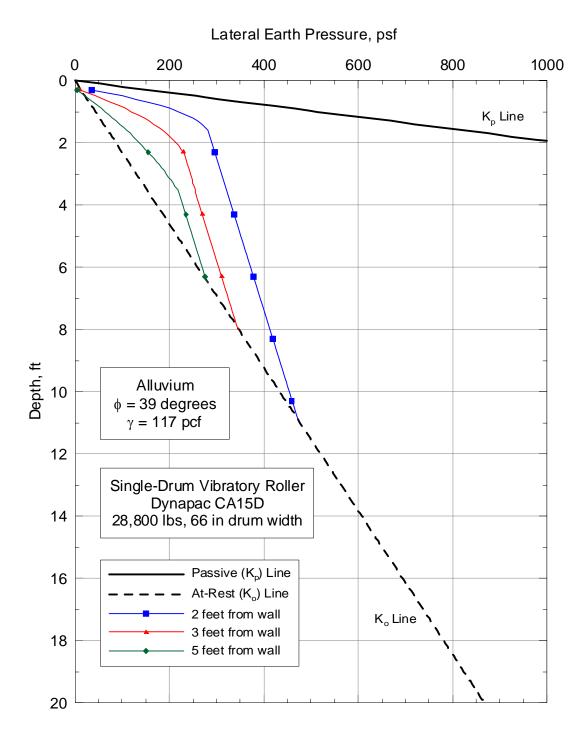


Figure 7-11. Compactor-induced pressures from roller compactor (Compactor model: Dynapac CA15D)

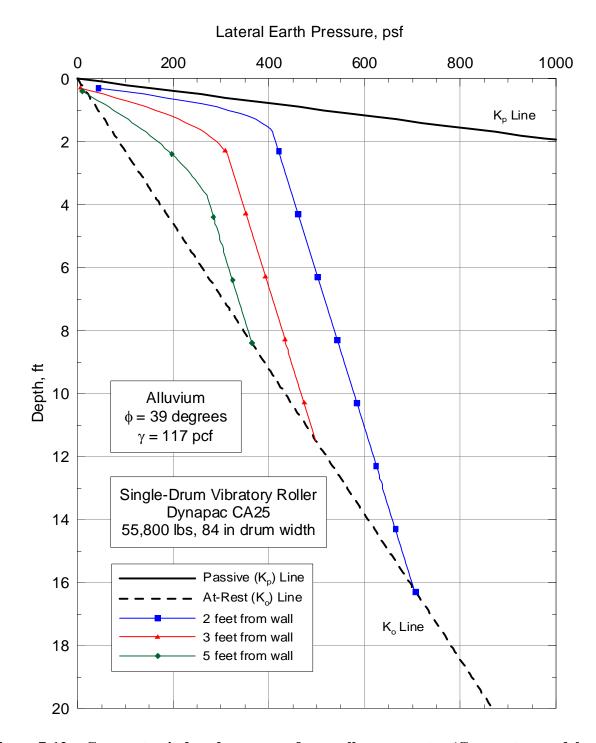


Figure 7-12. Compactor-induced pressures from roller compactor (Compactor model: Dynapac CA25)

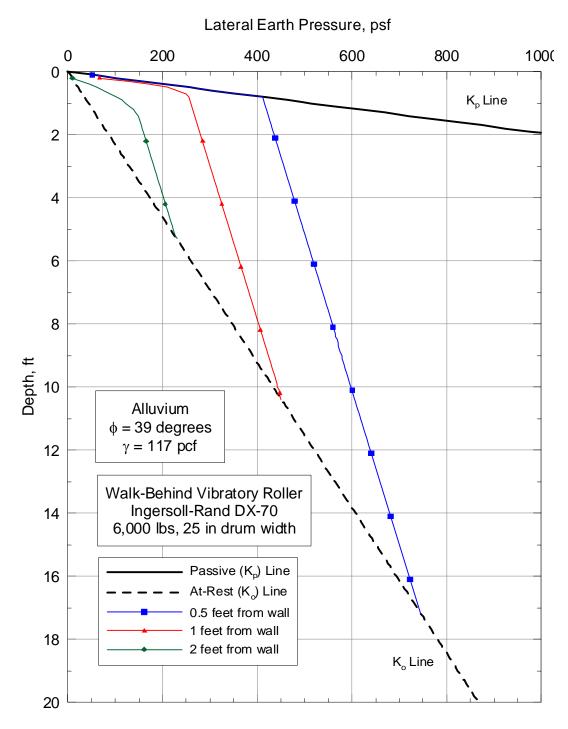


Figure 7-13. Compactor-induced pressures from roller compactor (Ingersoll-Rand DX-70).

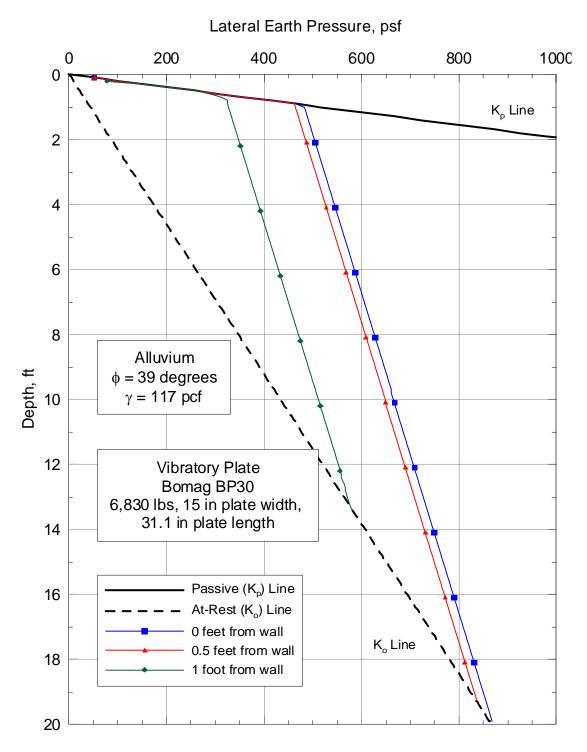


Figure 7-14. Compactor-induced pressures from plate compactor (Bomag BP30).

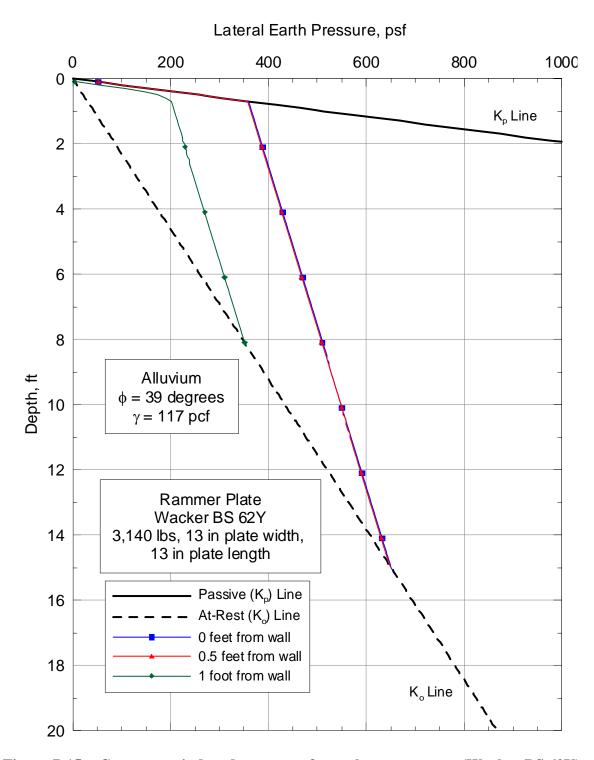


Figure 7-15. Compactor-induced pressures from plate compactor (Wacker BS 62Y).

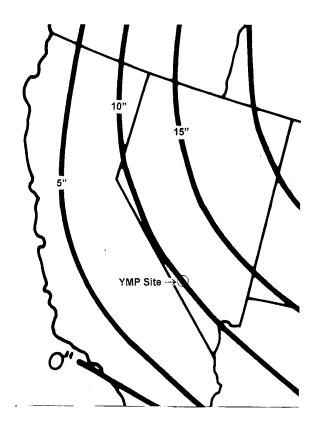


Figure 7-16. Extreme frost penetration (inches) at the North Portal Area (Figure III-1 BSC 2002b).

7.2 CONSTRUCTION CONSIDERATIONS

7.2.1 Stripping and Site Preparation

Portions of the site are currently covered with 5 to 22.4 feet of uncontrolled sand and gravel fill. All fill in the building areas should be removed down to the top of native alluvium. Any preexisting organic materials and roots, if any, encountered at the top of the native alluvium should be stripped at each of the structure sites. It is expected that no more than 6 inches of stripping of the original native surface would be needed to remove the organic materials and roots. In areas with preexisting heavy sagebrush growth, additional stripping may be required to remove the deep roots.

The excavated areas should extend outside the footing line for a distance equal to at least 1/2 the depth of the excavation up to a maximum of 5 feet outside the footing line.

The top 12 inches of the exposed subgrade surface should then be compacted to an in-place density of at least 95 percent of the maximum laboratory dry density as determined by ASTM D 1557.

The structural fill may consist of the excavated soil, or imported fill. Imported structural fill should consist of 5/8-inch minus crushed base course or 2-inch minus pit run gravel with less than 5 percent fines (minus U.S. No. 200 sieve size). All structural fill beneath or around structures should be compacted to an in-place density of at least 95 percent of the maximum laboratory dry density as determined by ASTM D 1557.

7.2.2 Foundations

All foundation should be buried a minimum of 2 feet below the ground surface.

7.2.3 Excavation, Backfill and Temporary Shoring

We recommend that all excavations be made as open excavations, with side slopes no steeper than 1.5 Horizontal to 1 Vertical (1.5H: IV). However, recognizing that some elements of certain structures may be as deep as 50 feet or more below the existing surface elevation, a combination of open cut and shoring may be necessary for those particular features. Temporary shoring to support these excavations may be designed based on the soil properties indicated in Table 2-1.

Consistent with conventional practice, actual temporary excavation slopes should be made the responsibility of the construction contractor. The construction contractor is able to observe the nature and conditions of the subsurface encountered and has the responsibility for methods, sequence, and schedule of construction. If instability is detected, slopes should be flattened or shored. All temporary excavation slopes should be accomplished in accordance with all local, state, and federal safety regulations. Excavation slopes and shoring may be designed using the soil properties shown in Table 2-1. Shoring systems, if used, should be monitored for vertical

and lateral movement during construction to confirm that movements are contained within allowable limits.

The granular soils observed in the explorations can be excavated using conventional equipment such as scrapers or rubber-tired or tracked hydraulic backhoes. Excavation in most of the site soils is not expected to require any unusual equipment or procedures. Any cobbles observed in the excavations should be removed from any excavated soils that will be used as backfill. No cemented layers were identified that would require special construction equipment or techniques.

7.2.4 Excavations for Underground Utilities

Backfill above and around underground utilities should be compacted to an in-place density of at least 95 percent of the maximum laboratory dry density as determined by ASTM D 1557. Moisture content of backfill materials should be within \pm 3 percent of optimum.

As an alternative to conventional trench backfilling, encasement of the conduit in controlled density fill (CDF) may be used. CDF used for pipe bedding or backfill should have a 28-day compressive strength between 50 and 200 psi.

Consistent with conventional practice, actual temporary excavation slopes for utility trenches should be made the responsibility of the construction contractor. The construction contractor is able to observe the nature and conditions of the subsurface materials encountered and has the responsibility for methods, sequence, and schedule of construction. If instability is detected, slopes should be flattened or shored. All temporary excavation slopes should be accomplished in accordance with local, state, and federal safety regulations.

7.2.5 Temporary and Permanent Slopes

Temporary cut slopes should be constructed with slopes no steeper than 1.5H: IV. Fill slopes should be no steeper than 2H: 1V. This recommendation is in conformance with the Project Design Documents (see BSC 2005, Section 4.2.1.2.7).

Permanent cut and fill slopes should be provided with erosion protection by placement of at least 3 inches of coarse concrete aggregate.

7.2.6 Compaction

All foundation stabilization, structural fill, utility bedding, and foundation and trench backfill materials should be compacted to an in-place density of at least 95 percent of the maximum laboratory dry density as determined by ASTM D 1557. Moisture content should be controlled to be within ± 3 percent of optimum.

In general, the thickness of backfill layers before compaction should not exceed 12 inches for heavy compactors and 8 inches for hand-operated mechanical compactors.

7.2.7 Suitability of On-site Materials

7.2.7.1 Structural Backfill

Based on field descriptions and laboratory testing of the alluvial materials encountered in the test pits performed at the surface facilities and the Fran Ridge borrow area (Sections 6.2.4 and 6.2.9 of BSC 2002a), these materials are suitable for use as structural backfill provided that material larger than 3 inches are removed and that a suitably cost-effective means is used to test the materials for quality control purposes. Backfill placed around structures should be placed in lifts not to exceed 12 inches loose depth for heavy compactors and 8 inches for hand-operated mechanical compactors, and compacted to an in-place density of at least 95 percent of the maximum laboratory dry density as determined by ASTM D 1557.

7.2.8 Concrete Aggregates

Based on gradation tests performed on the alluvial materials encountered in the test pits performed at the surface facilities and the Fran Ridge borrow area (Section 6.5.2 of BSC 2002a), materials encountered at the site are not suitable for use as concrete aggregates without processing. The unprocessed materials contain too many large size particles. Processing these deposits to produce acceptable concrete aggregate is expected to be cost-prohibitive. However, if ballast is also processed on site, the additional processing required for concrete aggregate may become more viable.

7.2.9 Volume Coefficients

Based on density test results compiled in BSC (2002b), Section I.2.1, the mean moist unit weight of the in-situ alluvium is between 114 and 117 pcf. The maximum dry unit weight for tests on Fran Ridge borrow material (also composed of alluvial soils) as reported in BSC (2002a), Section 8.1.1, was 114.5 pcf with a moisture content of 11%. Adjustments for the large particle sizes and for a moisture content one percentage point higher than the optimum resulted in a maximum estimated moist unit weight of 128 pcf (Section I.1.1 of BSC 2002b).

Therefore, assuming that the Fran Ridge material physical characteristics are similar to the insitu alluvium, the in-place relative compaction of the alluvium is estimated to be about 114/128 = 89% of its maximum value. Compaction of the excavated alluvium to 95% of its maximum dry density will result in a denser material that is smaller in volume. The difference involved with this process is therefore (114-128)/114 =-11%, or 11% percent shrinkage. Due to local variability in gravel content additional testing during construction will be necessary to determine the actual shrink or swell factors for the particular blend of materials.

7.2.10 Surface and Storm Water Drainage

Surface drainage should provide positive drainage of surface storm water away from the structures and pavement areas. We expect that storm water disposal may utilize conventional drywells installed within the alluvium. However, cementation in the alluvium may decrease the

effectiveness of this method and additional studies and analysis will be required to verify this if surface runoff is insufficient.

Infiltration testing is recommended for the alluvium. A factor of safety of at least 3.0 should be applied to the measured rate to accommodate plugging over time.

7.2.11 Septic System Drain Field

The septic system drain field should be designed in accordance the state and local requirements. The septic system should be designed using the average measured infiltration rate at 4 feet below the existing surface elevation in the alluvial materials. Current design standards allow septic systems to be designed based on actual infiltration rates without application of a factor of safety. Because of the expected heavy usage, provisions for reserve capacity should be included in the septic system drain field design.

7.2.12 Wet Weather Construction

Because of the granular nature of the soils at this site and the general environment of the site, wet weather construction should not be a major concern. Mitigation measures to reduce the potential impact of occasional storms would include providing positive drainage to direct storm water away from excavations and work zones. Effective maintenance of access roads and staging areas will also reduce the impact of an occasional storm.

7.2.13 Dewatering

Because of the depth of the groundwater, over 1000 feet below the ground surface elevation, dewatering is not a significant concern at this site.

7.3 ADDITIONAL INVESTIGATIONS/TESTING

The following is a list of items that will be required to finalize design for the YMP waste handing facilities at the North Portal area.

7.3.1 Additional Test Pits and Geologic Reconnaissance

It is recommended that an additional 5 to 6 shallow test pits be performed on Exile Hill to better characterize the slope for stability issues. This will include geologic reconnaissance and mapping of the slope area.

7.3.2 Additional Borings

Although there are numerous borings and test pits in the site vicinity, there are very few within the borders of the planned buildings. It is estimated that another 29 borings [each ~100 feet deep] will be required to provide sufficient coverage of the planned facilities. The borings should be performed using mud rotary, hollow-stem auger, or air-drill techniques. Sampling

would generally involve 3-inch diameter heavy-duty samplers along with 2-inch diameter SPT. Any encountered soft zones would be sampled with 3-inch diameter thin-walled Shelby tubes. Borings should extend about 15 feet into rock. Therefore, coring capability will also be needed. CPT is not an option due the amount of gravel present. The borings would be used to better define local stratigraphy for both static and dynamic analyses, and the depth of fill to be removed.

7.3.3 Laboratory Testing

Laboratory testing associated with the borings would consist of gradation, Atterberg limits, direct shear, moisture and density, relative density tests, and possibly large diameter triaxial testing.

7.3.4 CBR Testing

California bearing ratio tests are needed on the alluvium and anticipated fill sources for pavement design.

7.3.5 Field Plate Load Tests

Plate load tests should be conducted on undisturbed soils in the test pits to define the elastic parameters of the alluvium and the fill source.

7.3.6 Resistivity Testing

Field electrical resistivity tests should be performed on the alluvium and fill source materials.

7.3.7 Aggregate Testing

Qualification of on-site or local aggregate will require testing for use as backfill and under pavements. Required tests include specific gravity, absorption, degradation, and soundness.

7.3.8 Ballast Testing

Additional aggregate testing suites, as described in Section 7.3.7, would be needed to evaluate tunnel muck cuttings for use as ballast when suitable samples become available.

7.3.9 Chemical testing

Laboratory pH, chloride, sulphate, and resistivity tests will be needed to evaluate corrosion potential for metal pipes from alluvium and fill.

7.3.10 Field Infiltration Tests

Infiltration tests will be needed for design of on-site septic systems and storm water disposal.

7.3.11 Test Fill Program

A test fill program should be performed to evaluate the in-situ engineering properties of engineered fill, including its shear-wave and damping properties, and to determine the effect of construction equipment on the material. The test pad would also be used to establish relationships between the various density testing methods (i.e., nuclear, sand cone, and relative density).

7.3.12 Pavement Design

Design of temporary construction roads and operational pavements (and any special purpose roads, such as for heavy transport vehicles) will need to be provided when the pertinent additional field and laboratory tests are completed.

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APPENDIX A - SEISMIC WAVE VELOCITY

A1 Objective

The purpose of this analysis is to estimate representative shear-wave (V_s) and compression-wave (V_p) velocities for the soil and rock units present at the Yucca Mountain Project (YMP) site. The analysis is based on available seismic wave velocity data measured at the site and contained in BSC (2002a). Seismic wave velocities were obtained by the following seismic surveying methods: (1) downhole, (2) suspension P-S (OYO), and (3) spectral analysis of surface waves (SASW).

A2 Inputs

Direct input data used in the analysis herein are selected per Table 2 (summary of input data) of BSC (2002a). The seismic velocity data contained in BSC (2002a) are provided in tabular form consisting of V_s and V_p profiles at various depth intervals for different survey methods. The raw data from the surveying methods were not available for this calculation. Table A2-1 below lists data sources that were considered in this analysis:

Table A2-1. Tables and figures from previous reports providing seismic velocity data considered in the analysis (Data attached in Section A8.1).

Surveying Method	Source	Date of Surveys	Borings / Line surveys	Table/Figure	Data Tracking Number	Data
Downhole	BSC (2002a)	Oct. to Dec. 2000	RF#13 (two surveys), RF#14 through #26, #28, and #29	Tables 8 and 9	MO0111DVDWHBSC.001 MO0202WHBTMPKS.000 MO0110DVDBOREH.000 MO0202WAVEATD.000	V_s & V_p (1)
Suspension P-S (OYO)	BSC (2002a)	Sept. and Dec. 2000	RF#14 through #26, #28, and #29	Tables VII-2 and VII-3	MO0204SUSPSEIS.001	V_s & V_p (2)
SASW	BSC (2002a)	Summer 2000 and 2001	(3) Lines 1, 2, 4, 8, 10+37, 23, 29, 33, 32+35, and 34+36	Figures IX-1, IX-2, IX-4, IX-8, IX-10, IX-23, IX-29, IX-32, IX-33 and IX-34 (2 profiles)	MO0110SASWHBS.000	$V_s^{(1)}$

⁽¹⁾ Average velocities at various depth intervals

All data presented in Table A2-1 is provided in Section A8.1 of this calculation. Boring logs and soil contact depths provided in BSC (2002a) for all the locations listed Table A2-1 were used to match the soil layers with corresponding V_s and V_p values at depth. The tables providing the soil contact depths are contained in Section A8.2 of this calculation. The predominant soil layers identified in each boring are:

⁽²⁾ Average velocities by soil unit

⁽³⁾ Line surveys that were conducted at locations corresponding to nearby borings

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Existing Fill: Fill Alluvium: Qal

o Bedrock: Tmbt1 – Pre-Rainier Mesa Tuff bedded tuff

Tpki – Tuff unit "x"

Tpbt5 – Pre-tuff unit "x" bedded tuffs (also known as post-Tiva Canyon Tuff

bedded tuff)

Tpcrn – Tiva Canyon Tuff: crystal-rich member, nonlithophysal zone
Tpcpul – Tiva Canyon Tuff: crystal-poor member, upper lithophysal zone
Tpcpun – Tiva Canyon Tuff: crystal-poor member, middle nonlithophysal zone
Tpcpll – Tiva Canyon Tuff: crystal-poor member, lower lithophysal zone
Tpcpln – Tiva Canyon Tuff: crystal-poor member, lower nonlithophysal zone

A3 Background

Recent surveying investigations at the YMP site included measurements of V_s and V_p using three survey methods:

Downhole

Downhole surveys were performed at 16 boreholes (RF#13 through #26, #28, and #29). RF#13 was surveyed twice by downhole methods in 2000. Tables 8 and 9 of BSC (2002a) provide V_s and V_p data in terms of average seismic velocities at various depth intervals for each boring.

The locations of the borings where the downhole surveys were performed are shown in Figure A3-1.

Suspension log

Suspension log surveys were also performed at 16 boreholes (RF#13 through #26, #28, and #29). The receiver-to-receiver (RR) and source-to-receiver (SR) methods were both used in the suspension logging (except that only receiver-to-receiver was used for RF#13) to measure the shear-wave velocities. Tables VII-1, VII-2 and VII-3 from BSC (2002a) provide data in terms of seismic velocities averaged at each soil unit (determined from the geologic boring logs) for each boring. Per recommendations from BSC2002a, since more of the receiver-to-receiver seismic velocity data was missing, data from the SR method was used for evaluation of the suspension logging results.

The locations of the borings where the downhole surveys were performed are shown in Figure A3-1.

SASW

40 SASW surveys were performed at the site, of which 35 shear-wave velocity profiles were developed. The seismic velocities measured from the SASW surveys were determined from dispersion curves (surface wave velocity versus wavelength). Section 10.2.1.1 of BSC (2002a) presents 11 of these profiles corresponding to nearby boring locations (see Table A2-1). To simplify the analysis herein, only these profiles were used. Table A3-1 shows the profiles and their corresponding boring numbers.

The locations of the borings where the downhole surveys were performed are shown in Figure A3-2.

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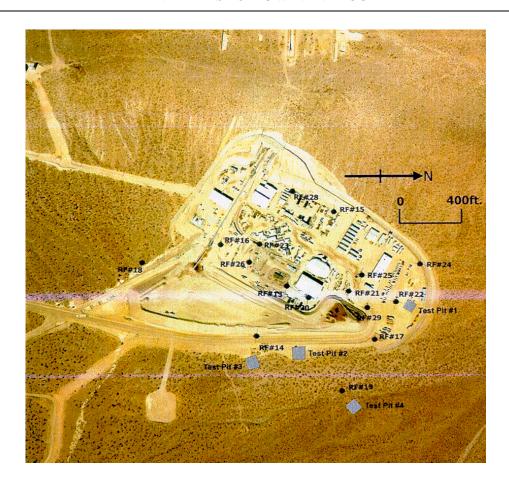


Figure A3-1. Locations of borings where downhole and suspension seismic surveys were conducted (Figure 2 of BSC 2002a, DTN:GS020383114233.001).

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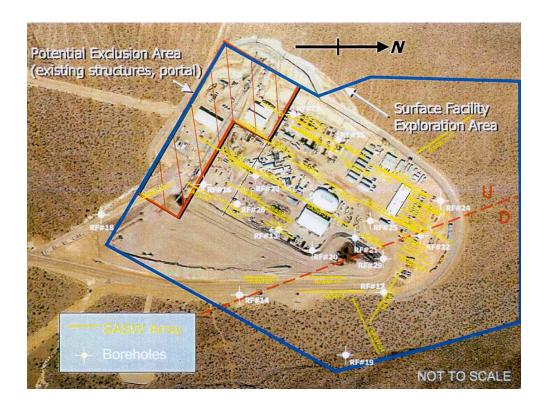


Figure A3-2. Locations of SASW seismic survey lines (Figure 43 of BSC 2002a).

Table A3-1. SASW Line Locations and Corresponding Borings.

SASW Line	Corresponding Boring (RF#)
1	13
2	21
4	26
8	28
10+37	15
23	22
29	16
33	23
32+35	23
34+36 (2 profiles)	17

Surveying information for each of the three methods are provided in BSC (2002a). Attachment VII of BSC2002a presents comparison figures showing seismic velocity versus depth profiles from the different survey methods.

Figure A3-3 and Figure A3-4 below show statistical values of V_s and V_p by lithostratigraphic unit measured from suspension surveys, respectively (the figures were taken from Figures 33 and 35 of BSC 2002a). Note that although Figure 33 states that the values are from source-to-receiver suspension surveys, it appears that the data is

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from receiver-to-receiver surveys (According to Table VII-2 of BSC 2002a, which shows statistics for suspension seismic source-to-receiver shear-wave velocities by lithostratigraphic unit, no measurements for Qal were made for RF#13, yet the figure below shows an average for that borehole for Qal. Table VII-1, on the other hand, which shows statistics for suspension seismic receiver-to-receiver shear-wave velocities by lithostratigraphic unit, does have a measurement for Qal for RF#13).

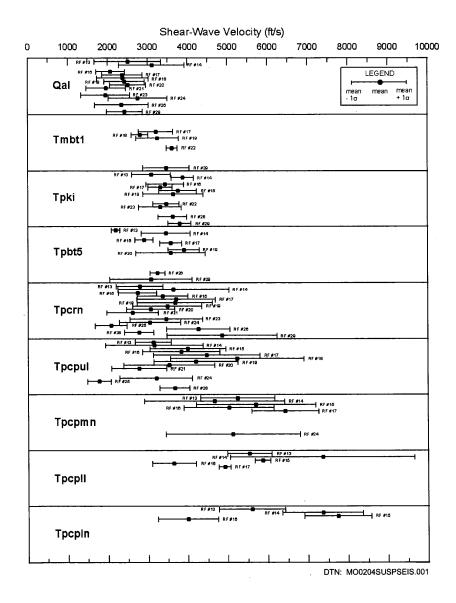


Figure A3-3. Statistical Values of Shear-Wave Velocity by Lithostratigraphic Unit from Source-to-Receiver Interval Suspension Surveys in Surface Facilities Area (Figure 33 of BSC 2002a, appears to represent values from receiver-to-receiver surveys).

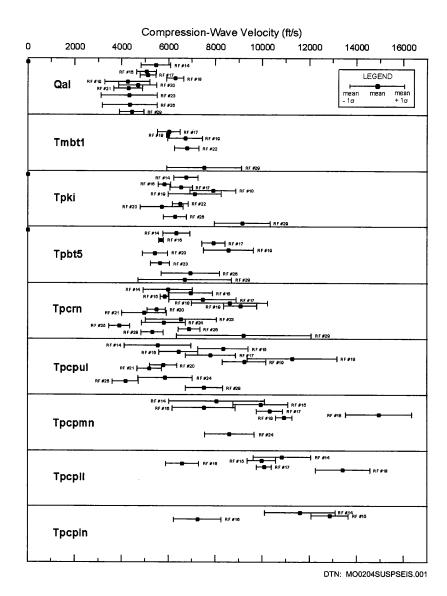


Figure A3-4. Statistical Values of Compression-Wave Velocity by Lithostratigraphic Unit from Source-to-Receiver Interval Suspension Surveys in Surface Facilities Area (Figure 35 of BSC 2002a).

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A downhole survey was performed at RF#13 in 1998. Previous surveys were also performed at the site in the mid 1980's. Downhole surveys were performed at borings designated as RF#3, 3B, #9 and #10. Laboratory sonic velocities were also measured at RF#9, #10, and #11. Because of the abundance of more current measurements, the data from these previous surveys are not considered in the analysis contained herein.

A4 Methodology

The data provided was analyzed separately for the alluvium, existing fill, and rock layers. In general, the seismic velocity values were averaged for each soil unit in each surveyed boring and for each surveying method. Statistical analysis (standard deviation and coefficient of variation) was also performed. The following steps were performed:

- 1. Where applicable, the seismic wave velocity profiles for each boring were superimposed over the geologic soil units.
- 2. Velocity values for each soil/rock layer were averaged where applicable using the following equation:

$$V_{average} = \frac{\sum_{i=1}^{n} V_i d_i}{\sum_{i=1}^{n} d_i}, \text{ where}$$
(A1)

 d_i = thickness of layer i in ft V_i = seismic velocity in layer i

* To estimate the average seismic wave velocity of alluvium, the layer was subdivided into four intervals: (1) 5-15 ft, (2) 15-30 ft, (3) 30-60 ft, and (4) 60-100 ft. Averages were determined for the existing fill and each bedrock unit. For the alluvium, an average was taken only if the sublayer thickness was at least half the amount of the depth interval.

In order to average the data, equal weight was given to each survey conducted.

A5 Assumptions

It is assumed that all data provided in the tables and figures referenced in Table A2-1 have been qualified for use in design analysis. It is assumed that all surveys performed in the surface facilities area are contained in BSC (2002a).

All of these assumptions are either sufficiently conservative or represent typical standards used in the industry and do not require further verification.

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A6 Calculations

The following sections describes the calculation approach to average the seismic wave velocities for the alluvium, existing fill, and bedrock, as outlined in Section A4. The methodology used to average the soil/rock units are essentially identical, with the exception that the alluvium is subdivided into 4 layers.

A6.1 Alluvium (Qal)

The seismic wave velocity data for the alluvium listed in Table A2-1 was used to develop a plot of seismic velocities versus depth for the 3 survey methods. The mid-depth of the measured values was used for the downhole and SASW surveys. Note that for this calculation, only SASW surveys that were conducted near borings were used (as shown in BSC 2002a). The data provided from the suspension logging surveys was an average of the entire alluvium (Qal) layer encountered for each boring (Tables VII-2 and VII-3 of BSC 2002a). The values were thus plotted against the mid-depth of the Qal layer for each boring. Figure A6-1 and Figure A6-2 show the profiles for both V_s and V_p values, respectively. Note that compression wave velocities are not measured by SASW surveys.

Following the methodology outlined in Section A4, EXCEL spreadsheets were used to conduct the analysis for each boring and are provided in Section A8.3 of this calculation. Table A6-1 shows the results from the analysis for shear wave velocity from downhole and SASW surveys. Table A6-2 presents both sets of averages from the downhole and SASW surveys. Figure A6-1 shows the results graphically.

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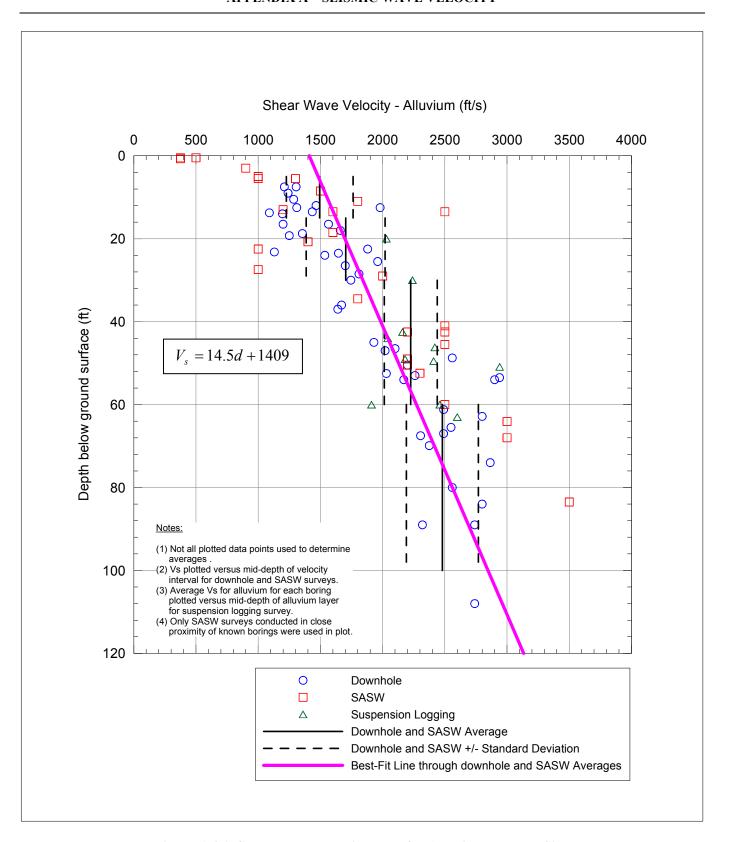


Figure A6-1. Shear Wave Velocity Data for Alluvium at YMP Site.

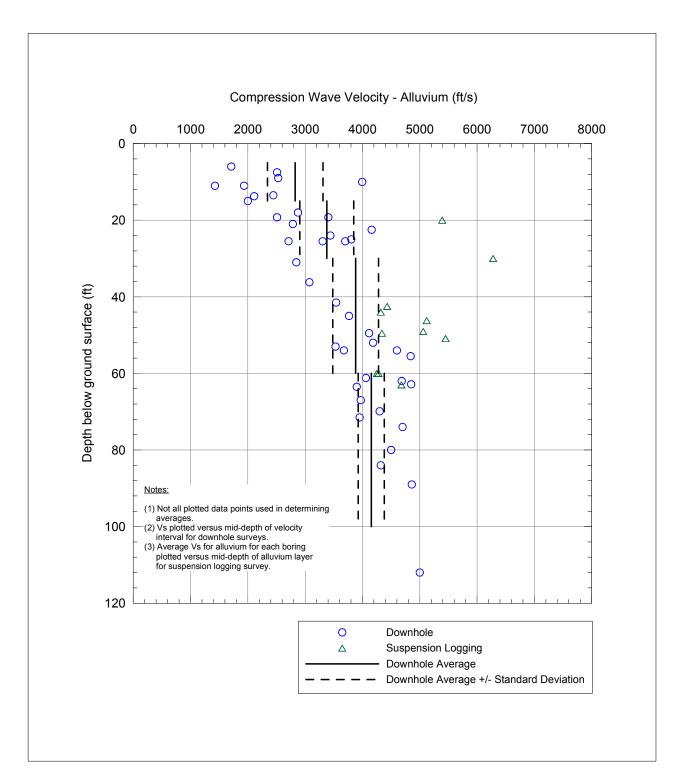


Figure A6-2. Compression Wave Velocity Data for Alluvium at YMP Site.

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Table A6-1. Computed alluvium shear wave velocity averages.

	SHEAR WAVE VELOCITY, Vs (FT/S)												
		Dow	nhole			SA	SW						
Boring				Interval /	Depth (ft)								
	1	2	3	4	1	2	3	4					
	5' - 15'	15' - 30'	30' - 60'	60' - 100'	5' - 15'	15' - 30'	30' - 60'	60' - 100'					
13 (survey 1)		1580	2030	2366		1453	2200	3192					
13 (survey 2)		1960	2384	2490									
14	1240	1700	2195	2375									
16		1533	2027			1000	2500						
17 (Profile 1)	1210	1880	2490	2490	1565	2300	2300	2300					
17 (Profile 2)					1480	1800	1800						
18	1435	1529	2162										
19	1285	1705	2157	2349									
20	1200	1528	2020	2595									
21	1310	1723	1930		1570	2140	2500						
22	1465	1906	2200	2200	2200	2200	2200	2200					
23		1886	2100			2000	2267						
24	1195	1467											
25	1645	1645	2638										
26		1745	2121	2550		1000	2600	3000					
28	1643				1800								
29	1660	1660	2119	2326									
# borings	11	15	14	9	5	8	8	4					
Avg. Vs (ft/s)	1390	1696	2184	2416	1723	1737	2296	2673					
St.Dev.,σ (ft/s)	189	156	195	125	292	526	251	496					
Coeff of var.,													
σ/Avg. Vs	0.14	0.09	0.09	0.05	0.17	0.30	0.11	0.19					

Table A6-2. Computed alluvium shear wave velocity averages of downhole and SASW surveys combined.

Interval	1	2	3	4
Depth (ft)	5 - 15	15 - 30	30 - 60	60 – 100
# of measurements	16	23	22	13
Average Vs, (ft/s)	1494	1710	2224	2495
St.Dev.,σ (ft/s)	268	322	218	295
Coeff of variation, σ/avg	0.18	0.19	0.10	0.12

A linear fit is plotted on Figure A6-1 through the 4 average velocity values reported in Table A6-2. The fitted equation was determined to be:

$$V_{s} = 14.5d + 1409 \tag{A2}$$

 V_s = shear wave velocity

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d = depth

Table A6-3 below presents the average results from the analysis for compression wave velocity from the downhole surveys for alluvium. Figure A6-2 shows the results graphically.

Table A6-3. Computed alluvium compression wave velocity averages.

	COMPRESSION WAVE VELOCITY, Vp (FT/S)												
			Dow	nhole									
	Boring		Interval /	Depth (ft)									
		1	2	3	4								
		5' - 15'	15' - 30'	30' - 60'	60' - 100'								
	13 (survey 1)		3746	4685	4685								
	13 (survey 2)		3700	3916	3970								
	14	2955	3805	4168	4300								
	16		3075	3667									
	17 (Profile 1)	2510	4160	4060	4060								
	18	3305	3305	3823									
	19	2748	3440	3797	3950								
	20	2470	3540	3540	4115								
	21	2845	2845	2951									
	22	2445	3141	4185	4185								
	23		3412	3765									
	24	2241	2785										
	25	2710	2710	4059									
	26		4115	4115	4115								
	28	3995											
	29	2875	2875	3595	4005								
-	# borings	11	15	14	9								
P	Avg. Vp (ft/s)	2827	3377	3880	4154								
St	.Dev.,σ (ft/s)	485	470	399	227								
	Coeff of var.,												
	σ/Avg. Vp	0.17	0.14	0.10	0.05								

Table A6-4 below shows average values obtained by source-to-receiver suspension surveys provided in BSC (2002a). BSC (2002a) reports final averages of V_s and V_p for all the borings (weighted by the number of measurements). An attempt to verify the suspension data failed to produce the same result as presented in Tables VII-2 and VII-3 of BSC (2002a). It is not specifically documented in BSC (2002a) how the statistical data was determined. Since the averages provided by BSC (2002a) are for the entire alluvium layer in each boring, the data could not be subdivided into the four depth intervals as was performed for the downhole and SASW surveys. The raw data for the suspension logging for borings RF#14 through #26, #28, and #29 were not available for this calculation for processing.

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Table A6-4. Seismic Wave Velocity Averages of Qal for Suspension Logging Surveys (as reported in BSC 2002a).

Boring	Source-t	o-receiver
	V _s (ft/s)	V_{p} (ft/s)
13	-	-
14	2940 ± 240	5450 ± 630
16	2180 ± 150	5060 ± 430
17	2420 ± 380	5120 ± 340
18	2240 ± 320	6280 ± 370
19	2460 ± 340	4250 ± 960
20	2600 ± 390	4680 ± 810
21	1910 ± 300	4280 ± 610
23	2040 ± 650	4320 ± 1200
24	2030	5390
26	2410 ± 620	4340 ± 1170
29	2160 ± 350	4430 ± 520
All	2040 ± 880	4660 ± 950

It can be seen from the above tables and figures that the shear and compression wave velocity of the alluvium generally increases with depth. The shear wave velocity results from the downhole, SASW, and suspension logging surveys show relatively good agreement with each other. However, compression wave velocity results are higher from the suspension logging than the downhole surveys. Since the averages provided for the suspension logging are for the entire alluvium layer in each boring, it cannot be determeined how the seismic velocity varies with depth by this method.

Since the provided averages from the suspension logging could not be checked, only the averages of V_p from the downhole surveys are used. The average seismic velocity range for the alluvium obtained from available data is estimated below (from downhole and SASW surveys):

Depth (ft)	Shear Wave Velocity ¹ (ft/s)	Compression Wave ² Velocity (ft/s)
5 – 15	$1{,}500\pm270$	$2,\!800\pm490$
15 – 30	$1{,}700\pm320$	$3,\!400\pm470$
30 – 60	$2,\!200\pm220$	$3,900 \pm 400$
60 - 100	$2,500 \pm 300$	$4,200 \pm 230$

¹ from downhole and SASW surveys

² from downhole surveys only

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Using the following equation, Poisson's ratio for the alluvium can be estimated:

$$\upsilon = \frac{2V_s^2 - V_p^2}{2V_s^2 - 2V_p^2} \tag{A3}$$

Using average values of the seismic velocities, a Poisson's ratio of 0.23 to 0.30 is estimated for the alluvium. If the V_p average of 4660 ft/s (Table A6-4) from the suspension logging surveys is used, the Poisson's ratio range is 0.30 to 0.44. These ranges are in good agreement with the BSC (2002a) reported value of 0.27 \pm 0.15 (Table VII-4).

A6.2 Existing Fill and Bedrock

Though the existing fill will be removed and thus is not necessary to consider, it is included in the analysis for completeness. It is difficult to estimate representative values of the bedrock units at the YMP site due to their varying thickness and depth locations. The cross-section in Figure 225 of BSC (2002a) shows the amount of dipping of the bedrock layers that exists at the YMP site. BSC (2002a) provides figures visually comparing the seismic wave velocities of these bedrock units.

The methodology used to average the alluvium was generally adopted to average the shear and compression wave velocities of the existing fill and rock layers, though unlike the alluvium, these materials were not subdivided into smaller intervals. The following rock layers for the downhole, SASW, and suspension logging surveys were used to compute the seismic wave averages:

Pre-Rainier Mesa Tuff bedded tuff Tmbt1 -Tpki – Tuff unit "x" Tpbt5 – Pre-tuff unit "x" bedded tuffs (also known as post-Tiva Canyon Tuff bedded tuff) Tpcrn – Tiva Canyon Tuff: crystal-rich member, nonlithophysal zone Tpcpul – Tiva Canyon Tuff: crystal-poor member, upper lithophysal zone Tpcpmn -Tiva Canyon Tuff: crystal-poor member, middle nonlithophysal zone Tpcpll – Tiva Canyon Tuff: crystal-poor member, lower lithophysal zone Tpcpln – Tiva Canyon Tuff: crystal-poor member, lower nonlithophysal zone

Computed averages were based solely on the extent of the measured seismic velocity profiles and the logged geologic borings provided in BSC (2002a), regardless of whether the surveyed profile extended through the entire rock layer or ended within the layer. Averages from the suspension logging surveys are provided in BSC (2002a) (shown in Section A8.1) for each rock unit and each boring were also used in this analysis. The averages obtained from the source-to-receiver method were used per BSC (2002a) recommendations.

The results of the shear wave velocity averages for each boring and the total averages for the data obtained from the downhole, SASW, and suspension logging surveys are shown in Table A6-5, Table A6-6 and Table A6-7, respectively. Table A6-8 shows the data from all the surveys (downhole, SASW, and suspension logging) averaged together. Table A6-9 and Table A6-10 show the results of the compression wave velocity averages from the downhole and suspension logging surveys for each boring, respectively. EXCEL spreadsheets were used to conduct the analysis for each boring and are provided in Section A8.3. Table A6-11 shows the data from the downhole and suspension logging surveys averaged together.

It is evident from the presented tables and above figures, that although it is unclear how the seismic velocity varies with depth within each bedrock unit, a notable increase in seismic velocity exists between the Tpcpul and Tpcpmn layers. Hence, using equation (A1), averages were determined for the rock layer from Tmbt1 to Tpcpul (upper rock) and from Tpcpmn to Tpcpln (lower rock). Thus for the following tables, averages were also computed for the "upper rock", "lower rock", and the entire rock layer (upper and lower rock combined).

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Table A6-5. Computed existing fill and bedrock shear wave velocity averages (downhole surveys).

	SHEAR WAVE VELOCITY, Vs (FT/S)													
						D	ownhole							
Boring						I	Material							
	Fill	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpul	Tpcpmn	Tpcpll	Tpcpln	All Rock	Upper	Lower		
											Rock ^a	Rock ^b		
13	909		2740	2740	2740	3110	5800	5800	5800	4080	2774	5800		
13	1090		2805	2810	3113	6490	6490	6490	6490	4746	3262	6490		
14			3091	2640	2640	4410	5000	5000	5000	3937	3484	5000		
15	1935				2363	3126	4053	5900	5900	3776	2838	5209		
16	836		2800	2800	2800	3143	5713	7000	7000	3745	2967	6349		
17		3134	3160	3160	3890	4393	4520			3598	3537	4520		
18		2900	3770		3525	4200	4200	4200		3901	3856	4200		
19		2740	3780	3780	3780	4100	4250			3537	3530	4250		
20	1200			2800	2800	2800				2800	2800			
21	1310				2500	2500				2431	2431			
22		3349	3393	3500	3500					3414	3414			
23	982		2865	2865	3416					3284	3284			
24	1195				2050	2070	2070			2063	2062	2070		
25	1645				2344	2100				2258	2258			
26	698		3677	3780	3780					3710	3710			
28	1305				2724	3300				2904	2904			
29		3237	3800	3800	3800					3457	3457			
# borings	11	5	11	11	17	13	9	6	5	17	17	9		
g. Vs (ft/s)	1191	3072	3262	3152	3045	3519	4677	5732	6038	3391	3092	4876		
ev.,σ (ft/s)	359	249	436	470	595	1203	1292	1010	755	702	519	1351		
Coeff of var.,														
σ/Avg. Vs	0.30	0.08	0.13	0.15	0.20	0.34	0.28	0.18	0.13	0.21	0.17	0.28		

^aTmbt1 to Tpcpul

Table A6-6. Computed existing fill and bedrock shear wave velocity averages (SASW surveys).

	SHEAR WAVE VELOCITY, Vs (FT/S)													
							SASW							
Boring		Material												
	Fill	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpul	Tpcpmn	Tpcpll	Tpcpln	All Rock	Upper	Lower		
											Rock ^a	Rock ^b		
13	848		3500	3500	3500					3500	3500			
15	1300				2803	3473	5000	5000		3615	3220	5000		
16	737			3000	3000					3000	3000			
17		2453	2700							2469	2469			
21	1160				2500					2500	2500			
22		2980								2980	2980			
23	1008		2500	2500	3360					3163	3163			
26	557		3000							3000	3000			
28	1040				2856	3200				2977	2977			
# borings	7	2	4	3	6	2	1	1	0	9	9	1		
Avg. Vs (ft/s)	950	2717	2925	3000	3003	3336	5000	5000		3023	2979	5000		
St.Dev.,σ (ft/s)	254	372	435	500	371	193	-	-		385	327	-		
Coeff of var.,														
σ/Avg. Vs	0.27	0.14	0.15	0.17	0.12	0.06	-	-		0.13	0.11	-		

^aTmbt1 to Tpcpul

^bTpcpmn to Tpcpln

^bTpcpmn to Tpcpln

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Table A6-7. Bedrock shear wave velocity averages as reported in BSC2002a (suspension logging surveys).

SHEAR WAVE VELOCITY, Vs (FT/S)														
					Suspe	nsion Log	ging, sourc	e-to-recei	ver					
Boring		Material State Table												
	Fill	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpul	Tpcpmn	Tpcpll	Tpcpln	All Rock	Upper	Lower		
											Rock ^a	Rock ^b		
14			3790	3420	3350	3130	4380	6280	7240	4391	3399	6268		
15					3360	4380	5410	6170	7160	5012	3995	6449		
16			3340	2190	3350	3620	4760	3770	4240	3687	3460	4382		
17		3240	3330	3660	3540	4200	6030	5140		3814	3547	5819		
18		2840	3360		3440	5640	7380	5430		3901	3374	6709		
19		3330	3390	3730	3360	3890	3590			3496	3494	3590		
20				2880	3170	3240				3189	3189			
21					2680	2760				2708	2708			
22		3710	3560							3667	3667			
23			3150	3110	3600					3496	3496			
24					3060	3270	4850			3452	3186	4850		
25					2010	2210				2086	2086			
26			3680	4040	3840					3742	3742			
28					2970	4450				3492	3492			
29		2160	3470	3800	3650	4650				2765	2765			
# borings	0	5	9	8	14	12	7	5	3	15	15	7		
vg. Vs (ft/s)	-	3056	3452	3354	3241	3787	5200	5358	6213	3526	3307	5438		
Dev.,σ (ft/s)	-	589	197	603	462	940	1229	1010	1709	692	475	1180		
Coeff of var.,														
σ/Avg. Vs	-	0.19	0.06	0.18	0.14	0.25	0.24	0.19	0.28	0.20	0.14	0.22		

^aTmbt1 to Tpcpul

Table A6-8. Computed existing fill and bedrock shear wave velocity averages of downhole, SASW, and suspension logging surveys.

Material	Fill	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpul	Tpcpmn	Tpcpll	Tpcpln	All Rock	Upper Rock ^a	Lower Rock ^b
# measurements	18	12	24	22	37	27	17	12	8	41	41	17
Avg. Vs (ft/s)	1097	3006	3277	3205	3113	3624	4912	5515	6104	3360	3146	5115
St.Dev.,σ (ft/s)	337	424	393	516	514	1033	1212	946	1081	656	475	1230
Coeff of var, σ/avg	0.31	0.14	0.12	0.16	0.17	0.29	0.25	0.17	0.18	0.20	0.15	0.24

^bTpcpmn to Tpcpln

^aTmbt1 to Tpcpul ^bTpcpmn to Tpcpln

Table A6-9. Computed existing fill and bedrock compression wave velocity averages (downhole surveys).

				COMF	PRESSION		ELOCITY	, Vp (FT/S	5)				
	Downhole												
Boring		Material											
	Fill	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpul	Tpcpmn	Tpcpll	Tpcpln	All Rock	Upper	Lowe	
											Rock ^a	Rock	
13		3746	4685	4685	4685	6748	9335	9335	9335	6785	4877	9335	
13		3700	3916	3970	5417	11180	11180	11180	11180	8201	5669	11180	
14	2955	3805	4168	4300	5900	7113	9203	11000	11000	7532	6352	10300	
15					4147	7403	12748	14000	14000	9084	6174	1353°	
16		3075	3667		4850	5864	8735	10000	10000	6267	5342	9360	
17	2510	4160	4060	4060	6731	9602	10210			6916	6699	10210	
18	3305	3305	3823		5881	7489	8300	8300		6643	6393	8300	
19	2748	3440	3797	3950	6350	6350	6350			5898	5894	6350	
20	2470	3540	3540	4115	4320	4320				4320	4320		
21	2845	2845	2951		4350	4850				4437	4437		
22	2445	3141	4185	4185	5560					5537	5537		
23		3412	3765		5167					5054	5054		
24	2241	2785			4878	4960	4960			4932	4927	4960	
25	2710	2710	4059		4328	4800				4495	4495		
26		4115	4115	4115	5750					5986	5986		
28	3995				4922	5640				5147	5147		
29	2875	2875	3595	4005	6040					5799	5799		
# borings	11	15	14	9	17	13	9	6	5	17	17	9	
g. Vp (ft/s)	2827	3377	3880	4154	5252	6640	9002	10636	11103	6061	5477	9281	
ev.,σ (ft/s)	485	470	399	227	778	1988	2354	1965	1785	1350	729	2545	
Coeff of var.,													
σ/Avg. Vp	0.17	0.14	0.10	0.05	0.15	0.30	0.26	0.18	0.16	0.22	0.13	0.27	

^aTmbt1 to Tpcpul

Table A6-10. Bedrock compression wave velocity averages as reported in BSC2002a (suspension logging).

				COMP	PRESSION	N WAVE N	/ELOCITY	, Vp (FT/S	3)					
	Suspension Logging, source-to-receiver													
Boring		Material Material												
	Fill	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpul	Tpcpmn	Tpcpll	Tpcpln	All Rock	Upper	Lower		
											Rock ^a	Rock ^b		
14			6750	6340	5980	5560	8060	10840	11610	7573	6060	10435		
15					6950	8340	9930	9970	12860	9358	7816	11539		
16			5830	5690	5840	6440	7520	6600	7240	6400	6127	7235		
17		6020	6550	7930	7460	7800	10320	10080		7273	6875	10263		
18		5960	7900		8620	11280	14960	13420		8344	7200	14430		
19		6710	7120	8570	9070	9240	10920			7935	7876	10920		
20				5420	5490	5790				5645	5645			
21					4960	5190				5041	5041			
22		6780								4817	4817			
23			5710	5640	6540					6348	6348			
24					5800	5860	8610			6279	5836	8610		
25					3910	4170				4009	4009			
26			6290	6940	6890					6495	6495			
28					5320	7530				6100	6100			
29		7530	9150	6690	9210					8172	8172			
# borings	0	5	8	8	14	11	7	5	3	15	15	7		
g. Vp (ft/s)	-	6600	6913	6653	6574	7018	10046	10182	10570	6653	6294	1049		
ev.,σ (ft/s)	-	643	1146	1131	1577	2061	2496	2439	2951	1463	1173	2273		
Coeff of var.,														
σ/Avg. Vp	-	0.10	0.17	0.17	0.24	0.29	0.25	0.24	0.28	0.22	0.19	0.22		

^aTmbt1 to Tpcpul

^bTpcpmn to Tpcpln

^bTpcpmn to Tpcpln

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Table A6-11. Computed existing fill and bedrock compression wave velocity averages of downhole and suspension logging surveys.

Material	Fill	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpul	Tpcpmn	Tpcpll	Tpcpln	All Rock	Upper Rock ^a	Lower Rock ^b
# measurements	11	20	22	17	31	24	16	11	8	32	32	19
Avg. Vp (ft/s)	2827	4183	4983	5330	5849	6813	9459	10430	10903	6338	5860	9810
St.Dev.,σ (ft/s)	485	1517	1663	1496	1360	1986	2394	2090	2094	1413	1034	2430
Coeff of var, σ/avg	0.17	0.36	0.33	0.28	0.23	0.29	0.25	0.20	0.19	0.22	0.18	0.25

^aTmbt1 to Tpcpul

Based on all the surveys, the following average ranges were estimated:

 V_s for upper rock: 3,100 \pm 480 ft/s V_p for upper rock: 5,900 \pm 1,030 ft/s

 V_s for lower rock: 5,100 \pm 1,230 ft/s V_p for lower rock: 9,800 \pm 2,430 ft/s

 V_s for entire rock: 3,400 ± 660 ft/s V_p for entire rock: 6,300 ± 1,410 ft/s

Using equation (A3), Poisson's ratio for the upper and lower rock layers are found to be similar. The range for the entire rock is 0.27 - 0.31. This is in relative good agreement with Figures 28 (downhole measurements) and 36 (suspension logging surveys) of BSC (2002a).

^bTpcpmn to Tpcpln

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APPENDIX A – SEISMIC WAVE VELOCITY

A7 Results/Conclusions

A simple analysis was performed to compute statistical values of the shear- and compression-wave velocities of the soil layers present at the Yucca Mountain Project site for three methods of seismic wave surveying. Data was provided from BSC (2002a).

Based on the comparisons made, the following average ranges of seismic velocities were estimated for the alluvium and bedrock materials:

• Alluvium

Depth (ft)	Shear Wave Velocity (ft/s)	Compression Wave Velocity (ft/s)
5 – 15	$1{,}500\pm270$	$2,\!800 \pm 490$
15 - 30	$1{,}700\pm320$	$3,\!400\pm470$
30 - 60	$2,\!200\pm220$	$3,\!900\pm400$
60 - 100	$2,500 \pm 300$	$4,200 \pm 230$

Bedrock

V_s for upper rock:	$3,100 \pm 480 \text{ ft/s}$
V _p for upper rock:	$5,900 \pm 1,030 \text{ ft/s}$
V _s for lower rock:	$5100 \pm 1,230 \text{ ft/s}$
V _p for lower rock:	$9,800 \pm 2,430 \text{ ft/s}$
V _s for entire rock:	$3,400 \pm 660 \text{ ft/s}$
V _p for entire rock:	$6,300 \pm 1,410 \text{ ft/s}$

The upper rock refers to the Tmbt1 to Tpcpul layers. The lower rock refers to the Tpcpun to Tpcpln layers. It should be noted that the analyses were based on averaging seismic wave velocities within each soil/rock unit.

A Poisson's ratio of 0.23 to 0.44 is estimated for the alluvium. A Poisson's ratio of 0.27 to 0.31 is estimated for the bedrock.

The averages do not take into account the influence of depth with seismic velocity. A review of the geologic conditions of the site shows that the bedrock unit layers may vary considerably in elevation at some locations. It should also be noted for simplification, that equal weight was given to each boring where a survey was conducted, regardless of the survey method or how many measurements were made within the soil/rock unit (in the case of the suspension logging surveys).

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A8 Attachments

The following sections contain data and spreadsheets used for the analyses:

- A8.1 Seismic wave data – contains all the seismic wave data and corresponding depths at which they were measured presented in BSC (2002a) that was used for the analyses.
- Soil contact depths contains all the soil contact depths from the boring logs presented in BSC A8.2 (2002a). These contact depths were superimposed onto the data from Section A8.1 in order to assign the appropriate seismic wave velocity values to their corresponding soil unit.
- EXCEL spreadsheets contains the averaging performed on the seismic wave velocity data for A8.3 each boring where data is available. Seismic wave velocity average = sum of (velocity \times thickness) / sum of thickness for each soil/rock unit.

A8.1 Seismic Wave Data (from BSC 2002a)

Shear wave velocity as reported in BSC (2002a) (downhole surveys)

Depth Range (ft) Velocity (ft/s) **RF#13** 3 - 10 750 (all boreholes surveyed by 10 - 251355 Redpath Geophysics unless 25 - 80 2030 otherwise indicated) 80 - 230 2740 230 - 345 5,800 ± Depth Range (ft) Velocity (ft/s) 1,090 **RF#13** 15 - 36 1,960 (GEOVision) 36 - 99 2,490 99 - 215 2810 215 - 345 6490 Depth Range (ft) Velocity (ft/s) 3 - 15 15 - 38 1700 RF#14 38 - 114 2375 114 - 165 3390 165 - 305 2640 305 - 520 5000 ± Depth Range (ft) Velocity (ft/s) 3 - 38 38 - 122 1935 **RF#15** 2700 ± 122 - 230 3380 230 - 320

5900

Table 8. WHB Area Downhole Shear-Wave Velocities

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Table 8. WHB Downhole Shear-Wave Velocities (continued)

	Depth Range (ft)	Velocity (ft/s)
	3 - 15	655
•	15 - 24	1130
RF#16	24 - 50	1640
	50 - 296	2800
	296 - 376	3540
	376 - 445	7000
	Depth Range (ft)	Velocity (ft/s)
	0 - 15	1210
RF#17	15 - 30	1880
(GEOVision)	30 - 100	2490
,	100 - 400	3160
	400 - 500	3890
	500 - 620	4520
	Depth Range (ft)	Velocity (ft/s)
	3 - 24	1435
DEMA	24 - 48	1670
RF#18	48 - 78	2900
	78 - 220	3860
	220 - 250	2400
	250 - 480	4200
	Depth Range (ft)	Velocity (ft/s)
	3 - 18	1285
	18 - 39	1810
RF#19	39 - 96	2305
	96 - 282	2740
	282 - 550	3780
	550 - 640	4250
	Depth Range (ft)	Velocity (ft/s)
RF#20	3 - 24	1200
RF#20	24 - 70	2020
	70 - 155	2800 ±
	Depth Range (ft)	Velocity (ft/s)
RF#21	3 - 20	1310
KF#Z1	20 - 84	1930
	84 - 185	2500 ±
	Depth Range (ft)	Velocity (ft/s)
	3 - 21	1465
RF#22	21 - 83	2200
	83 - 175	3540
	175 - 192	1400
	192 - 500	3500
	Depth Range (ft)	Velocity (ft/s)
	3 - 9	690
RF#23	9 - 21	1565
	21 - 72	2100
	72 - 110	2865
	110 - 155	3600
	Depth Range (ft)	Velocity (ft/s)
RF#24	3 - 18	1195
	18 - 33	1535
	33 - 260	2070

Table 8. WHB Downhole Shear-Wave Velocities (continued)

	Depth Range (ft)	Velocity (ft/s)
DE#05	3 - 37	1645
RF#25	37 - 86	2940
	86 - 155	2100
	Depth Range (ft)	Velocity (ft/s)
	3 - 12	425
RF# 26	12 - 46	1745
	46 - 95	2550
	95 - 260	3780
	Depth Range (ft)	Velocity (ft/s)
RF#28	3 - 10	1305
KF#20	10 - 39	1980
	39 - 95	3300
	Depth Range (ft)	Velocity (ft/s)
	3 - 33	1660
RF#29	33-75	2170
11/4/29	75-138	2560
	138-230	3320
	230 -405	3800

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Compression wave velocity as reported in BSC (2002a) (downhole surveys)

Table 9. WHB Area Downhole Compression-Wave Velocities

DE#46	Depth Range (ft)	Velocity (ft/s)		
RF#13 (all boreholes surveyed by	3-9	1455		
Redpath Geophysics unless	9 - 26	3405		
otherwise indicated)	26 - 226	4685		
Other wise indicated)	226 - 345	9335		
	Depth Range (ft)	Velocity (ft/s)		
	0 - 15	2110		
RF#13	15 - 36	3700		
(GEOVision)	36 - 99	3970		
	99 - 215	4900		
	215 - 345	11,180		
	Depth Range (ft)	Velocity (ft/s)		
	6 - 12	2530		
	12 - 38	3805		
RF#14	38 - 110	4300		
	110 - 304	5900		
	304 - 420	7500		
	420 - 520	11,000		
	Depth Range (ft)	Velocity (ft/s)		
	3 - 18	3215		
RF#15	18 - 39	3815		
Kr#15	38 - 133	4600 ±		
	133 - 210	9850		
	210 - 320	14,000 ±		
	Depth Range (ft)	Velocity (ft/s)		
	3 - 15	1590		
RF#16	15 - 50	3075		
KF#10	50 - 280	4850		
	280 - 376	6600 ±		
	376 - 445	10,000 ±		
	Depth Range (ft)	Velocity (ft/s)		
	0 - 15	2510		
RF#17	15 - 30	4160		
(GEOVision)	30 - 100	4060		
(100 - 400	5580		
	400 - 500	7190		
	500 - 620	10,210		
	Depth Range (ft)	Velocity (ft/s)		
	3 - 48	3305		
RF#18	48 - 78	4600		
	78 - 290	5850		
<u>L</u> .	290 - 390	7200		
	390 - 485	8300 ±		
	Depth Range (ft)	Velocity (ft/s)		
	3 - 9	1710		
RF#19	9 - 39	3440		
10, #10	39 - 104	3950		
	104 - 294	5000		
	294 - 640	6350		

Table 9. Downhole Compression Wave Velocities (concluded)

	Depth Range (ft)	Velocity (ft/s)		
RF#20	3 - 13	1935		
10 #20	13 - 70	3540		
	70 - 155	4320		
	Depth Range (ft)	Velocity (ft/s)		
RF#21	3-57	2845		
1	57 -120	3900		
	120-185	4850		
	Depth Range (ft)	Velocity (ft/s)		
RF#22	3 - 24	2445_		
	24 - 87	4185		
	87 - 505	5560		
	Depth Range (ft)	Velocity (ft/s)		
	3 - 18	2000		
RF#23	18 - 72	3765		
	72 - 120	4700		
	120 - 155	5500		
	Depth Range (ft)	Velocity (ft/s)		
RF#24	3 - 12	1425		
1	12 - 33	2785		
	33 - 260	4960		
	Depth Range (ft)	Velocity (ft/s)		
	3 - 41	2710		
RF#25	41 - 86	4840		
	86 - 105	3400		
	105 - 155	4800 ±		
	Depth Range (ft)	Velocity (ft/s)		
	3 - 10	840		
RF#26	10 - 95	4115		
	95 - 140	7030		
	140 - 260	5750		
	Depth Range (ft)	Velocity (ft/s)		
RF#28	3 - 39	3995		
	39 - 96	5640		
	Depth Range (ft)	Velocity (ft/s)		
	3 - 33	2875		
RF#29	33 - 75	3675		
	75 - 135	4500		
	135 - 405	6040		

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APPENDIX A – SEISMIC WAVE VELOCITY

Shear wave velocity as reported in BSC (2002a) (suspension surveys, source-to-receiver)

Table VII-2. Statistics for Suspension Seismic Receiver-to-Receiver Shear-Wave Velocities by Lithostratigraphic Unit

Borehole	Parameter	Fill	Qal	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpul	Tpcpmn	Tpcpll	Tpcpln
	Depth (ft)	0-12.5	12.5-98.0		98.0-164.4	164.4-169.3	169.3-219.1	219.1-231.5	231.5-286.7	286.7-300.9	300.9-350.1
	Mean Vs (ft/s)	NA	NA		NA	NA	NA	NA	NA	NA	NA.
RF#13	Median V _s (ft/s)	NA	NA.		NA						
(core)	Standard Deviation	NA	NA		NA						
	Coeff. of var. (%)*	NA	NA		NA						
	No. of meas.	0	0		0	0	0	. 0	0	0	0
	Depth (ft)		0-101.8		101.8-192.5	192.5-203.4	203.4-275.0	275.0-395.0	395.0-443.7	443.7-455.6	455.6-550.0
	Mean Vs (ft/s)		2940		3790	3420	3350	3130	4380	6280	7240
RF#14	Median V _s (ft/s)		2930		3830	3280	3220	3010	4310	6180	7330
(core)	Standard Deviation		242.3		247.7	392.3	539.7	1008.6	1370.3	978.5	716.1
	Coeff. of var. (%)		8.2		6.5	11.5	16.1	32.2	31.3	15.6	9.9
	No. of meas.		31	l .	56	6	44	73	30	7	55
	Depth (ft)	0-6.5					6.5-78.0	78.0-196.0	196.0-242.4	242.4-256.6	256.6-330.0
	Mean Vs (ft/s)	NA					3360	4380	5410	6170	7160
RF#15	Median V _s (ft/s)	NA					3230	4500	5120	6130	7180
(core)	Standard Deviation	NA					544.6	639.0	1357.9	433.5	468.9
	Coeff. of var. (%)	NA					16.2	14.6	25.1	7.0	6.5
	No. of meas.	0					31	72	28	9	37
	Depth (ft)	0-22.4	22.4-75.7		75.7-133.2	133.2-137.8	137.8-222.0	222.0-360.0	360.0-403.0	403.0-422.5	422.5-452.8
	Mean Vs (ft/s)	NA	2180		3340	2190	3350	3620	4760	3770	4240
RF#16	Median V _s (ft/s)	NA	2200		3400	2190	3300	3570	4810	3560	4180
(core)	Standard Deviation	NA	147.7		407.9	113.1	324.3	721.5	1075.6	570.9	541.6
	Coeff. of var. (%)	NA	6.8		12.2	5.2	9.7	20.0	22.6	15.2	12.8
	No. of meas.	.0	30		35	2	52	84	26	12	15
	Depth (ft)		0-92.4	92.4-287.2	287.2-348.4	348.4-368.9	368.9-478.0	478.0-587.3	587.3-637.6	637.6-653.2	653.2-667.8
	Mean Vs (ft/s)		2420	3240	3330	3660	3540	4250	5980	5240	NA
RF#17	Median V _s (ft/s)		2460	3160	3350	3620	3450	4200	6030	5140	NA
(core)	Standard Deviation		383.1	355.1	169.7	159.5	562.1	1053.9	616.4	156.9	NA
	Coeff. of var. (%)		15.9	11.0	5.1	4.4	15.9	24.8	10.3	3.0	NA
	No. of meas.		24	118	38	12	67	66	31	7	0
	Depth (ft)		0-60.0	60.0-65.0	65.0-204.0		204.0-292.0	292.0-425.0	425.0-470.0	470.0-493.6	
	Mean Vs (ft/s)		2240	2840	3360		3440	5640	7380	5430	
RF#18	Median V _s (ft/s)		2300	2830	3540		3300	5740	7360	5160	
(cuttings)	Standard Deviation		316.2	140.5	441.8		716.1	1337.7	1246.7	1619.0	
	Coeff. of var. (%)		14.1	4.9	13.2		20.8	23.7	16.9	29.8	
	No. of meas.		16	3	85		53	81	28	9	

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Table VII-2. Statistics for Suspension Seismic Source-to-Receiver Shear-Wave Velocities by Lithostratigraphic Unit (Continued)

Borehole	Parameter	Fill	Qal	Tmbt1	Tpki	Tpbt5	Tpcm	Tpcpul	Tpcpmn	Tpcpll	Tpcpln
	Depth (ft)		0-120.0	120.0-280.0	280.0-410.0	410.0-420.0	420.0-510.0		635.0-645.2		
	Mean Vs (ft/s)		2460	3330	3390	3730	3360	3890	3590		
RF#19	Median V _s (ft/s)		2440	3340	3410	3640	3270	3770	3600		
(cuttings)	Standard Deviation		339.4	318.7	412.5	238.3	483.2	688.2	55.7		
(cuttings)	Coeff. of var. (%)		13.8	9.6	12.2	6.4	14.4	17.7	1.6		
	No. of meas.		52	98	79	6	55	76	3		
	Depth (ft)	0-28.0	28.0-98.0			98.0-102.0	102.0-127.0	127.0-160.0			
	Mean Vs (ft/s)	2080	2600			2880	3170	3240			
RF#20	Median V _s (ft/s)	2080	2560		· · · · · · · · · · · · · · · · · · ·	2790	3200	3310			
(cuttings)	Standard Deviation	90.7	386.5			314.8	418.5	377.2			
(cuttings)	Coeff. of var. (%)	4.4	14.9			10.9	13.2	11.6			
	No. of meas.	4	43			3	15	15			
	Depth (ft)	0-5.0	5.0-115.0				115.0-165.0	6580.0-192.2			
	Mean Vs (ft/s)	NA NA	1910				2680	2760			
RF#21**	Median V _s (ft/s)	NA NA	1920		i		2670	2750			L
	Standard Deviation	NA NA	298.3				401.6	426.1			
(cuttings)	Coeff. of var. (%)	NA NA	15.6				15.0	15.4			
	No. of meas.	0	58				31	12			
	Depth (ft)		0-80.0	80.0-318.0	318 0-415 0	415.0-438.0	438.0-530.0	530.0-540.0			
	Mean Vs (ft/s)		NA	3710	3560	NA	NA	NA			
RF#22	Median V _s (ft/s)		NA NA	3720	3610	NA.	NA	NA			
(cuttings	Standard Deviation		NA.	118.4	214.1	NA	NA.	NA			
/core)			NA NA	3.2	6.0	NA	NA	NA			
•	Coeff. of var. (%)		0	51	47	0	0	0			
	No. of meas. Depth (ft)	0-12.0	12.0-76.0		76.0-92.0	92.0-95.0	95.0-159.1				
		1300	2040		3150	3110	3600				
RF#23	Mean Vs (ft/s) Median V _s (ft/s)	1300	1730		2990	3110	3550				
		NA	653.8		338.4	339.4	907.1				
(cuttings)	Standard Deviation	NA NA	32.1		10.8	10.9	25.2				
	Coeff. of var. (%)	NA.	32.1		9	2	36	<u> </u>			
	No. of meas.	0-10.0	10.0-30.0		1	——		110.0-230.0	230.0-268.0		
	Depth (ft)		2030			 	3060	3270	4850		
55464	Mean Vs (ft/s)	NA NA	2030		 		2960	3370	4650		
RF#24	Median V _s (ft/s)		NA NA				735.4	778.0	905.6		
(cuttings)	Standard Deviation	NA NA	NA NA	 	·	 	24.0	23.8	18.7		
ĺ	Coeff. of var. (%)	NA NA	1 NA 1			 	48	74	20		
	No. of meas.	0	1 1		<u> </u>	1	1 40			TN: MO0204	

DTN: MO0204SUSPSEIS.001

^{*}Note: The above table appears to be source-to-receiver, not receiver-to-receiver as indicated

APPENDIX A – SEISMIC WAVE VELOCITY

Table VII-2. Statistics for Suspension Seismic Source-to-Receiver Shear-Wave Velocities by Lithostratigraphic Unit (Continued)

Borehole	Parameter	Fill	Qal .	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpul	Tpcpmn	Tpcpii	Tpcpln
	Depth (ft)	0-10.0	10.0-70.0				70.0-125.0	125.0-159.0			
	Mean Vs (ft/s)	NA	NA				2010	2210			
RF#25	Median V _s (ft/s)	NA	NA				1990	2120			
(cuttings)	Standard Deviation	NA	NA				239.7	289.5			
	Coeff. of var. (%)	NA	NA				12.0	13.1			
	No. of meas.	0	0				26	16			
	Depth (ft)	0-14.0	14.0-85.0		85.0-204.0	204.0-211.0	211.0-264.9				
	Mean Vs (ft/s)	NA	2410		3680	4040	3840				
RF#26	Median V _s (ft/s)	NA	2180		3700	4030	3800				
(cuttings)	Standard Deviation	NA	623.0		243.7	591.7	336.4				
, ,	Coeff. of var. (%)	NA	25.8		6.6	14.7	8.8				
	No. of meas.	0	40		73	4	28				
	Depth (ft)	0-5.0	5.0-15.0			,	15.0-70.0	70.0-100.0			
	Mean Vs (ft/s)	NA	NA NA				2970	4450			
RF#28	Median V _s (ft/s)	NA	NA				3010	4710			
(cuttings)	Standard Deviation	NA	NA				202.5	814.5			
	Coeff. of var. (%)	NA	NA				6.8	18.3			
	No. of meas.	0	0				25	15			
	Depth (ft)		0-85.0	85.0-280.0		370.0-380.0					
	Mean Vs (ft/s)		2160	3470	3800	3650	4650				
RF#29	Median V _s (ft/s)		2070	3640	3900	3320	3920				
(cuttings)	Standard Deviation		351.3	576.5	295.7	1203.1	1568.0				
	Coeff. of var. (%)		16.2	16.6	7.8	33.0	33.7				
	No. of meas.		36	119	55	6	14				
	Mean Vs (ft/s)	1920	2040	3390	3440	3510	3300	3970	5460	5230	6790
	Median V _s (ft/s)	2010	2190	3450	3570	3530	3230	3810	5410	5210	7040
All tests	Standard Deviation	356.5	879.7	437.9	663.9	647.0	739.8	1227.4	1516.4	1308.5	1204.4
	Coeff. of var. (%)	18.5	43.1	12.9	19.3	18.4	22.4	31.0	27.8	25.0	17.7
	No. of meas.	5	419	389	489	41	534	575	166	44	107

DTN: MO0204SUSPSEIS.001

^{*} Coefficient of Variation (%) = 100*standard deviation / mean

^{**} In Assumption 4, Section 5, the contact between the Qal and Tpcrn is assumed to be at a depth of 70 feet. This table follows the geologic logs in Attachment I.

APPENDIX A – SEISMIC WAVE VELOCITY

Compression wave velocity as reported in BSC (2002a) (suspension surveys, source-to-receiver)

Table VII-3. Statistics for Suspension Seismic Source-to-Receiver Compression-Wave Velocities by Lithostratigraphic Unit

Borehole	Parameter	Fill	Qal	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpul	Tpcpmn	Tpcpli	Tpcpin
	Depth (ft)	0-12.5	12.5-98.0		98.0-164.4	164.4-169.3	169.3-219.1	219.1-231.5	231.5-286.7	286.7-300.9	300.9-350.1
	Mean Vp (ft/s)	NA NA	NA		NA	NA	NA	NA	NA	NA NA	NA
l, , l	Median Vp (ft/s)	NA	NA		NA						
RF#13 (core)	Standard Deviation	NA	NA		NA	NA	NA	NA	NA	NA	NA NA
	Coeff. of var. (%)*	NA	NA		NA	NA	NA	NA	NA	NA.	NA
	No. of meas.	0	0		0	0	0	0	0	0	0
	Depth (ft)		0-101.8		101.8-192.5	192.5-203.4		275.0-395.0			455.6-550.0
	Mean Vp (ft/s)		5450		6750	6340	5980	5560	8060	10840	11610
	Median Vp (ft/s)		5600		6810	6460	5760	5350	7260	10610	11560
RF#14 (core)	Standard Deviation		632.7		521.8	573.7	1057.8	1428.3	2042.1	1216.0	1493.3
	Coeff. of var. (%)		11.6		7.7	9.0	17.7	25.7	25.3	11.2	12.9
	No. of meas.		31		56	6	39	71	30	7	55
	Depth (ft)	0-6.5			-		6.5-78.0	78.0-196.0	196.0-242.4	242.4-256.6	
	Mean Vp (ft/s)	NA.					6950	8340	9930	9970	12860
	Median Vp (ft/s)	NA					6710	8080	9610	9880	12930
RF#15 (core)	Standard Deviation	NA					936.7	1066.7	1155.6	605.7	792.2
	Coeff, of var. (%)	NA					13.5	12.8	11.6	6.1	6.2
	No. of meas.	0					31	72	28	9	37
	Depth (ft)	0-22.4	22.4-75.7		75.7-133.2	133.2-137.8	137.8-222.0			403.0-422.5	
	Mean Vp (ft/s)	NA	5060		5830	5690	5840	6440	7520	6600	7240
	Median Vp (ft/s)	NA	5050		5800	5690	5800	6340	7410	6740	7630
RF#16 (core)	Standard Deviation	NA	427.3		270.3	91.9	184.7	833.4	1343.8	700.1	1014.9
	Coeff. of var. (%)	NA	8.5		4.6	1.6	3.2	12.9	17.9	10.6	14.0
	No. of meas.	0	30		35	2	52	84	26	12	15
	Depth (ft)		0-92.4	92.4-287.2	287.2-348.4						653.2-667.8
	Mean Vp (ft/s)		5120	6020	6550	7930	7460	7800	10320	10080	NA NA
DE#47 ()	Median Vp (ft/s)		5110	6070	6450	8010	7320	7840	10470	9960	NA
RF#17 (core)	Standard Deviation		342.8	483.7	480.8	492.6	1442.1	1071.3	556.9	317.4	NA
	Coeff. of var. (%)		6.7	8.0	7.3	6.2	19.3	13.7	5.4	3.1	NA
	No. of meas.		24	118	38	12	67	66	31	7	0
	Depth (ft)		0-60.0	60.0-65.0	65.0-204.0		204.0-292.0		425.0-470.0		
	Mean Vp (ft/s)		6280	5960	7900		8620	11280	14960	13420	
RF#18	Median Vp (ft/s)		6210	5930	8010		8060	11200	15260	12920	
(cuttings)	Standard Deviation		368.8	57.7	974.5		1618.3	1897.3	1409.9	1157.8	
	Coeff. of var. (%)		5.9	1.0	12.3		18.8	16.8	9.4	8.6	
	No. of meas.		16	3	85		53	81	28	9	l

DTN: MO0204SUSPSEIS.001

Table VII-3. Statistics for Suspension Seismic Source-to-Receiver Compression-Wave Velocities by Lithostratigraphic Unit (Continued)

Borehole	Parameter	Fill	Qai	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpul	Tpcpmn	Tpcpll	Tpcpln
	Depth (ft)		0-120.0	120.0-280.0	280.0-410.0	410.0-420.0	420.0-510.0				
	Mean Vp (ft/s)		4250	6710	7120	8570	9070	9240	10920		
RF#19	Median Vp (ft/s)		3800	6690	6860	8320	9080	9130	11030		
(cuttings)	Standard Deviation		962.5	736.1	1132.6	1064.5	701.9	928.9	342.7		
(cuttings)	Coeff. of var. (%)		22.7	11.0	15.9	12.4	7.7	10.1	3.1		
	No. of meas.		52	98	79	6	55	76	3		
	Depth (ft)	0-28.0	28.0-98.0			98.0-102.0	102.0-127.0	127.0-160.0			
	Mean Vp (ft/s)	3710	4680			5420	5490	5790			
RF#20	Median Vp (ft/s)	3710	4740			5230	5550	5690			
(cuttings)	Standard Deviation	239.8	813.8			536.8	389.7	581.4			
(cuttings)	Coeff. of var. (%)	6.5	17.4			9.9	7.1	10.0			
	No. of meas.	4	43			3	15	15			
	Depth (ft)	0-5.0	5.0-115.0				115.0-165.0				
	Mean Vp (ft/s)	NA	4280				4960	5190			
RF#21**	Median Vp (ft/s)	NA	4280				4870	5100			
(cuttings)	Standard Deviation	NA	609.4				958.9	529.0			
(outings)	Coeff, of var. (%)	NA	14.2				19.3	10.2			
	No. of meas.	0	58				31	12			
	Depth (ft)		0-80.0	80.0-318.0	318.0-415.0	415.0-438.0	438.0-530.0	530.0-540.0			
	Mean Vp (ft/s)		NA	6780	6510	NA	NA	NA			
RF#22	Median Vp (ft/s)		NA NA	6710	6520	NA	NA	NA			
(cuttings	Standard Deviation		NA	522.8	331.1	NA	NA _	NA			
/core)	Coeff. of var. (%)		NA	7.7	5.1	NA	NA	NA			
	No. of meas.		0	51	47	0	0	0			
	Depth (ft)	0-12.0	12.0-76.0		76.0-92.0	92.0-95.0	95.0-159.1				
	Mean Vp (ft/s)	5470	4320		5710	5640	6540				
RF#23	Median Vp (ft/s)	5470	4230		5270	5640	6560				
(cuttings)	Standard Deviation	NA	1200.4		922.1	410.1	1515.8				
(Coeff. of var. (%)	NA	27.8		16.1	7.3	23.2				
	No. of meas.	1	39		9	2	36				
	Depth (ft)	0-10.0	10.0-30.0				30.0-110.0		230.0-268.0		
	Mean Vp (ft/s)	NA	5390				5800	5860	8610		
RF#24	Median Vp (ft/s)	NA	5390			L	5620	5760	8520		
(cuttings)	Standard Deviation	NA	NA				935.2	1161.9	1057.4		
,	Coeff. of var. (%)	NA	NA				16.1	19.8	12.3		
	No. of meas.	0	1		1		48	74	20	L	L

DTN: MO0204SUSPSEIS.001

APPENDIX A – SEISMIC WAVE VELOCITY

Table VII-3. Statistics for Suspension Seismic Source-to-Receiver Compression-Wave Velocities by Lithostratigraphic Unit (Continued)

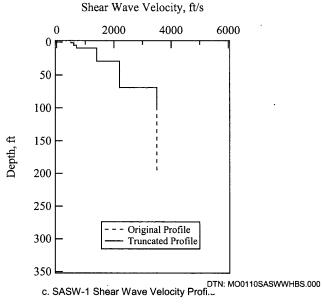
Borehole	Parameter	Fill	Qal	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpui	Tpcpmn	Tpcpll	Tpcpln
	Depth (ft)	0-10.0	10.0-70.0				70.0-125.0	125.0-159.0			
	Mean Vp (ft/s)	NA	NA				3910	4170			
RF#25	Median Vp (ft/s)	NA	NA	1			3930	4030			
(cuttings)	Standard Deviation	NA	NA				453.1	553.7			
	Coeff. of var. (%)	NA	NA				11.6	13.3			
	No. of meas.	0	0				26	16			
	Depth (ft)	0-14.0	14.0-85.0		85.0-204.0	204.0-211.0	211.0-264.9				
	Mean Vp (ft/s)	NA	4340		6290	6940	6890				
RF#26	Median Vp (ft/s)	NA	3840		6240	6660	6840				
(cuttings)	Standard Deviation	NA	1171.2		495.5	1241.1	472.3				
	Coeff. of var. (%)	NA	27.0		7.9	17.9	6.9				
	No. of meas.	0	33		73	4	28				
	Depth (ft)	0-5.0	5.0-15.0				15.0-70.0	70.0-100.0			
	Mean Vp (ft/s)	NA	NA .				5320	7530			
RF#28	Median Vp (ft/s)	NA	NA				5190	7650			
(cuttings)	Standard Deviation	NA	NA				477.3	799.1			
	Coeff. of var. (%)	NA	NA				9.0	10.6			
	No. of meas.	0	0				25	15			
	Depth (ft)		0-85.0	85.0-280.0	280.0-370.0	370.0-380.0	380.0-430.0				
	Mean Vp (ft/s)		4430	7530	9150	6690	9210				
RF#29	Median Vp (ft/s)		4500	7790	9310	5830	8070				
(cuttings)	Standard Deviation		524.7	1593.9	1185.3	1994.0	2866.4				
	Coeff. of var. (%)		11.9	21.2	13.0	29.8	31.1				
	No. of meas.		36	119	55	6	14				
	Mean Vp (ft/s)	4060	4660	6760	7100	7110	6730	7650	9990	9910	11430
	Median Vp (ft/s)	3820	4670	6550	6730	7060	6410	7470	9690	9990	12050
All tests	Standard Deviation	812.9	948.5	1168.9	1258.6	1394.2	1828.7	2310.9	2798.1	2540.1	2167.4
	Coeff. of var. (%)	20.0	20.4	17.3	17.7	19.6	27.2	30.2	28.0	25.6	19.0
	No. of meas.	5	363	389	477	41	529	573	166	44	107

DTN: MO0204SUSPSEIS.001

^{*} Coefficient of Variation (%) = 100*standard deviation / mean

** In Assumption 4, Section 5, the contact between the Qal and Tpcrn is assumed to be at a depth of 70 feet. This table follows the geologic logs in Attachment I.

Shear wave velocity (SASW surveys)



Location: SASW-1

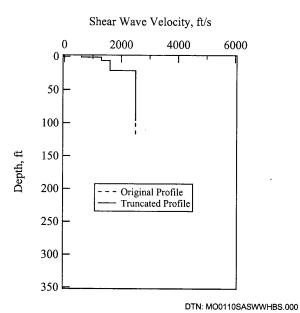
Layer No.	Thickness, ft	P-Wave Velocity, ft/s	S-Wave Velocity, ft/s	Poisson's Ratio***	Mass Density*** pcf
1	1	866	500	0.25	120
2	4	1039	600	0.25	120
3	4	1212	700	0.25	120
4	20	2425	1400	0.25	120
5	40	3810	2200	0.25	120
6	29*	6062	3500	0.25	80
7	102**	6062	3500	0.25	80

Figure IX-1. SASW-1 Results (continued)

DTN: MO0110SASWWHBS.000
* Vs profile truncated at 98 ft based on geological profile showing an offset fault beginning at a depth of approximately

^{**} Additional layering used in matching the theoretical dispersion curve to the complete experimental dispersion curve

^{***} Poisson's ratio and mass density from Wong (2002c, Appendix 1)



c. SASW-2 Shear Wave Velocity Profile

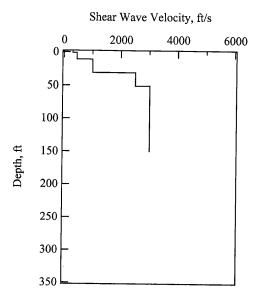
Location: SASW-2

Layer No.	Thickness, ft	P-Wave Velocity, ft/s	S-Wave Velocity, ft/s	Poisson's Ratio***	Mass Density*** pcf
1	1	1039	600	0.25	120
2	5	2252	1300	0.25	120
3	15	2771	1600	0.25	120
4	75*	4850	2500	0.25	120
5	24**	4850	2500	0.25	120

Figure IX-2. SASW-2 Results (continued)

DTN: MO0110SASWWHBS.000

* Vs profile truncated at 96 ft based on geological profile showing an offset fault beginning at a depth of approximately *** Poisson's ratio and mass density from Wong (2002c, Appendix 2)



DTN: MO0110SASWWHBS.000

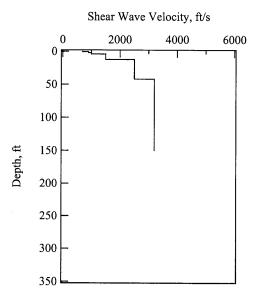
c. SASW-4 Shear Wave Velocity Profile

Location: SASW-4

Layer No.	Thickness, ft	P-Wave Velocity, ft/s	S-Wave Velocity, ft/s	Poisson's Ratio*	Mass Density* pcf
1	11	520	300	0.25	120
2	10	779	450	0.25	120
3	20	1732	1000	0.25	120
4	20	4330	2500	0.25	120
5	100	5196	3000	0.25	80

Figure IX-4. SASW-4 Results (continued)

DTN: MO0110SASWWHBS.000
* Poisson's ratio and mass density from Wong (2002c, Appendix 4)



DTN: MO0110SASWWHBS.000

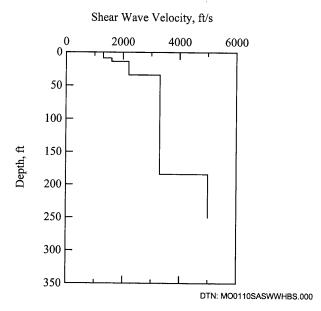
c. SASW-8 Shear Wave Velocity Profile

Location: SASW-8

Layer No.	Thickness, ft	P-Wave Velocity, ft/s	S-Wave Velocity, ft/s	Poisson's Ratio*	Mass Density* pcf
1	0.5	1212	700	0.25	120
2	1.5	1559	900	0.25	120
3	2.0	1732	1000	0.25	120
4	8	2598	1500	0.25	120
5	30	4330	2500	0.25	120
6	108	5543	3200	0.25	80

Figure IX-8. SASW-8 Results (continued)

DTN: MO0110SASWWHBS.000 * Poisson's ratio and mass density from Wong (2002c, Appendix 8)



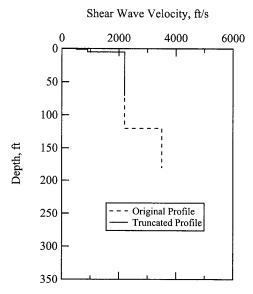
c. SASW-10+37 Shear Wave Velocity Profile

Location: SASW-10+37

Layer No.	Thickness, ft	P-Wave Velocity, ft/s	S-Wave Velocity, ft/s	Poisson's Ratio*	Mass Density*
1	1	2251	1300	0.25	120
2	8	2251	1300	0.25	120
3	55_	2771	1600	0.25	120
4	20	3810	2200	0.25	120
5	150	5716	3300	0.25	80
6	66	8660	5000	0.25	145

Figure IX-10. SASW-10+37 Results (continued)

DTN: MO0110SASWWHBS.000 * Poisson's ratio and mass density from Wong (2002c, Appendix 10)



DTN: MO0110SASWWHBS.000

c. SASW-23 Shear Wave Velocity Profile

Location: SASW-23

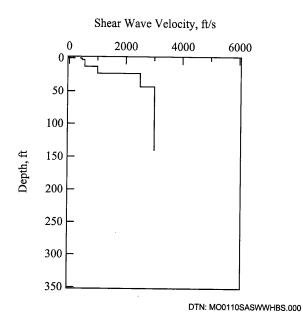
Layer No.	Thickness, ft	P-Wave Velocity, ft/s	S-Wave Velocity, ft/s	Poisson's Ratio***	Mass Density*** pcf
11	1	866	500	0.25	120
2	4	1559	900	0.25	120
3	65*	3810	2200	0.25	120
4	50**	3810	2200	0.25	120
5	60**	6062	3500	0.25	80

Figure IX-23. SASW-23 Results (continued)

DTN: MO0110SASWWHBS.000

* Vs profile truncated at 70 ft based on geological profile showing an offset fault beginning at a depth of approximately 70 ft.

** Additional layering used in matching the theoretical dispersion curve to the complete experimental dispersion curve *** Poisson's ratio and mass density from Wong (2002c, Appendix 23)



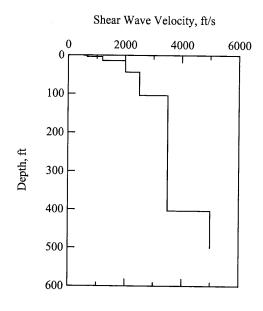
c. SASW-29 Shear Wave Velocity Profile

Location: SASW-29

Layer No.	Thickness,	P-Wave Velocity, ft/s	S-Wave Velocity, ft/s	Poisson's Ratio*	Mass Density*
1	0.5	692	400	0.25	120
_22	2	779	450	0.25	120
3	10	952	550	0.25	120
4	20	1732	1000	0.25	120
5	20	4330	2500	0.25	120
6	88	5196	3000	0.25	80

Figure IX-29. SASW-29 Results (continued)

DTN: MO0110SASWWHBS.000
* Poisson's ratio and mass density from Wong (2002c, Appendix 29)



DTN: MO0110SASWWHBS.000

c. SASW-32+35 Shear Wave Velocity Profile

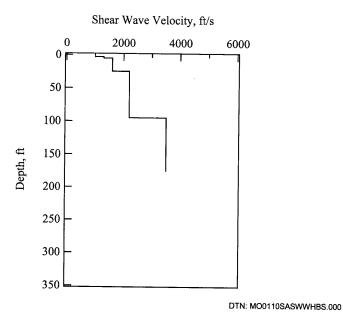
Location: SASW-32+35

Layer No.	Thickness, ft	P-Wave Velocity, ft/s	S-Wave Velocity, ft/s	Poisson's Ratio*	Mass Density* pcf
1	11	953	550	0.25	120
2	3	1126	650	0.25	120
3	10	2079	1200	0.25	120
4	30	3464	2000	0.25	120
5	60	4330	2500	0.25	120
6	300	5543	3500	0.25	80
77	96	8660	5000	0.25	145

DTN: MO0110SASWWHBS.000

Figure IX-32. SASW-32+35 Results (continued)

^{*} Poisson's ratio and mass density from Wong (2002c, Appendix 32)



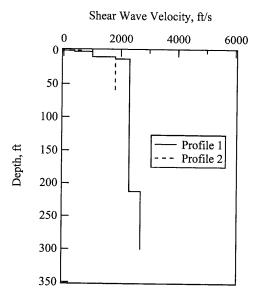
c. SASW-33 Shear Wave Velocity Profile

Location: SASW-33

Layer No.	Thickness, ft	P-Wave Velocity, ft/s	S-Wave Velocity, ft/s	Poisson's Ratio*	Mass Density*
1	3	1732	1000	0.25	120
2	2	2252	1300	0.25	120
3	20	2771	1600	0.25	120
4	70	3810	2200	0.25	120
5	80	6062	3500	0.25	80

Figure IX-33. SASW-33 Results (continued)

DTN: MO0110SASWWHBS.000
* Poisson's ratio and mass density from Wong (2002c, Appendix 33)



DTN: MO0110SASWWHBS.000

c. SASW-34+36 Shear Wave Velocity Profile

Location: SASW-34+36 Profile 1

Layer No.	Thickness, ft	P-Wave Velocity, ft/s	S-Wave Velocity, ft/s	Poisson's Ratio*	Mass Density* pcf
1	1.5	650	375	0.25	120
2	8	1732	1000	0.25	120
3	3	3117	1800	0.25	120
4	200	3983	2300	0.25	120
5	88	4676	2700	0.25	120

Location: SASW-34+36 Profile 2

Layer No.	Thickness, ft	P-Wave Velocity, ft/s	S-Wave Velocity, ft/s	Poisson's Ratio*	Mass Density* pcf
1	11	650	375	0.25	120
2	8	1732	1000	0.25	120
3	51	3117	1800	0.25	120

DTN: MO0110SASWWHBS.000

Figure IX-34. SASW-34+36 Results (continued)

^{*} Poisson's ratio and mass density from Wong (2002c, Appendix 34)

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APPENDIX A - SEISMIC WAVE VELOCITY

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A8.2 Soil Contact Depths (from BSC 2002a, based on DTN: GS030783114233.001)

Table 4. WHB Area Boreholes with Contact Depths and Total Depths in Feet

Borehole	Fill	Qal	Tmbt1	Tpki	Tpbt5	Tpcrn	Tpcpul	Tpcpmn	Tpcpll	Tpcpln	Total Depth (ft)
RF#14 (core)		0.0		101.8	192.5	203.4	275.0	395.0	443.7	455.6	550.0
RF#15 (core)	0.0					6.5	78.0	196.0	242.4	256.6	330.0
RF#16 (core)	0.0	22.4		75.7	133.2	137.8	222.0	360.0	403.0	422.5	452.8
RF#17 (core)		0.0	92.4	287.2	348.4	368.9	478.0	587.3	637.6	653.2	667.8
RF#18 (cuttings)		0	60	65		204	292	425	470		493.6
RF#19 (cuttings)		0	120	280	410	420	510	635			645.2
RF#20 (cuttings)	0	28			98	102	127				160.0
RF#21 (cuttings)	0	5				115 ⁽⁵⁾	165				192.2
RF#22 (cuttings/core)		0	80	318	415	438	530				540.6
RF#23 (cuttings)	0	12		76	92	95					159.1
RF#24 (cuttings)	0	10				30	110	230			268.0
RF#25 (cuttings)	0	10				70	125				159.0
RF#26 (cuttings)	0	14		85	204	211					264.9
RF#28 (cuttings)	0	5				15	70				99.8
RF#29 (cuttings)		0	85	280	370	380					430.0

^{*}Per Assumptions given in BSC 2002a and BSC 2002b, the following changes were implemented for the calculation herein:

- RF#20 Qal contact depth at 9ft
- RF#21 Qal contact depth at 70ft

Table 5. Revised Contact Depths and Total Depths in Feet in Borehole RF#13

	Borehole	Fill	Qal	Tmbt1	Tpki	Tpbt	Tpcrn	Трсри	Tpcpmn	Tpcpli	Tpcpln	Total
						5		.				Depth
L	RF#13 (cored)	0.0	12.5		98.0	164.4	169.3	219.1	231.5	286.7	300.9	350.1

Notes: Contacts are given as the depths in feet to the tops of the units.

A blank cell means that the unit was not encountered.

(per GS030783114233.001)

A8.3 EXCEL Spreadsheets

Attached are the EXCEL spreadsheets used to average the alluvium, existing fill, and rock materials for the analysis contained herein.

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole # 13 Redpath (Downhole) UT Austin (SASW)

Qal	Material	Depth	Thickness, d		Shear W	/ave Vel	ocity, Vs (1	ft/s)			ession Wa	
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Fill	1	1				500	500				
	Fill	3	2				600	1200				
	Fill	5	2	750	1500		600	1200		1455	2910	
	Fill	9	4	750	3000		700	2800		1455	5820	
	Fill	10	1	750	750		1400	1400		3405	3405	
	Fill	12.5	2.5	1355	3387.5	909	1400	3500	848	3405	8512.5	2173
	Qal	15	2.5	1355	3387.5		1400	3500		3405	8512.5	
	Qal	25	10	1355	13550		1400	14000		3405	34050	
	Qal	26	1	2030	2030		1400	1400		3405	3405	
	Qal	29	3	2030	6090		1400	4200		4685	14055	
2	Qal	30	1	2030	2030	1580	2200	2200	1453	4685	4685	3746
3	Qal	60	30	2030	60900	2030	2200	66000	2200	4685	140550	4685
	Qal	69	9	2030	18270		2200	19800		4685	42165	
	Qal	80	11	2030	22330		3500	38500		4685	51535	
4	Qal	98	18	2740	49320	2366	3500	63000	3192	4685	84330	4685
	Tpki	164.4	66.4	2740	181936	2740	3500	232400	3500	4685	311084	4685
	Tpbt5	169.3	4.9	2740	13426	2740	3500	17150	3500	4685	22956.5	4685
	Tpcrn	200	30.7	2740	84118		3500	107450	3500	4685	143830	
	Tpcrn	219.1	19.1	2740	52334	2740				4685	89483.5	4685
	Tpcpul	226	6.9	2740	18906					4685	32326.5	
	Tpcpul	230	4	2740	10960					9335	37340	
	Tpcpul	231.5	1.5	5800	8700	3110				9335	14002.5	6748
	Tpcpmn	286.7	55.2	5800	320160	5800				9335	515292	9335
	Tpcpll	300.9	14.2	5800	82360	5800				9335	132557	9335
	Tpcpln	345	44.1	5800	255780	5800				9335	411674	9335
	Tpcpln	350.1	5.1									
				ALL ROO UPPER ROO LOWER ROO		4080 2774 5800			3500 3500 NA			6785 4877 9335

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole # 13 GEOVision (Downhole)

Qal	Material	Depth	Thickness, d		Shear V	Vave Vel	ocity, Vs (f	t/s)			ession Wa ity, Vp (ft/	
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
		0					NA					
	Fill	12.5	12.5	1090	13625	1090				2110	26375	2110
1	Qal	15	2.5	1090	2725					2110	5275	
2	Qal	30	15	1960	29400	1960				3700	55500	3700
	Qal	36	6	1960	11760					3700	22200	
3	Qal	60	24	2490	59760	2384				3970	95280	3916
4	Qal	98	38	2490	94620	2490				3970	150860	3970
	Tpki	99	1	2490	2490					3970	3970	
	Tpki	100	1	2810	2810					4900	4900	
	Tpki	164.4	64.4	2810	180964	2805				4900	315560	4886
	Tpbt5	169.3	4.9	2810	13769	2810				4900	24010	4900
	Tpcrn	215	45.7	2810	128417					4900	223930	
	Tpcrn	219.1	4.1	6490	26609	3113				11180	45838	5417
	Tpcpul	231.5	12.4	6490	80476	6490				11180	138632	11180
	Tpcpmn	286.7	55.2	6490	358248	6490				11180	617136	11180
	Tpcpll	300.9	14.2	6490	92158	6490				11180	158756	11180
	Tpcpln	345	44.1	6490	286209	6490				11180	493038	11180
	Tpcpln	350.1	5.1									
				ALL ROO UPPER ROO LOWER ROO		4746 3262 6490			NA NA NA			8201 5669 11180

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole # 14 Redpath (Downhole)

Qal	Material	Depth	Thickness, d		Shear V	/ave Vel	ocity, Vs (f	t/s)			ession Wa ity, Vp (ft/s	
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Qal	3					NA					
	Qal	5	2	1240	2480							
	Qal	6	1	1240	1240							
	Qal	12	6	1240	7440					2530	15180	
1	Qal	15	3	1240	3720	1240				3805	11415	2955
2	Qal	30	15	1700	25500	1700				3805	57075	3805
	Qal	38	8	1700	13600					3805	30440	
3	Qal	60	22	2375	52250	2195				4300	94600	4168
4	Qal	100	40	2375	95000	2375				4300	172000	4300
	Qal	101.8	1.8	2375	4275					4300	7740	
	Tpki	110	8.2	2375	19475					4300	35260	
	Tpki	114	4	2375	9500					5900	23600	
	Tpki	165	51	3390	172890					5900	300900	
	Tpki	192.5	27.5	2640	72600	3091				5900	162250	5900
	Tpbt5	203.4	10.9	2640	28776	2640				5900	64310	5900
	Tpcrn	275	71.6	2640	189024	2640				5900	422440	5900
	Tpcpul	304	29	2640	76560					5900	171100	
	Tpcpul	305	1	2640	2640					7500	7500	
	Tpcpul	395	90	5000	450000	4410				7500	675000	7113
	Tpcpmn	420	25	5000	125000					7500	187500	
	Tpcpmn	443.7	23.7	5000	118500	5000				11000	260700	9203
	Tpcpll	455.6	11.9	5000	59500	5000				11000	130900	11000
	Tpcpln	520	64.4	5000	322000	5000				11000	708400	11000
	Tpcpln	550	30									
				ALL ROO	CK TOTAL	3937 3484			NA NA			7532 6352
				LOWER ROC		5000			NA			10300

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole #	15	Redpath	(Downhole)
		UT Austin	(SASW)

Qal	Material	Depth	Thickness, d		Shear W			Compression Wave Velocity, Vp (ft/s)				
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Fill	1	1				1300	1300				
	Fill	3	2				1300	2600				
	Fill	5	2	1935	3870		1300	2600		3215	6430	
	Fill	6.5	1.5	1935	2902.5	1935	1300	1950	1300	3215	4822.5	3215
	Tpcrn	9	2.5	1935	4837.5		1300	3250		3215	8037.5	
	Tpcrn	14	5	1935	9675		1600	8000		3215	16075	
	Tpcrn	15	1	1935	1935		2200	2200		3215	3215	
	Tpcrn	18	3	1935	5805		2200	6600		3215	9645	
	Tpcrn	30	12	1935	23220		2200	26400		3815	45780	
	Tpcrn	34	4	1935	7740		2200	8800		3815	15260	
	Tpcrn	38	4	1935	7740		3300	13200		3815	15260	
	Tpcrn	39	1	2700	2700		3300	3300		3815	3815	
	Tpcrn	60	21	2700	56700		3300	69300		4600	96600	
	Tpcrn	78	18	2700	48600	2363	3300	59400	2803	4600	82800	4147
	Tpcpul	100	22	2700	59400		3300	72600		4600	101200	
	Tpcpul	122	22	2700	59400	2700	3300	72600		4600	101200	
	Tpcpul	133	11	3380	37180		3300	36300		4600	50600	
	Tpcpul	184	51	3380	172380		3300	168300		9850	502350	
	Tpcpul	196	12	3380	40560	3126	5000	60000	3473	9850	118200	7403
	Tpcpmn	210	14	3380	47320		5000	70000		9850	137900	
	Tpcpmn	230	20	3380	67600		5000	100000		14000	280000	
	Tpcpmn	242.4	12.4	5900	73160	4053	5000	62000	5000	14000	173600	1274
	Tpcpll	250	7.6	5900	44840		5000	38000	5000	14000	106400	
	Tpcpll	256.6	6.6	5900	38940	5900				14000	92400	1400
	Tpcpln	320	63.4	5900	374060	5900				14000	887600	1400
	Tpcpln	330	10									
					K TOTAL	3776			3615			908
				UPPER ROC		2838			3220			6174
				LOWER ROC	K TOTAL	5209			5000			1353

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole #	16	Redpath	(Downhole)
		UT Austin	(SASW)

Qal	Material	Depth	Thickness, d		Shear W		Compression Wave Velocity, Vp (ft/s)					
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Fill	0.5	0.5				400	200				
	Fill	2.5	2				450	900				
	Fill	3	0.5				550	275				
	Fill	5	2	655	1310		550	1100		1590	3180	
	Fill	12.5	7.5	655	4912.5		550	4125		1590	11925	
	Fill	15	2.5	655	1637.5		1000	2500		1590	3975	
	Fill	22.4	7.4	1130	8362	836	1000	7400	737	3075	22755	2156
	Qal	24	1.6	1130	1808		1000	1600		3075	4920	
2	Qal	30	6	1640	9840	1533	1000	6000	1000	3075	18450	3075
	Qal	32.5	2.5	1640	4100		1000	2500		3075	7687.5	
	Qal	50	17.5	1640	28700		2500	43750		3075	53812.5	
	Qal	52.5	2.5	2800	7000		2500	6250		4850	12125	
3	Qal	60	7.5	2800	21000	2027	3000	22500	2500	4850	36375	3667
	Qal	75.7	15.7	2800	43960		3000	47100		4850	76145	
	Tpki	133.2	57.5	2800	161000	2800	3000	172500	3000	4850	278875	4850
	Tpbt5	137.8	4.6	2800	12880	2800	3000	13800	3000	4850	22310	4850
	Tpcrn	140.5	2.7	2800	7560		3000	8100	3000	4850	13095	
	Tpcrn	222	81.5	2800	228200	2800				4850	395275	4850
	Tpcpul	280	58	2800	162400					4850	281300	
	Tpcpul	296	16	2800	44800					6600	105600	
	Tpcpul	360	64	3540	226560	3143				6600	422400	5864
	Tpcpmn	376	16	3540	56640					6600	105600	
	Tpcpmn	403	27	7000	189000	5713				10000	270000	8735
	Tpcpll	422.5	19.5	7000	136500	7000				10000	195000	10000
	Tpcpln	445	22.5	7000	157500	7000				10000	225000	10000
	Tpcpln	452.8	7.8									
				ALL ROO UPPER ROO LOWER ROO		3745 2967 6349			3000 3000 NA			6267 5342 9360

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole # 17 GEOVision (Downhole) UT Austin (SASW)

Qal	Material	Depth	Thickness, d		Shear W	/ave Vel		•	ession Wa ity, Vp (ft/			
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Qal	1.5	1.5	1210	1815		375	562.5		2510	3765	
	Qal	5	3.5	1210	4235		1000	3500		2510	8785	
	Qal	9.5	4.5	1210	5445		1000	4500		2510	11295	
	Qal	12.5	3	1210	3630		1800	5400		2510	7530	
1	Qal	15	2.5	1210	3025	1210	2300	5750	1565	2510	6275	2510
2	Qal	30	15	1880	28200	1880	2300	34500	2300	4160	62400	4160
3	Qal	60	30	2490	74700	2490	2300	69000	2300	4060	121800	4060
4	Qal	92.4	32.4	2490	80676	2490	2300	74520	2300	4060	131544	4060
	Tmbt1	100	7.6	2490	18924		2300	17480		4060	30856	
	Tmbt1	212.5	112.5	3160	355500		2300	258750		5580	627750	
	Tmbt1	287.2	74.7	3160	236052	3134	2700	201690	2453	5580	416826	5521
	Tpki	300.5	13.3	3160	42028		2700	35910	2700	5580	74214	
	Tpki	348.4	47.9	3160	151364	3160				5580	267282	5580
	Tpbt5	368.9	20.5	3160	64780	3160				5580	114390	5580
	Tpcrn	400	31.1	3160	98276					5580	173538	
	Tpcrn	478	78	3890	303420	3890				7190	560820	6731
	Tpcpul	500	22	3890	85580					7190	158180	
	Tpcpul	587.3	87.3	4520	394596	4393				10210	891333	9602
	Tpcpmn	620	32.7	4520	147804	4520				10210	333867	10210
	Tpcpmn	637.6	17.6									
	Tpcpll	653.2	15.6									
	TpcpIn	667.8	14.6	ALL ROO	CK TOTAL	3598 3537			2469 2469			6916 6699
				LOWER ROC		4520			NA NA			10210

- 1. Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer
- 2. Qal is divided into 4 intervals (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole # 17

UT Austin (SASW)

Qal	Material	Depth	Thickness, d		Shear W			ession Wa ity, Vp (ft/s				
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Qal	1	1	NA			375	375				
	Qal	5	4				1000	4000				
	Qal	9	4				1000	4000				
1	Qal	15	6				1800	10800	1480			
2	Qal	30	15				1800	27000	1800			
3	Qal	60	30				1800	54000	1800			
4	Qal	92.4	32.4									
	Tmbt1	100	7.6									
	Tmbt1	287.2	187.2									
	Tpki	348.4	61.2									
	Tpbt5	368.9	20.5									
	Tpcrn	478	109.1									
	Tpcpul	587.3	109.3									
	Tpcpmn	637.6	50.3									
	Tpcpll	653.2	15.6									
	Tpcpln	667.8	14.6									
				ALL ROO UPPER ROO LOWER ROO		NA NA NA			NA NA NA			NA NA NA

- 1. Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer
- 2. Qal is divided into 4 intervals (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole # 18 Redpath (Downhole)

Qal	Material	Depth	Thickness, d		Shear Wave Velocity, Vs (ft/s)						ession Wa ity, Vp (ft/s	
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Qal	3	3				NA					
	Qal	5	2	1435	2870					3305	6610	
1	Qal	15	10	1435	14350	1435				3305	33050	3305
	Qal	24	9	1435	12915					3305	29745	
2	Qal	30	6	1670	10020	1529				3305	19830	3305
	Qal	48	18	1670	30060					3305	59490	
3	Qal	60	12	2900	34800	2162				4600	55200	3823
	Tmbt1	65	5	2900	14500	2900				4600	23000	4600
	Tpki	78	13	2900	37700					4600	59800	
	Tpki	100	22	3860	84920					5850	128700	
	Tpki	204	104	3860	401440	3770				5850	608400	5733
	Tpcrn	220	16	3860	61760					5850	93600	
	Tpcrn	250	30	2400	72000					5850	175500	
	Tpcrn	290	40	4200	168000					5850	234000	
	Tpcrn	292	2	4200	8400	3525				7200	14400	5881
	Tpcpul	390	98	4200	411600	4200				7200	705600	
	Tpcpul	425	35	4200	147000	4200				8300	290500	7489
	Tpcpmn	470	45	4200	189000	4200				8300	373500	8300
	Tpcpll	480	10	4200	42000	4200				8300	83000	8300
	Tpcpll	485	5							8300		
	Tpcpll	493.6	8.6									
				ALL ROO	K TOTAL	3901			NA			6643
				UPPER ROC		3856			NA			6393
				LOWER ROO	K TOTAL	4200			NA			8300
				•						-		

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole # 19 Redpath (Downhole)

Qal	Material	Depth	Thickness, d		Shear W	/ave Vel			ession Wa ity, Vp (ft/s			
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Qal	3	3				NA					
	Qal	5	2	1285	2570					1710	3420	
	Qal	9	4	1285	5140					1710	6840	
1	Qal	15	6	1285	7710	1285				3440	20640	2748
	Qal	18	3	1285	3855					3440	10320	
2	Qal	30	12	1810	21720	1705				3440	41280	3440
	Qal	39	9	1810	16290					3440	30960	
3	Qal	60	21	2305	48405	2157				3950	82950	3797
	Qal	96	36	2305	82980					3950	142200	
4	Qal	100	4	2740	10960	2349				3950	15800	3950
	Qal	104	4	2740	10960					3950	15800	
	Qal	120	16	2740	43840					5000	80000	
	Tmbt1	280	160	2740	438400	2740				5000	800000	5000
	Tpki	282	2	2740	5480					5000	10000	
	Tpki	294	12	3780	45360					5000	60000	
	Tpki	410	116	3780	438480	3780				6350	736600	6205
	Tpbt5	420	10	3780	37800	3780				6350	63500	6350
	Tpcrn	510	90	3780	340200	3780				6350	571500	6350
	Tpcpul	550	40	3780	151200					6350	254000	
	Tpcpul	635	85	4250	361250	4100				6350	539750	6350
	Tpcpmn	640	5	4250	21250	4250				6350	31750	6350
	Tpcpmn	645.2	5.2									
					CK TOTAL	3537			NA			5898
				UPPER ROC		3530			NA			5894
				LOWER ROO	K TOTAL	4250			NA			6350
										1		

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole # 20 Redpath (Downhole)

Qal	Material	Depth	Thickness, d		Shear Wave Velocity, Vs (ft/s)						ession Wa ity, Vp (ft/s	
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Fill	3	3				NA					
	Fill	5	2	1200	2400					1935	3870	
	Fill	9	4	1200	4800	1200				1935	7740	1935
	Qal	13	4	1200	4800					1935	7740	
1	Qal	15	2	1200	2400	1200				3540	7080	2470
	Qal	24	9	1200	10800					3540	31860	
2	Qal	30	6	2020	12120	1528				3540	21240	3540
3	Qal	60	30	2020	60600	2020				3540	106200	3540
	Qal	70	10	2020	20200					3540	35400	
4	Qal	98	28	2800	78400	2595				4320	120960	4115
1	Tpbt5	100	2	2800	5600					4320	8640	
	Tpbt5	102	2	2800	5600	2800				4320	8640	4320
	Tpcrn	127	25	2800	70000	2800				4320	108000	4320
	Tpcpul	155	28	2800	78400	2800				4320	120960	4320
	Tpcpul	160	5									
				ALL ROC UPPER ROC LOWER ROC		2800 2800 NA			NA NA NA			4320 4320 NA

- 1. Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer
- 2. Qal is divided into 4 intervals (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole #	21	Redpath	(Downhole)
		UT Austin	(SASW)

Qal	Material	Depth	Thickness, d		Shear V	ave Vel	ocity, Vs (f	t/s)			ession Wa ity, Vp (ft/s	
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Ανς
	Fill	1	1				600	600				
	Fill	3	2				1300	2600				
	Fill	5	2	1310	2620	1310	1300	2600	1160	2845	5690	284
	Qal	6	1	1310	1310		1300	1300		2845	2845	
1	Qal	15	9	1310	11790	1310	1600	14400	1570	2845	25605	284
	Qal	20	5	1310	6550		1600	8000		2845	14225	
	Qal	21	1	1930	1930		1600	1600		2845	2845	
2	Qal	30	9	1930	17370	1723	2500	22500	2140	2845	25605	284
	Qal	57	27	1930	52110		2500	67500		2845	76815	
3	Qal	60	3	1930	5790	1930	2500	7500	2500	3900	11700	295
	Qal	70	10	1930	19300		2500	25000		3900	39000	
	Tpcrn	84	14	1930	27020		2500	35000		3900	54600	
	Tpcrn	96	12	2500	30000		2500	30000		3900	46800	
	Tpcrn	100	4	2500	10000		2500	10000		3900	15600	
	Tpcrn	120	20	2500	50000		2500	50000	2500	3900	78000	
	Tpcrn	165	45	2500	112500	2500				4850	218250	435
	Tpcpul	185	20	2500	50000	2500				4850	97000	485
	Tpcpul	192.2	7.2									
				ALL ROO UPPER ROO LOWER ROO		2431 2431 NA			2500 2500 NA			443 443 NA

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

22	Redpath	(Downhole)
	UT Austin	(SASW)
	22	22 Redpath UT Austin

Qal	Material	Depth	Thickness, d		Shear W		Compression Wave Velocity, Vp (ft/s)					
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Qal	1	1				500	500				
	Qal	3	2				900	1800				
	Qal	5	2	1465	2930		900	1800		2445	4890	
1	Qal	15	10	1465	14650	1465	2200	22000	2200	2445	24450	2445
	Qal	21	6	1465	8790		2200	13200		2445	14670	
	Qal	24	3	2200	6600		2200	6600		2445	7335	
2	Qal	30	6	2200	13200	1906	2200	13200	2200	4185	25110	3141
3	Qal	60	30	2200	66000	2200	2200	66000	2200	4185	125550	4185
	Qal	70	10	2200	22000		2200	22000		4185	41850	
4	Qal	80	10	2200	22000	2200	2200	22000	2200	4185	41850	4185
	Tmbt1	83	3	2200	6600		2200	6600		4185	12555	
	Tmbt1	87	4	3540	14160		2200	8800		4185	16740	
	Tmbt1	100	13	3540	46020		2200	28600		5560	72280	
	Tmbt1	120	20	3540	70800		2200	44000		5560	111200	
	Tmbt1	175	55	3540	194700		3500	192500		5560	305800	
	Tmbt1	180	5	1400	7000	1400	3500	17500	2980	5560	27800	
	Tmbt1	192	12	1400	16800					5560	66720	
	Tmbt1	318	126	3500	441000	3349				5560	700560	5520
	Tpki	415	97	3500	339500	3393				5560	539320	5560
	Tpbt5	438	23	3500	80500	3500				5560	127880	5560
	Tpcrn	500	62	3500	217000	3500				5560	344720	
	Tpcrn	505	5							5560	27800	5560
	Tpcrn	530	25									
	Tpcpul	540.6	10.6									
				ALL ROO UPPER ROO LOWER ROO		3414 3414 NA			2980 2980 NA			5537 5537 NA

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole # 23 Redpath (Downhole) UT Austin (SASW)

Qal	Material	Depth	Thickness, d		Shear W	Vave Vel	ocity, Vs (1	ft/s)		Compression Wave Velocity, Vp (ft/s)		
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Fill	1	1				550	550				
	Fill	3	2				650	1300				
	Fill	4	1	690	690		650	650		2000	2000	
	Fill	5	1	690	690		1200	1200		2000	2000	
	Fill	9	4	690	2760		1200	4800		2000	8000	
	Fill	12	3	1565	4695	982	1200	3600	1008.3	2000	6000	200
	Qal	14	2	1565	3130		1200	2400		2000	4000	
	Qal	15	1	1565	1565		2000	2000		2000	2000	
	Qal	18	3	1565	4695		2000	6000		2000	6000	
	Qal	21	3	1565	4695		2000	6000		3765	11295	
2	Qal	30	9	2100	18900	1886	2000	18000	2000	3765	33885	341
	Qal	44	14	2100	29400		2000	28000		3765	52710	
3	Qal	60	16	2100	33600	2100	2500	40000	2267	3765	60240	376
	Qal	72	12	2100	25200		2500	30000		3765	45180	
	Qal	76	4	2865	11460		2500	10000		4700	18800	
	Tpki	92	16	2865	45840	2865	2500	40000	2500	4700	75200	470
	Tpbt5	95	3	2865	8595	2865	2500	7500	2500	4700	14100	470
	Tpcrn	100	5	2865	14325		2500	12500		4700	23500	
	Tpcrn	104	4	2865	11460		2500	10000		4700	18800	
	Tpcrn	110	6	2865	17190		3500	21000		4700	28200	
	Tpcrn	120	10	3600	36000		3500	35000		4700	47000	
	Tpcrn	155	35	3600	126000	3416	3500	122500		5500	192500	516
	Tpcrn	159.1	4.1				3500	14350	3360			
	•	404	244.9				3500	857150				
		500	96				5000	480000				
				ALL ROO UPPER ROO LOWER ROO		3284 3284 NA			3163 3163 NA			505 505 NA

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole # 24 Redpath (Downhole)

Qal	Material	Depth	Thickness, d		Shear W	Compression Wave Velocity, Vp (ft/s)						
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Fill	3	3				NA					
	Fill	5	2	1195	2390					1425	2850	
	Fill	10	5	1195	5975	1195				1425	7125	1425
	Qal	12	2	1195	2390					1425	2850	
1	Qal	15	3	1195	3585	1195				2785	8355	2241
	Qal	18	3	1195	3585					2785	8355	
2	Qal	30	12	1535	18420	1467				2785	33420	2785
	Tpcrn	33	3	1535	4605					2785	8355	
	Tpcrn	60	27	2070	55890					4960	133920	
	Tpcrn	100	40	2070	82800					4960	198400	
	Tpcrn	110	10	2070	20700	2050				4960	49600	4878
	Tpcpul	230	120	2070	248400	2070				4960	595200	4960
	Tpcpmn	260	30	2070	62100	2070				4960	148800	4960
	Tpcpmn	268	8									
				ALL ROO UPPER ROO LOWER ROO		2062			NA NA NA			4932 4927 4960

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

26

Redpath

(Downhole)

Borehole #

APPENDIX A - SEISMIC WAVE VELOCITY

			UT Austin	(SASW)								
Qal	Material	Depth	Thickness, d		Shear V		Compression Wave Velocity, Vp (ft/s)					
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Fill	1	1				300	300				
	Fill	3	2				450	900				
	Fill	5	2	465	930		450	900		840	1680	
	Fill	10	5	465	2325		450	2250		840	4200	
	Fill	11	1	465	465		450	450		4115	4115	
	Fill	12	1	465	465		1000	1000		4115	4115	
	Fill	14	2	1745	3490	698	1000	2000	557	4115	8230	2031
	Qal	15	1	1745	1745		1000	1000		4115	4115	
2	Qal	30	15	1745	26175	1745	1000	15000	1000	4115	61725	4115
	Qal	31	1	1745	1745		1000	1000		4115	4115	
	Qal	46	15	1745	26175		2500	37500		4115	61725	
	Qal	51	5	2550	12750		2500	12500		4115	20575	
3	Qal	60	9	2550	22950	2121	3000	27000	2600	4115	37035	4115
4	Qal	85	25	2550	63750	2550	3000	75000	3000	4115	102875	4115
	Tpki	95	10	2550	25500		3000	30000		4115	41150	
	Tpki	100	5	3780	18900	3780	3000	15000		7030	35150	
	Tpki	101	1	3780	3780		3000	3000	3000	7030	7030	
	Tpki	140	39	3780	147420					7030	274170	
	Tpki	204	64	3780	241920	3677				5750	368000	6097
	Tpbt5	211	7	3780	26460	3780				5750	40250	5750
	Tpcrn	260	49	3780	185220	3780				5750	281750	5750
	Tpcrn	264.9	4.9									

3000

3000

NA

5986

5986

NA

Notes:

ALL ROCK TOTAL 3710

UPPER ROCK TOTAL 3710

LOWER ROCK TOTAL NA

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

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APPENDIX A - SEISMIC WAVE VELOCITY

Borehole # 25 Redpath (Downhole)

Qal	Material	Depth	Thickness, d		Shear V	Compression Wave Velocity, Vp (ft/s)						
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Fill	3	3				NA					
	Fill	5	2	1645	3290					2710	5420	
	Fill	10	5	1645	8225	1645				2710	13550	2710
1	Qal	15	5	1645	8225	1645				2710	13550	2710
2	Qal	30	15	1645	24675	1645				2710	40650	2710
	Qal	37	7	1645	11515					2710	18970	
	Qal	41	4	2940	11760					2710	10840	
3	Qal	60	19	2940	55860	2638				4840	91960	4059
	Qal	70	10	2940	29400					4840	48400	
	Tpcrn	86	16	2940	47040					4840	77440	
	Tpcrn	100	14	2100	29400					3400	47600	
	Tpcrn	105	5	2100	10500					3400	17000	
	Tpcrn	125	20	2100	42000	2344				4800	96000	4328
	Tpcpul	155	30	2100	63000	2100				4800	144000	4800
	Tpcpul	159	4									
				ALL ROO UPPER ROO LOWER ROO					NA NA NA			449! 449! NA

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole :	#	28	Redpath UT Austin	(Downhole) (SASW)								
Qal	Material	Depth	Thickness, d		Shear W	ave Vel	ocity, Vs (1	ft/s)		Compr Veloc		
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Fill	0.5	0.5				700	350				
	Fill	2	1.5				900	1350				
	Fill	3	1				1000	1000				
	Fill	4	1	1305	1305		1000	1000		3995	3995	
	Fill	5	1	1305	1305	1305	1500	1500	1040	3995	3995	3995
	Qal	10	5	1305	6525		1500	7500		3995	19975	
	Qal	12	2	1980	3960		1500	3000		3995	7990	
1	Qal	15	3	1980	5940	1643	2500	7500	1800	3995	11985	3995
	Tpcrn	30	15	1980	29700		2500	37500		3995	59925	
	Tpcrn	39	9	1980	17820		2500	22500		3995	35955	
	Tpcrn	42	3	3300	9900		2500	7500		5640	16920	
	Tpcrn	60	18	3300	59400		3200	57600		5640	101520	
	Tpcrn	70	10	3300	33000	2724	3200	32000	2856	5640	56400	4922
'	Tpcpul	95	25	3300	82500	3300	3200	80000		5640	141000	
	Tpcpul	96	1				3200	3200		5640	5640	5640
	Tpcpul	99.8	3.8				3200	12160	3200			
		150	50.2				3200	160640				
				ALL ROO UPPER ROO LOWER ROO		2904 2904 NA			2977 2977 NA			5147 5147 NA

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX A - SEISMIC WAVE VELOCITY

Borehole :	#	29	Redpath UT Austin	(Downhole) (SASW)								
Qal	Material	Depth	Thickness, d		Shear V	Vave Vel	ocity, Vs (f	t/s)		Compression Wave Velocity, Vp (ft/s)		
Interval		(ft)	(ft)	Downhole	Vs x d	Avg	SASW	Vs x d	Avg	Downhole	Vs x d	Avg
	Qal	3	3									
	Qal	5	2	1660	3320					2875	5750	
1	Qal	15	10	1660	16600	1660				2875	28750	2875
2	Qal	30	15	1660	24900	1660				2875	43125	2875
	Qal	33	3	1660	4980					2875	8625	
3	Qal	60	27	2170	58590	2119				3675	99225	3595
	Qal	75	15	2170	32550					3675	55125	
4	Qal	85	10	2560	25600	2326				4500	45000	4005
	Tmbt1	100	15	2560	38400					4500	67500	
	Tmbt1	135	35	2560	89600					4500	157500	
	Tmbt1	138	3	2560	7680					6040	18120	
	Tmbt1	230	92	3320	305440					6040	555680	
	Tmbt1	280	50	3800	190000	3237				6040	302000	5645
	Tpki	370	90	3800	342000	3800				6040	543600	6040
	Tpbt5	380	10	3800	38000	3800				6040	60400	6040
	Tpcrn	405	25	3800	95000	3800				6040	151000	6040
	Tpcrn	430	25									
				ALL ROO UPPER ROO LOWER ROO		3457 3457 NA			NA NA NA			5799 5799 NA

^{1.} Average Seismic Velocity = Sum of (Velocity x Thickness) / Sum of Thickness for each soil/rock layer

^{2.} Qal is divided into 4 intervals - (1) 0-15', (2) 15-30', (3) 30-60', and (4) 60-100'

APPENDIX B - BEARING CAPACITY AND SETTLEMENT

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	Objective

B1 Objective

This calculation documents the alluvium bearing capacity and short-term settlement analyses for shallow footings and mat foundations at the surface facilities site area at the Yucca Mountain Project (YMP) site.

Design charts for allowable for foundation pressure for square and strip footings are provided. The recommended foundation pressures consider maximum allowable bearing capacity and maximum permissible foundation settlement.

Short-term settlement evaluations under the center and corner of mat foundations are also considered in these analyses.

B2 Inputs

The following input data is required to perform the analyses:

B2.1 Foundation Geometry

Footings with widths ranging from 2 to 30 feet and foundation embedment depths of 2, 4, and 6-feet are considered in the analyses for bearing capacity and settlement analyses of shallow footings.

Footing widths

$B_0 := 2ft$	Minimum f	ooting	width

 $\Delta B := 0.1 \text{ft}$ Footing width increment

B_f:= 30ft Maximum footing width

 $B_1 := B_0 + \Delta B$

 $B := B_0, B_1 ... B_f$ Footing width range

Embedment depths

d_f:= 2⋅ft, 4ft.. 6ft Depth of embedment range

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APPENDIX B - BEARING CAPACITY AND SETTLEMENT

A square 500 feet by 450 feet mat is considered in the bearing capacity and settlement analyses for mat foundations.

B2.2 Allowable Settlements

Maximum footing and mat foundation settlements of ½ and 1 inch are are considered in this calculation. A 300-year lifetime for the foundations is used to estimate long-term settlements.

$\delta_{\text{max}} := 0.5 \text{in}, 0.75 \text{in} 6.00 \text{in}$	Maximum settlement for calculations.
t := 100·year	Lifetime of structure for long-term settlement estimate (BSC 2004, Section 2.3.1)

B2.3 Soil Stratigraphy and Parameters

Based on BSC (2002a, Section 6.6.2) the subsurface conditions at the site consist of 5 to 28 feet of undocumented fill underlain by alluvial material. The surface facilities will be resting directly on the alluvial material. The undocumented fill will be removed from the WHB area. The alluvial material thickness varies from a few feet up to 120 feet. Bedrock is found beneath the surface deposits of fill and alluvium.

The groundwater table is located at a typical depth of 1,270 feet below the present ground surface (see BSC, 2002a, Section 6.6.3).

The following material parameters for the alluvium are considered in the bearing capacity and settlement analyses:

$\gamma := 114 \text{pcf}$	Moist density (see Table 11-1 of this report in Section 10.1.1.1)
$\phi_{eff} := 39 deg$	Equivalent effective friction angle (see Section 10.1.1.3 o this report)
ç := 0psf	Cohesion (see see Section 10.1.1.3 of this report)

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APPENDIX B - BEARING CAPACITY AND SETTLEMENT

The elastic settlements of shallow footings and mat foundations are evaluated with an alluvium Young's modulus profile that is obtained from the measurements of seismic shear wave velocities (see Appendix A of this report).

The average shear wave velocity and elastic modulus profiles are represented by the following best-fit equations:

Shear wave velocity profile

$m_0 := 14.4 \cdot \frac{1}{s}$	Slope of equation fit
$b := 1410 \frac{ft}{s}$	Intercept of equation fit
$V(z) := m_0 \cdot z + b$	Linear fit equation for shear wave velocity vs. depth; fitted from Figure A6-1.
v := 0.3	Poisson's ratio (Appendix A of this report)

Young's modulus profile

	The fitted shear wave velocity line to obtain
	Young's modulus is for small strains. A reduction
K := 0.1	factor, K, of 0.1 is applied to obtain Young's
	modulus for large strain conditions. As
	demonstrated in Figure B6.19, the factor is
	conservative for the expected range of strains
	(<1%).
	,

$$G_{max}(z) := V(z)^2 \cdot \frac{\gamma}{g}$$
 Shear modulus (at small strains) calculated from shear wave velocity.
$$E(z) := 2 \cdot K \cdot (1 + \nu) \cdot G_{max}(z)$$
 Young's modulus equation using linear fit shear wave velocity equation.

B2.4 Factor of Safety

A 3.0 factor of safety against bearing capacity failure of the alluvial material is implemented in the analyses to compute the allowable bearing capacity.

FS := 3.0 Factor of safety against bearing capacity failure

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APPENDIX B - BEARING CAPACITY AND SETTLEMENT

B3 Background

These analyses are the basis for recommendations and design guidelines for shallow footings and mat foundations for the surface facilities at the YMP site and for large mat foundations.

Ultimate bearing capacity values at the surface facilities area were previously presented in BSC (2002b, Section 9.2).

The current study presents shallow footings and mat foundations recommendations based on the material parameters presented in this report, Section 10.1.1. These recommendations are based on the field and laboratory test results reported in BSC (2002a, Section 6). These results include shear strength tests and in-situ shear wave velocity measurements in the alluvial material.

B4 Methodology

This section presents the methodology used to compute the bearing capacity and short-term settlement analyses for shallow footings and mat foundations.

B4.1 Foundation Pressures

The recommended foundation pressures for shallow footings is computed for square and strip footings and for different foundation embedment depths. These recommended pressures are limited by the following criteria:

- The recommended foundation pressure should not exceed the allowable foundation capacity that considers a factor of safety of 3.0 against the soil shear failure. This allowable value is computed using the general ultimate capacity equation reported in Bowles (1996, Table 4-1 and Table 4-5a).
- The induced footing settlements cause by the recommended foundation pressure should not exceed the maximum allowable foundation settlement. Elastic settlements are computed using the settlement analyses procedures proposed by Burland and Burbidge, and by Schmertmann et al. as reported in Terzaghi et al. (1996, Sections 50.2.5 and 50.2.6).

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APPENDIX B - BEARING CAPACITY AND SETTLEMENT

B4.2 Short-term Settlements for Shallow Footings

Short-term settlements of shallow foundations are computed for square and strip footings using the Burland and Burbidge, and the Schmertmann et al. methods as presented in Terzaghi et al. (1996, Sections 50.2.5 and 50.2.6). Both methods use elastic theory to evaluate immediate settlements.

The Burland and Burbidge method is based on field measurements of foundation settlements. It uses the soil average standard penetration test blow count (N_{60}) values to estimate the soil's vertical compression. The Schmertmann et al. method is based on field measurements of vertical strain beneath shallow footings. It uses the elastic soil modulus to estimate settlements.

The following discussion describes the methodology used to obtain the N_{60} values and the elastic modulus for the alluvial material to be used as input parameters in the short-term settlement estimates.

N₆₀

 N_{60} results on the alluvial material are reported in only one of the exploration boreholes drilled in the WHB area. The reported values are unrealistically high and, therefore, are not used in the settlement analyses.

As an alternative to determine the N_{60} values for the alluvial material, two different procedures that correlate N_{60} values with experimental soil parameters were reviewed. The soil parameters reviewed in these correlations are as follows:

Using the correlations presented in Seed and Idris (1970) and Seed et al. (1986), N_{60} values for the alluvial material were evaluated using the extensive seismic shear wave velocity measurements performed at the site (see BSC 2002a, Section 6; and Appendix A of this report). The estimated N_{60} values with these correlations were unrealistically high for the given velocity measurements and thus a not used in the settlement analyses.

 N_{60} values for the alluvial material were correlated to the internal friction angle of the alluvium. The basis for the internal friction angle was from relative density measurements discussed in Section 9.1.1. The relationship proposed by Peck et al. (1974, page 310), is used to correlate N_{60} values with internal friction angle. These values were used in the short-term settlement analyses.

Young's modulus

Estimate of the soil's Young's modulus are obtained from the seismic shear wave velocity measurements performed at the site (see Appendix A of this report). The average shear wave velocity profile adopted in this calculation is presented in Section B2.3 of this report.

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APPENDIX B - BEARING CAPACITY AND SETTLEMENT

B4.3 Elastic Settlements for Mat Foundation

Settlements of a mat foundation on the alluvial sand were determined using elastic theory.

The stress profile under the mat was computed using a Boussinesq equation for a uniform vertical load. The incremental strain profile under the mat was computed using an iterative procedure that accounted for the degradation of Young's modulus with strain. In the iterative procedure, an initial small-strain Young's modulus was determined from the shear wave velocity profile presented in Section B2.3.

The shear modulus degradation curve for sands (Seed et al, 1986) was used to represent the Young's modulus degradation behavior of the alluvial material. For the purpose of the analysis herein, this assumption is considered conservative.

B4.4 Long-term Settlements

The Burland and Burbidge procedure was implemented to compute the long-term settlements of footings (see Terzaghi et al, 1996, Section 50.2.5). This method estimates settlements based on the soil standard penetration test blow count (N_{60}) values.

B5 Assumptions

It is conservatively assumed that bedrock is very deep and that it has no effect on the bearing capacity and settlement analyses for shallow footings and mat foundations.

Additionally, the Young's modulus, E, is assumed to degrade the same as the shear modulus, G, for sands. This yields conservative results since Poisson's ratio does not remain constant with strain. It is also assumed that there is no rock strain for the mat analysis.

No eccentric or inclined loading is considered in the analyses.

The preconsolidated characteristics of the alluvial material due to the removal of the overlaying undocumented fill is not considered in the short-term settlement analyses. This is a conservative assumption.

A 300-year lifetime for the footing structures is assumed in the long-term settlements calculations (Subsurface Facility Description Document, BSC, 2004a, Section 2.3.1).

All of these assumptions are either sufficiently conservative or represent typical standards used in the industry and do not require further verification.

B6 Calculations

Calculations were performed using Mathcad and EXCEL on a stand-alone PC. The PC is networked for printing and file storage but the programs used are loaded on the PC. These programs started and operated normally during calculation preparation.

The allowable bearing capacity results consider an adequate margin of safety against bearing capacity failure with associated tolerable footing settlement. The following schematic (Figure B6-1) for a shallow footing presents the definitions of the different symbols used in the bearing capacity and short-term settlement analyses:

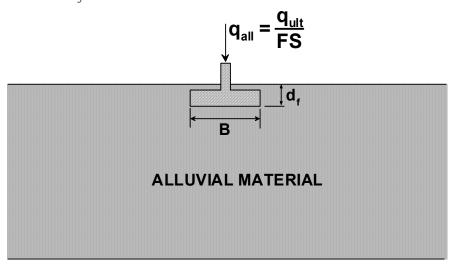


Figure B6-1. Schematic for shallow footing.

B6.1 Bearing Capacity for Shallow Footings

The bearing capacity of shallow footings was computed using the general ultimate capacity equation reported in Bowles (1996, Table 4-1 and Table 4-5a).

Effective overburden pressure

Check values

$$q(d_f) := d_f \gamma$$
 $q(2ft) = 228 psf$

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APPENDIX B - BEARING CAPACITY AND SETTLEMENT

Bearing capacity factors

Check values

$$N_q(\phi) := e^{\pi \cdot tan(\phi)} \cdot tan \left(45 deg + \frac{\phi}{2}\right)^2$$

$$N_q(0deg) = 1$$

$$N_{\gamma}(\phi) := 2 \cdot (N_{q}(\phi) + 1) \cdot \tan(\phi)$$

$$N_{\gamma}(0\text{deg}) = 0$$

$$N_{c}(\phi) := \begin{cases} \pi + 2 & \text{if } \phi = 0 \\ \left(N_{q}(\phi) - 1\right) \cdot \cot(\phi) & \text{otherwise} \end{cases}$$

$$N_{c}(0\deg) = 5.142$$

Shape factors

Check values

Square footings

$$s_{q}$$
 square $(\phi) := 1 + tan(\phi)$

$$s_{q_square}(0deg) = 1$$

$$s_{\gamma \text{ square}} := 0.6$$

$$s_{\gamma_square} = 0.6$$

$$s_{c_square}(\phi) := 1 + \frac{N_q(\phi)}{N_c(\phi)}$$

$$s_{c_square}(0deg) = 1.194$$

$$s_{q \text{ strip}} := 1$$

$$s_{q \text{ strip}} = 1$$

$$s_{\gamma \text{ strip}} := 1$$

$$s_{\gamma_strip} = 1$$

$$s_{c_strip} := 1$$

$$s_{c \text{ strip}} = 1$$

Ultimate bearing capacity

Square footings

$$q_{ult_square} \Big(B, d_f, c, \phi, \gamma \Big) \coloneqq c \cdot N_c \Big(\phi \Big) \cdot s_{c_square} \Big(\phi \Big) + q \Big(d_f \Big) \cdot N_q \Big(\phi \Big) \cdot s_{q_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot s_{\gamma_square} \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot S_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot S_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot S_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) \cdot S_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot N_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot M_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot M_\gamma \Big(\phi \Big) + 0.5 \cdot \gamma \cdot B \cdot M_\gamma \Big(\phi \Big) + 0.5$$

Strip footings

$$q_{ult_strip}\big(B, d_f, c, \phi, \gamma\big) := c \cdot N_c\big(\phi\big) \cdot s_{c_strip} + q\big(d_f\big) \cdot N_q\big(\phi\big) \cdot s_{q_strip} + 0.5 \cdot \gamma \cdot B \cdot N_\gamma\big(\phi\big) \cdot s_{\gamma_strip}$$

Allowable bearing capacity

Square footings

$$\begin{aligned} q_{all_square} \Big(B, d_f, c, \phi, \gamma \Big) \coloneqq \frac{q_{ult_square} \Big(B, d_f, c, \phi, \gamma \Big)}{FS} \end{aligned}$$

$$q_{all_strip} \Big(B, d_f, c, \phi, \gamma \Big) \coloneqq \frac{q_{ult_strip} \Big(B, d_f, c, \phi, \gamma \Big)}{FS}$$

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APPENDIX B - BEARING CAPACITY AND SETTLEMENT

Results

Figure B6-2 presents the allowable bearing capacities for square and strip footings. Results for 2-foot and 6-foot foundation embedment depths are presented in these figures

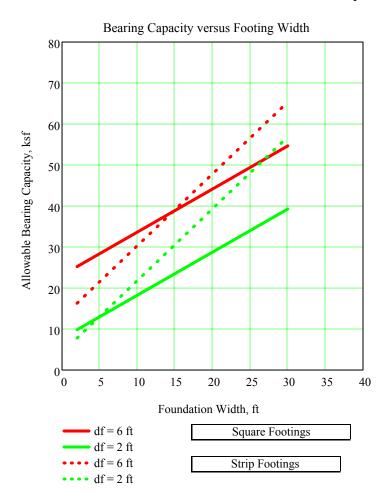


Figure B6-2. Allowable bearing pressure versus foundation width for square and strip footings

B6.2 Short-term Settlements for Shallow Footings

Short-term settlements of shallow foundations are computed for square and strip footings using the Burland and Burbidge, and Schmertmann et al. methods as presented in Terzaghi et al. (1996, Sections 50.2.5 and 50.2.6). Both methods use elastic theory to evaluate immediate settlements..

Burland and Burbidge (Terzaghi et al. 1996, Section 50.2.5) Method

 N_{60}

The following equation correlates N_{60} values with ϕ . This equation is the regression curve to the chart presented by Peck et al. (1974, page 310).

Note: the computed N_{60} values are bounded to a maximum value of 60 blows per foot and a minimum value of 3 blows per foot.

$$N_{60}(\phi) := \begin{bmatrix} .0027305858 - 17.924589 \cdot \frac{\phi}{\deg} + 1.4246932 \cdot \left(\frac{\phi}{\deg}\right)^2 \dots & \text{if } \phi \ge 28 \deg \\ + -.03770745 \cdot \left(\frac{\phi}{\deg}\right)^3 + .00035020841 \cdot \left(\frac{\phi}{\deg}\right)^4 \end{bmatrix}$$

Effective preconstruction pressure at the footing base

Check value

$$\sigma_{\text{VO}}(d_{\text{f}}) := d_{\text{f}} \gamma$$

$$\sigma_{\text{VO}}(1 \, \text{ft}) = 114 \, \text{psf}$$

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APPENDIX B - BEARING CAPACITY AND SETTLEMENT

Zone of footing influence

The following equation corresponds to Equation 50.6 presented by Terzaghi et al. (1996, page 395).

Check value

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$$Z_{I}(B) := \left(\frac{B}{m}\right)^{0.75} m$$

$$Z_{I}(10ft) = 2.307 \text{ m}$$

Average coefficient of vertical compression

The following equation corresponds to Equation 50.7 presented by Terzaghi et al. (1996, page 395).

Check value

$$m_{V}(\phi) := \frac{1.7}{N_{60}(\phi)^{1.4}} MPa^{-1}$$

$$m_{V}(\phi_{eff}) = 0.0093 \,\mathrm{MPa}^{-1}$$

Foundation length-to-width ratio

The following values are derived from Equation 50.14 presented by Terzaghi et al. (1996 page 397).

Square Footings

$$S_{c_sq} := 1$$

$$S_{c_st} := 1.56$$

Immediate settlement equation for square and strip

The following equations correspond to Equations 50.11a and 50.11b presented by Terzaghi et al. (1996,page 396). Equation 50.11a is applicable for foundation pressures greater than the effective preconsolidation pressure. Equation 50.11b is applicable for foundation pressures less than the effective preconsolidation pressure.

Square footings

$$\begin{split} S_{c1_sq} &(\mathtt{B}, \mathtt{d}_f, \mathtt{c}, \phi, \gamma) := & \begin{bmatrix} Z_I (\mathtt{B}) \cdot m_V (\phi) \cdot \left[\mathtt{q}_{all_square} (\mathtt{B}, \mathtt{d}_f, \mathtt{c}, \phi, \gamma) \ ... \right] \cdot S_{c_sq} & \text{if } \mathtt{q}_{all_square} (\mathtt{B}, \mathtt{d}_f, \mathtt{c}, \phi, \gamma) > \sigma_{vo} (\mathtt{d}_f) \\ & + \left(\frac{2}{3} \cdot \sigma_{vo} (\mathtt{d}_f) \right) \cdot (-1) \\ & \frac{1}{3} \cdot Z_I (\mathtt{B}) \cdot m_V (\phi) \cdot \mathtt{q}_{all_square} (\mathtt{B}, \mathtt{d}_f, \mathtt{c}, \phi, \gamma) \cdot S_{c_sq} & \text{otherwise} \\ \end{split}$$

$$\begin{split} S_{c1_st}\big({}_{B},{}_{d_f},{}_{c},\phi,\gamma\big) := & \begin{bmatrix} Z_I\big({}_{B}\big) \cdot m_V\big(\phi\big) \cdot \left[q_{all_strip}\big({}_{B},{}_{d_f},{}_{c},\phi,\gamma\big) \dots \right] \cdot S_{c_st} & \text{if } q_{all_strip}\big({}_{B},{}_{d_f},{}_{c},\phi,\gamma\big) > \sigma_{VO}\big({}_{d_f}\big) \\ + - \left(\frac{2}{3} \cdot \sigma_{VO}\big({}_{d_f}\big)\right) \cdot (-1) & \end{bmatrix} \cdot S_{c_st} & \text{if } q_{all_strip}\big({}_{B},{}_{d_f},{}_{c},\phi,\gamma\big) > \sigma_{VO}\big({}_{d_f}\big) \\ \frac{1}{3} \cdot Z_I\big({}_{B}\big) \cdot m_V\big(\phi\big) \cdot q_{all_strip}\big({}_{B},{}_{d_f},{}_{c},\phi,\gamma\big) \cdot S_{c_st} & \text{otherwise} \end{split}$$

Schmertmann (Terzaghi et al. 1996, Section 50.2.6) Method

Embedment correction factor (regression equation)

This equation is the regression curve to the chart presented in Figure 50.10 by Terzaghi et al. (1996, Section 50.2.6).

$$C_1\!\left(B,d_f\!\right) \coloneqq \frac{1.0561309 + 0.66610907 \cdot\!\left(\frac{d_f}{B}\right)}{1 + 1.2514064 \cdot\!\left(\frac{d_f}{B}\right) - 0.0024535149 \cdot\!\left(\frac{d_f}{B}\right)^2}$$

$$C_1(B, d_f) := min(1, C_1(B, d_f))$$

Bound C_1 to a maximum value of 1

Strain influence equations for square and strip footings

These Equations correspond to the curves presented in Figure 50.9 presented by Terzagh et al. (1996, Section 50.2.6) for square (L/B = 1) and strip (L/B > 10) footings. L is the footing length.

Square footings

$$I_{z_sq}(z,B,d_f) := \begin{bmatrix} \frac{4}{5 \cdot B} \cdot \left(z - d_f\right) + \frac{1}{5} & \text{if } z \le \left(d_f + \frac{B}{2}\right) \\ \frac{-2}{5 \cdot B} \cdot \left(z - d_f\right) + \frac{4}{5} & \text{otherwise} \end{bmatrix}$$

$$I_{z_st}(z,B,d_f) := \begin{bmatrix} \frac{4}{5 \cdot B} \cdot (z - d_f) + \frac{1}{5} & \text{if } z \le \left(d_f + \frac{B}{2}\right) \\ \frac{-6}{35 \cdot B} \cdot (z - d_f) + \frac{24}{35} & \text{otherwise} \end{bmatrix}$$

Immediate settlement equations

These equations represent the continuous form of Equations 50.23a and 50.23b presented Terzaghi et al. (1996, Section 50.2.6).

Square footings

$$\mathrm{S}_{c2a_sq}\!\!\left(\mathrm{B}\,,\mathrm{d}_f\!\right) \coloneqq \int_{\mathrm{d}_f}^{\mathrm{d}_f\!+\!2\mathrm{B}} \frac{\mathrm{I}_{z_sq}\!\!\left(z,\mathrm{B}\,,\mathrm{d}_f\!\right)}{\mathrm{E}(z)}\,\mathrm{d}z$$

$$\mathbf{S}_{c2\ sq}\!\!\left(\mathbf{B},\mathbf{d}_f\!,\mathbf{c},\!\phi,\!\gamma\right)\coloneqq\mathbf{C}_1\!\!\left(\mathbf{B},\mathbf{d}_f\!\right)\!\cdot\!\left(\mathbf{q}_{all\ square}\!\!\left(\mathbf{B},\mathbf{d}_f\!,\mathbf{c},\!\phi,\!\gamma\right)-\sigma_{vo}\!\!\left(\mathbf{d}_f\!\right)\!\right)\!\cdot\!\mathbf{S}_{c2a\ sq}\!\!\left(\mathbf{B},\mathbf{d}_f\!\right)$$

$$\mathrm{S}_{c2a_st}\!\!\left(\mathrm{B},\mathrm{d}_f\!\right) \coloneqq \int_{\mathrm{d}_f}^{\mathrm{d}_f\!+\!4\mathrm{B}} \frac{\mathrm{I}_{z_st}\!\!\left(\mathrm{z},\mathrm{B},\mathrm{d}_f\!\right)}{\mathrm{E}(\mathrm{z})}\,\mathrm{d}\mathrm{z}$$

$$\mathbf{S}_{c2_st}\!\!\left(\mathbf{B},\mathbf{d}_f\!,\!c,\phi,\gamma\right) \coloneqq \mathbf{C}_1\!\!\left(\mathbf{B},\mathbf{d}_f\!\!\right)\!\cdot\!\left(\mathbf{q}_{all_strip}\!\!\left(\mathbf{B},\mathbf{d}_f\!,\!c,\phi,\gamma\right) - \sigma_{vo}\!\!\left(\mathbf{d}_f\!\right)\!\right)\!\cdot\!\mathbf{S}_{c2a_st}\!\!\left(\mathbf{B},\mathbf{d}_f\!\right)$$

$$S_{c2} \text{ sq}(5\text{ft}, 0\text{ft}, c, \phi_{eff}, \gamma) = 0.104 \text{ in}$$
 Check value

$$S_{c2_st}(5ft, 0ft, c, \phi_{eff}, \gamma) = 0.313 \text{ in}$$
 Check value

Results

Figures B6-3 and B6-4 present settlement estimates versus allowable bearing capacities 1 square and strip footings, respectively. Settlements are evaluated with the Burland and Burbidge, and Schmertmann Methods. Results for 2 and 6-feet foundation embedment depths are presented in these figures.

For ploting purposes let:

$$\begin{split} & q_{all_sq_6ft}(B) \coloneqq q_{all_square} \Big(B, 6ft, c, \phi_{eff}, \gamma \Big) \\ & q_{all_sq_2ft}(B) \coloneqq q_{all_square} \Big(B, 2ft, c, \phi_{eff}, \gamma \Big) \\ & q_{all_st_6ft}(B) \coloneqq q_{all_strip} \Big(B, 6ft, c, \phi_{eff}, \gamma \Big) \\ & q_{all_st_2ft}(B) \coloneqq q_{all_strip} \Big(B, 2ft, c, \phi_{eff}, \gamma \Big) \end{split}$$

in Figures B6-3 and B6-4 below.

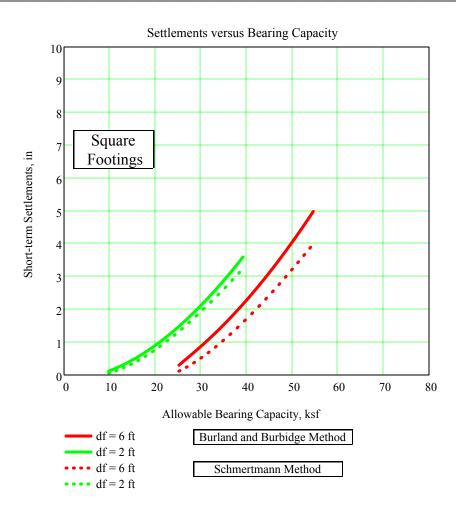


Figure B6-3. Short-term settlement estimates versus allowable bearing capacities for square footings

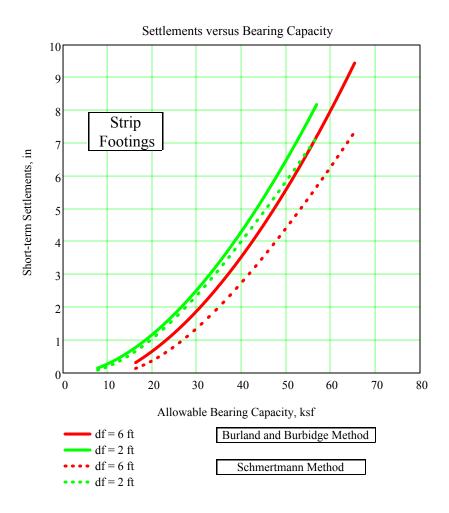


Figure B6-4. Short-term settlement estimates versus allowable bearing capacities for str footings

B6.3 Foundation Pressure Considering a Maximum Allowable Short-term Settlement $(S_c = \delta_{max})$

The allowable foundation pressure is constrained to a pressure that produces a footing maximum allowable short-term settlement, δ_{max} . This capacity is computed using the methods proposed by Burland and Burbidge, and Schmertmann et al. as reported in Terzaghi et al. (1996, Sections 50.2.5 and 50.2.6).

Burland and Burbidge (Terzaghi et al. 1996, Section 50.2.5) Method

The following equations correspond to Equations 50.11a and 50.11b presented by Terzaghi et al. (1996), page 396.

Square footings

$$\begin{split} q_{\delta max_c1_sq}({}_{B},{}_{d_f},{}_{c},{}_{\phi},{}_{\gamma},{}_{\delta}{}_{max}) \coloneqq & \left| \frac{\delta_{max}}{Z_I({}_{B})\cdot m_V({}_{\phi})\cdot S_{c_sq}} + \frac{2}{3}\cdot \sigma_{VO}({}_{d_f}) \right. \\ & \left| \frac{3\cdot \delta_{max}}{Z_I({}_{B})\cdot m_V({}_{\phi})\cdot S_{c_sq}} \right. \\ & \left| \frac{\sigma_{VO}({}_{d_f})}{\sigma_{VO}({}_{\phi})\cdot S_{c_sq}} \right| + \frac{2}{3}\cdot \sigma_{VO}({}_{d_f}) \\ & \left| \frac{\sigma_{VO}({}_{d_f})}{\sigma_{VO}({}_{\phi})\cdot S_{c_sq}} \right| \\ & \left| \frac{\sigma_{VO}({}_{d_f})}{\sigma_{VO}({}_{\phi})\cdot S_{c_sq}} \right| \\ & \left| \frac{\sigma_{VO}({}_{d_f})}{\sigma_{VO}({}_{\phi})\cdot S_{c_sq}} \right| \\ & \left| \frac{\sigma_{VO}({}_{d_f})}{\sigma_{VO}({}_{\phi})\cdot S_{c_sq}} \right| \\ & \left| \frac{\sigma_{VO}({}_{d_f})}{\sigma_{VO}({}_{\phi})\cdot S_{c_sq}} \right| \\ & \left| \frac{\sigma_{VO}({}_{\phi})}{\sigma_{VO}({}_{\phi})\cdot S_{c_sq}} \right| \\ & \left|$$

$$\begin{aligned} q_{\delta max_c1_st}\!\!\left(_{B,\,d_f,\,c,\,\phi}\,,_{\gamma}\,,_{\delta_{max}}\right) &:= & \left| \frac{\delta_{max}}{Z_I\!\!\left(_{B}\right)\!\cdot m_V\!\!\left(_{\phi}\right)\!\cdot S_{c_st}} + \frac{2}{3}\!\cdot \!\sigma_{VO}\!\!\left(_{d_f}\!\!\right) \;\; \text{if} \;\; q_{all_strip}\!\!\left(_{B,\,d_f,\,c,\,\phi}\,,_{\gamma}\right) > \sigma_{VO}\!\!\left(_{d_f}\!\!\right) \\ & \frac{3\!\cdot \!\delta_{max}}{Z_I\!\!\left(_{B}\right)\!\cdot m_V\!\!\left(_{\phi}\right)\!\cdot S_{c_st}} \;\; \text{otherwise} \end{aligned} \right.$$

$$q_{\delta max_c1_sq}\!\!\left(5\mathrm{ft},0\mathrm{ft},c,\phi_{eff},\gamma,0.5\mathrm{in}\right) = 20.825\,\mathrm{ksf} \quad \blacktriangleleft \qquad \quad \text{Check value}$$

$$q_{\delta max_c1_st}(5ft, 0ft, c, \phi_{eff}, \gamma, 0.5in) = 13.35 \text{ ksf}$$
 Check value

Schmertmann (Terzaghi et al. 1996, Section 50.2.6) Method

These equations represent the continuous form of Equations 50.23a and 50.23b presented by Terzaghi et al. (1996, Section 50.2.6).

Square footings

$$\mathsf{q}_{\delta max_c2_sq}\!\!\left(\mathsf{B},\mathsf{d}_f,\delta_{max}\!\right) \coloneqq \frac{\delta_{max}}{C_1\!\left(\mathsf{B},\mathsf{d}_f\!\right)\!\cdot\!\left(\mathsf{S}_{c2a_sq}\!\!\left(\mathsf{B},\mathsf{d}_f\!\right)\!\right)} + \sigma_{vo}\!\!\left(\mathsf{d}_f\!\right)$$

Strip footings

$$\mathsf{q}_{\delta max_c2_st}\!\!\left(\mathsf{B},\mathsf{d}_f,\delta_{max}\!\right) \coloneqq \frac{\delta_{max}}{\mathsf{C}_1\!\left(\mathsf{B},\mathsf{d}_f\!\right)\!\cdot\!\left(\mathsf{S}_{c2a_st}\!\!\left(\mathsf{B},\mathsf{d}_f\!\right)\right)} + \sigma_{vo}\!\!\left(\mathsf{d}_f\!\right)$$

$$q_{\delta max_c2_sq}(5ft, 0ft, 0.5in) = 25.391 \text{ ksf}$$
 Check value

$$q_{\delta max_c2_st}(5ft, 0ft, 0.5in) = 14.017 \text{ ksf}$$
 Check value

Results

Figures B6-5 and B6-6 present the maximum foundation pressure versus foundation width for square and strip footings, respectively. Settlements are evaluated with the Burland ar Burbidge, and Schmertmann methods. Results for 2-foot and 6-foot foundation embedment depths are presented in these figures.

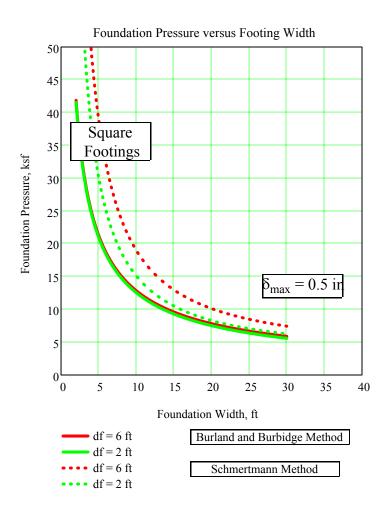


Figure B6-5. Foundation pressure versus foundation width for square footings considering a maximum allowable foundation settlement of 0.5 in

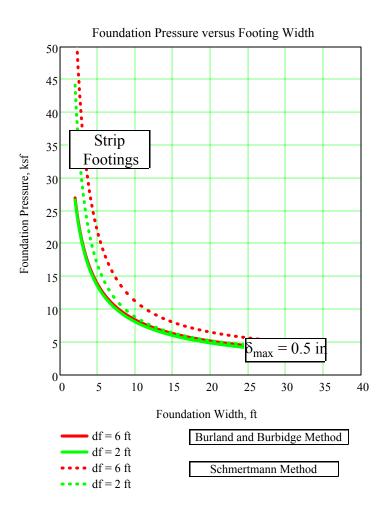


Figure B6-6. Foundation pressure versus foundation width for strip footings considering a maximum allowable foundation settlement of 0.5 in

B6.4 Design Foundation Pressure

The design foundation pressure is computed as the minimum of the allowable bearing capacity or the foundation pressure as determined above from Sections B6.1 and B6.3.

The maximum foundation pressure for design is further limited by a cutoff value. This value corresponds to the minimum pressure of the values determined in Sections B6.1 and B6.3 for a 2-foot wide footing.

Burland and Burbidge (Terzaghi et al. 1996) method

Square footings

$$\begin{aligned} \mathsf{q}_{fp_c1_sq}\!\!\left(B,\mathsf{d}_f,c,\phi,\gamma,\delta_{max}\right) \coloneqq & \left| \begin{array}{l} \mathsf{q}_{all_square}\!\!\left(B,\mathsf{d}_f,c,\phi,\gamma\right) & \text{if } S_{c1_sq}\!\!\left(B,\mathsf{d}_f,c,\phi,\gamma\right) \leq \delta_{max} \\ \mathsf{q}_{\delta max_c1_sq}\!\!\left(B,\mathsf{d}_f,c,\phi,\gamma,\delta_{max}\right) & \text{otherwise} \end{array} \right. \end{aligned}$$

$$\begin{aligned} \mathsf{q}_{fp_c1_st}\!\!\left(B\,,\mathsf{d}_f,c\,,\phi,\gamma,\delta_{max}\right) \coloneqq & \left| \begin{array}{l} \mathsf{q}_{all_strip}\!\!\left(B\,,\mathsf{d}_f,c\,,\phi,\gamma\right) & \text{if } S_{c1_st}\!\!\left(B\,,\mathsf{d}_f,c\,,\phi,\gamma\right) \leq \delta_{max} \\ \mathsf{q}_{\delta max_c1_st}\!\!\left(B\,,\mathsf{d}_f,c\,,\phi,\gamma,\delta_{max}\right) & \text{otherwise} \end{array} \right. \end{aligned}$$

$$\begin{aligned} q_{fp_c1_st}\big(B, d_f, c, \phi, \gamma, \delta_{max}\big) &:= & \begin{aligned} q_{fp_c1_st0}\big(d_f, c, \phi, \gamma, \delta_{max}\big) &\text{ if } q_{fp_c1_st}\big(B, d_f, c, \phi, \gamma, \delta_{max}\big) > q_{fp_c1_st0}\big(d_f, c, \phi, \gamma, \delta_{max}\big) \\ q_{fp_c1_st}\big(B, d_f, c, \phi, \gamma, \delta_{max}\big) &\text{ otherwise} \end{aligned}$$

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$$q_{fp_c1_sq}(20ft, 0ft, c, \phi_{eff}, \gamma, 0.5in) = 2.103 \text{ ksf}$$
 Check value

Schmertmann (Terzaghi et al. 1996) method

Square footings

$$\begin{aligned} \mathsf{q}_{fp_c2_sq}\!\!\left(B,\mathsf{d}_f,c,\phi,\gamma,\delta_{max}\right) \coloneqq & \left[\begin{array}{l} \mathsf{q}_{all_square}\!\!\left(B,\mathsf{d}_f,c,\phi,\gamma\right) & \text{if } S_{c2_sq}\!\!\left(B,\mathsf{d}_f,c,\phi,\gamma\right) \leq \delta_{max} \\ \mathsf{q}_{\delta max_c2_sq}\!\!\left(B,\mathsf{d}_f,\delta_{max}\right) & \text{otherwise} \end{array} \right. \end{aligned}$$

$$\begin{aligned} q_{fp_c2_sq}\big(B,d_f,c,\phi,\gamma,\delta_{max}\big) &:= & \begin{aligned} q_{fp_c2_sq0}\big(d_f,c,\phi,\gamma,\delta_{max}\big) & \text{ if } q_{fp_c2_sq}\big(B,d_f,c,\phi,\gamma,\delta_{max}\big) > q_{fp_c2_sq0}\big(d_f,c,\phi,\gamma,\delta_{max}\big) \\ q_{fp_c2_sq}\big(B,d_f,c,\phi,\gamma,\delta_{max}\big) & \text{ otherwise} \end{aligned}$$

$$\begin{aligned} \mathsf{q}_{fp_c2_st}\!\!\left(B,\mathsf{d}_f,\!c,\phi,\gamma,\delta_{max}\right) \coloneqq & \left| \begin{array}{l} \mathsf{q}_{all_strip}\!\!\left(B,\mathsf{d}_f,\!c,\phi,\gamma\right) & \text{if } S_{c2_st}\!\!\left(B,\mathsf{d}_f,\!c,\phi,\gamma\right) \leq \delta_{max} \\ \mathsf{q}_{\delta max_c2_st}\!\!\left(B,\mathsf{d}_f,\!\delta_{max}\right) & \text{otherwise} \end{array} \right. \end{aligned}$$

$$\begin{aligned} q_{fp_c2_st}\big(B, d_f, c, \phi, \gamma, \delta_{max}\big) &:= & \left[q_{fp_c2_st0}\big(d_f, c, \phi, \gamma, \delta_{max}\big) & \text{if } q_{fp_c2_st}\big(B, d_f, c, \phi, \gamma, \delta_{max}\big) > q_{fp_c2_st0}\big(d_f, c, \phi, \gamma, \delta_{max}\big) \\ q_{fp_c2_st}\big(B, d_f, c, \phi, \gamma, \delta_{max}\big) & \text{otherwise} \end{aligned} \right]$$

$$q_{fp\ c2\ sq}(20\text{ft}, 0\text{ft}, c, \phi_{eff}, \gamma, 0.5\text{in}) = 2.103 \text{ ksf}$$
 Check value

$$q_{fp_c2_st}(20ft, 0ft, c, \phi_{eff}, \gamma, 0.5in) = 3.505 \text{ ksf}$$
 Check value

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Results

Figures B6-7 and B6-8 present the design foundation pressure versus foundation width for square and strip footings, respectively. Settlements are evaluated with the Burland and Burbidge, and Schmertmann methods. Results for 2-foot and 6-foot foundation embedment depths are presented in these figures.

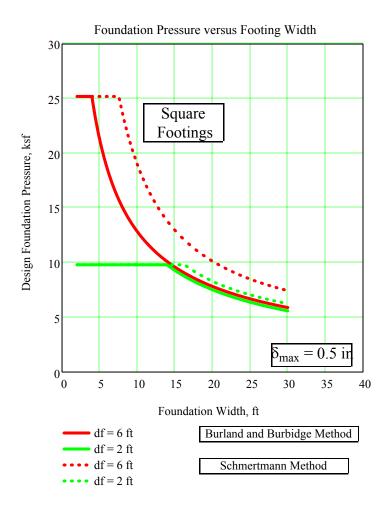


Figure B6-7. Design foundation pressure versus foundation width for square footings considering a maximum allowable foundation settlement of 0.5 in

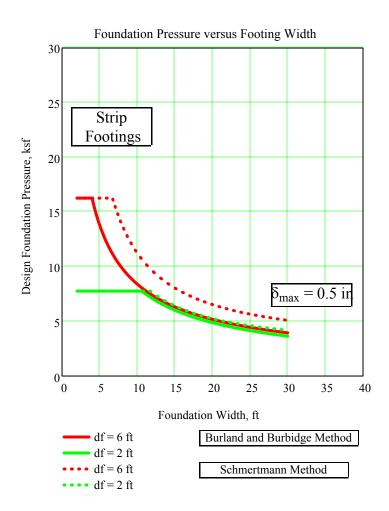


Figure B6-8. Design foundation pressure versus foundation width for strip footings considering a maximum allowable foundation settlement of 0.5 in

B6.5 Settlements for Different Foundation Pressures

The short-term settlements for different foundation pressures are computed using the procedures by Burland and Burbidge, and Schmertmann et al. as reported in Terzaghi et al. (1996, Sections 50.2.5 and 50.2.6).

The following bearing pressure range is considered in the analyses:

$$q_{bn} := 0.5ksf, 0.6ksf...40ksf$$

Bearing pressure range

Burland and Burbidge (Terzaghi et. al 1996) method

The following equations correspond to Equations 50.11a and 50.11b presented by Terzaş et al. (1996, Section 50.2.5).

Square Footings

$$\begin{split} S_{bp_c1_sq}\!\!\left(B, d_f, \phi, q_{bp}\right) &:= & \left| Z_I\!\!\left(B\right) \cdot m_v\!\!\left(\phi\right) \cdot \!\left(q_{bp} - \frac{2}{3} \cdot \sigma_{vo}\!\!\left(d_f\!\!\right)\right) \cdot S_{c_sq} \right. \text{ if } q_{bp} > \sigma_{vo}\!\!\left(d_f\!\!\right) \\ & \frac{1}{3} \cdot Z_I\!\!\left(B\right) \cdot m_v\!\!\left(\phi\right) \cdot q_{bp} \cdot S_{c_sq} \right. \text{ otherwise} \end{split}$$

Strip Footings

$$\begin{split} \mathbf{S}_{bp_c1_st}\!\!\left(\mathbf{B},\mathbf{d}_f,\phi,\mathbf{q}_{bp}\!\right) &:= & \left| Z_{I}\!\!\left(\mathbf{B}\right)\!\cdot\!\mathbf{m}_{v}\!\!\left(\phi\right)\!\cdot\!\!\left(\mathbf{q}_{bp} - \frac{2}{3}\!\cdot\!\sigma_{vo}\!\!\left(\mathbf{d}_f\!\right)\right)\!\cdot\!\mathbf{S}_{c_st} \quad \mathrm{if} \ \mathbf{q}_{bp} > \sigma_{vo}\!\!\left(\mathbf{d}_f\!\right) \right| \\ & \frac{1}{3}\!\cdot\!Z_{I}\!\!\left(\mathbf{B}\right)\!\cdot\!\mathbf{m}_{v}\!\!\left(\phi\right)\!\cdot\!\mathbf{q}_{bp}\!\cdot\!\mathbf{S}_{c_st} \quad \mathrm{otherwise} \end{split}$$

Check values

$$\begin{array}{c|c} S_{bp_c1_st} \Big(5ft, 0ft, \phi_{eff}, q_{bp} \Big) = & q_{bp} = \\ \hline 0.019 & in & 0.5 \\ \hline 0.022 & 0.026 & 0.7 \\ \end{array}$$

Schmertmann (Terzaghi et. al 1996) method

These equations represent the continuous form of Equations 50.23a and 50.23b presented by Terzaghi et al. (1996, Section 50.2.6).

Square Footings

$$S_{bp_c2a_sq}\!\!\left(B,d_f\!\right) := \int_{d_f}^{d_f\!+2B} \frac{I_{z_sq}\!\!\left(z,B,d_f\!\right)}{E(z)}\,dz$$

$$\mathbf{S}_{bp_c2_sq}\big(\mathbf{B},\mathbf{d}_f,\mathbf{q}_{bp}\big)\coloneqq \mathbf{C}_1\big(\mathbf{B},\mathbf{d}_f\big)\cdot \big(\mathbf{q}_{bp}-\sigma_{vo}\big(\mathbf{d}_f\big)\big)\cdot \mathbf{S}_{bp_c2a_sq}\big(\mathbf{B},\mathbf{d}_f\big)$$

Strip Footings

$$S_{bp_c2a_st}(B,d_f) := \int_{d_f}^{d_f+4B} \frac{I_{z_st}(z,B,d_f)}{E(z)} dz$$

$$S_{bp} c_{2} st(B, d_{f}, q_{bp}) := C_{1}(B, d_{f}) \cdot (q_{bp} - \sigma_{vo}(d_{f})) \cdot S_{bp} c_{2a} st(B, d_{f})$$

Check values

$$\begin{array}{c} S_{bp_c2_sq} \big(5\, \mathrm{ft}, 0\, \mathrm{ft}, q_{bp} \big) = & q_{bp} = \\ \hline 0.01 & in & 0.5 \\ \hline 0.012 & 0.014 & 0.7 \\ \end{array}$$
 ksf

$$\begin{array}{ccc} S_{bp_c2_st} \Big(5ft, 0ft, q_{bp} \Big) = & & q_{bp} = \\ \hline 0.018 & & & & 0.5 \\ \hline 0.021 & & & & 0.6 \\ \hline 0.025 & & & & 0.7 \\ \end{array}$$
 ksf

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Results

Figures B6-9 through B6-12 present the estimated settlements versus foundation pressure for square and strip footings. Settlements are evaluated with the Burland and Burbidge, and Schmertmann methods. Figures B6-9 and B6-10 present the results for square and strip footings with 6-foot foundation embedment depth, respectively. Figures B6-11 and B6-12 present the results for square and strip footings with 2-foot foundation embedment depth, respectively.

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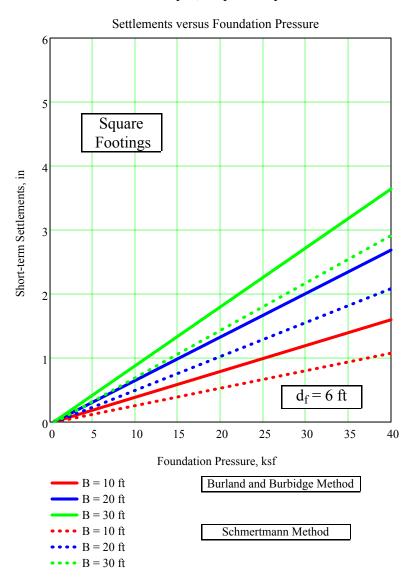


Figure B6-9. Short-term settlements versus foundation pressure for square footings

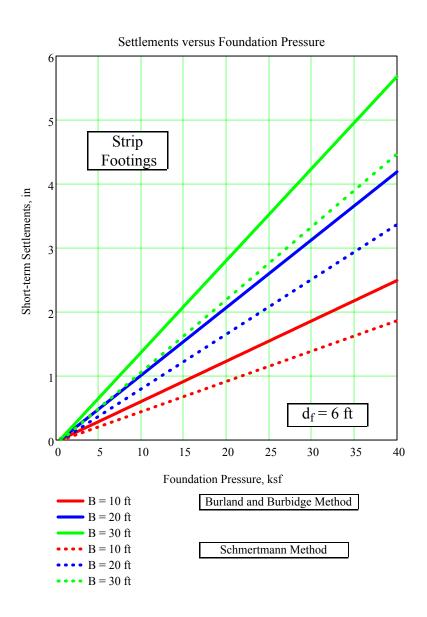


Figure B6-10. Short-term settlements versus foundation pressure for strip footings

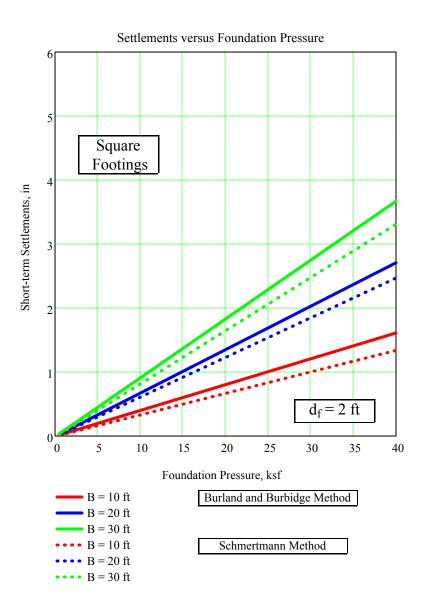


Figure B6-11. Short-term settlements versus foundation pressure for square footings

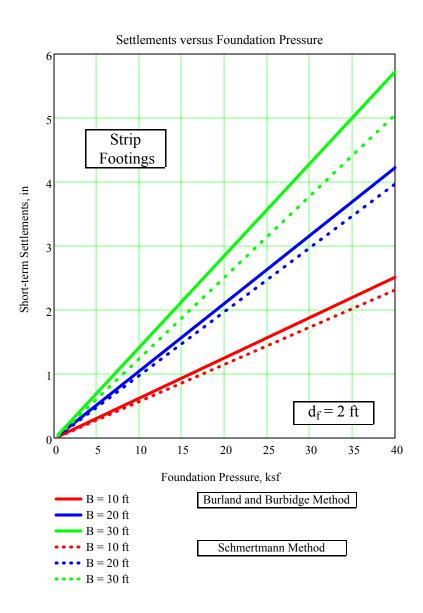


Figure B6-12. Short-term settlements versus foundation pressure for strip footings

A comparison of the above methods show similar results for the design pressure. Results from the Schmertmann method are adopted since more data from the project (shear wave velocity) is available for this method. The Burland and Burbidge method uses an N_{60} value, which was derived from relative density measurements. The design pressure calculated by the Schmertmann method is limited for larger footing sizes for conservatism.

Figures B6-13 through B6-16 present our recommendations to the project for allowable fouundation pressures and immediate settlements.

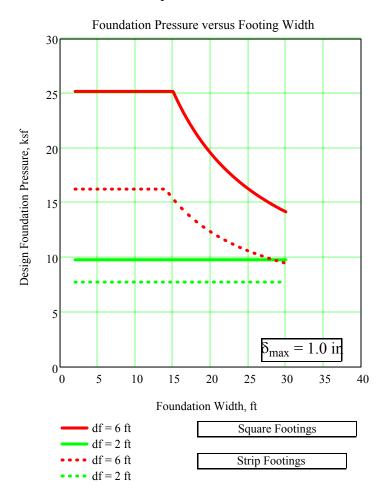


Figure B6-13. Design foundation pressure versus foundation width for square and strip footings considering a maximum allowable foundation settlement of 1.0 in

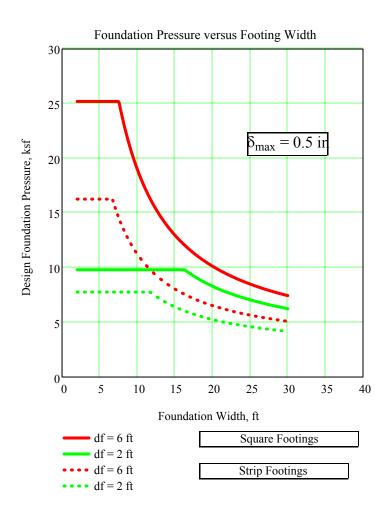


Figure B6-14. Design foundation pressure versus foundation width for square and strip footings considering a maximum allowable foundation settlement of 0.5 in

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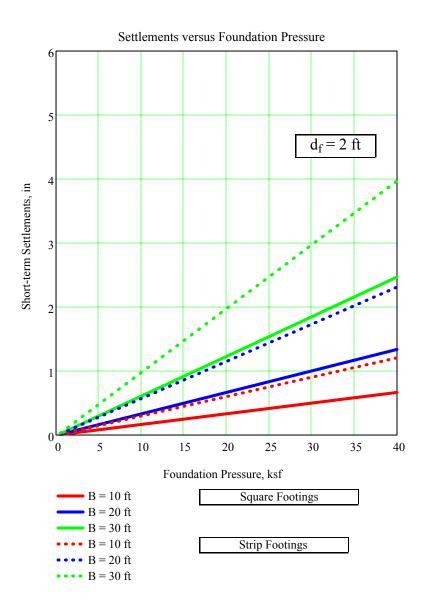


Figure B6-15. Short-term settlements versus foundation pressure for strip footings

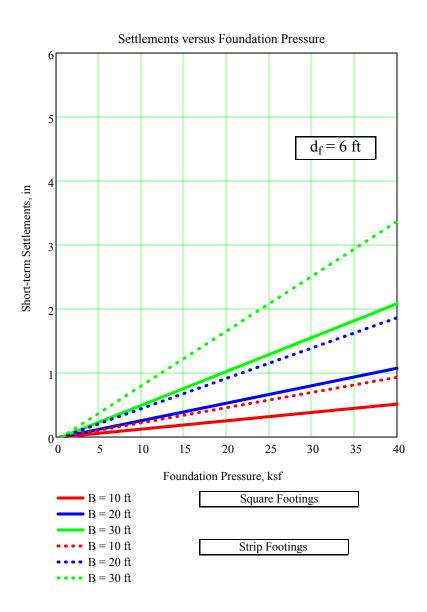


Figure B6-16. Short-term settlements versus foundation pressure for strip footings

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B6.6 Long-term Settlements

The Burland and Burbidge procedure was implemented to compute the footings long-term settlement (see Terzaghi et al, 1996, Section 50.2.5). This method estimates settlements based on the soil standard penetration test blow count (N_{60}) values.

Compression strain

Square footings

$$\begin{split} \epsilon_{\text{c_sq}}\!\!\left(\text{B}\,,\text{d}_{\text{f}},\text{c}\,,\phi,\gamma\right) &\coloneqq \left| \frac{1.4}{N_{60}\!\left(\phi\right)^{1.4}} \text{ if } q_{\text{all_square}}\!\!\left(\text{B}\,,\text{d}_{\text{f}},\text{c}\,,\phi,\gamma\right) > \sigma_{\text{vo}}\!\!\left(\text{d}_{\text{f}}\right) \\ &\frac{1}{3} \cdot \frac{1.4}{N_{60}\!\left(\phi\right)^{1.4}} \text{ if } q_{\text{all_square}}\!\!\left(\text{B}\,,\text{d}_{\text{f}},\text{c}\,,\phi,\gamma\right) \leq \sigma_{\text{vo}}\!\!\left(\text{d}_{\text{f}}\right) \end{split}$$

Strip footings

$$\epsilon_{\text{c_st}}\!\!\left(\text{B}, \text{d}_f, \text{c}, \phi, \gamma\right) \coloneqq \left| \frac{1.4}{N_{60}(\phi)^{1.4}} \right| \text{if } q_{\text{all_strip}}\!\!\left(\text{B}, \text{d}_f, \text{c}, \phi, \gamma\right) > \sigma_{\text{vo}}\!\!\left(\text{d}_f\right)$$

$$\frac{1}{3} \cdot \frac{1.4}{N_{60}(\phi)^{1.4}} \quad \text{if } q_{\text{all_strip}}\!\!\left(\text{B}, \text{d}_f, \text{c}, \phi, \gamma\right) \leq \sigma_{\text{vo}}\!\!\left(\text{d}_f\right)$$

$$\varepsilon_{c_st}(5ft, 0ft, c, \phi_{eff}, \gamma) = 7.647 \times 10^{-3}$$
 Check value

Secondary compression strain index

Square footings

$$\varepsilon_{\alpha \text{ sq}}(B, d_f, c, \phi, \gamma) := 0.02 \cdot \varepsilon_{c \text{ sq}}(B, d_f, c, \phi, \gamma)$$

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Strip footings

$$\begin{split} \epsilon_{\alpha_st}\!\!\left(B,d_f,c,\phi,\gamma\right) &:= 0.02 \cdot \epsilon_{c_st}\!\!\left(B,d_f,c,\phi,\gamma\right) \\ \epsilon_{\alpha_sq}\!\!\left(5\mathrm{ft},0\mathrm{ft},c,\phi_{eff},\gamma\right) &= 1.529 \times 10^{-4} \end{split} \qquad \qquad \text{Check value} \\ \epsilon_{\alpha} \ \ \mathrm{st}\!\!\left(5\mathrm{ft},0\mathrm{ft},c,\phi_{eff},\gamma\right) &= 1.529 \times 10^{-4} \end{split} \qquad \qquad \qquad \text{Check value}$$

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Long-term settlement equation

Square footings

$$S_{c3_sq}\!\!\left(B,d_f\!,c,\phi,\gamma\right) \coloneqq \epsilon_{\alpha_sq}\!\!\left(B,d_f\!,c,\phi,\gamma\right) \cdot Z_I\!\!\left(B\right) \cdot log\!\!\left(\frac{\frac{t}{year}day}{1 \cdot day}\right)$$

Strip footings

$$\mathbf{S}_{c3_st}\!\!\left(\mathbf{B},\mathbf{d}_f,\mathbf{c},\boldsymbol{\phi},\!\boldsymbol{\gamma}\right) \coloneqq \epsilon_{\alpha_st}\!\!\left(\mathbf{B},\mathbf{d}_f,\mathbf{c},\boldsymbol{\phi},\!\boldsymbol{\gamma}\right)\!\cdot\!Z_{I}\!\!\left(\mathbf{B}\right)\!\cdot\!\log\!\left(\frac{\frac{t}{year}day}{1\cdot day}\right)$$

$$S_{c3_st}(5ft, 0ft, c, \phi_{eff}, \gamma) = 0.017 \text{ in}$$
 Check value

Results

Figures B6-17 presents the estimated long-term settlements versus foundation pressure for square and strip footings and embedment depth considered herein. Settlement are evaluated with the Burland and Burbidge method.

APPENDIX B - BEARING CAPACITY AND SETTLEMENT

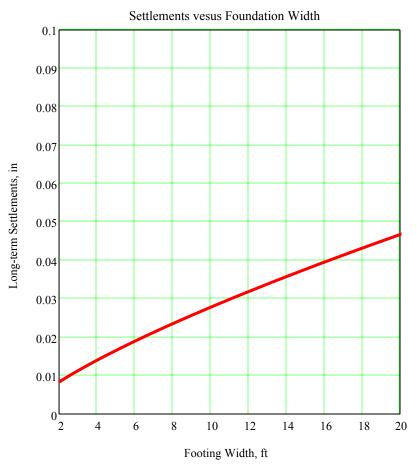


Figure B6-17. Long-term settlements versus footing width for square and strip footings and embedment depth considered herein

<u>Units:</u>

$$kPa = 1000 \cdot Pa \qquad psf = \frac{lbf}{ft^2} \qquad pcf = \frac{lbf}{ft^3} \qquad year = 365 \cdot day$$

$$MPa = 10^3 \cdot kPa \qquad ksf = \frac{1000lbf}{ft^2} \qquad tsf = \frac{2000lbf}{ft^2} \qquad fps = \frac{ft}{s}$$

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B6.7 Elastic Settlement for Mat Foundation

Elastic settlements are computed based on a uniform vertical stress distribution, representative average shear wave velocities, and modulus degradation curves for sands. The settlements are determined for uniform vertical loads of 3, 5, and 7 ksf. The following are performed for the computation:

- **Alluvium Thickness** Divide the alluvium layer (120 feet thick) into 1 ft sublayers (h1, h2...h_i), where i = sublayer number. Since the mat thickness used in the analysis is assumed to be 3 feet, subtract 3 feet from the top portion of the alluvium.
- Vertical Stress Distribution, σ_z Compute the vertical stress distribution below the mat (corner and center) for the entire alluvium layer. For a uniform load on a rectangular mat (beneath the mat corner), use the following equation from pp. 54 of Poulos and Davis (1991):

$$\sigma_z = \frac{q}{2\pi} \left[\tan^{-1} \frac{\ell b}{zR_3} + \frac{\ell bz}{R_3} \left(\frac{1}{R_1^2} + \frac{1}{R_2^2} \right) \right], \text{ where}$$

$$z = depth$$

$$\ell = \frac{L}{2} \text{ (for distribution at center of foundation)}$$

$$\ell = L \text{ (for distribution at corner of foundation)}$$

$$b = \frac{B}{2} \text{ (for distribution at center of foundation)}$$

$$b = B \text{ (for distribution at corner of foundation)}$$

$$R_1 = (\ell^2 + z^2)^{1/2}$$

$$R_2 = (b^2 + z^2)^{1/2}$$

$$R_3 = (\ell^2 + b^2 + z^2)^{1/2}$$

Multiply Eq. (B1) by 4 for the stress distribution at the center of foundation. Figure B6-18 below shows the stress distributions for the 3 uniform vertical loads.

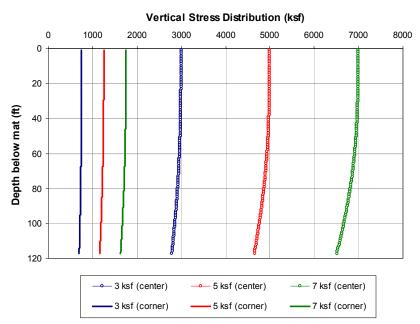


Figure B6-18. Vertical stress distribution versus depth for vertical loads of 3, 5, and 7 ksf.

• Modulus Degradation Curves – Select appropriate modulus degradation curves (G/G_{max} versus shear strain, γ_t) to be used to determine the strains induced in the alluvium layer during vertical loading. Dynamic testing was performed on one reconstituted alluvium sample in BSC (2002a). The modulus degradation curve obtained from the testing closely follows the mean curve from Seed and Idriss (1970) for sands as shown below:

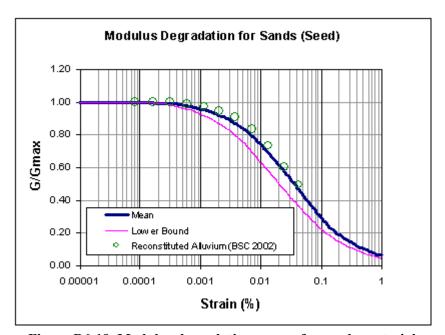


Figure B6-19. Modulus degradation curves for sandy material.

The lower bound curve from Seed and Idriss (1970) is included in the analyses for conservatism.

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• Shear Wave Velocity, Vs – Select representative shear wave velocity values for the alluvium layer to be used for the analyses. Table B6-1 (data determined in Appendix A) summarizes the lower bound (mean minus one standard deviation) and mean V_s values used at different depths in the alluvium for the analysis:

Table B6-1. Average shear way	ve velocity values	(computed in .	Appendix A	1).
-------------------------------	--------------------	----------------	------------	-------------

Depth from ground surface (ft)	Lower bound (ft/s)	Mean (ft/s)
0-15	1,200	1,500
15-30	1,400	1,700
30-60	2,000	2,200
60-120	2,200	2,500

• Young's Modulus of Elasticity, E and axial strain, ε_a – Use the vertical stress distribution (3, 5, and 7 ksf), the modulus degradation curves (mean and lower bound), and shear wave velocity averages (mean and lower bound) to determine Young's Modulus and the amount of axial strain induced in the alluvium layer.

The modulus degradation curves are modified to show elastic modulus versus axial strain. It is assumed that the shear modulus degradation relationship, G/G_{max} is analogous to the elastic modulus degradation, E/E_{max} . This is a conservative assumption since it is known that the elastic modulus degrades less than the shear modulus. Calculate dynamic G_{max} from the shear wave velocity values using:

$$G_{\text{max}} = \frac{{V_s}^2 \gamma}{\varphi} \tag{B2}$$

where $\gamma = 114$ pcf (unit weight of alluvium). Using this, the degradation curves can be modified to show G versus γ_t for each velocity average. E can then be determined by:

$$E = 2G(1+\upsilon) \tag{B3}$$

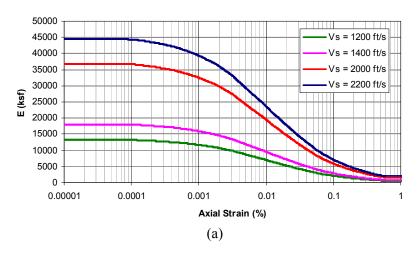
where $\upsilon = 0.3$ (Poisson's ratio of alluvium). The shear strain, γ_t , can be expressed as axial strain, ε_a , by the following relationship (Equation 11 of Vucetic and Dobry 1986):

$$\varepsilon_a = \frac{\gamma_t}{1.73} \tag{B4}$$

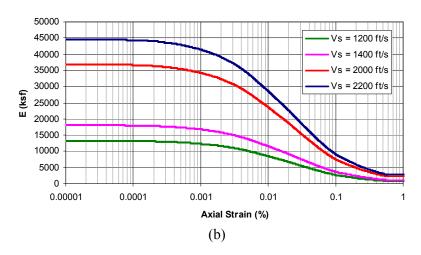
Using (B3) and (B4), the degradation curves can be modified to show E versus ε_a . The following curves for combinations of mean and lower bound values of modulus degradation curves and shear wave velocity value are generated:

APPENDIX B - BEARING CAPACITY AND SETTLEMENT

Lower Bound G/Gmax Curve and Lower Bound Vs



Mean G/Gmax Cruve and Lower Bound Vs



Mean G/Gmax and Mean Vs

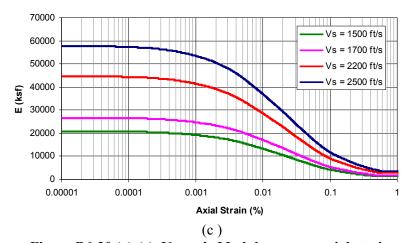


Figure B6-20 (a)-(c). Young's Modulus versus axial strain.

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Using the appropriate curve, an initial axial strain can be used to determine the corresponding E. The new axial strain can then be computed using:

$$\varepsilon_a = \frac{\sigma_z}{E} \tag{B5}$$

where σ_z is computed in (B1) for 3, 5, and 7 ksf vertical loading. The new strain can then be used with the curves to determine a new E. This iterative process using (B5) is continued until the axial strain converges, which represents the amount of strain induced in the alluvium due to the vertical loading.

• **Settlement** – Compute the total settlement of the alluvium from the final axial strains by summing the settlements in each alluvium layer using:

$$Settlement = \sum_{i=1}^{120} \varepsilon_i h_i$$
 (B6)

The calculations are performed for each vertical load case (3, 5, and 7 ksf) for the following bound conditions of modulus degradation and shear wave velocity (Table B6-2):

Table B6-2. Shear wave velocity and modulus degradation curve bound conditions used in analysis.

Shear wave velocity	Modulus degradation
Lower	Lower
Lower	Mean
Mean	Mean

Table B6-3 shows a sample EXCEL spreadsheet calculation (center of the mat foundation under 5 ksf loading using mean values of the shear wave velocity and modulus degradation curve for sands).

The results of the analyses (center and corner of the mat for different shear wave velocity and modulus degradation bound conditions and for various loadings) are shown in Table B6-4. A summary of the expected elastic settlements is shown in Section B7 of this calculation. Because of the conservatism in assuming that Young's modulus, E, degrades the same as the shear modulus for sands, the calculated settlements may be unrealistically high. Hence, for the summary table in Section B7, the settlements computed using the lower bounds of the shear wave velocity and modulus degradation are not used.

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Table B6-3. Example EXCEL spreadsheet to calculate elastic settlement.

MEAN VELOCITY PROFILE MEAN SEED AND IDRISS (1970) CURVE

B = 450 B/2 = ft 225 L = L/2 = 250 500 ft ft q = 5000 psf $\gamma =$ 114 pcf υ= 0.3

SETTLEMENT TOTAL	0.55	in	
STRAIN LEVEL	0.15	%	

Depth from bottom			distribut			Initial st	rains fror	n Vs			Iterative E	process	
	Δz	R1	•	•	σZ	1/2	0	Гm					C - 11
of mat		R1	R2	R3		Vs	Gmax	Emax	γ initial	ε a_initial	static	ε a_final	Sett.
(ft)	(ft)				(psf)	(ft/s)	(ksf)	(ksf)	(%)	(%)	(ksf)	(%)	(in)
0		250	225	336									
1	1	250	225	336	5000	1500	7966	20711	0.024	0.014	3326	0.15	0.018
2	1	250	225	336	5000	1500	7966	20711	0.024	0.014	3326	0.15	0.018
3	1	250	225	336	5000	1500	7966	20711	0.024	0.014	3326	0.15	0.018
4	1	250	225	336	5000	1500	7966	20711	0.024	0.014	3326	0.15	0.018
5	1	250	225	336	5000	1500	7966	20711	0.024	0.014	3326	0.15	0.018
6	1	250	225	336	5000	1500	7966	20711	0.024	0.014	3326	0.15	0.018
7	1	250	225	336	5000	1500	7966	20711	0.024	0.014	3326	0.15	0.018
8	1	250	225	336	5000	1500	7966	20711	0.024	0.014	3326	0.15	0.018
9	1	250	225	336	5000	1500	7966	20711	0.024	0.014	3326	0.15	0.018
10	1	250	225	336	5000	1500	7966	20711	0.024	0.014	3326	0.15	0.018
11	1	250	225	337	5000	1500	7966	20711	0.024	0.014	3326	0.15	0.018
12	1	250	225	337	5000	1500	7966	20711	0.024	0.014	3326	0.15	0.018
13	1	250	225	337	4999	1700	10232	26602	0.019	0.011	6381	0.08	0.009
14	1	250	225	337	4999	1700	10232	26602	0.019	0.011	6381	0.08	0.009
15	1	250	225	337	4999	1700	10232	26602	0.019	0.011	6381	0.08	0.009
16	1	251	226	337	4999	1700	10232	26602	0.019	0.011	6381	0.08	0.009
17	1	251	226	337	4999	1700	10232	26602	0.019	0.011	6381	0.08	0.009
18	1	251	226	337	4998	1700	10232	26602	0.019	0.011	6381	0.08	0.009
19	1	251	226	337	4998	1700	10232	26602	0.019	0.011	6381	0.08	0.009
20	1	251	226	337	4998	1700	10232	26602	0.019	0.011	6381	0.08	0.009
21	1	251	226	337	4997	1700	10232	26602	0.019	0.011	6381	0.08	0.009
22	1	251	226	337	4997	1700	10232	26602	0.019	0.011	6381	0.08	0.009
23	1	251	226	337	4997	1700	10232	26602	0.019	0.011	6381	0.08	0.009
24	1	251	226	337	4996	1700	10232	26602	0.019	0.011	6394	0.08	0.009
25	1	251	226	337	4996	1700	10232	26602	0.019	0.011	6394	0.08	0.009
26	1	251	226	337	4995	1700	10232	26602	0.019	0.011	6394	0.08	0.009
27	1	251	227	337	4994	1700	10232	26602	0.019	0.011	6394	0.08	0.009
28	1	252	227	338	4994	2200	17135	44552	0.011	0.006	21057	0.02	0.003
29	1	252	227	338	4993	2200	17135	44552	0.011	0.006	21057	0.02	0.003
30	1	252	227	338	4992	2200	17135	44552	0.011	0.006	21057	0.02	0.003
31	1	252	227	338	4992	2200	17135	44552	0.011	0.006	21057	0.02	0.003
32	1	252	227	338	4991	2200	17135	44552	0.011	0.006	21057	0.02	0.003
33	1	252	227	338	4990	2200	17135	44552	0.011	0.006	21057	0.02	0.003
34	1	252	228	338	4989	2200	17135	44552	0.011	0.006	21057	0.02	0.003
35	1	252	228	338	4988	2200	17135	44552	0.011	0.006	21102	0.02	0.003
36	1	253	228	338	4987	2200	17135	44552	0.011	0.006	21102	0.02	0.003
37	1	253	228	338	4986	2200	17135	44552	0.011	0.006	21102	0.02	0.003
38	1	253	228	338	4985	2200	17135	44552	0.011	0.006	21102	0.02	0.003
39	1	253	228	339	4984	2200	17135	44552	0.011	0.006	21191	0.02	0.003
40	1	253	229	339	4982	2200	17135	44552	0.011	0.006	21191	0.02	0.003
41	1	253	229	339	4981	2200	17135	44552	0.011	0.006	21191	0.02	0.003
42	1	254	229	339	4980	2200	17135	44552	0.011	0.006	21191	0.02	0.003
43	1	254	229	339	4978	2200	17135	44552	0.011	0.006	21191	0.02	0.003
44	1	254	229	339	4977	2200	17135	44552	0.011	0.006	21191	0.02	0.003
45	1	254	229	339	4975	2200	17135	44552	0.011		21191		0.003
	-								• • • •				

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ĺ	46	1	254	230	339	4974	2200	17135	44552	0.011	0.006	21191	0.02	0.003
	47	1	254	230	340	4972	2200	17135	44552	0.011	0.006	21191	0.02	0.003
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	50	1	255	230	340	4966	2200	17135	44552	0.011	0.006	21236	0.02	0.003
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	52 52	1	255	231	340	4962	2200	17135	44552	0.011	0.006	21236	0.02	0.003
	53 54	1 1	256 256	231 231	340 341	4960 4958	2200 2200	17135 17135	44552 44552	0.011 0.011	0.006 0.006	21236 21236	0.02 0.02	0.003
	5 5	1	256	232	341	4956	2200	17135	44552	0.011	0.006	21236	0.02	0.003
	56	1	256	232	341	4953	2200	17135	44552	0.011	0.006	21236	0.02	0.003
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	59	1	257	233	341	4946	2500	22127	57531	0.009	0.005	32279	0.02	0.002
	60	1	257	233	342	4943	2500	22127	57531	0.009	0.005	32279	0.02	0.002
	61 62	1 1	257 258	233 233	342 342	4940 4938	2500 2500	22127 22127	57531 57531	0.009 0.009	0.005 0.005	32279 32279	0.02 0.02	0.002 0.002
	63	1	258	234	342	4935	2500	22127	57531	0.009	0.005	32279	0.02	0.002
	64	1	258	234	342	4932	2500	22127	57531	0.009	0.005	32337	0.02	0.002
	65	1	258	234	343	4929	2500	22127	57531	0.009	0.005	32337	0.02	0.002
	66	1	259	234	343	4926	2500	22127	57531	0.009	0.005	32337	0.02	0.002
	67	1	259	235	343	4922	2500	22127	57531	0.009	0.005	32337	0.02	0.002
	68	1	259	235	343	4919	2500	22127	57531	0.009	0.005	32337	0.02	0.002
	69 70	1	259	235	343	4916	2500	22127	57531	0.009	0.005	32337	0.02	0.002
	70 71	1 1	260 260	236 236	344 344	4912 4909	2500 2500	22127 22127	57531 57531	0.009	0.005 0.005	32380 32380	0.02 0.02	0.002 0.002
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	77	1	262	238	345	4886	2500	22127	57531	0.008	0.005	32424	0.02	0.002
	78 70	1	262	238	345	4882	2500	22127	57531	0.008	0.005	32424	0.02	0.002
	79 80	1 1	262 262	238 239	345 346	4878 4873	2500 2500	22127 22127	57531 57531	0.008 0.008	0.005 0.005	32424 32481	0.02 0.02	0.002 0.002
	81	1	263	239	346	4869	2500	22127	57531	0.008	0.005	32481	0.02	0.002
	82	1	263	239	346	4864	2500	22127	57531	0.008	0.005	32481	0.01	0.002
	83	1	263	240	346	4860	2500	22127	57531	0.008	0.005	32539	0.01	0.002
	84	1	264	240	347	4855	2500	22127	57531	0.008	0.005	32539	0.01	0.002
	85	1	264	241	347	4851	2500	22127	57531	0.008	0.005	32539	0.01	0.002
	86	1	264	241	347	4846	2500	22127	57531	0.008	0.005	32539	0.01	0.002
	87 88	1 1	265	241	347 348	4841 4836	2500 2500	22127 22127	57531 57531	0.008	0.005	32582 32582	0.01 0.01	0.002 0.002
	89	1	265 265	242 242	348	4831	2500	22127	57531	0.008 0.008	0.005 0.005	32582	0.01	0.002
	90	1	266	242	348	4826	2500	22127	57531	0.008	0.005	32582	0.01	0.002
	91	1	266	243	348	4821	2500	22127	57531	0.008	0.005	32640	0.01	0.002
	92	1	266	243	349	4815	2500	22127	57531	0.008	0.005	32640	0.01	0.002
	93	1	267	243	349	4810	2500	22127	57531	0.008	0.005	32640	0.01	0.002
	94	1	267	244	349	4805	2500	22127	57531	0.008	0.005	32683	0.01	0.002
	95 06	1	267	244	349	4799	2500	22127	57531	0.008	0.005	32683	0.01	0.002
	96 97	1 1	268 268	245 245	350 350	4793 4788	2500 2500	22127 22127	57531 57531	0.008 0.008	0.005 0.005	32683 32741	0.01 0.01	0.002 0.002
	98	1	269	245	350	4782	2500	22127	57531	0.008	0.005	32741	0.01	0.002
	99	1	269	246	351	4776	2500	22127	57531	0.008	0.005	32741	0.01	0.002
	100	1	269	246	351	4770	2500	22127	57531	0.008	0.005	32799	0.01	0.002
	101	1	270	247	351	4764	2500	22127	57531	0.008	0.005	32799	0.01	0.002
	102	1	270	247	351	4758	2500	22127	57531	0.008	0.005	32799	0.01	0.002
	103	1	270	247	352	4752	2500	22127	57531	800.0	0.005	32842	0.01	0.002
	104 105	1 1	271	248	352	4746 4739	2500	22127	57531 57531	0.008	0.005	32842	0.01	0.002
	105 106	1	271 272	248 249	352 353	4739	2500 2500	22127 22127	57531 57531	0.008 0.008	0.005 0.005	32842 32871	0.01 0.01	0.002 0.002
	107	1	272	249	353	4726	2500	22127	57531	0.008	0.005	32871	0.01	0.002
	108	1	272	250	353	4720	2500	22127	57531	0.008	0.005	32957	0.01	0.002
	109	1	273	250	354	4713	2500	22127	57531	0.008	0.005	32957	0.01	0.002
	110	1	273	250	354	4707	2500	22127	57531	0.008	0.005	32957	0.01	0.002
	111	1	274	251	354	4700	2500	22127	57531	0.008	0.005	33015	0.01	0.002
	112	1	274	251	354	4693	2500	22127	57531	0.008	0.005	33015	0.01	0.002
	113 114	1 1	274 275	252 252	355 355	4686 4679	2500 2500	22127 22127	57531 57531	0.008 0.008	0.005 0.005	33087 33087	0.01 0.01	0.002 0.002
	115	1	275	252	355	4679	2500	22127	57531	0.008	0.005	33130	0.01	0.002
	116	1	276	253	356	4665	2500	22127	57531	0.008	0.005	33130	0.01	0.002
	117	1	276	254	356	4657	2500	22127	57531	0.008	0.005	33130	0.01	0.002

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Table B6-4. Results of elastic settlement analyses.

Load	V _s	G / G _{max}	Depth	V _s		Settlem	ent Under (Center of M	/lat		Settleme	ent Under C	Corner of N	l at
(ksf)	Bound	Bound	(ft)	(ft/s)	Sett (in)	Emax (ksf)	ε _{a_initial} (%)	E final (ksf)	ε _{a_final} (%)	Sett (in)	Emax (ksf)	ε _{a_initial} (%)	E final (ksf)	ε _{a_final} (%)
			0 - 12	1200		13255	0.01	1028	0.29		13255	0.00	6776	0.01
	Lower	Lower	13 - 27	1400	0.8	18042	0.01	2733	0.11	0.1	18042	0.00	10806	0.01
	LOWEI	LOWEI	28 - 57	2000	0.0	36820	0.00	14505	0.02	0.1	36820	0.00	28364	0.00
			58 - 117	2200		44552	0.00	20465	0.01		44552	0.00	35739	0.00
			0 - 12	1200		13255	0.01	2352	0.13		13255	0.00	8929	0.01
3	Lower	Mean	13 - 27	1400	0.4	18042	0.01	5535	0.05	0.0	18042	0.00	13459	0.01
3	LOWEI	Mean	28 - 57	2000	0.4	36820	0.00	21176	0.01	0.0	36820	0.00	31670	0.00
			58 - 117	2200		44552	0.00	28145	0.01		44552	0.00	39305	0.00
			0 - 12	1500		20711	0.01	7693	0.04		20711	0.00	16038	0.00
	Mean	Mean	13 - 27	1700	0.2	26602	0.01	12520	0.02	0.0	26602	0.00	21773	0.00
	IVICALI	IVICALI	28 - 57	2200	0.2	44552	0.00	28045	0.01	0.0	44552	0.00	39305	0.00
			58 - 117	2500		57531	0.00	39920	0.01		57531	0.00	52057	0.00
			0 - 12	1200		13255	0.02	587	0.85		13255	0.01	4494	0.03
	Lower	Lower	13 - 27	1400	2.9	18042	0.02	798	0.63	0.1	18042	0.00	8093	0.02
	Lower	Lower	28 - 57	2000	2.9	36820	0.01	7767	0.06	0.1	36820	0.00	23879	0.01
			58 - 117	2200		44552	0.01	12638	0.04		44552	0.00	31149	0.00
			0 - 12	1200		13255	0.02	818	0.61		13255	0.01	7082	0.02
5	Lower	Mean	13 - 27	1400	1.6	18042	0.02	2208	0.23	0.1	18042	0.00	11217	0.01
5	Lowei	Mean	28 - 57	2000	1.6	36820	0.01	14626	0.03	0.1	36820	0.00	28954	0.00
			58 - 117	2200		44552	0.01	21269	0.02		44552	0.00	36521	0.00
			0 - 12	1500		20711	0.01	3326	0.15		20711	0.00	13629	0.01
	Maan	Maan	13 - 27	1700	0.5	26602	0.01	6381	0.08	0.1	26602	0.00	19079	0.01
	Mean	Mean	28 - 57	2200	0.5	44552	0.01	21057	0.02	0.1	44552	0.00	36509	0.00
			58 - 117	2500		57531	0.00	32279	0.02		57531	0.00	49051	0.00
			0 - 12	1200		13255	0.03	587	1.19		13255	0.01	2879	0.06
	1	1	13 - 27	1400	4.5	18042	0.02	798	0.88	0.2	18042	0.01	6013	0.03
	Lower	Lower	28 - 57	2000	4.5	36820	0.01	4199	0.17	0.2	36820	0.00	20760	0.01
			58 - 117	2200		44552	0.01	7565	0.09		44552	0.00	27219	0.01
			0 - 12	1200		13255	0.03	818	0.86		13255	0.01	5421	0.03
7	Lauran	Maan	13 - 27	1400	3.0	18042	0.02	1114	0.63	0.1	18042	0.01	9476	0.02
′	Lower	Mean	28 - 57	2000	3.0	36820	0.01	8721	0.08	0.1	36820	0.00	26333	0.01
			58 - 117	2200		44552	0.01	15186	0.05		44552	0.00	33784	0.01
			0 - 12	1500		20711	0.02	1587	0.44		20711	0.00	11693	0.01
	Moon	Moor	13 - 27		26602	0.00	16906	0.01						
	Mean	Mean	28 - 57	2200	1.3	44552	0.01	14951	0.05	0.1	44552	0.00	33739	0.01
			58 - 117	2500		57531	0.01	25750	0.03		57531	0.00	46352	0.00

Notes:

- 1. Assume 120 ft thick Alluvium layer
- 2. Assume that upper 3 ft of Alluvium will be removed for mat thickness (120 3 = 117 total feet of alluvium)
- 3. Assume Poisson's ratio of Alluvium = 0.3
- 4. Assume unit weight of soil = 114 pcf
- 5. Mat dimensions, B = 450 ft, L = 500 ft
- 6. G/Gmax for Seed and Idriss (1970) curve assumed constant for strains > 1%
- 7. Iterate strains until difference < 0.001%

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APPENDIX B – BEARING CAPACITY AND SETTLEMENT

B7 Results and Conclusion

The following figures and table summarize the results of the bearing capacity and settlement analyses contained herein:

Description Figure . B7-1 Allowable foundation pressure for square and strip footings on alluvium vs. foundation width and foundation embedment (1-inch design settlement). B7-2 Allowable foundation pressure for square and strip footings on alluvium vs. foundation width and foundation embedment (1/2-inch design settlement). B7-3 Immediate settlements for different widths of square and strip footings on alluvium vs. foundation pressure $(d_f = 2 \text{ ft})$. B7-4 Immediate settlements for different widths of square and strip footings on alluvium vs. foundation pressure ($d_f = 6$ ft). B7-5 Long-term settlements for square and strip footings with different depths of foundation embedment. B7-6 Elastic settlement of mat foundation (3 ksf vertical load). B7-7 Elastic settlement of mat foundation (5 ksf vertical load). B7-8 Elastic settlement of mat foundation (7 ksf vertical load). Table Description B7-1 Results of elastic settlement of 500' × 450' mat foundation analyses

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APPENDIX B – BEARING CAPACITY AND SETTLEMENT

Figures B7-1 through B7-4 pertain to bearing capacity and immediate settlement calculations for shallow square and strip footings. For these figures, the results from the Schmertmann method are used since more data (shear wave velocity) is available for this method. The Burland and Burbidge (Terzaghi et al. 1996) method, which is based on blow counts, was not considered reliable. Few blow counts were recorded at the site and due to the high gravel content of the alluvium, are not representative of the more compressible matrix material.

Figure B7-5 presents the long-term settlements evaluation for square and strip footings using the Burland and Burbidge (Terzaghi et al. 1996) method.

Figures B7-6 to B7-8 show the variation with depth of percent of total settlement and percent strain for elastic settlements in the center and corner of a mat foundation. A summary of the predicted total and maximum differential elastic settlements (center and corner of mat foundation) is shown in Table B6-1. The predicted settlements are considered to be very conservative due to the assumption that Young's modulus degrades the same as the shear modulus for sands (alluvium). In actuality, the predicted settlements should be less. Additionally, for the elastic settlement analyses, the stiffness of the mat foundation is not considered to redistribute the loads.

The results of the analyses contained herein appear reasonable for the design of foundations for the expected loading at the Yucca Mountain Project.

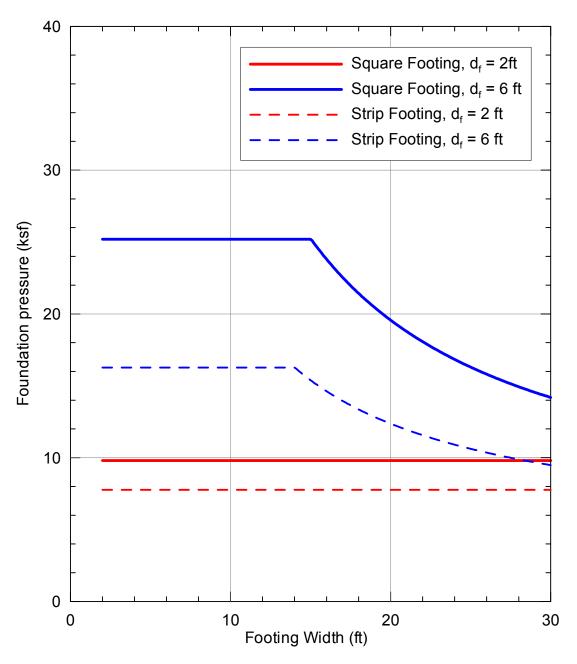


Figure B7-1. Allowable foundation pressure for square and strip footings on alluvium vs. foundation width and foundation embedment (1-inch design settlement).

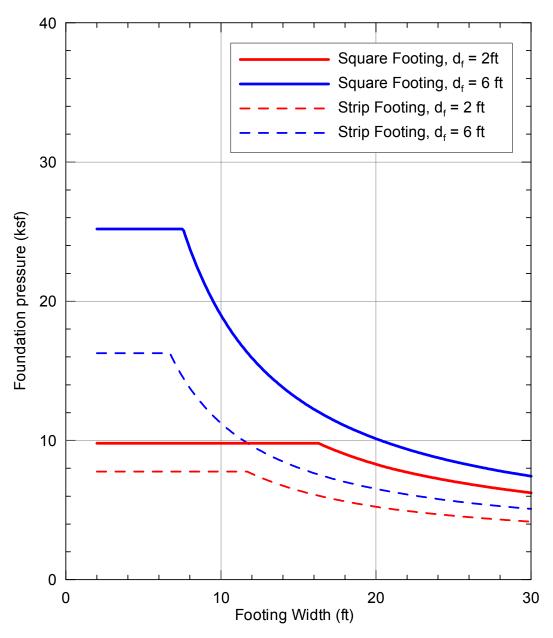


Figure B7-2. Allowable foundation pressure for square and strip footings on alluvium vs. foundation width and foundation embedment (½-inch design settlement).

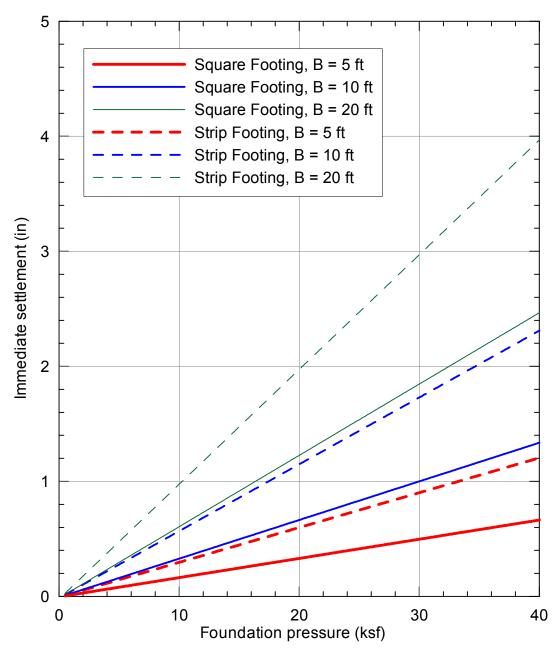


Figure B7-3. Immediate settlements for different widths of square and strip footings on alluvium vs. foundation pressure ($d_f = 2 \text{ ft}$)

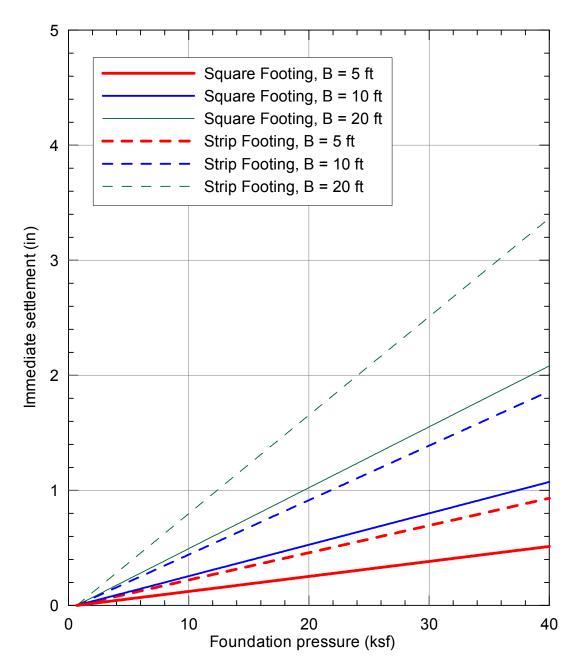


Figure B7-4. Immediate settlements for different widths of square and strip footings on alluvium vs. foundation pressure ($d_f = 6$ ft).

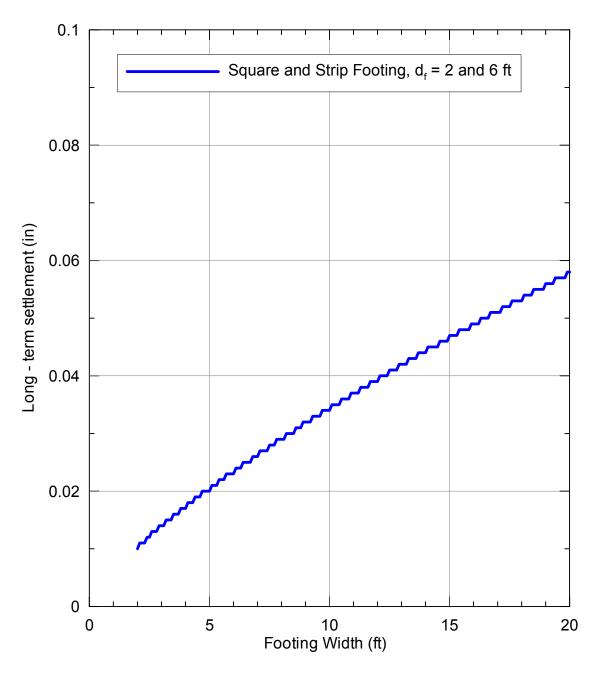
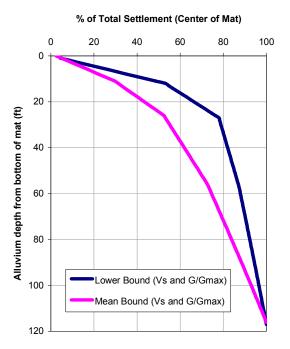


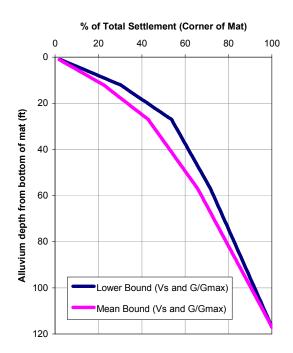
Figure B7-5. Long-term settlements for square and strip footings and different depths of foundation embedment.

APPENDIX B - BEARING CAPACITY AND SETTLEMENT

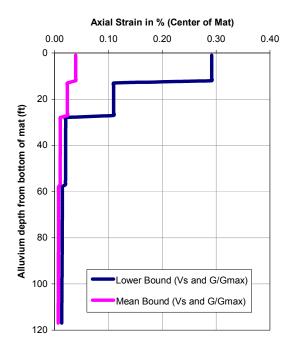
MAT DIMENSIONS: B = 450 FT, L = 500 FT, ASSUME THICKNESS = 3 FT

PERCENT OF TOTAL SETTLEMENT WITH DEPTH FOR 3 KSF LOAD





PERCENT STRAIN WITH DEPTH FOR 3 KSF LOAD



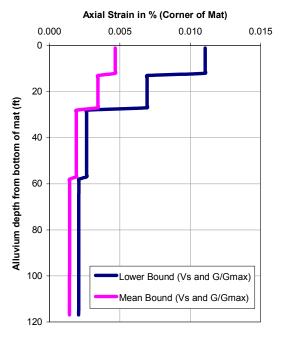


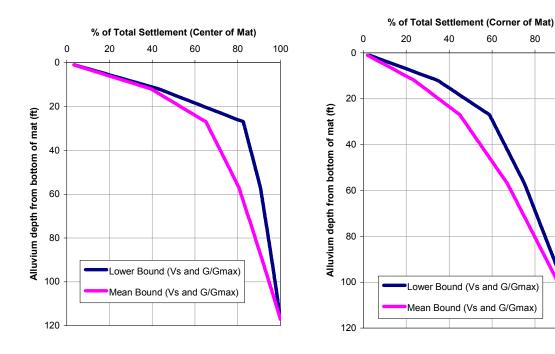
Figure B7-6. Elastic settlement of mat foundation (3 ksf vertical load).

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APPENDIX B - BEARING CAPACITY AND SETTLEMENT

MAT DIMENSIONS: B = 450 FT, L = 500 FT, ASSUME THICKNESS = 3 FT PERCENT OF TOTAL SETTLEMENT WITH DEPTH FOR 5 KSF LOAD



PERCENT STRAIN WITH DEPTH FOR 5 KSF LOAD

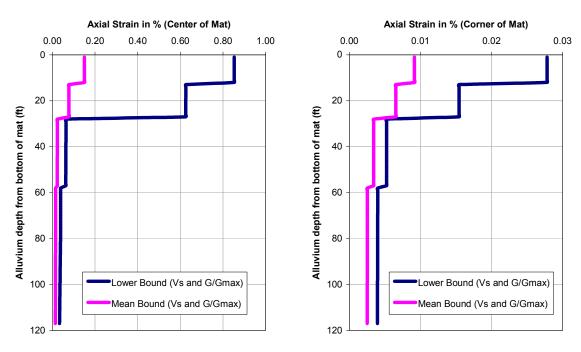
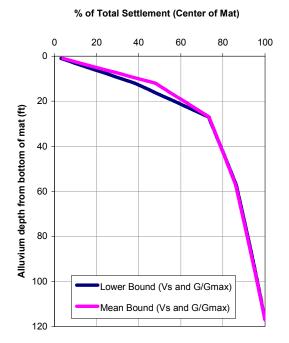


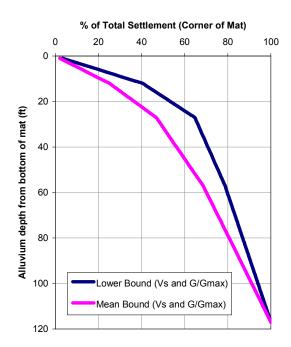
Figure B7-7. Elastic settlement of mat foundation (5 ksf vertical load).

APPENDIX B - BEARING CAPACITY AND SETTLEMENT

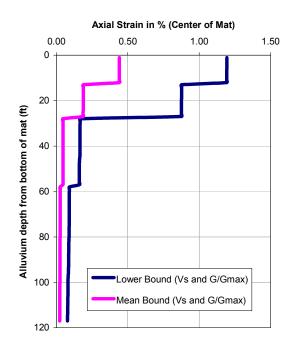
MAT DIMENSIONS: B = 450 FT, L = 500 FT, ASSUME THICKNESS = 3 FT

PERCENT OF TOTAL SETTLEMENT WITH DEPTH FOR 7 KSF LOAD





PERCENT STRAIN WITH DEPTH FOR 7 KSF LOAD



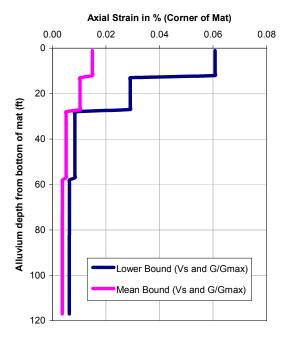


Figure B7-8. Elastic settlement of mat foundation (7 ksf vertical load).

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Table B7-1. Results of elastic settlement of 500' × 450' mat foundation analyses.

Load (ksf)	Total S	Settlement	Maximum Differential
	Center of Mat	Corner of Mat	Corner to Center of Mat (340ft)
3	0.2 - 0.4 in	Negligible	0.4 in
5	0.5 – 1.6 in	~ 0.1 in	1.5 in
7	1.3 - 3 in	~ 0.1 in	3 in

APPENDIX C – LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

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APPENDIX C – LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

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C1 Objective

The purpose of this analysis is to estimate the potential lateral pressures acting at the waste handling surface facilities at Yucca Mountain for yielding and non-yielding walls under static and dynamic conditions. Lateral pressures due to roller and plate compactors and surcharge loads, and lateral pressures acting on temporary shoring are also considered.

Calculations for the resistance to lateral loads resulting from passive resistance or base friction are also performed.

C2 Inputs

Table C2-1 below lists the parameters used in the analysis contained herein (as determined in Section 10 of this study). Although engineered backfill may be used locally at the site (and is stronger than alluvium), the properties of the alluvium are used in the calculations of this analysis for conservatism. A horizontal seismic coefficient, k_h , and Poisson's ratio, ν , are necessary to determine the dynamic lateral earth pressures and, hence, are also listed below. A coefficient of horizontal acceleration, k_h , of 1 g is used in the analysis so that it may be scaled for any selected peak ground acceleration, PGA.

Table C2-1. Parameter Inputs

Parameter	Alluvium	Engineered Fill
Friction angle, φ	39 deg	42 deg
Unit weight, γ	117 pcf	127 pcf
Horizontal seismic coefficient, k _h	1.0*	-
Poisson's ratio, υ	0.3	-

^{*}to be scaled for any selected PGA

Several input parameters are needed in order to estimate the lateral earth pressures created from compaction equipment acting on the soil. Table C2-2 below lists the input parameters used in the analysis herein. If a plate compactor is considered, the width and length of the particular equipment is needed for the analysis and thus is also shown in the table below.

Table C2-2. Compaction Equipment Inputs (Duncan et al. 1991).

Com Name	Type	Static & Dynamic Force (lbf)	Roller Width (in)	Plate Width (in)	Plate Length (in)	Compactor Distance from Wall (ft)
Dynapac CA15D	Single-drum vibratory roller	28,800	66	-	-	2, 3, 5
Dynapac CA25	Single-drum vibratory roller	55,800	84	-	-	2, 3, 5
Ingersoll-Rand DX-70	Walk-behind vibratory roller	6,000	25	-	-	0.5, 1, 2
Bomag BP30	Vibratory plate	6,830	-	15	31.1	0, 0.5, 1
Wacker BS62Y	Rammer plate	3,140	-	13	13	0, 0.5, 1

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APPENDIX C – LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

C3 Background

The surface of the WHB area is currently covered by generally 5 to 9 feet of existing fill. One isolated location recorded 28 ft of fill, but there is some doubt as to its validity (see Section 5, Assumption 10 of BSC 2002b). The fill is underlain by approximately 10 to 120 feet of alluvium BSC (2002a). It is understood that all of the existing fill is to be removed, and the WHB facility will lie directly on the alluvium.

At this time, a 55-foot below-grade pool is planned to be constructed within the wet-process building. Upon completion of its walls, backfill will be placed against it. Hence, stresses induced by compaction equipment must be considered in calculating the earth pressures acting on this wall.

Due to continuous changes in design, other walls (yielding and non-yielding) may potentially be constructed at the YMP site.

C4 Methodology

The following sections outline the methods used and provide the theory and references that are adopted for the analysis.

C4.1 Static lateral earth pressures

The static analysis is based on the Rankine theory (Fang 1991) for determining earth pressures acting on a wall. Lateral at-rest (for non-yielding walls) and active (for yielding walls) earth pressure forces for a vertical wall with horizontal backfill are determined.

C4.2 Dynamic lateral earth pressures (yielding walls)

The seismic analysis for yielding walls is based on simplified methods to determine the dynamic active earth pressure force. The simplified method developed by Seed and Whitman (1970) is used in this analysis to determine the seismic active pressure force increment.

C4.3 Dynamic lateral earth pressures (non-vielding walls)

Procedures outlined in Section 3.5 of ASCE 4-98 are followed to determine the seismic stress increment acting on non-yielding walls. The analysis does not include the dynamic contribution due to surcharge loads.

C4.4 Surcharge pressures

Static lateral surcharge pressures for non-yielding walls are determined based on elastic solutions. Equations used for various surcharge loads for yielding walls (USN 1986) are shown in Figure C7-2 and Figure C7-3 of this appendix. Live loads are not considered in the analysis.

C4.5 Compaction-induced pressures

Procedures outlined in Duncan and Seed (1986), Duncan et al. (1991), and USN (1986) are followed to determine the additional lateral earth pressures that will develop due to various types of compaction equipment. A comparison with the method outlined in USN (1986) is also performed.

APPENDIX C – LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

C4.6 Temporary shoring pressure

The pressure acting on temporary shoring during excavation of the alluvium is estimated using Figure 12.22e of Fang (1991) for soldier piles.

C4.7 Resistance to lateral loads

Sliding friction is estimated based on Table 1 of USN (1986) and per recommendations in BSC (2002a). The Rankine theory is used to estimate passive resistance.

C5 Assumptions

The following assumptions are made in the analysis:

- Walls are to be vertical with a horizontal backfill.
- o Groundwater is deep enough that it will not affect the lateral earth pressures.
- o Bedrock is deep enough that it will not affect the lateral earth pressures.
- o Wall friction is conservatively assumed to be zero.

All of these assumptions are either sufficiently conservative or represent typical standards used in the industry and do not require further verification.

C6 Calculations

All calculations are conducted using the computer program Mathcad. The Mathcad worksheets containing the calculations are all located in Section C8 of this calculation. The following sections outline the procedures performed.

C6.1 Static lateral earth pressures

The following equations are used to determine the static earth pressure coefficients, K, for various conditions:

$$K_o = 1 - \sin \phi$$
 At-rest (C1)

$$K_A = \tan\left(45 - \frac{\phi}{2}\right)^2 \qquad \text{Active} \tag{C2}$$

$$K_P = \tan\left(45 + \frac{\phi}{2}\right)^2$$
 Passive (C3)

The distributed pressure and resultant force of each condition are calculated using the following equations:

$$p = K\gamma H$$
 Distributed pressure (C4)

$$P = K\gamma \frac{H^2}{2}$$
 Resultant force (C5)

where,

H = height of wall

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APPENDIX C – LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

Pressure distribution diagrams are shown in Section C7 of this calculation. Table C6-1 below shows the computed static earth pressure coefficients using the properties of the alluvium. Equivalent fluid weights, $K\gamma$, is multiplied by the wall height, H, to determine lateral earth pressures.

Condition	Earth Pressure Coefficient, K					
	Alluvium	Engineered Fill				
At-Rest, K _o	0.37	0.33				
Active, K _A	0.23	0.20				
Passive, K _P	4.4	5.0				

Table C6-1. Earth Pressure Coefficients.

C6.2 Dynamic Lateral Pressures

Using the simplified method developed by Seed and Whitman (1970), the seismic active earth pressure increment coefficient for a yielding wall is calculated using the following equation:

$$\Delta K_{AE} = \frac{3}{4} k_h \tag{C6}$$

As stated in Section C2, a coefficient of horizontal acceleration, k_h , of 1 g is used in the analysis so that it may be scaled to any given PGA. The distributed pressure and resultant force increment are calculated using the following equations:

$$\Delta p_{ae} = \Delta K_{AE} \gamma H$$
 Distributed pressures (C7)

$$\Delta P_{AE} = \Delta K_{AE} \gamma \frac{H^2}{2}$$
 Resultant force (C8)

Seed and Whitman (1970) suggest that the component of the resultant force may be taken to act at approximately 0.6H above the wall base. The sum of the initial static active earth pressure force (equation C5), and the dynamic active earth pressure force increment (equation C8) produces the dynamic lateral pressure for a yielding wall:

$$P_{AE} = \Delta P_{AE} + P_A \tag{C9}$$

For non-yielding walls, procedures outlined in Section 3.5 of ASCE 4-98 are followed to determine the incremental stresses developed due to seismic loading. A conservative estimate of the dynamic soil pressures may be obtained from Figure 3500-1 of ASCE 4-98 shown as Figure C6-1 below.

APPENDIX C - LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

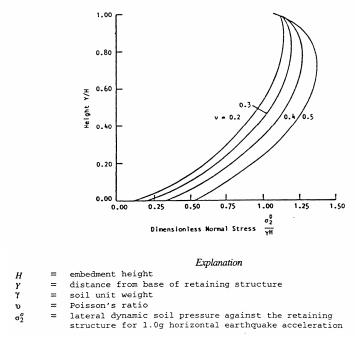


Figure C6-1. Variation of normal dynamic soil pressures for the elastic solution.

Assuming H = 50ft, γ = 117 pcf, and υ = 0.3 (Section 2) for the alluvium, the seismic pressure is determined using Figure C6-1. The pressure can then be scaled to any given coefficient of horizontal acceleration. For the above parameters, Figure C6-2 shows a plot of the seismic pressure coefficient scaled to 1g acceleration versus the unit height of a non-yielding wall. Note that the analysis does not include the dynamic contribution due to surcharge loads.

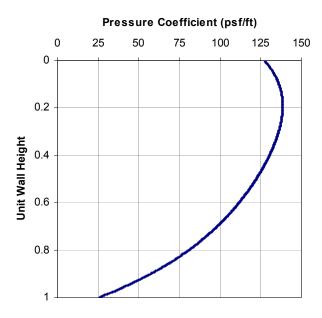


Figure C6-2. Seismic pressure coefficient scaled to 1g versus unit height for non-yielding walls (per ASCE 4-98).

APPENDIX C – LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

C6.3 Surcharge pressures

Static surcharge pressures for non-yielding walls may be calculated as $K_o \times q$ where K_o is the static earth pressure coefficient at rest (0.37), and q is the surcharge load to be applied. The pressure distribution diagrams are shown in Section 7 of this calculation. For yielding walls, refer to the schematic recommendations provided in Section C7 of this calculation. The analysis does not include live loads.

C6.4 Compaction-Induced Pressures

The procedures outlined in Duncan and Seed (1986) are followed to determine the incremental horizontal stresses due to compaction. The equation for the incremental horizontal pressure due to a point load ($\Delta \sigma_h$) presented in Poulos and Davis (1991) is used and modified to taken into account either a roller or plate compactor, the compactor distance from the wall, roller width, plate area, and friction angle of the soil.

Duncan et al. (1991) is used to select various compaction equipment (summarized in Table C2-2). Results are shown as lateral earth pressures due to compaction versus depth. For the analysis, the lateral pressure increment due to compaction is determined and limited to not exceed the passive earth pressures. The pressure increment linearly increases from the depth where it intersects the passive pressure line or where it the pressure is locally at a maximum value near the surface, whichever is larger, until it converges with the at-rest soil pressure line.

The calculations and equations used are provided in Section C8 of this calculation. To avoid redundancy, only one sample calculation for a roller compactor is shown. A check of the results against recommendations from USN (1986) is also included. Figure C6-3 shown below from the USN (1986) manual is used for the check.

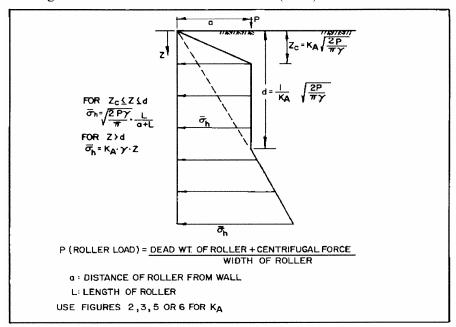


Figure C6-3. Design pressure envelope for non-yielding walls with compaction effects (Figure 13 from USN 1986).

Results of the analysis are shown in Section C7 of this calculation.

APPENDIX C – LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

C6.5 Temporary Shoring Pressure

The pressure of the alluvium acting on temporary shoring provided by soldier piles are estimated using Figure 12.22e in Fang (1991) for sands. The pressure is considered to be uniform acting on the full height of the shoring wall and is expressed by:

$$p_{ts} = 0.65K_{A}\gamma H = 17.5H \tag{C10}$$

C6.6 Resistance to Lateral Loads

Resistance to lateral loads can be developed from passive pressure against the vertical face of the sub-grade walls and footings, and from the friction against the base.

C6.6.1 Passive Pressures

The coefficient for resistance developed from passive pressures was calculated in Section C6.1. The distributed passive pressure is calculated to be:

$$p_p = K_p \gamma H = 515H \tag{C11}$$

C6.6.2 Interface Friction Coefficient

The interface resistance between the soil and structures placed in it is a function of the soil and the structure. Typically, the interface friction coefficient, f, is estimated to be equal to $\tan \phi$, where ϕ is the internal friction angle of the soil. Other adjustments, based on the structural material type and a factor of safety, FS, are also included in the final design value.

The recommended interface friction coefficient between alluvium and concrete is derived from consideration of the soil internal friction angle determined in Section 9 of this study and recommended typical values of interface friction angles published in the literature as described below.

• Internal friction angle, ϕ (see Section C2) = 39 deg

• Estimated base friction, f = $\tan 39 \deg = 0.81$

Bowles 1996 recommends $f = tan(\phi)$. This corresponds to f = 0.81.

USN 1986 (pp. 7.2-121) recommends for cofferdam allowable design to use f = 0.5 on smooth rock, or $tan(\phi)$ otherwise. The worst case is 0.5 (for steel acting against soil). USN 1986 (pp. 7.2-63) recommends ultimate interface friction coefficients between mass concrete and the following soils:

• Clean gravel, gravel sand mixtures and coarse sand 0.55 to 0.6

• Clean fine to med. sand, silty med. to coarse sand, silty or clayey gravel 0.45 to 0.55

Clean fine sand, silty or clayey fine to med. sand
 0.35 to 0.45

The alluvium materials at the site consist of coarse sand and gravel. Hence, the average ultimate interface friction coefficient between mass concrete and the alluvial material is estimated to be about 0.55.

It is recommended to use **0.81** as the ultimate friction coefficient for the alluvium. The ultimate interface resistance for engineered fill is calculated in the same fashion as the alluvium, except that the internal friction angle is 42 degrees. The ultimate interface friction coefficient for the engineered fill is determined to be **0.91**. For engineered

APPENDIX C – LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

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fill, a reduced value corresponding to a factor of safety of at least 1.5, should be used when determining the overall resistance against sliding for a structural element.

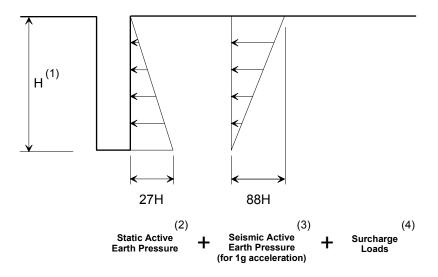
C7 Results / Conclusions

C7.1 Lateral Earth Pressures on Yielding Walls:

The combined lateral earth pressures acting on a yielding wall are as follows:

Static Active + Seismic Active Increment + Static Surcharge

Figure C7-1, Figure C7-2, and Figure C7-3 show the pressure distribution sketch.

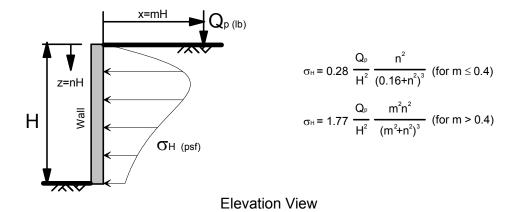


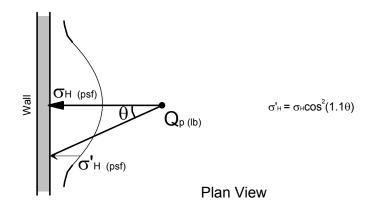
Notes:

- (1) Height of wall, H, is presented in feet.
- (2) Static active earth pressure for alluvium: $K_A = 0.23$, $\gamma = 117$ pcf.
- (3) Seismic active earth pressure for alluvium based on Seed and Whitman (1970) simplified method where $K_h = 1g$ (to be scaled by actual peak ground acceleration, PGA).
- (4) Surcharge loads are shown in next figure.
- (5) Pressures are presented in psf.

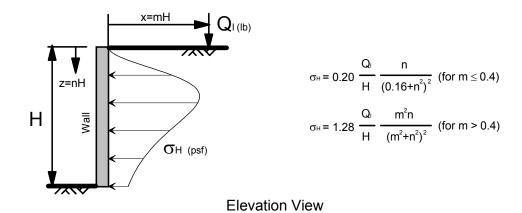
Figure C7-1. Pressure Distribution Sketch for Yielding Walls (not to scale)

APPENDIX C - LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS





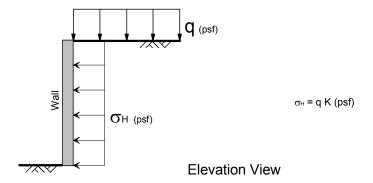
Lateral Pressure due to Point Load



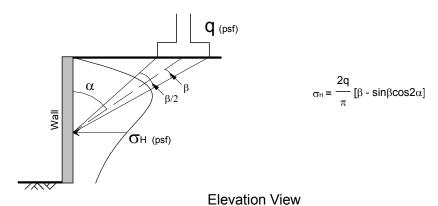
Lateral Pressure due to Line Load

Figure C7-2. Surcharge loads for yielding walls (taken from USN 1986).

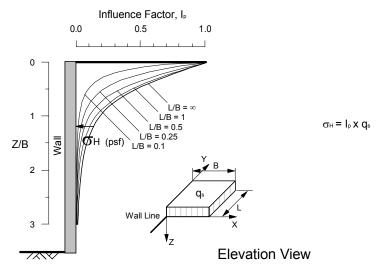
APPENDIX C – LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS



Lateral Pressure due to Uniform Surcharge



Lateral Pressure due to Strip Load



Lateral Pressure due to Footing

Figure C7-3. Surcharge loads for yielding walls continued (taken from USN 1986)

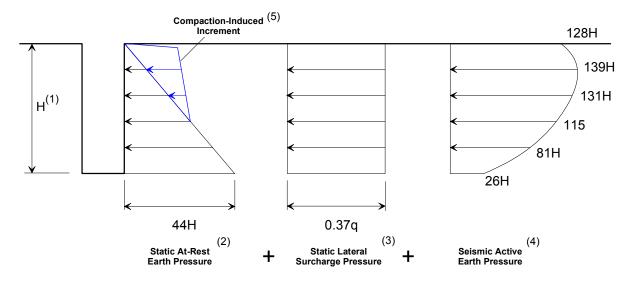
APPENDIX C - LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

C7.2 Lateral Earth Pressures on Non-Yielding Walls

The combined lateral earth pressures acting on a non-yielding wall are as follows:

Static At-Rest + Compaction-Induced Increment + Static Surcharge + Seismic Active

The pressure distribution sketch for non-yielding walls is shown in Figure C7-4 through Figure C7-9.



Notes:

- (1) Height of wall, H, is presented in feet.
- (2) Static at-rest earth pressures for alluvium: $K_0 = 0.37$, $\gamma = 117$ pcf.
- (3) Static lateral surcharge pressure based on K_0q , where q is surcharge to be determined.
- (4) Seismic active earth pressure based on methods from ASCE 4-98, where k_h = 1g (to be scaled by actual peak ground acceleration, PGA); does not include the dynamic contribution due to surcharge loads.
- (5) Compaction-induced pressure increments for specific compaction equipment provided in the next following figures.
- (6) Pressures are presented in psf.

Figure C7-4. Pressure Distribution Sketch for Non-Yielding Walls (not to scale)

APPENDIX C – LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

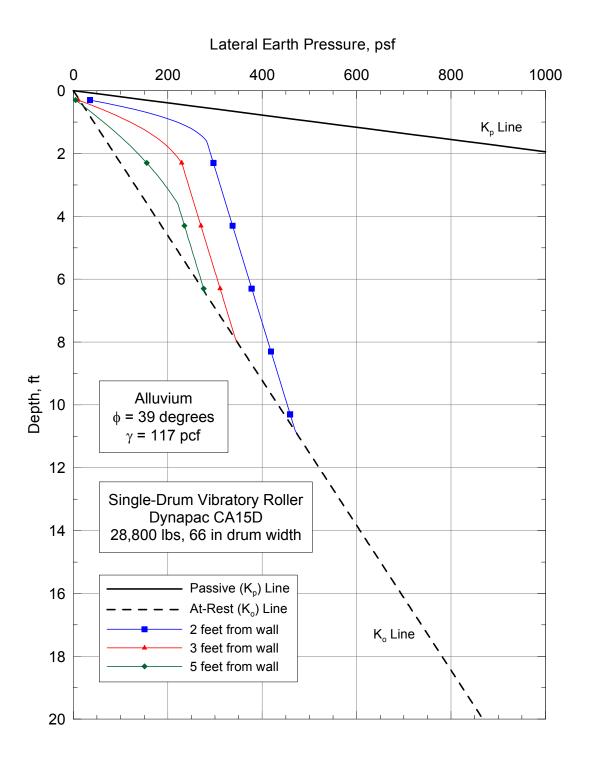


Figure C7-5. Compactor-induced pressures from roller compactor (Dynapac CA15D).

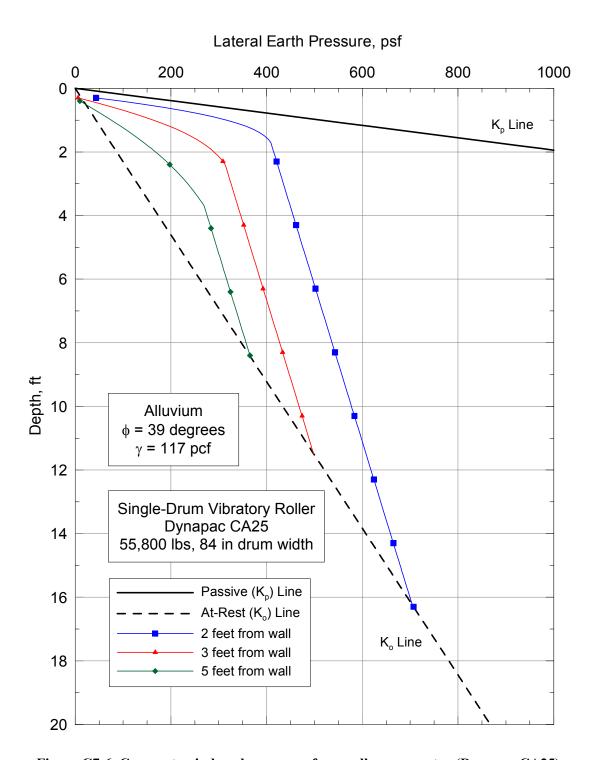


Figure C7-6. Compactor-induced pressures from roller compactor (Dynapac CA25).

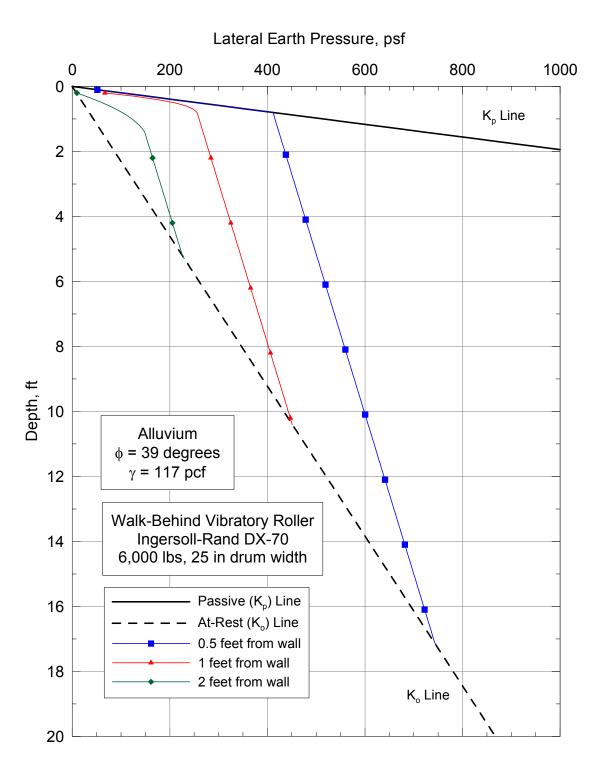


Figure C7-7. Compactor-induced pressures from roller compactor (Ingersoll-Rand DX-70).

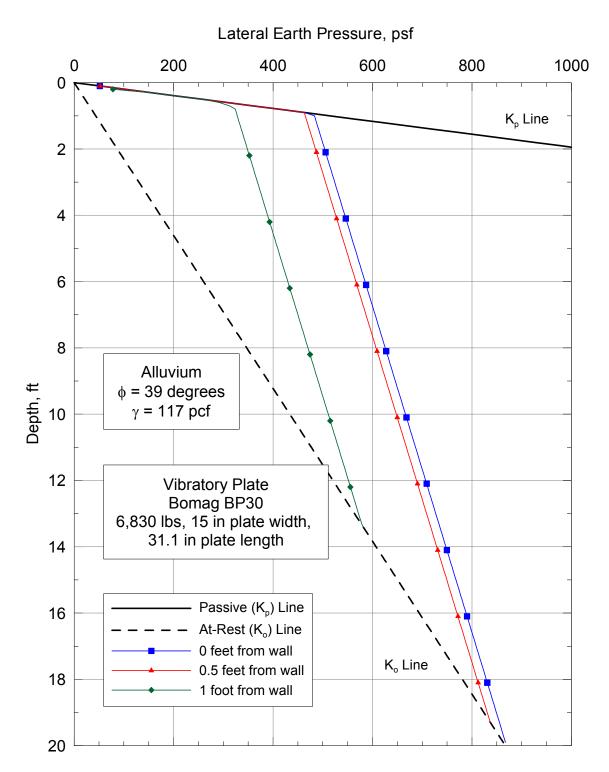


Figure C7-8. Compactor-induced pressures from plate compactor (Bomag BP30).

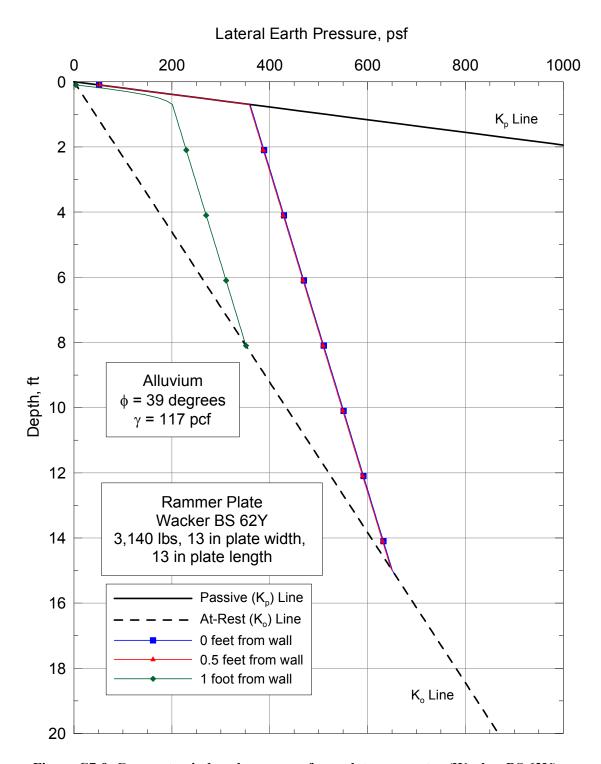


Figure C7-9. Compactor-induced pressures from plate compactor (Wacker BS 62Y).

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C7.3 Temporary Shoring Pressure

The pressure of the alluvium acting on temporary shoring provided by soldier piles is estimated to be 17.5H.

C7.4 Resistance to Lateral Loads

The coefficient for resistance developed from passive pressures was calculated in Section C6.1. The passive pressure against the vertical face of the sub-grade walls and footings is calculated to be 515H.

The average interface friction coefficient between mass concrete and the alluvium or potential engineered fill is estimated to be **0.5**, where tan $\delta = 0.5$. An appropriate factor of safety should be applied to this value.

C8 MathCad Worksheets

APPENDIX C - LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

MathCad Calculations

Alluvium parameters

 $\phi_{alluv} := 39 deg$ Friction angle

 $\gamma_{alluv} := 117 pcf$ Unit weight

v := 0.3 Poisson's ratio

Static Lateral Pressures

For Non-Yielding Walls:

At Rest Pressures (based on alluvium properties)

 $K_o := 1 - \sin(\phi_{alluv})$

 $K_0 = 0.37$

Static At-Rest Earth Pressure Coefficient

 $p_r(H) := K_o \cdot \gamma_{alluv} \cdot H$

Distributed Static At-Rest Earth Pressure

 $p_r(H) = 43.37 \,H \cdot \frac{psf}{ft}$

 $P_R(H) := K_0 \cdot \gamma_{alluv} \cdot \frac{H^2}{2}$

Resultant Static At-Rest Earth Force

For Yielding Walls:

· Active Pressures (based on alluvium properties)

 $K_A := tan \left(45 deg - \frac{\phi_{alluv}}{2} \right)^2$

 $K_A = 0.23$

Static Active Earth Pressure Coefficient

 $\mathsf{p}_{\mathsf{a}}(\mathsf{H}) \coloneqq \mathsf{K}_{\mathsf{A}} \cdot \gamma_{\mathsf{alluv}} \cdot \mathsf{H}$

Distributed Static Active Earth Pressure

 $p_a(H) = 26.618 \, H \cdot \frac{psf}{ft}$

 $P_{A}(H) := K_{A} \cdot \gamma_{alluv} \cdot \frac{H^{2}}{2}$

Resultant Static Active Earth Force

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Dynamic Lateral Pressures (yielding walls)

Active Pressures

$$k_h := 1$$

$$\Delta K_{AE} := \frac{3}{4} \cdot k_h$$

 $\Delta K_{A\,F} = 0.75 \label{eq:delta-K}$ Coefficient

$$\Delta p_{ae}(H) := \Delta K_{AE} \cdot \gamma_{alluv} \cdot H$$

Distributed Seismic Active Earth Pressure Increment

Seismic Active Earth Pressure Increment

$$\Delta p_{ae}(H) = 87.75 \, H \cdot \frac{psf}{ft}$$

$$\Delta P_{\text{AE}}(H) := \Delta K_{\text{AE}} \cdot \gamma_{\text{alluv}} \cdot \frac{H^2}{2}$$

Resultant Seismic Active Earth Pressure Force Increment. It is suggested that the component may be taken to act at approximately 0.6H per Seed and Whitman (1970).

$$P_{AE}(H) := \Delta P_{AE}(H) + P_{A}(H)$$

Sum of initial static active earth pressure force and dynamic active earth pressure force increment

Dynamic Lateral Pressures (nonyielding walls)

$$a := 1$$
 Acceleration [g], to be multiplied by kh

$$d := 0 \text{ ft}, 0.1 \text{ ft... } H$$

Coefficients for ASCE 4-98 seismic stresses:

$$M := \begin{pmatrix} 1.0829167 & 0.070869084 & -3.1836133 & 3.5952709 & -2.0641442 \\ 1.0888187 & 1.1176702 & -4.0053697 & 4.333532 & -2.3203657 \\ 1.0968336 & 1.7075112 & -5.3728278 & 5.6727378 & -2.7717642 \\ 1.0788775 & 2.2549514 & -5.719958 & 5.1033643 & -2.1980003 \\ \end{pmatrix}$$

$$y(\text{eqtn}, x) := M_{\text{eqtn}, 0} + M_{\text{eqtn}, 1} \cdot x + M_{\text{eqtn}, 2} \cdot x^2 + M_{\text{eqtn}, 3} \cdot x^3 + M_{\text{eqtn}, 4} \cdot x^4$$

$$eqtn_0 := \begin{bmatrix} trunc \left(\frac{v}{0.1} \right) - 2 & \text{if } v \ge 0.2 \\ 0.2 & \text{otherwise} \end{bmatrix}$$
 eqtn₁ := eqtn₀ + 1

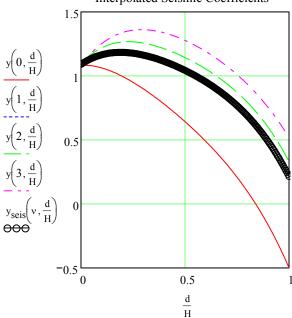
APPENDIX C - LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

$$eqtn_0 = 1 eqtn_1 = 2$$

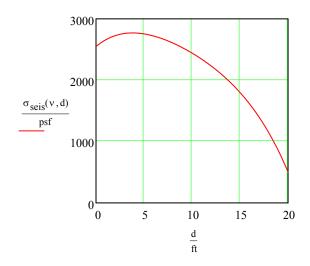
$$y_{seis}(v,d) := \frac{v - \left(eqtn_0 + 2\right) \cdot 0.1}{.1} \cdot \left(y\left(eqtn_1, d\right) - y\left(eqtn_0, d\right)\right) + y\left(eqtn_0, d\right)$$

The interpolated seismic coefficients per ASCE 4-98 are shown below:

Interpolated Seismic Coefficients



The seismic pressure increment are calculated fror $\sigma_{seis}(v, x) := y_{seis}\left(v, \frac{x}{H}\right) \cdot \gamma_{alluv} \cdot H \cdot a$



$$\sigma_{\text{seis}}(v, 0H) = 127.392 \text{ H} \cdot \frac{\text{psf}}{\text{ft}}$$

$$\sigma_{\text{seis}}(v, .2H) = 138.422 \text{ H} \cdot \frac{\text{psf}}{\text{ft}}$$

$$\sigma_{\text{seis}}(v, .4H) = 130.218 \text{ H} \cdot \frac{\text{psf}}{\text{ft}}$$

$$\sigma_{\text{seis}}(v,.6H) = 111.479 \text{ H} \cdot \frac{\text{psf}}{\text{ft}}$$

$$\sigma_{\text{seis}}(v, .8H) = 80.48 \, H \cdot \frac{\text{psf}}{\text{ft}}$$

$$\sigma_{\text{seis}}(v, H) = 25.071 \text{ H} \cdot \frac{\text{psf}}{\text{ft}}$$

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APPENDIX C - LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

Compaction-Induced Lateral Earth Pressures (Duncan and Seed 1986 and Duncan et al. 1991 procedure)

Methodology:

- 1. Solve Bousinesq stress due to load
- 2. Reduce Item 1 using factor, F, and add to Ko stress
- 3. Limit Item 2 so as to not exceed Kp stress
- 4. Find depth to peak stress
- 5. Smooth relationiship below peak Bousinesq stress

Input:

Example using Dynapac CA15D

P := 28800 lbf Static + dynamic force of compactor

 $CHD := 0.01 \, ft$ Closest distance from compactor edge to wall

 $\phi := 39 \text{deg}$ Internal friction angle of alluvium

 $\gamma := 117 pcf$ Unit weight of alluvium

Type := "roller" Type of analysis (plate or roller), **use lower case**

width := 66in Compactor width length := 31.1in Compactor length

 $\chi := \frac{4 - 3\sin(\phi)}{8 - 4\sin(\phi)}$ Poisson's ratio per Duncan et. al (1991)

<u>Calculations</u>:

Roller Calcs:
$$\mathbb{R}(x,y,z) := \sqrt{x^2 + y^2 + z^2}$$

$$\Delta\sigma_h(x,y,z) := \frac{P}{2\cdot\pi} \left[\frac{3\cdot\left(x^2+y^2\right)\cdot z}{R(x,y,z)^5} - \frac{1-2\cdot\nu}{R(x,y,z)^2+z\cdot R(x,y,z)} \right]$$
 Equation 2.2b from pp. 16 of Poulos and Davis (1991)
$$\nu = 0.385$$

Bousinesq stress due to compaction:

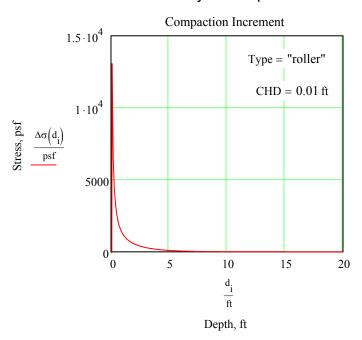
$$\Delta\sigma(d) := \begin{bmatrix} \frac{2}{\text{width}} \left(\int_{\text{CHD}}^{\text{CHD+width}} \Delta\sigma_h(x,0 \text{ft},d) \ dx \right) & \text{if Type} = \text{"roller"} \\ \frac{2}{\text{width} \cdot \text{length}} \left(\int_{\text{CHD}}^{\text{CHD+width}} \int_{\text{CHD}}^{\frac{\text{length}}{2}} \Delta\sigma_h(x,y,d) \ dy \ dx \right) & \text{otherwise} \end{bmatrix}$$

APPENDIX C - LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

$$i := 0 .. \frac{H}{0.1 ft}$$

$$d_i := i \cdot 0.1 ft$$

The unmodified stresses due only to compaction are shown below:



This stress increment is modified per Duncan et al. (1986):

$$\alpha := 0.7794423 - 0.51338219 \cdot e^{-19.574578 \cdot \sin(\phi)^{4.9554863}} \qquad \qquad \alpha = 0.708$$

$$F := \frac{5^{\alpha}}{4} - 0.25$$

$$K_p := tan \left(45 deg + \frac{\phi}{2} \right)^2$$
 passive pressure $K_p = 4.395$

$$K_0 := 1 - \sin(\phi)$$
 at-rest pressure $K_0 = 0.371$

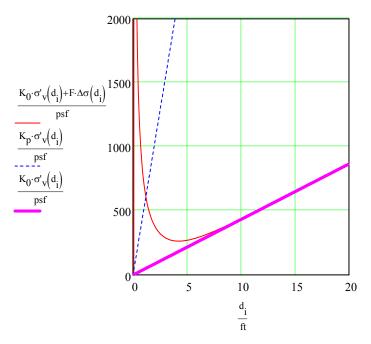
$$K_2 := K_0 \cdot (1 - F)$$
 $K_2 = 0.174$

$$\sigma'_{V}(d) := \gamma \cdot d$$

APPENDIX C - LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

Limit stress in upper portion of wall to passive pressure

$$\begin{split} \sigma'_h(d) := & \left| \begin{matrix} K_0 \cdot \sigma'_v(d) + F \cdot \Delta \sigma(d) & \mathrm{if} \ \left(K_0 \cdot \sigma'_v(d) + F \cdot \Delta \sigma(d) \right) \leq K_p \cdot \sigma'_v(d) \\ K_p \cdot \sigma'_v(d) & \mathrm{otherwise} \end{matrix} \right| \end{split}$$



$$F = 0.531$$

$$K_0 = 0.371$$

$$K_p = 4.395$$

$$K_2 = 0.174$$

Find critical depth where stress, $\left.\sigma'\right|_{h}(d)$, is a maximum off the Ko-line

$$k_i := \sigma'_h(d_i) - K_0 \cdot \sigma'_v(d_i)$$

k_max is the maximum stress increment off the Ko-line

$$depth := .2ft$$

$$d_c := root(max(k) - \sigma'_h(depth) + K_0 \cdot \sigma'_v(depth), depth)$$

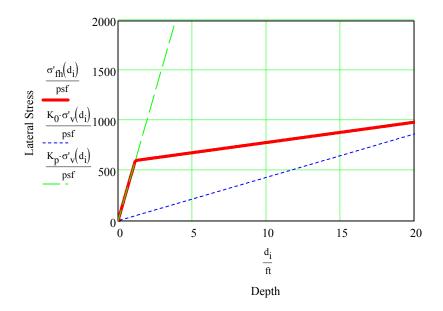
$$d_c = 1.174 \, ft$$
 Critical depth

The total of static and compaction stresses for the wall are determined as follows (note: stress must not go below Ko line):

$$\begin{split} \sigma'_{fh}(\textbf{d}) &:= \left[\begin{array}{l} \sigma'_h(\textbf{d}) & \text{if } \textbf{d} \leq \textbf{d}_c \\ \text{otherwise} \\ \left[\sigma'_h \! \! \left(\textbf{d}_c \right) + \textbf{K}_2 \cdot \sigma'_v \! \! \left(\textbf{d} - \textbf{d}_c \right) \right. & \text{if } \left(\sigma'_h \! \! \left(\textbf{d}_c \right) + \textbf{K}_2 \cdot \sigma'_v \! \! \left(\textbf{d} - \textbf{d}_c \right) \right) \geq \textbf{K}_0 \cdot \sigma'_v (\textbf{d}) \\ \left[\textbf{K}_0 \cdot \sigma'_v (\textbf{d}) \right. & \text{otherwise} \end{split} \right] \end{split}$$

APPENDIX C - LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

The combined static and compaction stresses are shown below:



$\sigma'_{fh}(d_i) =$	
-2.029·10 ⁴	psf
51.427	
102.855	
154.282	
205.709	
257.136	
308.564	
359.991	
411.418	
462.846	
514.273	
565.7	
604.084	
606.117	
608.149	
610.182	

Check results against NavFac DM7.02 (USN 1986) Using equations from Figure 13:

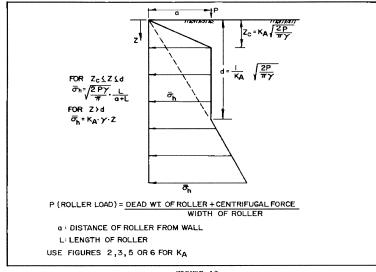


FIGURE 13 Horizontal Pressure on Walls from Compaction Effort

APPENDIX C - LATERAL EARTH PRESSURES AND RESISTANCE TO LATERAL LOADS

Redefine some variables to correspond to NavFac:

$$K_a := \frac{1}{K_p}$$

$$P_{\text{width}} = \frac{P}{\text{width}}$$

$$z_i := d_i$$
 $a := 0$ ft

$$K_a = 0.228$$

$$K_{a} := \frac{1}{K_{p}}$$

$$P_{a} := \frac{P}{\text{width}}$$

$$K_{a} = 0.228$$

$$P = 5.236 \times 10^{3} \frac{\text{lbf}}{\text{ft}}$$

$$z_{cr} \coloneqq K_a \sqrt{\frac{2 \cdot P}{\pi \cdot \gamma}} \qquad \qquad \text{Critical depth}$$

$$z_{cr} = 1.214 \, ft$$

$$d := \frac{1}{K_a} \sqrt{\frac{2 \cdot P}{\pi \cdot \gamma}}$$

Depth where compaction effects merge with pressure line

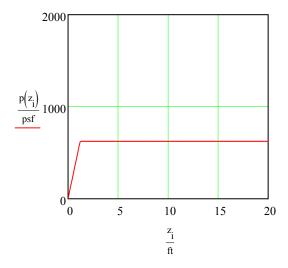
$$d = 23.462 \, ft$$

$$p(z) := \left[\left(\frac{2 \cdot P \cdot \gamma}{\pi} \cdot \frac{\text{length}}{a + \text{length}} \right) \cdot \left(\frac{z}{z_{cr}} \right) \text{ if } z < z_{cr} \right]$$
otherwise
$$\left[\sqrt{\frac{2 \cdot P \cdot \gamma}{\pi} \cdot \frac{\text{length}}{a + \text{length}}} \right] \text{ if } z_{cr} \le z \le d$$

$$K_{a} \cdot \gamma \cdot z \text{ if } z > d$$

$$\sqrt{\frac{2 \cdot P \cdot \gamma}{\pi}} \cdot \frac{\text{length}}{a + \text{length}} \quad \text{if} \quad z_{cr} \le z \le c$$

$$K_a \cdot \gamma \cdot z \quad \text{if} \quad z > d$$



Solution assumes that the compactor is used at the wall (distance from wall = 0')

This matches relatively well with the solution obtained from the Duncan et al. (1986) and (1991) solution.

$$psf = \frac{lbf}{f^2}$$
 $pcf = \frac{lbf}{f^3}$

$$pcf \equiv \frac{lbf}{a^3}$$