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ENGINEERING GEOLOGY OF THE NORTHERN ESTANCIA  
VALLEY, NORTH-CENTRAL NEW MEXICO

By

Don Barrie

N.M.I.M.T.  
LIBRARY  
SOCORRO, N.M.

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New Mexico Institute of Mining and Technology

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## ABSTRACT

The study area occupies approximately 655 mi<sup>2</sup> and is located in the northern part of the Estancia Valley. Rocks of Precambrian to Tertiary age are present. Late Cenozoic valley-fill material is up to 405 ft thick and consists of alluvial and lacustrine deposits. Historically, seismicity has been low; there are no known active faults present. Depth to ground water ranges from 30 to over 350 ft.

Clayey, lacustrine soil contains varying amounts of kaolinite, illite, smectite, and mixed-layer illite-smectites. The plasticity of lacustrine soil is probably more strongly affected by the amount of clay-size material present than by clay mineralogy.

The results of various soil index tests and field observations are used to assess five important engineering concerns:

- (1) Conditions favorable for liquefaction are not present in the study area. Soils are not typically cohesionless, the water table is too deep, and predicted ground accelerations likely to be caused by earthquakes are too low.
- (2) Generally, the shrink-swell potential of soil units is low, although expansive soils occur locally.
- (3) Soils most likely to be collapsible include the granular varieties of older alluvium.
- (4) Granular soils are fairly well suited as subgrade (natural substratum) for road construction purposes. Cohesive soils are less suitable, although typically high N-values indicate that they can probably furnish adequate support for most traffic loads.
- (5) Generally, both granular and cohesive soils exhibit good bearing capacity for foundations due to cementation by secondary calcium carbonate. Consolidation of saturated, lacustrine soil is not a critical concern due to its low compressibility. Foundation settlement related to ground water withdrawal is not currently a concern.

Rock index test results and field observations are used to assess four additional engineering concerns:

- (1) Most surface rock in the study area is not rippable, whereas lower-strength rocks occurring primarily in the shallow subsurface are rippable.
- (2) The shrink-swell potential of rock units is minimal; however, weakly consolidated Dockum Group siltstone and mudstone are locally expansive.
- (3) The overall resource potential for crushed rock aggregate, building stone, sand, gravel, and caliche, is high and should be adequate to support most future construction projects.
- (4) For lower-strength rocks, allowable contact pressures imposed by large foundation loads may be limited by rock strength and rock quality. In contrast, contact pressures on higher-strength rocks are limited only by rock quality.

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## INTRODUCTION AND BACKGROUND

### NMBMMR SSC Project

In December 1985 the New Mexico Bureau of Mines and Mineral Resources (NMBMMR) produced a report on the Estancia Valley in north-central New Mexico (NMBMMR Staff, 1985). They concluded that the site appeared suitable for construction of the Superconducting Super Collider (SSC), an oval-shaped particle accelerator approximately 53 mi in circumference and 12 ft in diameter. A preliminary geotechnical characterization of the Estancia Valley site begun in June 1986 and completed in January 1987, encountered no "fatal flaws" that would preclude siting the SSC in the northern Estancia Valley (Johnpeer et al., 1987). A further study documenting ground subsidence potential, tunneling conditions, and water quality and availability was conducted between January and May 1987. The scope of work was interdisciplinary and included: aerial photograph interpretation, engineering geologic mapping, drilling, reviewing published seismicity data, ground motion monitoring, shallow trenching, downhole geophysical logging, surface geophysical surveys, aerial flyovers, and laboratory testing. The results of this effort were summarized in volume 3 of the eight-volume SSC proposal submitted to the U.S. Department of Energy (DOE) for review September, 1987. As a project research assistant, the writer was involved principally with the laboratory testing of soil and rock samples and also participated in sample collecting, engineering geologic mapping, logging of

geotechnical boreholes and trenches, and aerial reconnaissance. The final SSC report (New Mexico SSC Proposal, 1987, v. 3) and accompanying laboratory data released a large amount of new information. However, due to time constraints and also to the objectives of the study, emphasis was placed primarily on identifying any geotechnical conditions that would adversely affect the Estancia Valley site as a possible location for the SSC.

### Purpose and Scope of Present Study

The main objective of this study is to investigate in detail some of the important engineering (geotechnical) properties of soil and rock units within the northern Estancia Valley. This objective was met by analyzing the abundant laboratory data that became available as a result of NMBMMR studies and by completing additional laboratory tests beyond the scope of the NMBMMR effort. This study assigns average engineering properties to soil and rock units present in the study area. Most importantly, this information was used to evaluate a number of important engineering concerns. It is hoped that this study will aid planners, contractors, and engineers in any future development of the northern Estancia Valley.

### LOCATION AND EXTENT

The study area is located in the northern part of the Estancia Valley in

central New Mexico (Fig. 1). It occupies an area of approximately 655 mi<sup>2</sup> and is bounded on the north by 35°15' N latitude, on the south by 34°50' N latitude, on the east by 105°49' W longitude, and on the west by 106°15' W longitude. From north to south the study area is 26.5 miles long and from east to west it is 24.7 miles wide. Most of it lies within Santa Fe and Torrance Counties, although a small portion extends into Sandoval and Bernalillo Counties toward the west. The town of Moriarty, located in the south-central part of the study area, is approximately 35 miles east of Albuquerque.

## TOPOGRAPHY

The Estancia Valley is an elliptical, topographically closed basin of approximately 2,000 mi<sup>2</sup> located near the geographic center of New Mexico (Fig. 1). To the north the valley is bounded by a broad saddle at about 6,500 ft in elevation, which separates the Estancia Valley from the Galisteo Valley. To the west the valley is bordered by the Sandia and Manzano Mountains. South Mountain and San Pedro Mountain border the valley to the northwest. Chupadera Mesa borders the valley to the south; the Pedernal Hills and Rattlesnake Hill border it to the east.

The study area ranges in elevation from 8,748 ft at the top of South Mountain to about 6,128 ft along the southern boundary near McIntosh (Fig. 2). Except for local relief due to drainages and minor hills, the central part of the valley floor in

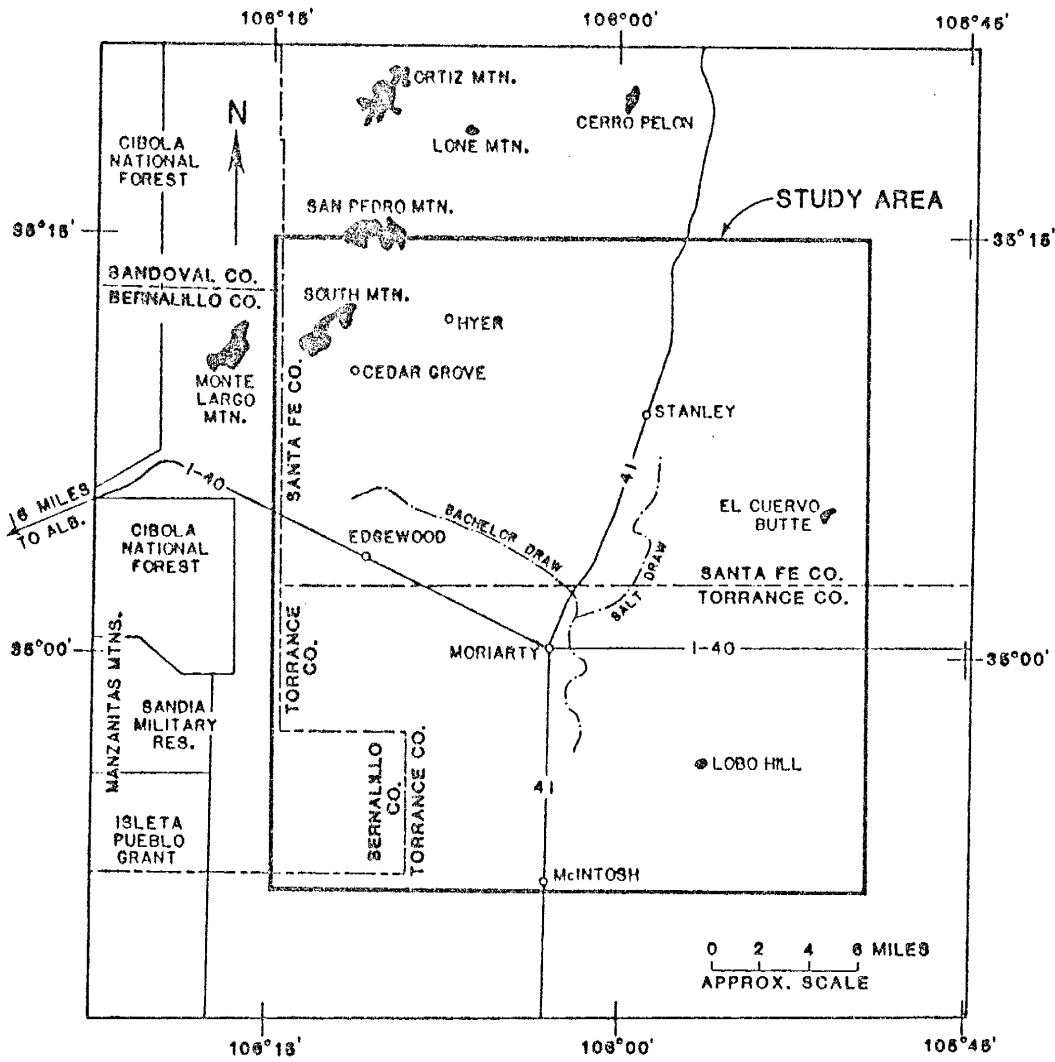
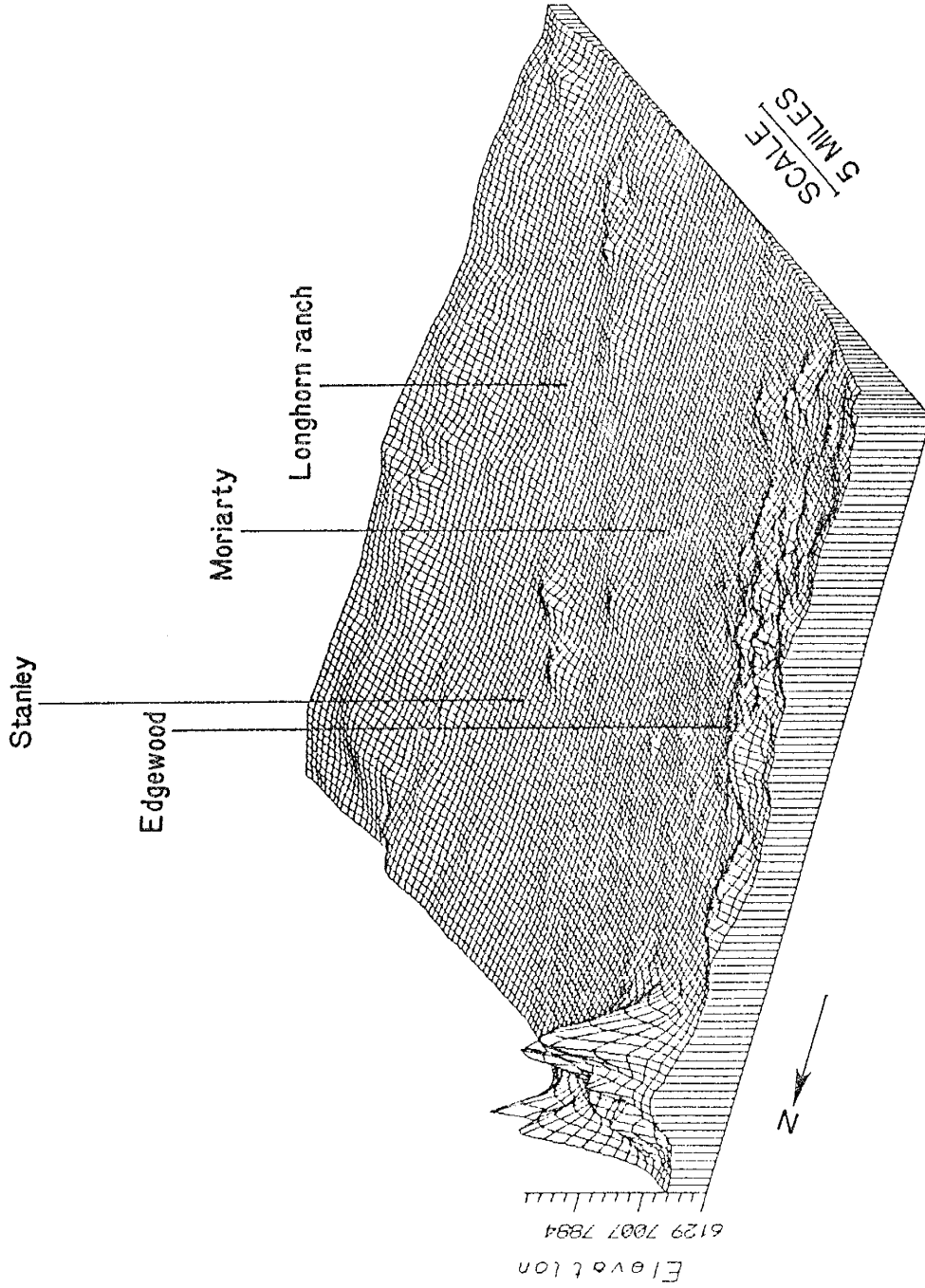


FIGURE 1--Location map (from Johnpeer et al., 1987).



(5)

FIGURE 2--Three-dimensional topographic plot of the study area (vertically exaggerated).



the study area is relatively flat and slopes toward the south at approximately 15 ft per mile. Westward from the break in slope above the valley floor, slopes rise at about 40 to 50 ft per mile toward the base of the Manzano and Sandia Mountains. Eastward, slopes rise between about 40 and 100 ft per mile toward the Pedernal Hills.

## SETTING

### Climate

Rainfall data collected from various locations in the Estancia Valley display annual averages between 11.2 and 13.7 inches (Gabin and Lesperance, 1977). July and August are the wettest months, typically receiving between 2 to 3 inches each month. Most of this moisture falls during brief, heavy thundershowers. The amount of precipitation increases rapidly with elevation, rising to over 22 inches per year at higher mountain elevations such as at the summit of South Mountain (Soil Conservation Service, 1970).

Temperatures within the valley vary greatly both on a diurnal and an annual scale. The daily temperature range throughout the year is usually between 30° and 40° F, and the annual range is as much as 100° to 120° F (Soil Conservation Service, 1970). The monthly mean temperature during the winter months (December and January) is about 30° to 35°F. During the summer months (June, July, and August) it is about 65° to 70° F (New Mexico SSC Proposal, 1987, v. 3).

### Vegetation and Land Use

Native plants within the study area consist mainly of various grasses with some pinon and juniper (Soil Conservation Service, 1975). Grazeland predominates in the northern and eastern parts of the study area. Irrigated cropland predominates in the central part, where alfalfa, corn, potatoes, sugar

beets, pinto beans, barley, wheat, and pumpkin are grown (Soil Conservation Service, 1970). Approximately 85% of the land in the study area is privately owned. State and federal land leases account for the remaining 15% (Fig. 3). Population centers are clustered in the central and western parts of the study area in McIntosh, Moriarty, Stanley, and Edgewood. Land uses adjacent to the study area include Cibola National Forest along the Sandia and Manzano Mountains, the Isleta Pueblo land grant to the southwest, and the Sandia Military Reservation to the west (Fig. 1).

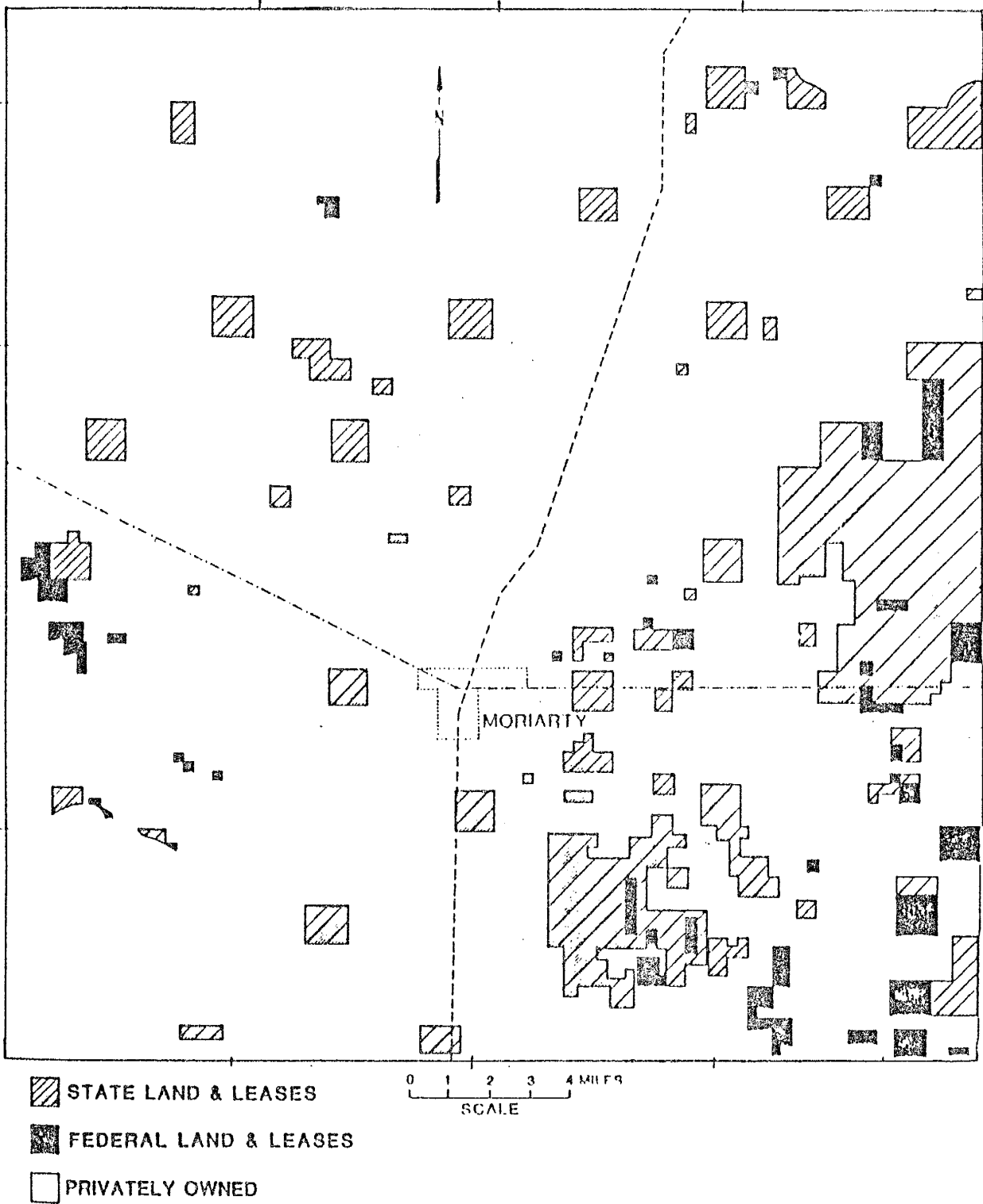


FIGURE 3--Land ownership (from Johnpeer et al., 1987).

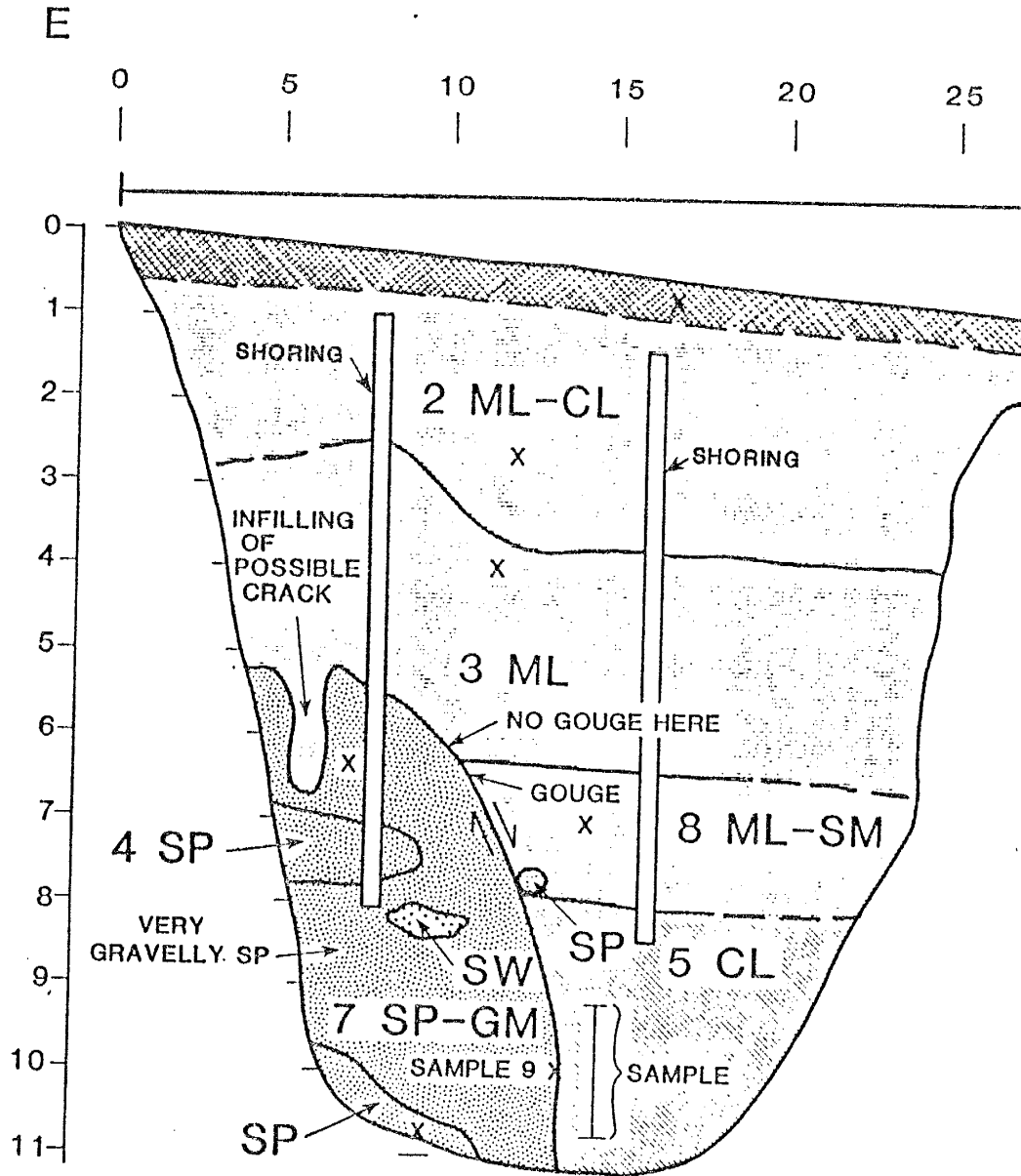


FIGURE 8--Partial log of backhoe trench 10 (BH-10, Appendix II) showing previously unmapped fault (USCS designations from ASTM D2487-85).

## HYDROLOGIC SETTING

Due to its topographic closure the Estancia Valley has no external drainage. Recharge by precipitation is the source of all ground water (Jenkins, 1982). Based on 14 well hydrographs (New Mexico SSC Proposal, 1987, v. 3) the average rate of decline of the water table has been approximately 1 ft per year (since 1955).

## Principal Aquifers

Valley fill is the principal aquifer in the main part of the Estancia Valley. As stated previously, the valley fill is at least 405 ft thick near Stanley and consists of four units: (1) Quaternary-Tertiary alluvium, (2) older (Quaternary) alluvium, (3) lacustrine deposits, and (4) younger (Holocene) alluvium. The lacustrine deposits generally confine ground water to the underlying alluvium (Jenkins, 1982), although the interbedded sands present in the lake deposits cause perched water locally. The quality of water in the valley fill varies with locality. West of NM-41, in Torrance County (Fig. 1), water from wells in the valley fill is satisfactory for domestic use. East of NM-41, water quality is poorer. Most of the water in this area is unsatisfactory for domestic use, and some of it is unsatisfactory for irrigation and undesirable for stock animals (Smith, 1957).

The arkosic limestone member of the Madera Formation is the principal aquifer in the western part of the study area. Much of the ground-water recharge

for the study area occurs in the Madera Formation of the Manzano Mountains to the west (Jenkins, 1982). As an aquifer the Madera is less reliable for yield and permanence than the valley fill (Smith, 1957). The Madera is a heterogeneous, fractured-rock aquifer in which availability of ground water is site dependent. Permeability and porosity are secondary, controlled mainly by fractures, joints, and solution cavities. Well yields often vary greatly over short distances because saturated fractures are not evenly distributed throughout the rock (Jenkins, 1982). High-yield wells in the Madera, capable of producing over 1,000 gpm, occur locally. Dry wells occur also. Ground water is available in sufficient quantities for domestic use from most of the area underlain by the Madera Formation (Jenkins, 1982).

Sandstone of the Glorieta Formation is the principal aquifer in the central and east-central parts of the study area. Permeability of the Glorieta is generally low, although locally, well yields of more than 3,000 gpm have been reported (Smith, 1957). Water from the Glorieta is used for irrigation, stock and for domestic purposes.

The Yeso Formation is the main aquifer in the areas east of Moriarty and Lobo Hill (Fig. 1). Well yields from the Yeso in this area are typically about 15 gpm, although yields of over 3,000 gpm exist locally where the Yeso is fractured (Smith, 1957). Permeability and porosity in the Yeso are mostly in sandstone, although some water wells produce from cavities in limestone (Titus, 1969). Water quality from the Yeso varies from satisfactory to unsatisfactory for domestic

use and irrigation. Generally, water from the Yeso is satisfactory for stock (Smith, 1957).

#### Depth to Ground water

A generalized depth to ground water map, based on 68 well measurements, is shown for the central part of the study area in Figure 9. Water depths were taken from selected well logs on file at the office of the State Engineer (written comm., 1987), from data provided by Robert White of the USGS (written comm., 1987), and from geotechnical borehole logs (Appendix I).

Depth to ground water is dependent both upon the potentiometric surface and the surface topography. Generally, depth to ground water in the area shown in Figure 9 increases with elevation so that it is deeper in the northern part of this area than in the southern part. Ground-water depths are also greater along the margins this area than in the central part of the area. Depths to ground water range from about 30 ft east-southeast of Moriarty to more than 359 ft about 7.5 mi north-northwest of Edgewood.

Figure 10 is a contour map of the regional potentiometric surface for the same area shown in Figure 9. Figure 11 is a three-dimensional view of this potentiometric surface. These two figures show that the potentiometric surface is generally flat in the central part of the area shown and slopes gently toward the south and inward from the margins of the area. This is consistent with Smith's (1957) observation that water entering upland recharge areas flows toward lower discharge areas in the southern part of the Estancia Valley. The southwest part of



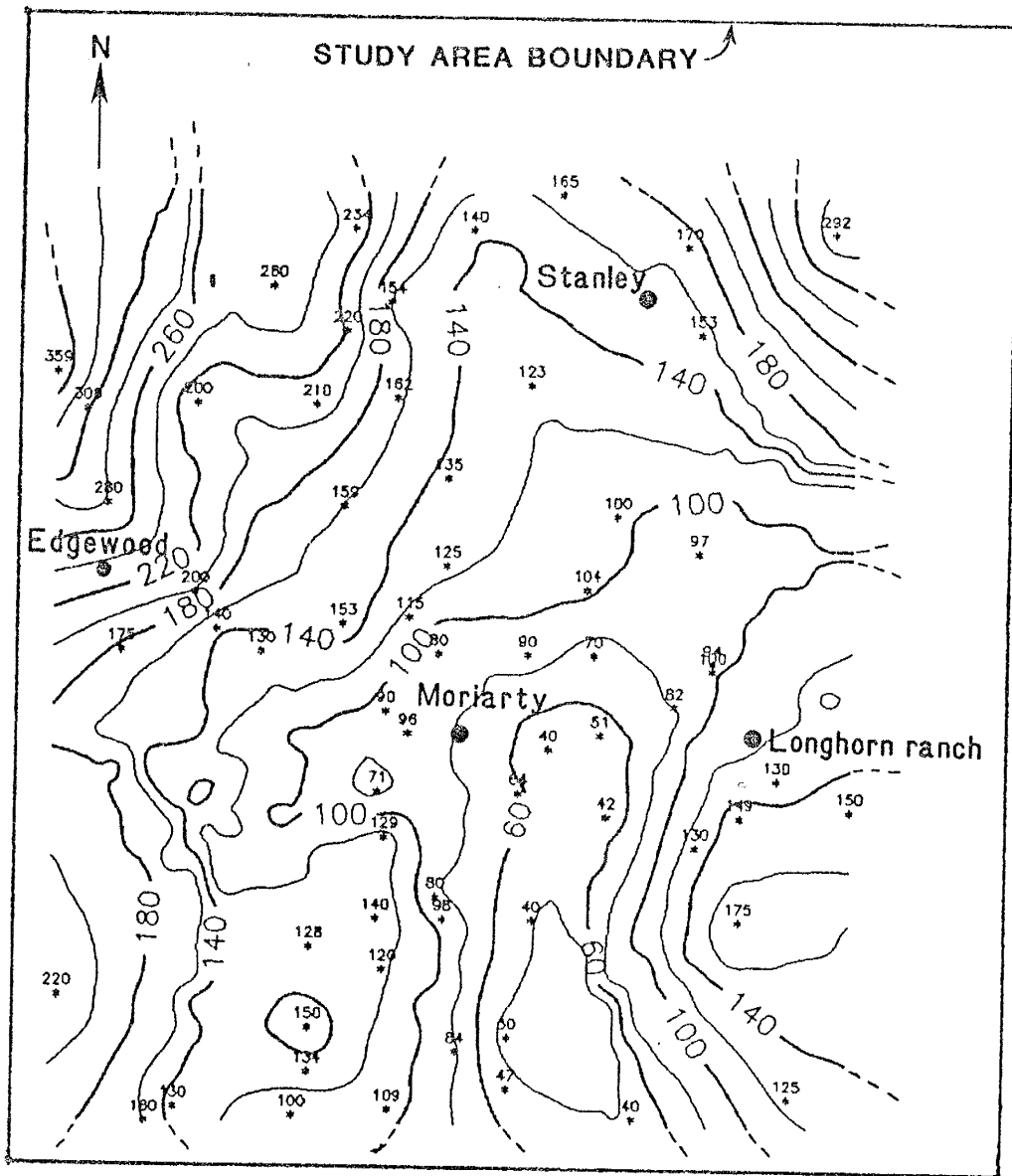


FIGURE 9--Contour map of depth to water.

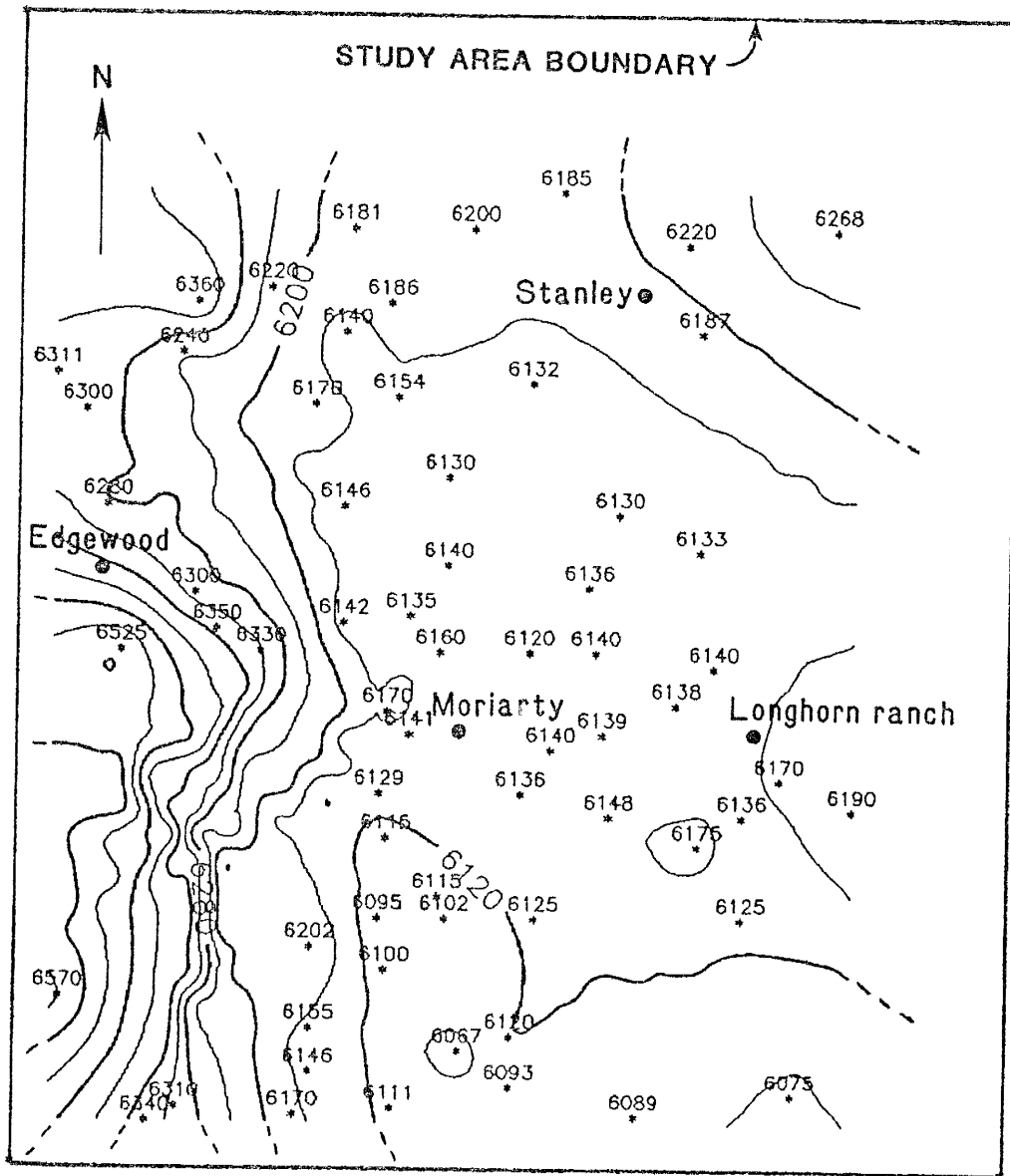


FIGURE 10--Contour map of elevation of the regional potentiometric surface.

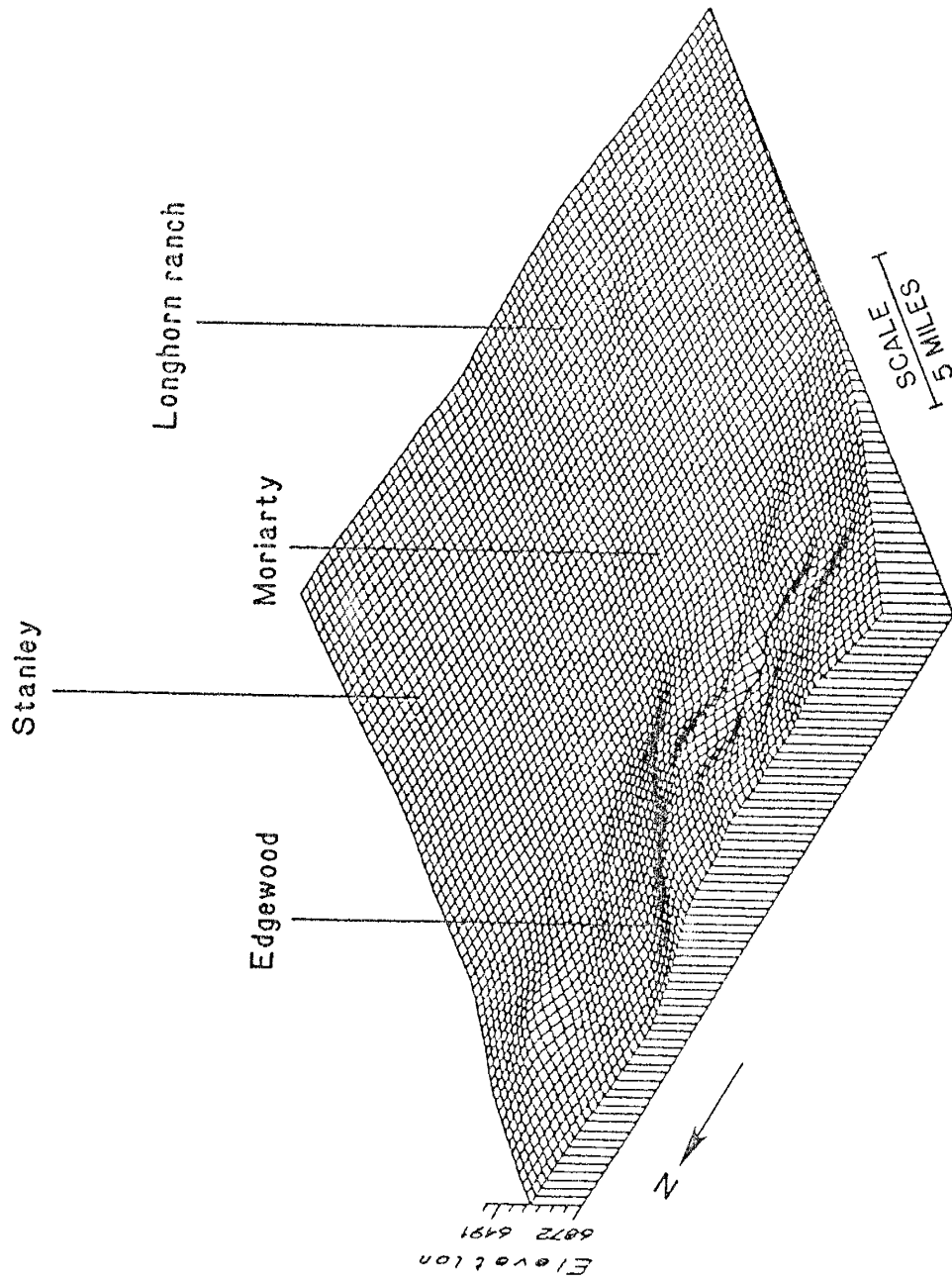


FIGURE 11--Three-dimensional view of the regional potentiometric surface shown in Figure 10 (vertically exaggerated).

the area shown in Figures 10 and 11 is probably elevated because this area is adjacent to the major recharge area of the Manzano Mountains to the west. It is noted that Figures 9, 10, and 11 are generalized and do not always accurately reflect local groundwater conditions. Nonetheless, these figures are useful because they indicate regional hydrologic trends.

## GEOTECHNICAL CHARACTERIZATION OF SOIL UNITS

## Introduction

In this section the engineering properties of soil units in the study area are considered. The construction and maintenance of engineered structures such as building foundations, tunnels, highways, and embankments depends primarily on the physical characteristics of underlying soil or rock units. As used in this study, *soil* is defined as a naturally occurring, unconsolidated superficial deposit overlying bedrock (International Conference of Building Officials, 1982) and includes material both above and below the root zone.

In this section soils are classified according to two systems specifically designed for engineering use. These include the Unified Soil Classification System (USCS) and the American Association for State Highway and Transportation Officials system (AASHTO). These systems are most commonly used in engineering practice because they provide a systematic method of categorizing soils according to their engineering behavior. Only a particle-size distribution (sieve analysis) and the Atterberg limits are necessary to completely classify a soil according to these systems. Knowing the USCS and/or AASHTO classification of a soil allows one to make initial estimates of its engineering properties.

Further geotechnical characterization of soil units in the study area involved additional testing. Moisture content determinations, calculation of liquidity

indexes, and standard penetration tests were also done. Unconfined compressive strength and shear strength measurements were made in backhoe trenches with a pocket penetrometer and a Torvane shear device, respectively. X-ray diffraction (XRD) analyses were done on selected lacustrine clay (Qld) samples.

With the exception of XRD analyses, all of the above tests are easily and quickly done, and offer an efficient, economical means of obtaining important information regarding the engineering behavior of soils. These tests are known as index tests, and the test results are index parameters. These parameters are quite useful because they correlate with the engineering properties of soils. Both the engineering index parameters and the engineering properties of a soil (i.e., unconfined compressive strength, shear strength, relative density, compression index, etc.) are affected by the same things. These include: clay mineralogy, moisture content, secondary cement, particle size, and gradation. Index tests are more time- and cost-effective than more elaborate tests such as the triaxial shear and consolidation (oedometer) tests; therefore, more data can be obtained quicker and cheaper. The results, however, are only approximate and in most cases cannot serve as a substitute for more elaborate testing, although they can sometimes be used in preliminary design work. Index testing is therefore more appropriate for initial feasibility studies than for detailed, specific site investigations. Index-test results can help to outline critical engineering concerns. Because this study is meant to be a broad, reconnaissance-type investigation,

index testing is relied upon heavily.

The objectives of this section are twofold: (1) to generally characterize soil units in the study area in terms of several important engineering properties, and (2) to utilize this information in assessing five basic engineering concerns. These concerns are: liquefaction potential, expanse potential, collapse potential, suitability of various soils for road construction, and certain foundation design considerations. Following a discussion of sampling methods and test procedures, the test results are presented. Finally and foremost, test results are used in evaluating the above five engineering concerns.

#### Sampling Methods

Three types of soil samples were collected for this investigation: (1) geotechnical drill hole samples, (2) backhoe trench samples, and (3) samples obtained from hand-dug pits and man-made cuts. Forty-one samples collected from hand-dug pits were sampled at regular intervals along topographic survey lines (SL-1 to SL-3, Plate 1). Figure 12 shows the locations of sample collection sites. Both disturbed and undisturbed samples were collected from drill holes, although time constraints prevented extensive testing of undisturbed samples. Disturbed samples were obtained with a standard 18-inch split-spoon sampler, and undisturbed samples were obtained with standard 3-inch diameter galvanized, thin-walled Shelby tubes.

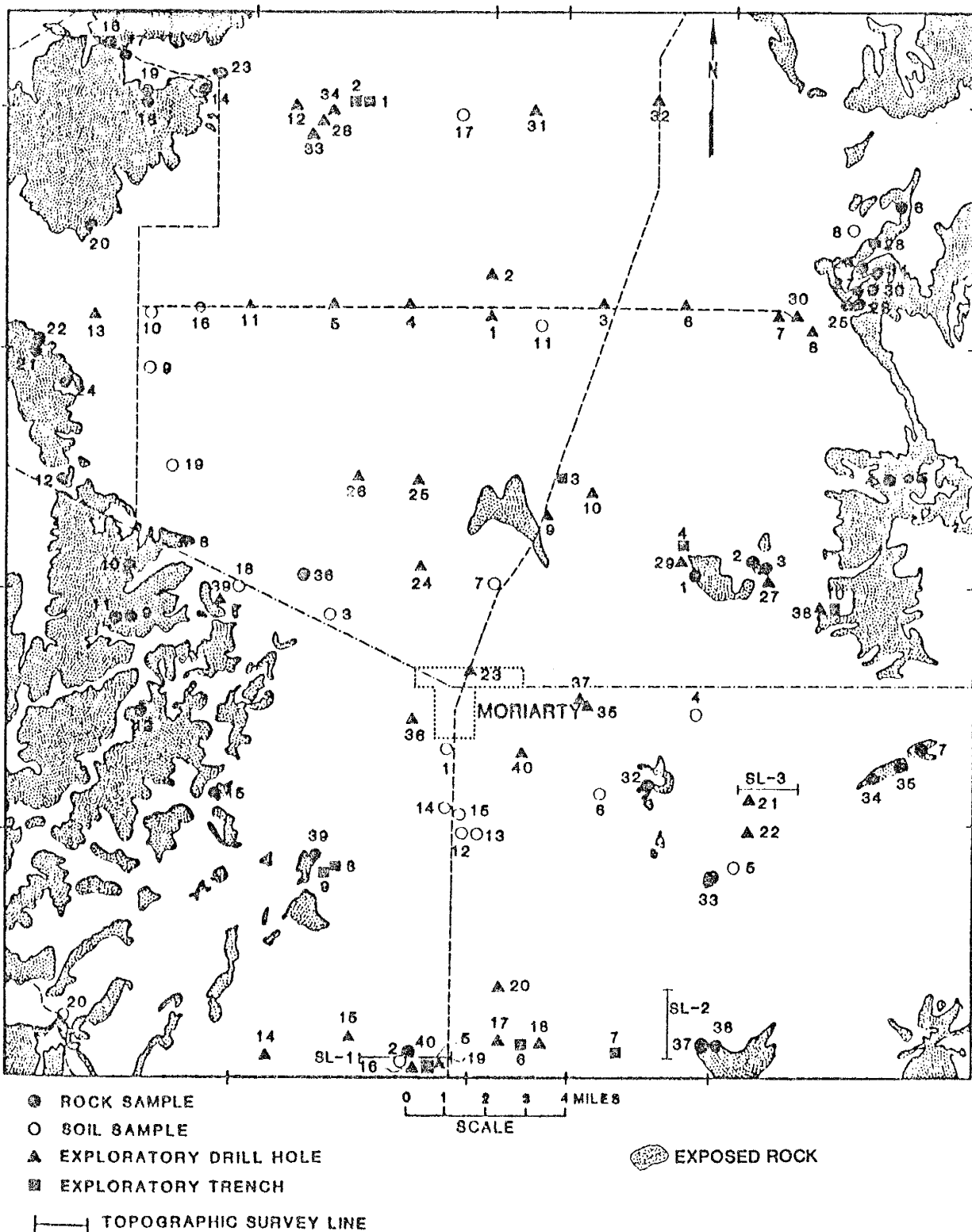


FIGURE 12--Activity location map (after New Mexico SSC Proposal, v. 3, 1987).



## Test Procedures

### Gradation (particle-size distribution)

Particle-size distribution and Atterberg limits are used for classification in the Unified and AASHTO systems. Two hundred forty-five of the 313 soil samples collected for this investigation were analyzed to determine their particle-size distributions. The procedure used generally conforms to ASTM designations D421-85, D422-72, and D1140-71 and consisted of passing a representative soil sample through a series of 8-inch diameter, standard sieves. Because of high percentages of fines (particles smaller than 0.075 mm) it was necessary in most cases to wash samples through sieves with distilled, deionized water. For the purpose of this investigation only the particle-size distribution of the gravel- and sand-size portion of soil samples (particles larger than 0.075 mm) was determined.

### Atterberg limits

The Atterberg limits of a soil correspond to moisture contents at certain critical stages of soil behavior. The liquid limit (LL) marks the boundary between plastic and liquid behavior, whereas the plastic limit (PL) denotes the boundary between semi-solid and plastic behavior. The shrinkage limit (SL) denotes the boundary between semi-solid and solid behavior and corresponds to the moisture content below which no further volume decrease occurs (Fig. 13). Mean shrinkage limit values were estimated using a procedure outlined by Holtz and

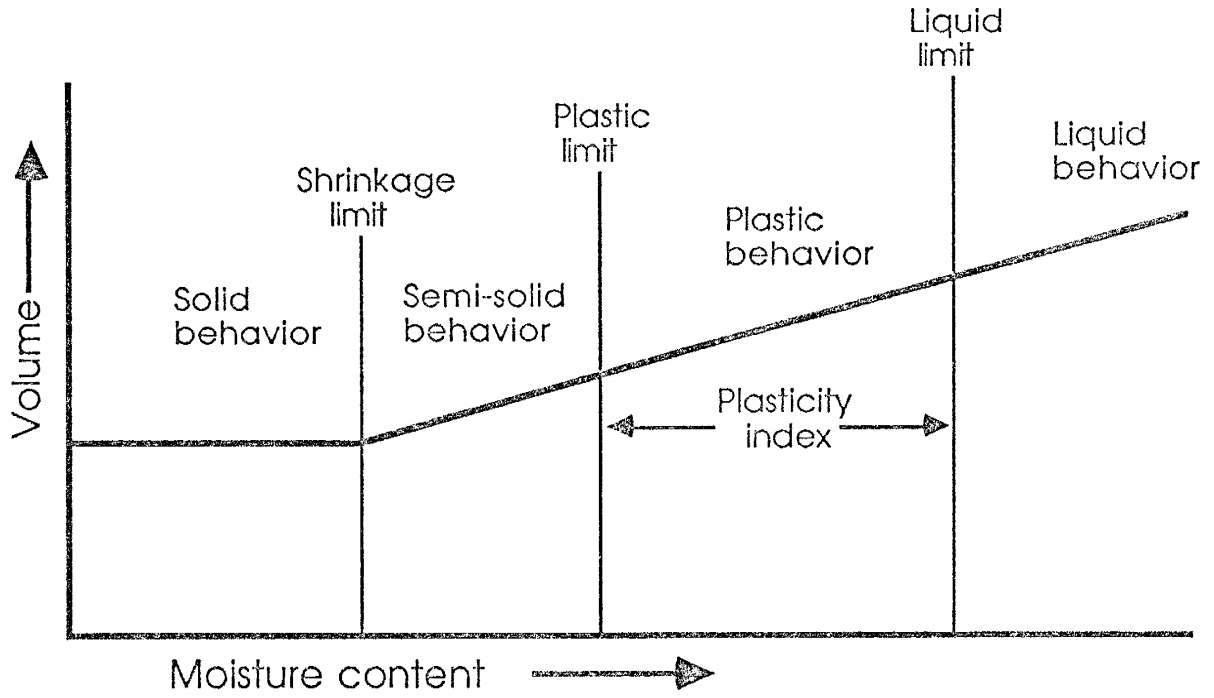


FIGURE 13--Graphical representation of Atterberg limits.

Kovacs (1981). The plasticity index (PI), equal to the liquid limit minus the plastic limit, specifies the range over which the soil behaves as a plastic. Highly plastic soils have high plasticity indexes. Liquid limits, plastic limits, and plasticity indexes of 228 soil samples were determined using procedures outlined by Lambe (1951) and by ASTM designation D4318-84.

The Atterberg limits are particularly useful index parameters which have been found to correlate with important engineering properties such as shrink-swell potential and compressibility.

#### Moisture content

The moisture content of a soil specimen is defined as the percentage ratio of the weight of water in the specimen to the weight of the dry specimen. Moisture content determinations were made on 246 soil samples following a procedure outlined by ASTM designation D2216-80.

#### Liquidity index

Although the natural moisture content of a soil is itself a useful index property, the relationship between soil moisture content and the Atterberg limits is of greater significance because it provides an estimate of how a soil is likely to behave when sheared. This relation is expressed by the liquidity index (LI), which is defined by the following formula given by Holtz and Kovacs (1981):

$$LI = (w_n - PL)/PI$$

where  $w_n$  is the natural moisture content, PL is the plastic limit, and PI is the plasticity index of the sample. If the liquidity index (LI) of a soil is greater than zero, the soil is likely to behave as a liquid when sheared or disturbed. Similarly, LI values between zero and one, and less than zero indicate plastic and solid behavior, respectively. Liquidity indexes were calculated from moisture content and Atterberg limits data for 126 samples. The Liquidity indexes of nonplastic ( $PI=0$ ) soils were not determined.

#### Standard penetration tests (SPT)

One hundred eighty-seven standard penetration tests (ASTM D1586-84) were conducted. The test involved driving a standard split-spoon sampler a distance of 18 inches into the undisturbed soil at the bottom of a drill hole using a 140-lb hammer free-falling from a height of 30 inches. The number of hammer blows required to drive the sampler the final 12 inches is designated the N-value. Although many factors affect the reproducibility of the test (i.e., overburden pressure, variations in hammer free-fall distance, improper seating of sampler, etc.) the standard penetration test is currently the most popular and economical means to obtain subsurface information (Bowles, 1982). Empirical correlations between N-value and important strength properties make this test quite useful in engineering practice.

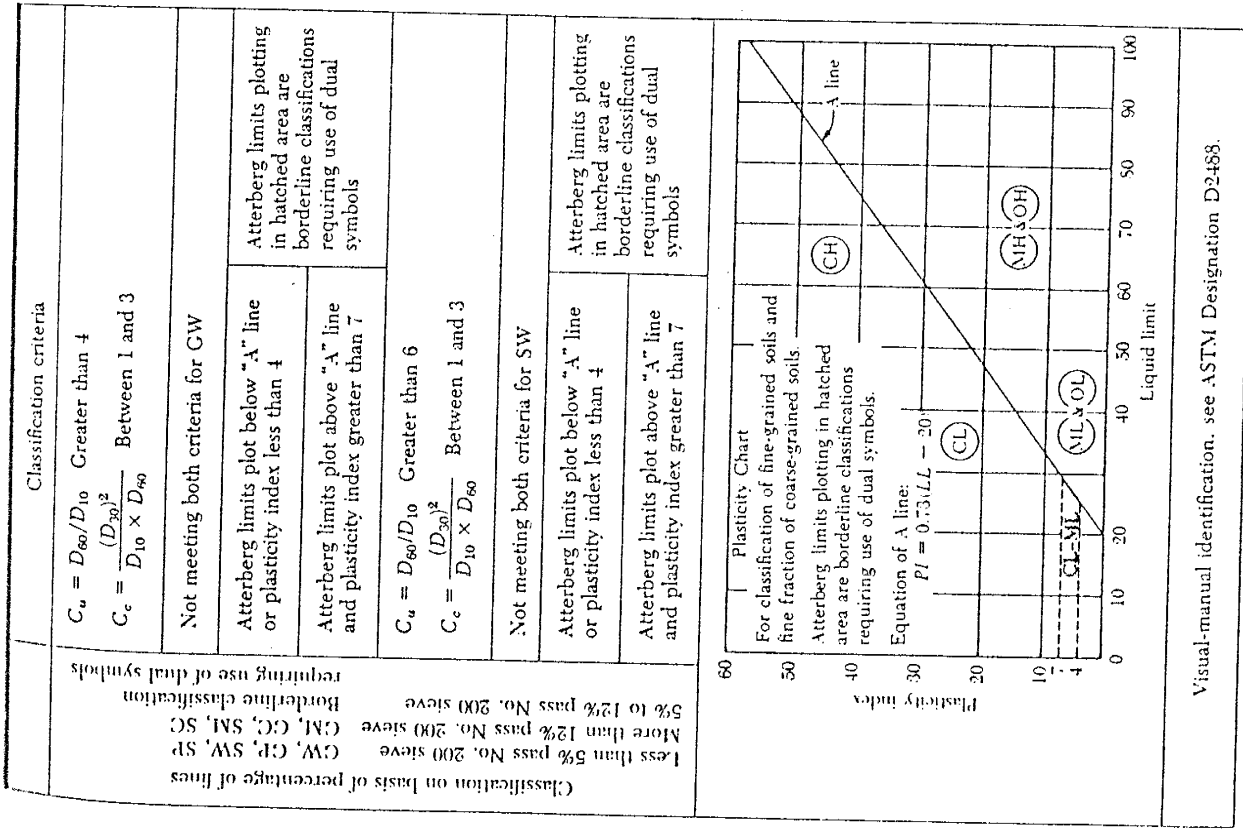
### X-ray diffraction (XRD)

Clay mineralogy can have significant effects on the physical properties of a soil, particularly if the soil is fine grained. In order to identify the relative abundance of important clay mineral groups present in the lacustrine unit (Qld) semiquantitative x-ray diffraction analyses were conducted on 16 disturbed soil samples collected from drill hole 20 (Plate 1), which penetrated the lacustrine unit. Oriented, sedimented slides were prepared and run on a Rigaku x-ray diffractometer following procedures developed by Austin (written comm., 1987). Semiquantitative analyses (accurate to 1 part in 10) based on x-ray diffractogram peak heights were done using the method described by Austin (written comm., 1987) to determine the relative abundance of kaolinite, illite, smectite, chlorite, and illite-smectite mixed-layer clay mineral groups present.

### Soil classification

Both the Unified and AASHTO systems were used in this study to classify soils. In the Unified Soil Classification System (ASTM D2487-85), soils are either coarse- or fine-grained. Coarse-grained soils are divisible into gravel and sand and further divisible into eight groups based on grading and Atterberg limits. Fine-grained soils include silts and clays and are divisible into six groups based on Atterberg limits. Two additional divisions are made for highly organic soils. Figure 14 shows the general criteria used for assigning USCS group names.

The AASHTO classification system (ASTM D3282-83), used for determining



Major divisions	Group symbols	Typical names
Gravels 50% or more of coarse fraction retained on No. 4 sieve	Clean Gravels	CW Well-graded gravels and gravel-sand mixtures, little or no fines
		GP Poorly graded gravels and gravel-sand mixtures, little or no fines
	Gravels with Fines	GM Silty gravels, gravel-sand-silt mixtures
		GC Clayey gravels, gravel-sand-clay mixtures
Sands More than 50% of coarse fraction passes No. 4 sieve	SW Well-graded sands and gravelly sands, little or no fines	
	SP Poorly graded sands and gravelly sands, little or no fines	
Sands with Fines	SM Silty sands, sand-silt mixtures	
	SC Clayey sands, sand-clay mixtures	
Sils and Clays Liquid limit 50% or less	ML Inorganic silts, very fine sands, rock flour, silty or clayey fine sands	
	CL Inorganic clays of low to medium plasticity, gravelly clays, sandy clays, silty clays, lean clays	
	OL Organic silts and organic silty clays of low plasticity	
Sils and Clays Liquid limit greater than 50%	MH Inorganic silts, micaceous or diatomaceous fine sands or silts, elastic silts	
	CH Inorganic clays of high plasticity, fat clays	
	OH Organic clays of medium to high plasticity	
Highly Organic Soils	PT Peat, muck, and other highly organic soils	

FIGURE 14--Unified Soil Classification System (USCS; from Das, 1985).

the relative quality of soils for use in road construction, is shown in Figure 15. Soils are divided into seven groups (A-1 to A-7) and several subgroups. Gravels and sands are classified as groups A-1 to A-3, whereas silts and clays fall into groups A-4 to A-7. Highly organic soils are placed in a separate group (A-8). To evaluate the relative quality of different soils within the same group, a number denoted as the group index (GI) is given in parentheses after the AASHTO group or subgroup designation. The group index is calculated from the following equation:

$$GI = (F - 35)[0.2 + 0.005(LL - 40)] + 0.01(F - 15)(PI - 10)$$

where GI is the group index, F is the weight percent passing the No. 200 sieve, LL is the liquid limit, and PI is the plasticity index. The group index (GI) is used in conjunction with other information (such as estimated traffic load) in the design of highway pavement thicknesses (Oglesby and Hicks, 1982).

## Test Results

### Engineering index parameters

Table 1 is a summary of the mean values and ranges of six important engineering index parameters for soil units in the study area. Common USCS and AASHTO designations are also given. Samples from each soil unit are divided into cohesive and granular varieties. Cohesive soils consist predominantly of low plasticity clays (CL) with varying amounts of sand and silty clays (CL-ML). Common granular soils include silty sands (SM) and clayey

General classification	Granular materials (35% or less of total sample passing No. 200)						
Group classification	A-1		A-3	A-2-4	A-2		
	A-1-a	A-1-b			A-2-5	A-2-6	A-2-7
Sieve analysis (percent passing) No. 10 No. 40 No. 200	50 max. 30 max. 15 max.	50 max. 25 max.	51 min. 10 max.	35 max.	35 max.	35 max.	35 max.
Characteristics of fraction passing No. 40 Liquid limit Plasticity index	6 max.		NP	40 max. 10 max.	41 min. 10 max.	40 max. 11 min.	41 min. 11 min.
Usual types of significant constituent materials	Stone fragments, gravel and sand		Fine sand	Silty or clayey gravel and sand			
General subgrade rating	Excellent to good						

General classification	Silt-clay materials (More than 35% of total sample passing No. 200)			
Group classification	A-4	A-5	A-6	A-7 A-7-5* A-7-6†
Sieve analysis (percent passing) No. 10 No. 40 No. 200	36 min.	36 min.	36 min.	36 min.
Characteristics of fraction passing No. 40 Liquid limit Plasticity index	40 max. 10 max.	41 min. 10 max.	40 max. 11 min.	41 min. 11 min.
Usual types of significant constituent materials	Silty soils		Clayey soils	
General subgrade rating	Fair to poor			

\*For A-7-5,  $PI \leq LL - 30$

†For A-7-6,  $PI > LL - 30$

FIGURE 15--AASHTO classification system (from Das, 1985).



TABLE 1--Summary of engineering index parameters for soils.

	Clay		Old		Ceo		QIa	
	cohesive*	granular#	cohesive*	granular#	cohesive*	granular#	cohesive*	granular#
mean liquid limit (LL) range (min.:max)	31 24:43	28 -	33 20:79	28 19:42	35 24:60	29 19:39	34 23:53	27 19:36
mean plasticity index (PI) range (min.:max)	11 7:18	4 -	15 4:52	7 2:28	16 4:39	8 3:18	14 4:31	7 1:18
estimated mean shrinkage limit (SL)	18	23	15	19	16	19	17	19
mean standard penetration (N) range (min.:max)	15 -	20 14:34	26 5:56	28 7:56	33 7:>50	29 7:>50	31 11:59	37 6:>50
mean moisture content (%) range (min.:max)	9 6:12	8 3:12	16 4:40	11 3:21	12 7:22	8 2:20	12 4:21	8 2:19
mean liquidity index (LI) range (min.:max)	-1.2 -1.9:-0.05	-3 -	-0.4 -2.7:0.71	-2.5 -5:-0.28	-0.8 -2.5:-0.59	-2.1 -6:-0.22	-1.1 -3.8:0.17	-2.5 -10:-0.2
common USCS designations	CL CL-ML	SM	CL CL-ML	SM ML	CL	SM SC	CL-ML CL	SM SM-SC SC
common AASHTO designations	A-4(2)	A-1-a(1) A-4(0)	A-6(10) A-4(2)	A-4(1) A-2-4(0)	A-4(3) A-6(9)	A-4(1) A-2-4(0)	A-6(10) A-4(1) A-7-6(33)	A-4(1) A-2-4(0)

\*i.e., clays.

#i.e., sands and silts.

sands (SC). Silts and sandy silts (ML) are less common and are grouped with granular soils. Clean sands and gravels (SP, SW, GP) occur locally in young alluvium (Qay) as stream channel deposits and as beach and bar deposits associated with lacustrine sediments (Titus, 1969). Common AASHTO soil designations include A-1-a (sandy and gravelly stream channel, beach, and bar deposits), A-2-4 and A-4 (silty, granular soils), and A-4 and A-6 (cohesive soils). A-4 soils are grouped under both cohesive and granular designations in Table 1, depending on whether their USCS equivalents are clays or silts and sands.

Mean liquid limit (LL) values for various soil units range from 27 to 35 and are based on soils exhibiting plasticity (i.e.,  $PI \geq 0$ ). Cohesive soils within each unit exhibit liquid limits slightly higher than granular soils (Table 1). Liquid limits greater than 50 are uncommon, although they do occur locally in cohesive varieties of all units except young alluvium (Qay).

Mean plasticity index (PI) values range from 4 to 16 (Table 1). Cohesive soils exhibit higher mean PI values than granular soils. PI values greater than 30 are uncommon but occur locally in all cohesive soil units except young alluvium (Qay).

Mean shrinkage limit (SL) values range from 15 to 23 and are higher for granular soils than for cohesive soils (Table 1). In all cases, the estimated mean shrinkage limits are 1 to 3 percent below mean plastic limits (mean PL = mean LL minus mean PI).

Mean standard penetration test (N) values for lacustrine deposits (Qld), older

alluvium (Qao), and Tertiary-Quaternary alluvium (QTa) range from 26 to 37 and are generally higher than N-values obtained in young alluvium (Qay; Table 1). This difference is attributable to accumulation of pedogenic calcium carbonate in older units. Actual mean N-values for these units are probably higher than indicated in Table 1 because standard penetration tests are terminated when N-value reaches 50 blows, and actual N-values obtained for all three of these units often are in excess of 50. No significant statistical difference in mean N-values exists for cohesive and granular soils within a given unit.

Mean moisture contents are typically between 8 and 16 percent for various soil units (Table 1). Cohesive soils exhibit slightly higher moisture content than granular soils. Except for several samples obtained from the lacustrine unit (Qld) all samples were collected from above the water table. Liquidity index (LI) values reflect this; they are almost exclusively negative for all soil units sampled, indicating that the natural moisture content of these soils are generally below their plastic limits. Therefore, soil units should fail by fracture if sheared or otherwise disturbed.

#### Clay mineralogy

The engineering behavior of clayey soils is strongly influenced by clay mineralogy. Soil plasticity, in particular, is often dependent on the types of clay minerals present. Potentially problematic, highly plastic soils, for example, typically contain an abundance of montmorillonite. Kaolinitic soils, by contrast,

typically exhibit lower plasticity (Holtz and Kovacs, 1981). In an attempt to determine if a relationship exists between clay mineralogy and the plasticity characteristics of clayey, lacustrine soils in the study area, semiquantitative x-ray diffraction (XRD) analyses and plasticity index (PI) determinations were made on 16 disturbed samples collected from the lacustrine unit (Qld). Samples were obtained from drill hole 20 (Plate 1), located approximately 6 mi south and 1 mi east of Moriarty, at sample intervals of 6 to 18 inches.

Table 2 compares the relative abundance of kaolinite, illite, smectite (Na and/or Ca montmorillonite), and illite-smectite mixed-layer clays determined from XRD analyses with plasticity index (PI) values. None of the samples tested contains measurable amounts of chlorite. Examination of Table 2 indicates no systematic relationship between clay mineralogy and plasticity index (PI). For example, the sample obtained from a depth interval of 18-19.5 ft exhibits a relatively high PI of 35 and contains only 2 parts in 10 smectite and 4 parts in 10 illite-smectite mixed-layer clays, both of which tend to exhibit higher plasticity than kaolinite and illite. Other samples containing up to 3 and 4 parts in 10 smectite and up to 4 parts in 10 illite-smectite mixed-layer clays exhibit much lower PI values. The sample obtained from a depth interval of 12-13 ft, for example, contains 4 parts in 10 smectite and 2 parts in 10 illite-smectite mixed-layer clays, and is nonplastic. Similarly, the sample obtained from a depth interval of 29-30.5 ft contains 4 parts in 10 smectite and 3 parts in 10 illite-smectite mixed-layer clays, and exhibits a PI of only 14.

TABLE 2--Comparison between clay mineralogy and plasticity index for selected samples obtained from drill hole 20 (DH-20).

Drill Hole 20 Depth (ft.)	Clay mineral abundance (parts in ten)				Plasticity Index (PI)
	Kaolinite	Illite	Smectite	Mixed layers	
0-1.5	2	4	1	3	16
1.5-3	3	3	1	4	15
3-4.5	3	2	3	2	7
4.5-6	3	2	2	3	14
6-7.5	3	6	1	0	19
7.5-9	4	3	1	1	12
9-10.5	2	2	3	3	17
10.5-12	3	1	3	3	12
12-13*	3	1	4	2	NP
13-13.5*	4	3	1	2	17
13.5-15	2	3	1	4	26
15-16.5	2	3	2	3	30
18-19.5#	2	2	2	4	35
19.5-21#	2	1	3	4	27
21-22.5#	2	2	3	4	21
29-30.5	2	1	4	3	14

\*indicates 2 samples taken from same 18" interval.  
 #Data obtained from Karen Brown (written comm., 1987).

This lack of correlation between clay mineralogy and plasticity is probably due to other factors such as particle-size distribution that influence plasticity more strongly. The relative amount of clay-size material present in these lacustrine soils likely determines their plasticity characteristics rather than the clay mineralogy of these soils. Soils exhibiting higher PI values almost certainly contain a higher percentage of clay-size material than soils with lower PI values. Skempton (1953) found that the plasticity index (PI) of a soil linearly increases with the percent clay-size fraction (percent finer than 0.02 mm by weight) present in it. Hydrometer analyses were attempted in order to confirm this relationship for lacustrine soils. Unfortunately, even after samples were centrifuged and treated with dispersant, flocculation of clay-sized particles rendered hydrometer tests invalid. To summarize, the plasticity of lacustrine soils is probably more strongly affected by the amount of clay-size material present than by clay mineralogy.

### Engineering Applications

#### Liquefaction potential

Ground motion during earthquakes may cause a loss of soil strength resulting from elevated pore pressures. This in turn can result in damage to engineered structures. The process leading to such a loss of strength is called soil liquefaction. Saturated granular soils without cohesive fines are most susceptible to liquefaction. The greater the content of clay and other fine particles contributing to plasticity, the less the susceptibility to sudden pore pressure

build-up (National Academy of Sciences, 1985).

Liquefaction is not a major concern in the study area due to the lack of saturated, cohesionless soils and the low level of seismicity. Soils in the study area generally contain an abundance of fines (silt and clay), as indicated in **Table 1** and Appendix III. Clays (CL), silty clays (CL-ML), silty sands (SM), and clayey sands (SC) are quite common, whereas cohesionless sands and gravels (SP, SW, GP, GW) occur only locally and are only locally saturated. Only lacustrine soils (Qld) that occur below the 6,225 ft contour are saturated to near the ground surface and therefore offer any potential for liquefaction. These soils consist of interbedded sands (thin and cohesionless), plastic silts, silty clays, and clays up to approximately 100 ft thick. Saturated, cohesionless sands within this unit are typically only a few feet thick, however, and are not considered thick enough to be susceptible to liquefaction (New Mexico SSC Proposal, 1987, v. 3). Furthermore, soils most susceptible to liquefaction generally occur where ground water is 30 ft deep or less (National Academy of Sciences, 1985). Depth to water in the study area at its shallowest point is 30 ft and is typically 100 to 300 ft (Fig. 9).

Although liquefaction effects have occurred during earthquakes of Modified Mercalli intensity VI, liquefaction becomes common at intensity VII if susceptible deposits are present (National Academy of Sciences, 1985). The largest historic event in the region was an intensity VII event that occurred near Cerrillos, located about 35 miles north of the Moriarty, in 1918. This event caused shaking in

Moriarty of modified Mercalli intensities estimated to be only III-IV. No reports of liquefaction are associated with this event, and no evidence of liquefaction has been found in the study area (New Mexico SSC Proposal, 1987, v. 3).

Typical ground acceleration associated with earthquakes that have caused liquefaction has been estimated to be about 10 percent of gravity (National Academy of Sciences, 1985). According to Algermissen et al., (1982) the maximum probable horizontal ground acceleration likely to occur in the study area (in rock, with 90 percent probability of not being exceeded in 250 years) is about 6 percent of gravity. This suggests that earthquakes which could potentially cause liquefaction do not occur very frequently in the study area.

To summarize, conditions favorable for liquefaction are not present in the study area. Soils are generally not cohesionless, the water table is too deep, and predicted ground accelerations likely to be caused by earthquakes are too low.

#### Expanse (shrink-swell) potential

Expansive soils are cohesive soils that shrink or swell excessively with changes in moisture content. Whether or not a soil is expansive depends upon several factors. Of greatest importance is the difference between the natural moisture content and the equilibrium moisture content that prevails both during and after construction of a structure. If the equilibrium moisture content during or after construction is substantially greater than the natural moisture content before construction, a potentially expansive soil may swell, causing damage to the overlying foundation. Similarly, if the equilibrium moisture content during or after



construction is less than the natural moisture content before construction, shrinkage that can also cause foundation damage can result. Clay mineralogy strongly affects expansive potential. Soils containing an abundance of montmorillonite or illite-smectite mixed-layer clays high in smectite (i.e., some bentonites) tend to undergo greater volume changes than soils containing predominantly illite, kaolinite, or chlorite. Density, moisture, and soil structure also affect expansive potential. Soils of high density typically swell more than low-density soils; dry soils swell more than moist soils; and remolded soils swell more than undisturbed soils (Costa and Baker, 1981).

The identification of potentially expansive soils involves correlations between direct measurement of volume changes and various engineering index parameters. Peck et al. (1974) suggested that potentially expansive soils exhibit plasticity index (PI) values of 10 or more, whereas Costa and Baker (1981) stated that PI values greater than 15 usually indicate a swelling problem may exist. In a review of expansive soils, Snethen (1986) stated that liquid limits greater than 50 and PI values greater than 25 are reasonable lower limits for indicating potential problems. Gromko (1974) summarized various methods of predicting the degree of expansion from engineering parameters such as the shrinkage limit (SL), plasticity index (PI), and colloid content. Three of these systems are shown in Table 3. In general, three criteria are necessary for potentially damaging swelling to occur. These are (1) the presence of montmorillonite (smectite) in the soil; (2) a natural moisture content near the plastic limit (PL); and (3) an available source of

TABLE 3--Summary of engineering index systems used to estimate degree of expansion (after Gromko, 1974).

Shrinkage limit (SL) (percent)	Degree of expansion	Shrinkage index (=LL-SL) (percent)	Degree of expansion
<10	Critical	0-20	Low
10-12	Marginal	20-30	Medium
>12	Noncritical	30-60	High
		>60	Very high
Swell potential (S)* (percent)	Total expansion (1-psi load), air dry to saturated (percent)	Shrinkage limit (SL)#	Degree of expansion
0-1.5	0-10	<11	Very high
1.5-5	10-20	7-12	High
5-25	20-35	10-16	Medium
>25	>35	>15	Low
Colloid content (percent minus 0.01 mm)#	Plasticity index (PI)#	Shrinkage limit (SL)#	Degree of expansion
>28	>35	<11	Very high
20-31	25-41	7-12	High
13-23	15-28	10-16	Medium
<15	<18	>15	Low

\*Swell potential (S) =  $60K(PI)\exp(2.44)$ ; where  $K = 0.000036$ , and  $PI$  = plasticity index.

#Soil properties should not be used separately; all three must be considered to arrive at the estimated degree of expansion from air dry to saturated conditions.

water (Gromko, 1974).

Granular soils in the study area are generally not expansive, as indicated by mean PI values of less than 10 and estimated mean shrinkage limits in excess of 12 (Table 1). Estimates of the expanse potential for cohesive (clayey) soils in the study area range from low to medium, depending upon which engineering parameters are used to make the estimate. Cohesive soils in the study area typically exhibit PI values corresponding to the lower limit of those exhibited by expansive soils, as discussed above and in Table 3 (scheme 2). The mean natural moisture content of cohesive, lacustrine soils (Qld) is 16 percent, which is within 2 percent of their mean plastic limit (18), suggesting they might be expansive; however, the estimated mean shrinkage limit (SL) of cohesive lacustrine soils is 15 and their mean PI value is also 15, suggesting that these soils are not highly expansive according to schemes 1 and 3 in Table 3. Expansive soils typically exhibit shrinkage limits less than 10 (Table 3, scheme 1). Highly expansive clays, although uncommon, occur locally in lacustrine deposits (Qld), older alluvium (Qao), and Quaternary-Tertiary alluvium (QTa), as evidenced by liquid limits above 50 and PI values greater than 25 (Table 1; Appendix III). Table 3 (scheme 2) gives formulas whereby the swell potential (S) and the shrinkage index (SI) can be calculated from the PI. The mean swell potentials of lacustrine (Qld) and older alluvial (Qao) soils are equal to 1.6 and 1.9 percent, respectively, indicating that these soils exhibit a medium degree of expansion. Their mean shrinkage indexes, however, are equal to 18 and 19,

respectively, indicating that these soil units exhibit a low degree of expansion.

The Soil Conservation Service (1970, 1975) has delineated two potentially expansive soil localities in the study area (Plate 1). One area occurs in the west-central part of the study area where clayey horizons exhibiting high plasticity have developed in residuum weathered from limestone bedrock of the Madera Formation. This soil is typically only about 3.5 feet thick, however, and therefore should not pose any significant threat to the stability of small structures such as single family houses, provided that the foundations of such structures are designed properly. Foundation footings should be seated below the depth to which highly expansive clays occur, and the floors of the overlying structures should be suspended above a crawl space. These clays are generally thin enough that this can probably be accomplished economically. As a precaution against potentially damaging soil heave, slab-type foundations that rest directly on the ground should possibly be avoided in this area, or, alternatively, specially designed to withstand soil heave. The economic feasibility of specially designed slab foundations will depend on the results of detailed, site-specific tests used to estimate the magnitude of swell (such as the oedometer test--ASTM D4546-85).

The other area of potentially expansive soil occurs in the northeast part of the study area (Plate 1) in association with floodplain deposits derived from sandstone, siltstone, and mudstone of the Dockum Group. The depth to bedrock underlying these deposits is typically greater than 5 ft. These deposits are of very limited aerial extent and can easily be avoided when selecting construction sites

in the area. Overall, the natural expansive potential of soil units in the study area is considered to be low. With the exception of younger alluvium (Qay), soil units typically contain an abundance of secondary (pedogenic) calcium carbonate, which inhibits volume changes. In view of the typically low moisture content of soils, swelling (and subsequent shrinkage) can nonetheless occur where excess moisture is allowed to accumulate. This is particularly true in lowlying, lacustrine soils consisting of thick, clayey sequences. Leaky water lines and drains, sprinkler systems, and septic tanks associated with various structures can greatly increase soil moisture and could therefore lead to soil heave. Providing adequate surface drainage is the simplest method of controlling soil moisture. This can be achieved by grading the ground surface downward from the structure. Typically, a 1 to 3 percent slope provides proper drainage (Snethen, 1986). Other simple, preventive measures include subsurface drains, proper maintenance of water lines, special watering procedures for vegetation adjacent to structures, and locating septic tanks, vegetation, and sprinkler systems away from structures.

Compaction control is another method of controlling soil heave which could develop in association with construction on clayey, lacustrine soil. Expansive clays heave much less when compacted to low densities at high moisture contents. Gizienski and Lee (1965), for example, found that when a test soil was compacted to about 4.5 percent above optimum moisture content, its swell was negligible for any degree of compaction. Compaction at 3 percent below

optimum, however, caused excessive swell even for low degrees of compaction. Compaction methods that induce kneading (rather than static compaction methods) have also been found to reduce heave on wetting. This is due to the development of a more dispersed arrangement of soil particles (Gromko, 1974).

#### Collapse potential

Collapsible, or hydrocompactive soils are those that exhibit a tendency to compact with the addition of water. Such soils occur throughout arid to semi-arid regions of the southwestern and central U.S. and in many places consist of silty and clayey sands (SM, SC) deposited by mud and debris flows on alluvial fans. They are typically Holocene ( $\leq 10,000$  yrs) in age, exhibit low natural moisture contents, have low dry densities, and have never been wetted subsequent to deposition. When these soils are wetted they tend lose their dry strength and densify. Potential sources of wetting include disruption of natural runoff due to grading, roof runoff, leaky utility lines, and lawn watering (Johnpeer et al., 1987).

Correlations between soil collapsibility and various engineering parameters such as dry density, moisture content, Atterberg limits, and grain-size distribution have achieved limited success (Reimers, 1986). A modified form of the consolidation test has been used more successfully in predicting collapse potential (Jennings and Knight, 1975). In the test an undisturbed sample is placed in a consolidometer and incrementally loaded to its in-situ overburden stress. The sample is then flooded, and the amount of consolidation upon wetting

is measured. The change in void ratio of the sample due to wetting gives a qualitative evaluation of collapse potential (Jennings and Knight, 1975).

Due to the absence of available modified consolidation data for soil units in the study area, the susceptibility of soil units to collapse is difficult to estimate. Lacustrine soils (Qld) in the study area are probably not susceptible to collapse because they have been wetted many times subsequent to their deposition. Soils most likely to collapse include granular, uncemented older alluvium (Qao) and Tertiary-Quaternary alluvium (QTa). Granular soils within these units have low natural moisture content and consist of silty, clayey, and silty-clayey sands (SM, SC, SM-SC; Table 1). N-values within these soils average 29 (Qao) and 37 (QTa), respectively, suggesting that these units are relatively dense. This implies that collapsible soils within these units are not common, because collapsible soils typically exhibit low relative densities. An important factor regarding the collapsibility of cemented soils is the sensitivity to dissolution of their intergranular cement (Reginatto, 1971). The dissolution of calcium carbonate due to local use of groundwater (for irrigation, lawn watering, etc.) is not a concern. Research needs to be conducted to better understand the effects of increased water acidity on soil collapsibility.

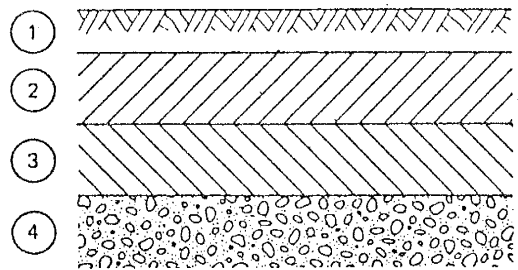
#### Suitability of various soils for road construction

The suitability of various soil units in the study area for road construction purposes is best assessed by considering their AASHTO designations. The

AASHTO soil classification system is designed specifically for this purpose. Figure 16 defines the terms relating to pavement systems used in this section. Table 1 indicates the general subgrade rating of the various AASHTO group or subgroup designations. Sandy and gravelly soils (groups A-1 to A-3) are considered excellent to good for use in subgrades (the native soil upon which the road is constructed) because they are less compressible than silty and clayey soils (groups A-4 to A-7) and can be compacted to higher densities, thereby giving them better bearing capacity. Soils falling into groups A-2-6, A-2-7, and A-4 to A-7 range in quality as subgrade from good to poor. Fair and poor subgrades typically require a layer of subbase material or an increased thickness of base course over that required for better subgrades in order to furnish adequate support. The mean group index (GI), shown in parentheses after the AASHTO group or subgroup designations for soil units in the study area (Table 1), is useful for approximate within-group evaluations. Under average conditions of good drainage and adequate compaction, the supporting ability (bearing capacity) of a subgrade varies as the inverse of its group index (GI); that is, a group index of zero indicates a relatively good subgrade material, whereas a group index of 20 or greater indicates a very poor subgrade (ASTM, 1987).

The granular varieties of soil units within the study area appear to be fairly well suited as subgrade. A-2-4 and A-4 soils are common; A-4 soils exhibit group index (GI) values of zero or one, indicating their overall suitability as subgrade (Table 1). Cohesive soils in the study area are less suitable as subgrade.





- ① Surface coarse: 20-23 cm Portland cement or 2-8 cm asphaltic concrete.
- ② Base coarse: 5-10 cm asphaltic concrete, 15-30 cm sand-gravel base, 20-30 cm soil-ement, or 15-20 cm asphalt stabilized sand.
- ③ Subbase (this layer may be omitted): 15-30 cm sand-gravel.
- ④ Subgrade: The natural soil at a site. The top 0.15-0.5 m is usually compacted prior to the placement of the other layers of the pavement.

FIGURE 16--Definitions of terms relating to pavement systems (after Holtz and Kovacs, 1981).

AASHTO designation A-6 is common, and the group index value of this designation is typically fairly high (9 or greater); however, A-4 designations exhibiting low group indexes are also common in the cohesive varieties of soil units (Table 1). Furthermore, the average N-value of cohesive soils within approximately 5 feet of the surface is 23, indicating that these soils can probably furnish adequate support for most traffic loads.

Suitable base coarse and subbase material can be obtained from rock and soil units exhibiting high-quality aggregate resource potential, as discussed in the aggregate resource potential section of this study and as shown in Table 11 (p.98 ). Crushed rock of excellent to good quality for base coarse and subbase can be obtained from the Madera and San Andres Formations, Tertiary monzonite, and Precambrian schist and quartzite. Good quality sand and gravel for base coarse and subbase can be obtained from beach and bar deposits associated with ancestral Lake Estancia. Lower quality sand and gravel generally requiring beneficiation is obtainable from young alluvium, older alluvium, and Quaternary-Tertiary alluvium.

#### Foundation design considerations

The purpose of this section is to discuss briefly several geotechnical and geological concerns that directly affect the design of foundations on soils in the study area. Generally, the probable performance of a proposed foundation can be evaluated with respect to two types of unsatisfactory behavior: bearing

capacity failure and excessive settlement (Peck et al., 1974). Bearing capacity failure results when the stress imparted to the underlying soil or rock by the foundation exceeds the inherent supporting ability of the soil or rock, and a local shear failure results. Alternatively, foundation settlement caused by consolidation of the underlying soil or rock due to the weight of the foundation and the overlying structure may be so great or uneven as to cause damage to the structure.

Foundations are typically designed to minimize the potential for soil-bearing capacity failures and excessive settlements. Obviously, a proposed foundation design must take into account the engineering properties of the soil or rock beneath the foundation.

In general, both granular and cohesive soils in the study area exhibit good bearing capacity, as evidenced by typically high N-values for both granular and cohesive soils. Average N-values for various soil units range from 15 to 37, with N-values greater than 50 (refusal) common in all units except young alluvium (Table 1, Appendix IV). Figures 17 and 18 show correlations between N-value and relative density or shear strength, for granular and cohesive soils, respectively. Based on the N-value versus compactness relationship shown in Figure 17, granular soils in the study area range from medium dense to very dense (excluding young alluvium). Approximately half of these soils are dense to very dense (Appendix IV). As shown in Figure 17, the relative density of these soils typically exceeds 70 percent (based on typical N-values of 30 or greater); therefore, their bearing capacity is estimated to be high. It is emphasized,

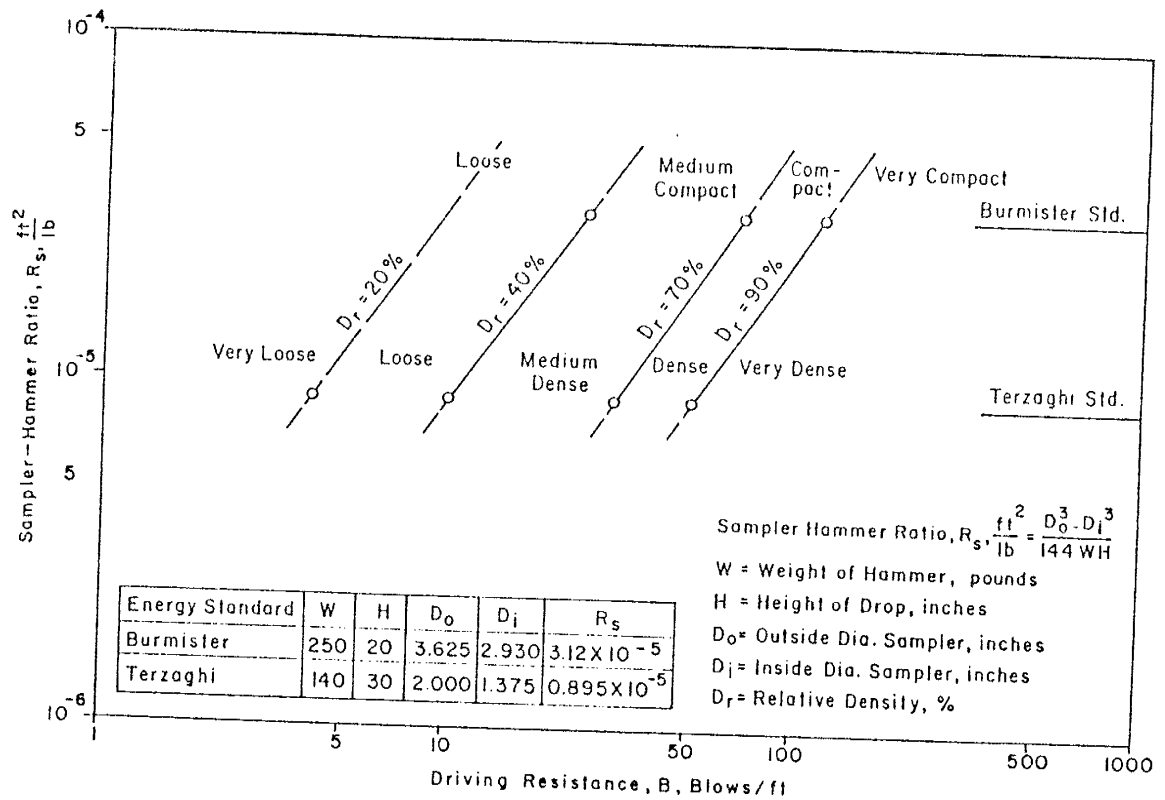


FIGURE 17--Standard penetration (N) value vs. compactness--cohesionless soils (from Lowe and Zaccheo, 1975).

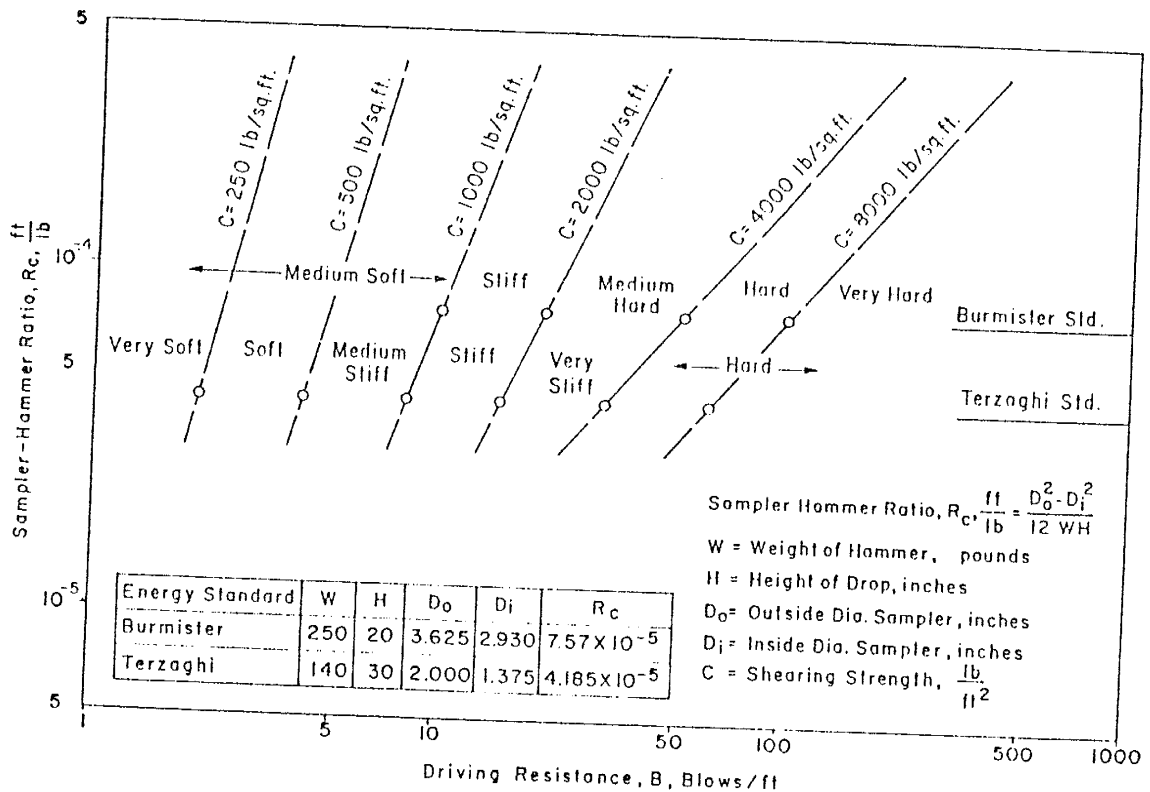


FIGURE 18--Standard penetration (N) value vs. consistency--cohesive soils (from Lowe and Zaccheo, 1975).

however, that estimates of relative density based on N-value are very approximate. The relationship between N-value and relative density strictly applies only to uncemented, cohesionless sands and silts.

The consistency of cohesive soils ranges from medium stiff to hard (Fig. 18). Medium stiff soils are exceedingly rare, whereas very stiff to hard soils are common (Appendix IV). Table 4 shows the mean undrained shear strengths of cohesive soils in the study area based on pocket penetrometer, Torvane, and N-value data. Although these various estimates of shear strength are only approximate, they agree with one another fairly well, indicating that cohesive soils in the study area generally possess high shear strength and good bearing capacity.

Time-dependent settlement (consolidation) of cohesive soils is generally of more concern than immediate settlement due to elastic deformation. This is because consolidation settlement can be several times greater than the immediate settlement (Das, 1985). The amount of consolidation settlement likely to occur in a clay layer of a given thickness and compressibility due to a specified, foundation-induced stress increase can be calculated from the results of a standard consolidation test (ASTM D2435-80). For general purposes, the compressibility of a saturated, cohesive soil can be estimated from the liquid limit (LL) by the following equation (Terzaghi and Peck, 1967):

$$C_c = 0.009(LL - 10)$$

where  $C_c$  is the compression index and LL is the liquid limit. The larger the  $C_c$ ,

TABLE 4--Shear strength estimates for cohesive soils based on pocket penetrometer, Torvane, and standard penetration test (SPT) data.

Method used	Approximate mean undrained shear strength (tons/sq ft)	Number of tests done	Remarks
Pocket penetrometer	1.9	11	Measures unconfined compressive strength, $q$ ; undrained shear strength = $q/2$ (Holtz and Kovacs, 1981); $q$ values shown on trench logs (appendix II).
Torvane	2.3	4	Torvane shear strength values shown on trench logs (appendix II).
Standard penetration (SPT)	2	83	Shear strength based on approximate correlation between $N$ value and shear strength shown in figure 18.

the more compressible the soil. The liquid limit of a saturated, cohesive soil sample from drill hole 35 (Plate 1) was determined to be 39, which gives a compression index ( $C_c$ ) of 0.26 from the above equation. Comparison of this  $C_c$  value with some typical  $C_c$  values shown in Table 5 suggests that saturated, cohesive soils in the study area are not highly compressible. Foundation settlement caused by consolidation of saturated, lacustrine clays in the study area can therefore probably be controlled with standard foundation designs such as reinforced concrete slabs and spread footings. Furthermore, saturated, cohesive soils occur only in the lacustrine unit (Qld; Plate 1) below a depth of 30 ft. Consolidation settlement should therefore not be a concern for small to medium-size structures constructed on shallow, raft or footing foundations.

Finally, ground settlement due to drawdown of groundwater, although it may be a future concern, is not presently a problem. Hydrographs indicate that groundwater levels have been declining at a rate of about 1 ft per year since 1948 (New Mexico SSC Proposal, v. 3; 1987), although there are no reports of related subsidence. Throughout most of the study area, depth to the regional groundwater table is high, typically between 100 and 300 feet, and the well-cemented nature of most near-surface soils inhibits settlement due to groundwater withdrawal at depth. Only the central and south-central parts of the study area may possibly experience future settlement problems as ground water continues to decline. These are areas where cohesive, somewhat compressible lacustrine deposits are the thickest. Lacustrine soils might subside as much as a

TABLE 5--Typical values of the compression index  $C_c$  (after Holtz and Kovacs, 1981).

Soil	$C_c$
Normally consolidated medium and sensitive clays	0.2-0.5
Chicago silty clay (CL)	0.15-0.3
Boston blue clay (CL)	0.3-0.5
Vicksburg buckshot clay (CH)	0.5-0.6
Swedish medium sensitive clays (CL-CH)	1-3
Canadian Leda clays (CL-CH)	1-4
Mexico City clay (MH)	7-10
Organic clays (OH)	4 and up
Peats (Pt)	10-15
Organic silt and clayey silts (ML-MH)	1.5-4.0
San Francisco Bay Mud (CL)	0.4-1.2
San Francisco Old Bay clays (CH)	0.7-0.9
Bangkok clay (CH)	0.4



few inches in the next 50 years if water-level declines continue at their present rate (R. Raymond, written comm., 1987). Such declines are treatable using conventional foundation designs.

## GEOTECHNICAL CHARACTERIZATION OF ROCK UNITS

### Introduction

Characterization of rock for engineering purposes is necessary to obtain an estimate of how a given rock unit or rock mass will behave when used in construction. The properties of rock to be used for this purpose must be specified in terms that can be incorporated into foundation designs and construction techniques. To this end, many field and laboratory tests have been proposed, ranging from simple and inexpensive ones to more elaborate, costly ones requiring much time and effort.

For the purpose of this investigation, a number of field and laboratory tests were done on various rock units, including point load strength index (PLSI), strength anisotropy index (SAI), dry density, relative hardness, rock quality designation (RQD), percent recovery, and x-ray diffraction (XRD). Los Angeles wear values given by the New Mexico Highway Department (1975) and chemical analyses presented in the New Mexico SSC Proposal (1987, v. 3) were also utilized. The advantages of these tests are twofold: (1) they have been found to be quite useful in evaluating a variety of engineering concerns, and (2) they are, for the most part, easily and quickly done, which makes them both time- and cost-effective. Following an explanation of test procedures, test results are presented and discussed. Finally, the engineering applications section, which is of major importance to this thesis, is presented. In this section four important

engineering concerns are evaluated for various rock units, including (1) basic foundation design considerations, (2) expanse (shrink-swell) potential, (3) rippability, and (4) aggregate resource potential.

### Test Procedures

#### Point load strength index

The point load strength test is a relatively quick, simple, and inexpensive method of determining rock strength and classifying rocks for engineering purposes (Read et al., 1980). When first introduced, the point load strength test was used mainly to predict uniaxial compressive strength (UCS), which was formerly the established test for general-purpose rock strength classification. Point load strength now often replaces UCS in this role because when properly conducted it is as reliable as UCS and much quicker to measure (ISRM Commission on Testing Methods, 1985). It is an indirect tensile test; a concentrated vertical applied load creates a horizontal tensile stress and failure occurs by splitting along one or more planes parallel to the direction of load application (Boisen, 1977). Figure 19 shows the point load testing device used in this investigation.

Three types of point load strength tests were conducted: (1) diametral tests, (2) axial tests, and (3) block tests. Diametral tests were conducted parallel to the diameter of core segments in a direction parallel to bedding. Axial tests were conducted parallel to the central axis of core segments in a direction

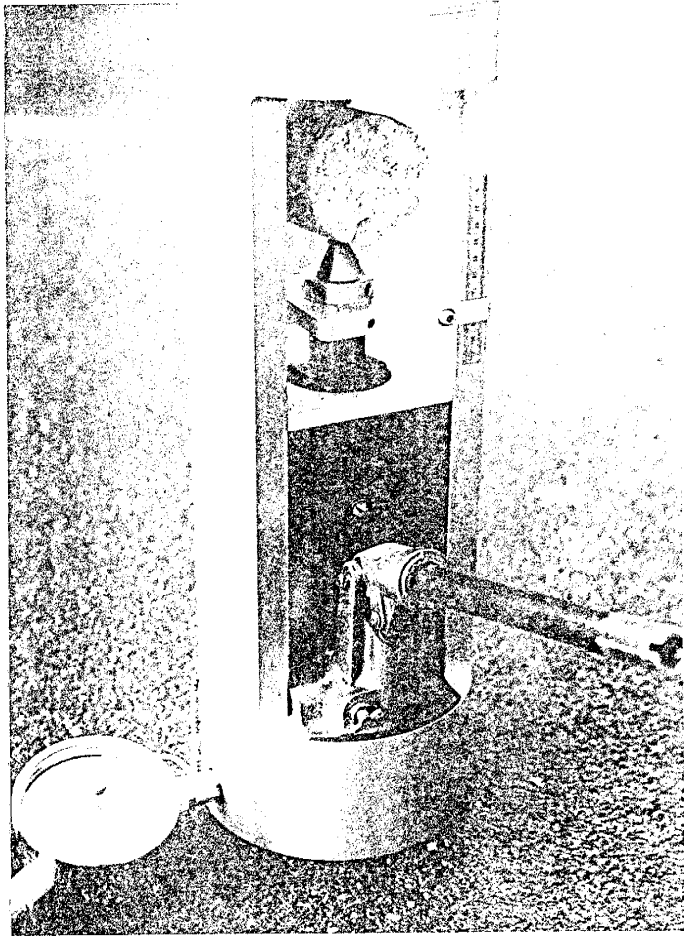


FIGURE 19--Point load strength testing device, model PIL-5.

100000

perpendicular to bedding. Block tests were carried out on regular blocks cut from outcrop specimens perpendicular to bedding or foliation. A total of 312 point load strength index (PLSI) tests were done for this investigation.

Point load strength index (PLSI) values shown in Tables 6 and 7 are probably higher than the in-situ PLSI values of these rock units because core segments and outcrop samples were air-dried before testing, and point load strength index varies with moisture content. Oven-dried specimens, for example, are typically much stronger than moist ones (ISRM Commission on Testing Methods, 1985). In this study, testing of core segments at their natural moisture content could not be carried out because drilling fluids were used in obtaining the core. Outcrop samples were cut into blocks before testing with a water-lubricated saw. Therefore, specimens were thoroughly air-dried for a minimum of 1 week before testing so that moisture content would not vary greatly among samples.

Although intended as an index test for strength classification of rock material, the point load strength test has also been used to predict other strength parameters. Broch and Franklin (1972) and Franklin (1977), for example, have proposed that the unconfined compressive strength (UCS) of a rock material is equal to 24 times the size-corrected point load strength index (PLSI). The size-corrected point load strength index (PLSI) is defined as the point load strength that would have been measured by diametrically testing a standard-size core segment 50 mm in diameter (ISRM Commission on Testing Methods, 1975). Pells (1975) and Read et al. (1980), however, have shown that the suggested

conversion factor of 24 cannot be universally applied. In tests on many different rock types this conversion factor has been found to vary between 15 and 50 (ISRM Commission on Testing Methods, 1985). Therefore, correlations with UCS are only approximate.

#### Strength anisotropy index

Certain rock types, particularly bedded sedimentary rocks and schistose metamorphic rocks, exhibit a higher point load strength index (PLSI) when tested perpendicular to the plane of greatest anisotropy (i.e., bedding and foliation) than when tested parallel to the plane of greatest anisotropy. Strength estimates of such rock types may therefore be misleading if rock specimens are tested in only one direction. In order to assess the likelihood of this condition for rock units in the study area, mean strength anisotropy index (SAI) values for various rock units were determined and are discussed in the results section. Strength Anisotropy Index (SAI) is defined as the ratio of point load strength index (PLSI) values measured perpendicular and parallel to bedding or foliation as determined from axial and diametral tests, respectively. A number of core samples that had previously been tested diametrically were also tested axially for the purpose of determining their SAI.

#### Dry density and relative hardness

Dry densities were obtained by measuring the volume of an oven-dried

specimen and then determining its weight. The relative hardness of a rock material is an estimate of how easily it can be scratched. Relative hardness designations range from very soft (can be scratched with a fingernail) to very hard (cannot be scratched with a knife).

#### Rock quality designation (RQD) and percent recovery

Both of these measurements provide estimates of in-situ rock quality. RQD is considered the better estimate and is defined as the percentage ratio of the total length of core segments at least 4 inches long to the total length of core barrel advance. Breaks obviously caused by drilling are ignored. Percent recovery is defined as the percentage ratio between the length of core recovered and the total length of core barrel advance, regardless of the number of fractures.

#### X-ray diffraction (XRD)

Semiquantitative x-ray diffraction analyses were conducted on samples of Dockum Group rocks in order to determine their clay mineralogy. Test procedures used are the same as those used for XRD analysis of lacustrine soil samples. An outline of these procedures is given in the test procedures section of the previous chapter (p.44).

### Test Results

Table 6 lists mean point load strength index (PLSI) values, strength anisotropy index (SAI) values, dry density measurements, and relative

TABLE 6--Summary of rock test data for subsurface (core) samples.

Geologic Unit	Point Load Strength Index PLSI (MPa)* Mean#: Range#	PLSI Test Type**	Mean Strength Anisotropy Index SAI	Strength Designation##	Mean Dry Density (PCF)	URCS Density Designation###	Relative Hardness***
Pm	4.8: 3.2-6.3	D	1.0	very high	168	very dense	moderately hard-medium
Pg	4.1: 2.6-6.0	D	1.1	very high	153	dense	moderately hard-medium
Py	1.4: 1.2-2.0	D	1.7	high	139	light	medium-very soft
Rd	0.9: 0.7-1.0	D	2.1	medium	149	medium dense	soft-very soft
Km	0.5: 0.4-0.5	D	2.4	medium	133	light	very soft

\*1MPa = 10.44 ton/square ft.

#Extreme values deleted.

\*\*D = Diametrical test.

##After Broch and Franklin (1972).

###After Williamson, 1984.

\*\*\*Hardness ranges from very soft (can be scratched with fingernail) to moderately hard (can be scratched with knife).



hardnesses determined for various rock types from selected core segments. Mean PLSI strength designations and Unified Rock Classification System (URCS) density designations are after Broch and Franklin (1972) and Williamson (1984), respectively. Limestone of the Madera Formation (Pm) exhibits the highest mean PLSI of all rock units tested diametrically (parallel to bedding). PLSI values range from 3.2-6.3 MPa and average 4.8 MPa. A mean strength anisotropy index (SAI) of 1.0 indicates that Madera rocks exhibit similar PLSI values when tested both parallel and perpendicular to bedding. Therefore, engineering assessments that are based on the mean PLSI of this unit should not be influenced by directional considerations such as bedding attitude. For example, the rippability of the Madera, which can be estimated from its point load strength, is equally low for both flat-lying and steeply dipping rock masses. The same holds true for the Glorieta Sandstone, which also exhibits a high mean point load strength index (4.1 MPa) and a low mean strength anisotropy index (1.1; Table 6).

Table 6 indicates that dry density and relative hardness appear to correlate generally with point load strength index. The harder and more dense the rock, the higher its size-corrected point load strength index (PLSI). This suggests that dry density and relative hardness are also useful for estimating rock substance strength.

Rocks of the Yeso Formation (Py), the Dockum Group (Trd), and the Mancos Shale (Km) exhibit lower PLSI values and higher SAI values than rocks of the

Madera and Glorieta Formations. PLSI values for Yeso rocks range from 1.2-2.0 MPa, averaging 1.4 MPa. The low density of Yeso rocks (Table 6) may result from small, open cavities and unfilled fractures, as indicated by logs for drill holes 29 and 37 (Appendix I). Rocks of the Dockum Group (Trd) have a PLSI range between 0.7 and 1.0 MPa and an average PLSI of 0.9 MPa. Although Dockum Group rocks exhibit lower PLSI values and lower hardness designations than Yeso rocks, they are more dense (Table 6). This is an exception to the observation noted above that the more dense a rock, the higher its point load strength. The Mancos Shale (Km) exhibits the lowest mean PLSI value, the highest mean SAI value, the lowest dry density, and the lowest relative hardness of all rocks tested. Core segments are easily scratched with a fingernail and show a well-developed fissility parallel to bedding. For rock units tested both diametrically and axially, higher SAI values correspond to lower PLSI values; thus, weaker rocks are more anisotropic with regard to point load strength than stronger rocks, probably because they often break quite easily parallel to bedding. The in-situ substance strength of these units as measured by their PLSI will therefore be affected more strongly by bedding attitude in both subcrop and outcrop than will higher-strength units such as the Madera Formation and the Glorieta Sandstone.

Mean PLSI values for rocks of the Dockum Group (Trd), the Yeso Formation (Py), and the Abo Formation (Pa) are less representative than mean PLSI values determined for other rock types because these three units are highly variable with

regard to lithology. Each unit contains relatively competent sandstones interbedded with less competent siltstones and mudstones. Generally, PLSI values for sandstones are higher than those for siltstones and mudstones. Dockum Group rocks clearly show this trend. Axial PLSI values determined on core segments consisting predominantly of siltstone and mudstone average 1.8 MPa, whereas axial PLSI values determined on Dockum Group outcrop specimens consisting primarily of sandstone average 4.4 MPa. In addition, mean PLSI values for the Abo and Yeso Formations (Tables 6 and 7) were determined only from specimens of sandstone and are probably higher than PLSI values for interbedded siltstones and mudstones, which were not tested.

Table 7 lists mean PLSI values, PLSI ranges, dry densities, and relative hardnesses of block specimens from outcrop. Precambrian quartzite (pCq) from Lobo Hill exhibits the highest mean axial PLSI value of all rock types tested (12.3 MPa), whereas rocks of the Dockum Group (Trd) exhibit the lowest PLSI value (4.4 MPa). Dry densities for Madera, Glorieta, and Dockum Group rocks shown in Table 7 generally agree with those values obtained in Table 6. PLSI values recorded in Table 7 are greater than those shown for the same rock units in Table 6. This is due in part to the testing method used. PLSI values in Table 7 were obtained by testing rock specimens perpendicular to bedding or foliation. PLSI values shown in Table 6, however, were obtained by diametrically testing core segments parallel to bedding. For Dockum Group rocks, PLSI values shown in

TABLE 7--Summary of rock test data for surface (outcrop) samples.

Geologic Unit	Point Load Strength Index PLSI (MPa)* Mean#: Range#	PLSI Test Type**	Strength Designation##	Mean Dry Density (PCF)	URCS Density Designation###	Relative Hardness***
pCq	12.3: 10.6-13.4	BP	extremely high	165	very dense	very hard
Pa	10.7: 10.4-11.0	BP	extremely high	154	dense	moderately hard
Pb/Psa	9.8: 2.7-12.4	BP	very high	157	dense	hard-soft
Tm	8.4: 2.3-17.1	BP	very high	154	dense	hard-soft
Pm	6.5: 4.6-8.9	BP	very high	168	very dense	moderately hard
pCs	6.0: 5.7-6.4	BP	very high	161	very dense	hard
Pg	5.8: 3.6-9.0	BP	very high	146	medium dense	medium-hard
Rd	4.4: 2.7-7.5	BP	very high	148	medium dense	moderately hard-medium

\*1MPa = 10.44 tons/sq ft.

#Extreme values deleted.

\*\*BP = Block test; samples tested perpendicular to plane of greatest anisotropy (ie-bedding, foliation).

##After Broch and Franklin (1972); strength designations range from extremely low to extremely high.

###After Williamson (1984); density designations range from very light to very dense.

\*\*\*Hardness ranges from very soft (can be scratched with fingernail) to very hard (cannot be scratched with knife).

Table 6 were obtained primarily from siltstone and mudstone, whereas PLSI values shown in Table 7 were obtained from more competent sandstone.

While point load strength index provides a means of estimating the substance strength of intact rock, rock quality designation (RQD) and percent recovery provide a better estimate of in-situ rock quality. Table 8 lists mean RQD values, mean percent recovery, and their ranges for eight holes drilled at various locations. Core of NX size was obtained from drill holes 27, 30, and 39. Core from other drill holes is of larger (3-inch) diameter, and RQD values could not be obtained (RQD only applies to core of NX size). Mean RQD values for drill holes 27, 30, and 39 indicate rock of fair to poor quality in these regions, according to the classification scheme shown in Table 9; however, information is too sparse to estimate the overall quality of rock units. Furthermore, as indicated by the wide ranges of percent recovery for various rock units, rock quality varies with location. RQD values for Glorieta rocks in the vicinity of drill hole 27 indicate rocks of fair-poor quality, although high recovery ratios of Glorieta rocks penetrated by drill hole 29 indicate excellent quality rocks in this area. Variable RQD values obtained in slightly fractured to very fractured limestone of the Madera Formation penetrated by drill hole 39 indicate good to very poor quality rock. Similarly, variable RQD values obtained for Glorieta Sandstone penetrated by drill hole 27 are also attributable to local fracturing of this unit. Recovery ratios for Dockum Group rocks (drill holes 30 and 34) vary from 50-100 percent, and recovery ratios for Yeso rocks (drill holes 29 and 37) vary from 27-93 percent,

TABLE 8--In-situ rock quality data for subsurface (core) samples.

Drill Hole Number	Core Interval (ft)*	Geologic Unit	Mean RQD	RQD Range	Mean %Rec	%Rec Range	Description
DH-27	3-149	Pg	38%	12-70%	74%	55-98%	gray to white quartz sandstone; vertical, iron-stained fractures; local copper staining; brecciated and highly weathered zones.
DH-29	20-230	Pg	-	-	96%	90-100%	gray to white quartz sandstone with vertical, iron-stained fractures cemented by calcite.
DH-29	280-355	Py	-	-	37%	27-48%	red sandstone, siltstone, and mudstone; local clay seams and quartz-filled fractures
DH-30	58-89		66%	0-100%	81%	50-100%	red sandstone and mudstone with reduction spots and clay partings.
DH-32	140-155	Km	-	-	100%	100%	black to gray, fissile, silty mudstone dipping 25°-42°; numerous calcite veins crosscutting and parallel to bedding; pyrite crystals.
DH-34	180-195		-	-	100%	100%	red sandstone, siltstone, and fissile mudstone dipping at approx. 10-30°; mudstone contains expansive clays (sample barrel forced open upon recovery).
DH-37	120-135	Py	-	-	93%	93%	red, vuggy sandstone, siltstone, and mudstone; calcite-filled cavities; calcite-filled and unfilled vertical and horizontal fractures.
DH-39	14.5-103		46%	0-81%	89%	54-100%	slightly fractured to very fractured gray, fossiliferous limestone; minor shaley partings.

\*ft below surface.

TABLE 9--Relation between RQD and in-situ rock quality (from Peck et al., 1974).

RQD (%)	Rock Quality
90-100	excellent
75-90	good
50-75	fair
25-50	poor
0-25	very poor

indicating rock of variable quality.

## Engineering Applications

### Rippability

Read et al. (1980) proposed that point load strength index (PLSI) values can be used to estimate rock mass rippability. They noted that for particular tractor and ripper arrangements, and PLSI values up to 2.5 MPa, rock in the Melbourne, Australia, area broke readily regardless of the nature of rock mass defects. Above 2.5 MPa, however, rippability depended increasingly on the orientation, spacing, and interconnectedness of the defects. Table 10, based upon rippability designations suggested by Read et al. (1980), relates rippability and mean PLSI values for rock units within the study area. As indicated by Table 10, most surface rock in the study area exhibits PLSI values much higher than 2.5 MPa, and, in addition, it is not favorably jointed; therefore, it is not rippable in most cases. Sandstone of the Dockum Group (Trd) may be an exception, however. As shown in Figure 20, closely spaced, horizontal joints may render Dockum Group sandstone locally rippable. Subsurface samples of Mancos Shale (Km) and Dockum Group siltstone and mudstone (Trd) exhibit PLSI values below 2.5 MPa, and are therefore considered rippable. Subsurface samples of Yeso sandstone (Py) from drill holes 29 and 37 also exhibit low enough strength to be rippable; however, higher strength Yeso limestone exposed elsewhere (i.e., on El Cuervo Butte) indicates that the Yeso is not rippable everywhere.



TABLE 10--Relation between point load strength index (PLSI) and rippability (after Read et al., 1980).

Rock Unit	Mean Point Load Strength Index PLSI (MPa)*	Lithology(s)	PLSI Test Type**	Occurrence of Samples Tested	Rippability Designation Usually Rippable by
Km	0.5	shale	D	subcrop	D7#
Rd	0.9	siltstone and mudstone	D	subcrop	D7#
Py	1.4	sandstone	D	subcrop	D7#
Rd	4.4	sandstone	B	outcrop	D9# with favorable defect pattern
Pg	5.8/4.1	sandstone	B/D	outcrop/subcrop	not rippable
pCs	6.0	schist	B	outcrop	not rippable
Pm	6.5/4.8	limestone	B/D	outcrop/subcrop	not rippable
Tm	8.4	monzonite	B	outcrop	not rippable
Pb/Psa	9.8	limestone and sandstone	B	outcrop	not rippable
Pa	10.4	sandstone	B	outcrop	not rippable
pCq	12.3	quartzite	B	outcrop	not rippable

\*1 MPa = 10.44 tons/sq ft.

\*\*D = Diametral test; B = Block test.

#Caterpillar® tractor.

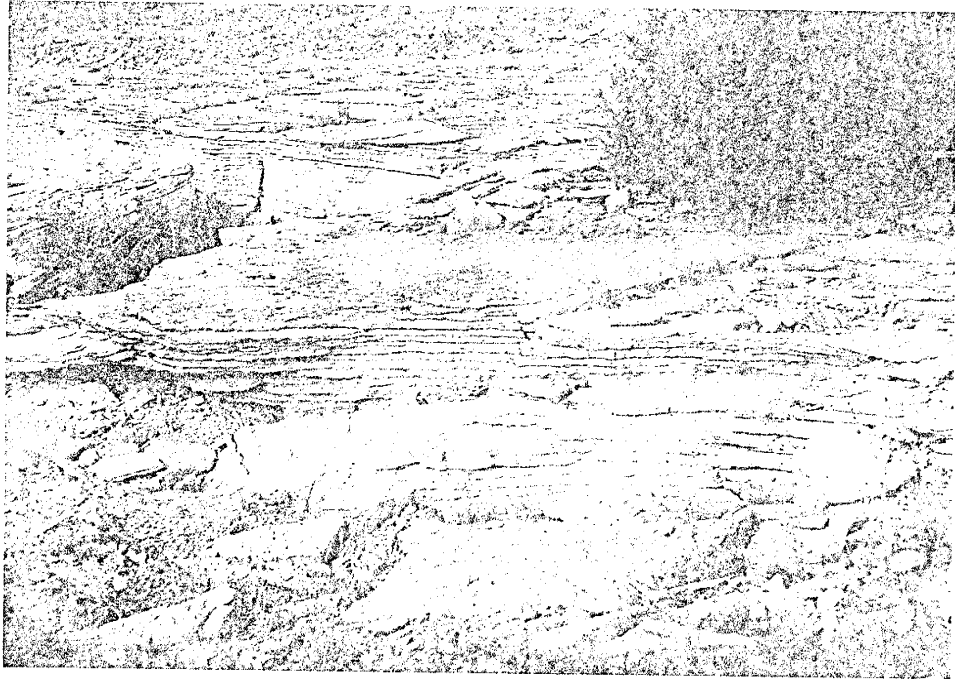


FIGURE 20--Closely spaced, horizontal joints in Dockum Group sandstone.

### Expanse (shrink-swell) potential

Generally, the swelling capacity of rock units in the study area is estimated to be minimal; however, both field and laboratory evidence indicate that siltstone and mudstone lithologies of the Dockum Group may be expansive locally. As noted in Table 8, core segments of Dockum Group rock obtained from drill hole 37 forced open the core barrel as the rock absorbed water used for drilling. Also, highly expansive clays derived from Dockum Group rocks have been identified in the northeast part of the study area by the Soil Conservation Service (1975). Semiquantitative x-ray diffraction (XRD) analysis of two Dockum Group samples from drill hole 37 indicates that potentially expandable mixed-layer illite-smectites represent a significant fraction (about 4 parts in 10) of the clay mineral portion of these rocks. Less competent lithologies within the Dockum Group may therefore swell if the equilibrium moisture content after construction is allowed to become higher than the natural moisture content of these rocks before and during construction.

### Aggregate Resource Potential

Aggregate resources are materials that can be used as riprap, ballast, subbase, fill, or as an additive to cement or asphalt. The American Society for Testing and Materials (ASTM, 1987) has established several important criteria by which the quality of aggregate can be assessed. The plasticity index (PI; ASTM

D4318-84), gives an indication of clay content and should be very low or equal to zero for high-quality aggregate. Materials with a high PI tend to absorb large amounts of water and undergo large volume changes, which inhibit aggregate performance.

Another common property used to assess aggregate quality is its resistance to abrasion as measured by the Los Angeles (LA) wear test (ASTM C131). In this test, a steel drum is loaded with about 5 kg of oven-dried aggregate of a specified grading and with a specified number of iron balls. The drum is rotated at 30-33 RPM for 500 revolutions. The wear caused by the tumbling and dropping of the aggregate and the balls in the drum is determined from the weight percentage of material which, after the test, will pass a No. 12 (0.066 in) sieve. General practice dictates that aggregate to be used in concrete should exhibit LA wear values less than about 40 to 50 percent. Los Angeles wear values for various aggregates have been found to correlate with their service behavior in concrete; i.e., low LA abrasion values indicate better performance. Such tests have also indicated that the lower the wear, the higher the flexural and compressive strength of the concrete (Troxell et al., 1968).

A third test used to assess aggregate quality is soundness loss (ASTM C88), which measures resistance to weathering. Samples are alternately saturated with sodium and magnesium sulphate and dried for a specified number of cycles. The amount of spalling or disintegration is a measure of the soundness. Weight loss is usually specified to be between 0 and 16 percent (Costa and Baker,

1981), although higher soundness loss values can occur. Other important procedures used to assess aggregate quality include classification of aggregate under the Unified and AASHTO systems (ASTM D2487-85 and D 3282-83), specific gravity and absorption (ASTM C127 and C 128), potential reactivity (ASTM C289), and petrographic examination.

The most common aggregate resources in the study area are limestone, quartzite, sand, gravel, and caliche. Rock resources will be discussed first, followed by a consideration of sand, gravel, and caliche resources.

## Rock

Limestone of the Madera Formation is the most important aggregate resource because of its high quality and availability. Typically low Los Angeles wear values of about 20-25, low soundness loss values, and low PI values indicate that limestone quarried from the Madera is well suited for aggregate and can be utilized for all road-building purposes, including base courses and surface courses. Shaly horizons within the Madera are rarely numerous or thick enough to adversely affect aggregate quality (New Mexico Highway Department, 1975). The lower gray limestone member of the Madera has been reported to be cherty, (Kottlowski, 1961) although it is not known whether chert accumulation is generally sufficient to adversely affect aggregate quality. A chemical analysis of the Madera Limestone indicates that it is low in silica, magnesium and iron oxide, and contains sufficient calcium to be usable in cement manufacture (New Mexico

SSC Proposal, v. 3; 1987). Other uses for the Madera include crushed stone for aggregate, road metal, ballast, stucco, soil conditioner, and dimension or building stone (McLemore, 1984).

Although of limited aerial extent within the study area, limestone of the San Andres Formation (Psa) also represents a high-quality aggregate resource. Los Angeles wear and soundness loss values for the San Andres Limestone are generally similar to those of the Madera Limestone. According to the New Mexico Highway Department (1975), minor gypsum, sandstone, and siltstone interbeds can be avoided if loader operators exercise reasonable care. Thin limestone units within the Yeso Formation are unacceptable for most highway construction use because of low resistance to weathering (New Mexico Highway Department, 1975). These units may, however, offer some potential for local aggregate use in light-duty roads.

Limestones of the Madera and San Andres Formations are also well suited for use in bituminous mixtures such as asphalt. Due to electropositive surface charges, the best adhesion between bitumin and aggregate is obtained with strongly electropositive rock such as limestone (Orchard, 1976).

Tertiary monzonite (Tm) and Precambrian schist (pEs) are also generally good aggregate materials. When crushed to a specified size gradation, they are well suited for concrete mixes and road construction purposes.

Other rock types within the study area are generally unsuitable for aggregate. Sandstones of the Glorieta, Bernal, Yeso, and Abo Formations and

(34)

the Dockum Group are unsuitable as aggregate material because they contain iron oxide and calcite cement which can cause staining and solution weathering (Costa and Baker, 1981). Finer-grained lithologies, including Dockum Group, Yeso, and Abo siltstones and mudstones, and the Mancos Shale, are unsuitable as aggregate material because they exhibit low strength and high plasticity indexes.

Dimension or building stone can be obtained from Precambrian quartzite and the Madera limestone. Precambrian quartzite suitable for dimension stone occurs on Lobo Hill (Fig. 1); however, production has been limited because of remoteness from potential markets (McLemore, 1984).

#### Sand, gravel, and caliche

For aggregate resource purposes sand is usually defined as a granular material ranging in size from 2 mm to 0.075 mm. Gravel includes material coarser than 2 mm. Caliche is loosely defined as the "B" horizon of a soil profile enriched by the accumulation of secondary calcium carbonate (Lovelace, 1972). Sand, gravel, and caliche are low-value high-bulk commodities which must generally be mined close to construction sites. Cities, highways, and other areas of intense construction will therefore dictate the general location of excavations. Sand, gravel, and caliche are basic construction materials that are used primarily as aggregate for concrete, as permeable subgrades, in bituminous mixes, and for foundation support. Caliche is also used in cement manufacture, provided the

magnesium oxide (MgO) content is less than 10 percent (Bateman, 1956).

Sand and gravel of fair to poor quality can be obtained from Quaternary-Tertiary alluvium (QTa), older alluvium (Qao), and younger alluvium (Qay). These units typically contain excessive amounts of silt, clay, and calcium carbonate ( $\text{CaCO}_3$ ), and in most cases beneficiation will be required in order to remove deleterious materials and obtain gradation characteristics suitable for road construction, as specified by ASTM D448-86.

The lacustrine unit (Qld) is generally unsuitable as aggregate due to high amounts of fines and lack of abundant gravel; however, higher quality sand and gravel deposits related to late Pleistocene Lake Estancia are exploitable where they have accumulated in sufficient quantity as bars and spits. These deposits have been mapped by the Soil Conservation Service (1970) and are shown on Plate 1. Such deposits are potentially good sources of sand and gravel because in places they contain clean, well-graded to poorly-graded gravel and sand suitable for concrete as well as for road construction purposes. One such deposit which has been actively mined occurs approximately 2 miles south of Moriarty (Fig. 21). Soil samples S-12 and S-13 (Plate 1) were collected at this site. This deposit consists of clean, pebble-size gravel and interbedded poorly-graded sand.

Caliche deposits associated with Quaternary-Tertiary alluvium (QTa) occur locally in the north-central and east-central parts of the study area (Plate 1). These deposits have also been mapped by the Soil Conservation Service (1970;





FIGURE 21--A typical bar deposit mined for aggregate.

Plate 1) and are a potentially good source of poorly consolidated, nodular caliche that can be crushed and utilized for road construction. According to Gillette (1934) the best caliches for road construction are those that have low liquid limits and low plasticity indexes. Samples tested by the New Mexico Highway Department (1975) lack plasticity ( $PI = 0$ ) and are therefore well-suited for road construction. To date, caliche has not been mined extensively in the study area; it is at present used for maintenance of pre-existing roads (New Mexico Highway Department, 1975). Additional laboratory testing and further exploration are necessary to better characterize the engineering properties and available reserves of caliche in the study area.

A special problem relating to aggregate utilization results from lack of water for washing gravel to be used as concrete aggregate (New Mexico Highway Department, 1975). Quantities of water required for washing gravel on large projects should be obtainable in most cases from wells.

Table 11 is a summary of available aggregate resources in the study area. Generally, the overall resource potential for crushed rock aggregate, dimension stone, sand, gravel, and caliche is high and should be adequate to supply most foreseeable future construction needs.

#### Foundation design considerations

Since most intact, unweathered rocks are much stronger than concrete, rock strength does not typically influence the design of foundations. Madera and

TABLE 11--Summary of aggregate resources.

## Rock

Unit	Lithology	Suitability as Aggregate	Uses/Notes
	limestone	excellent-good	cement, concrete, road construction, dimension stone
Psa	limestone	excellent-good	cement, concrete, road construction
pCq	quartzite	excellent-good	concrete, road construction, dimension stone
pCs	schist	good	concrete, road construction
Tm	monzonite	good	concrete, road construction
Pg	sandstone	unsuitable	low quality due to intergranular, calcite cement
Pb	sandstone, limestone	unsuitable	insufficient quantity to be exploitable
	sandstone, mudstone	unsuitable	plasticity of material is too high
Py	sandstone, siltstone, mudstone	unsuitable	unsuitable due to high plasticity and intergranular cement
Pa	sandstone siltstone, mudstone	unsuitable	unsuitable due to high plasticity and intergranular cement
Km	shale	unsuitable	plasticity of material is too high

## Nonrock

Unit	Deposit Type	Suitability as Aggregate	Uses
Qay	alluvial	fair-poor	concrete, road construction (requires beneficiation)
Qld	bar and spit	good	concrete, road construction
Qld	lacustrine	unsuitable	plasticity of material is too high
Qao	alluvial	fair-poor	concrete, road construction (requires beneficiation)
QTa	alluvial	fair-poor	concrete, road construction (requires beneficiation)
QTa	caliche	good-fair	concrete, road construction

Glorieta rocks, for example, generally exhibit very high point load strength indexes (PLSI), and the unconfined compressive strength (UCS) of these rock units as predicted by their point load strength greatly exceeds that of concrete, even by conservative estimates. The UCS of an intact core segment of Glorieta Sandstone from drill hole 27 (Plate 1) was measured at 92.5 MPa (966 tons/sq ft; New Mexico SSC Proposal, 1987, v. 3). In comparison, the unconfined compressive strength of concrete used for construction of most footings and piers ranges from 17.2 to 34.5 MPa (180-360 tons/sq ft; Peck et al., 1974). Allowable contact pressures beneath foundations will therefore be governed exclusively by settlement associated with rock defects rather than by rock strength. The same holds true for all other rock units generally exhibiting high PLSI values, including Precambrian rock (pCs and pCq), Tertiary monzonite (Tm), and rock of the Bernal and San Andres Formations (Pb/Psa).

Point load strength index (PLSI) values determined on core segments of shale, siltstone, mudstone, and sandstone from the Mancos Shale (Km), the Dockum Group (Trd), and the Yeso Formation (Py) are much lower than for other rock types. Pells (1975) found that for mudstones subjected to axial point load strength tests, the predicted unconfined compressive strength (UCS) of these rocks (based on a conversion factor of 24 proposed by Broch and Franklin, 1972) closely matched actual UCS values. Estimates of UCS values for shales, siltstones, and mudstones of the Mancos Shale (Km), and the Dockum Group (Trd), based on axial PLSI values and a more conservative conversion factor of

20, range from 22.8 to 35.2 MPa (238 to 367 tons/sq ft). In addition, the unconfined compressive strengths of two Yeso core segments tested were measured at 229 and 236 tons/sq ft (New Mexico SSC Proposal, v. 3; 1987). As stated above, the unconfined compressive strength (UCS) of concrete ranges from 17.2 to 34.5 MPa (180-360 tons/sq ft). UCS values of Mancos Shale, Dockum Group siltstone and mudstone, and Yeso sandstone thus may approximate that of concrete. Contact pressures imposed by large foundation loads could therefore be limited by rock strength as well as by rock defects, particularly if loads of high intensity are applied by a small number of piers or piles. Rocks of the Mancos Shale, Dockum Group, and Yeso Formation may therefore require more conservative foundation designs involving distribution of foundation loads among a greater number of piers or piles. Where rocks are unjointed and unfractured, anchoring foundations in more competent rock units may allow substantial increases in contact pressure. This is not applicable to fractured or jointed rocks, however, where contact pressure is limited by the compressibility of the rocks due to defects rather than to rock strength (Peck et al., 1974).

In summary, the predicted unconfined compressive strength of the Mancos Shale (Km), Dockum Group siltstones and mudstones (Trd), and Yeso sandstone (Py) may, by conservative estimates, be low enough that both rock strength and compressibility could limit allowable bearing pressures and thus affect the design of foundations on these units.

## CONCLUSIONS

Based upon field and laboratory test results, the following conclusions can be drawn from this investigation:

1) The engineering behavior of soils is improved by the presence of secondary (pedogenic) calcium carbonate. By acting as a natural soil cement, calcium carbonate increases soil bearing capacity, lowers soil expansion and soil liquefaction potential, and minimizes settlement due to ground-water withdrawal. The stability and economy of foundations constructed on soils is therefore improved. In addition, pedogenic calcium carbonate deposits can be mined locally as an aggregate resource for use in road construction.

2) Problem soils are uncommon. Expansive soils occur locally; however, they are generally thin. Where thicker, they are of such limited aerial extent that they can easily be avoided when selecting construction sites.

Soil and hydrologic conditions necessary for liquefaction are not present in the study area. Cohesionless sands and gravels occur only locally, the water table, in most places, is deep, and high earthquake-induced ground accelerations are unlikely.

Consolidation of saturated, lacustrine clay under foundation loads could possibly lead to settlement of structures in the south-central part of the study area; however, the compressibility of this clay is not exceedingly great. Therefore, structures can be designed to withstand such settlement.

3) Some rock types (the Mancos Shale and Dockum Group siltstones and mudstones) are rippable with a D7-type tractor-ripper arrangement regardless of joint spacing and orientation, due to their low strength. In contrast, the rippability of certain higher-strength rock types (Dockum Group sandstone) is determined only by joint spacing and orientation, not by rock strength. Dockum Group sandstone is locally rippable with a D9-type tractor-ripper arrangement.

4) Depending upon rock type, allowable contact pressures on rocks may be affected either solely by rock quality, or by both rock quality and rock strength. Most rock units exhibit predicted uniaxial compressive strength (UCS) values well in excess of concrete; therefore, only rock quality (and not rock strength) is a critical consideration in the design of foundations on most rock units. The Mancos Shale, Dockum Group siltstones and mudstones, and Yeso sandstone, however, exhibit predicted UCS values approximating that of concrete. For these units, both rock strength (because it is low) and rock quality may affect the design of foundations, particularly where very high foundation loads are involved.

5) Aggregate resources are adequate to supply most future construction needs. Lacustrine beach and bar deposits provide a good source of sand and gravel for concrete manufacture and for all road construction purposes. Lower quality sand and gravel generally requiring beneficiation can also be obtained from young alluvium, older alluvium, and Quaternary-Tertiary alluvium. Limestone from the Madera and San Andres Formations is of excellent to good quality and can also be used for concrete manufacture and for all road

construction purposes. Precambrian quartzite and schist from the Lobo Hill area and Tertiary monzonite from South Mountain are sources of building stone and crushed rock aggregate.

The northern Estancia Valley is geotechnically well suited to support future development and urbanization. Soil conditions are favorable and adequate aggregate resources exist. In addition, there is currently a wealth of geotechnical information available on the area in the form of geologic maps, soil surveys, water well logs, boring logs, trench logs, and laboratory test data for soil and rock units. Considerable savings in future foundation exploration programs can be realized if planners, contractors, and engineers review this information before more detailed, site-specific investigations are made. Information gathered from such sources, which include this investigation, can provide insight into the types of problems likely to be encountered during future construction projects.



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APPENDIX I

Geotechnical drill hole logs



THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	'N' VALUE	STD. PEN. TEST	LABORATORY DATA							USCS SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION
				MOISTURE (% of dry wt.)	DRY DENSITY (lbs./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	SAMPLE TYPE U=Undisturbed D=Disturbed	GRAVEL/SAND/FINES (%)				
6267	0-2	15	6	NA	26	19	D	CL-ML				Gravel/Sand/Fines (%)	
6262	5-6½	8	6	NA	19	16	D	ML				Silty clay 0/46/54	
6259	8-8½	25	2	NA	NP	NP	D	SW-SM				Silt with sand 1/27/72	
6253	14-15½	20	10	NA	26	19	D	CL-ML				Well graded sand with silt 0/90/10	
												Silty clay 0/40/60	

NOTES:

NV = no value  
NP = non-plastic

BORING LOG  
NEW MEXICO SSC PROPOSAL JULY 31, 1987

BORING: SSC-DH-1  
DATE DRILLED: 9/22/86  
EQUIPMENT USED: SIMCO 2800 HS  
LOCATION: NE¼, NE¼, Sec 36, T10N, R8E  
ELEVATION: 6267  
TOTAL DEPTH: 20 feet

THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	LABORATORY DATA					USC9 SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION Gravel/Sand/Fines (%)
		WATER VALUE	STANDARD PEN. TEST	MOISTURE (% of dry wt.)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT			
6276	4-5 1/2	6	7	NA	21	NP	D	SM	
6271	9-10 1/2	27	16	NA	53	22	D	CH	Silty sand 0/79/21
6261	19-20 1/2	26	12	NA	25	19	D	SM/SC	Clay 0/3/97
6251	29-30 1/2	27	9	NA	30	9	D	SC	Silty sand/clayey sand 0/64/36
6241	39-40 1/2	25	13	NA	28	16	D	CL	Clayey sand 0/71/29
									Clay 0/41/59

NOTES: Don King Ranch.  
Heavy rains.

NV - no value  
NP - non-plastic

BORING: SSC-DH-2  
DATE DRILLED: 9/23/86  
EQUIPMENT USED: SINCO 2800 HS  
LOCATION: NE 1/4 NE 1/4 SEC 25 T11N R8E  
ELEVATION: 6280  
TOTAL DEPTH: 40.5 feet

THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	"N" VALUE	STD. PEN. TEST	LABORATORY DATA					SAMPLE TYPE <small>U=Undisturbed D=Disturbed</small>	USCS SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION Gravel/Sand/Fines (%)
				MOISTURE (% of dry wt.)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT					
6309	9-10 <sup>1</sup> / <sub>2</sub>	21	12	NA	38	NP	D	SM			Silty sand 0/67/33	
6299	19-20 <sup>1</sup> / <sub>2</sub>	REF	4	NA	NP	NP	D	SP			Poorly-graded sand with gravel 24/76/0	
6289	29-30 <sup>1</sup> / <sub>2</sub>	30	9	NA	29	15	D	CL			Sandy lean clay 0/39/61	
6273	45	--	12	NA	32	17	D	SC			Clayey sand 0/66/34	

BORING: SSC-DH-3  
 DATE DRILLED: 9/24/86  
 EQUIPMENT USED: SIMCO 2800 HS  
 LOCATION: SW<sup>1</sup>/<sub>4</sub>, SE<sup>1</sup>/<sub>4</sub>, Sec 28, T11N, R9E  
 ELEVATION: 6318  
 TOTAL DEPTH: 45.5 feet

NOTES: REF = refusal

NV = no value  
 NP = non-plastic

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	"N" VALUE	STD. PEN. TEST	LABORATORY DATA					USCS SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION
				MOISTURE (% of dry wt.)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTICITY LIMIT	SAMPLE TYPE			
6314	4-5	14	10	NA	26	17	D	SC		Gravel/Sand/Fines (%)	
6309	9-10	9	11	NA	23	NP	D	SM		Clayey sand 1/69/30	
6299	19-20	29	7	NA	25	21	D	SM/SC		Silty sand 0/72/28	
6289	29-30	39	13	NA	32	17	D	SC		Silty sand/clayey sand 0/69/31	
										Clayey sand 0/60/40	

NOTES: On Highway 471, 2½ miles west of Don King Ranch.  
 Drilled by G. J.

NV = no value  
 NP = non-plastic

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

BORING: SSC-DH-4  
 DATE DRILLED: 9/25/86  
 EQUIPMENT USED: SIMCO 2800 HS  
 LOCATION: SE¼ SE¼ SEC27 T11N R8E  
 ELEVATION: 6318  
 TOTAL DEPTH: 30 feet

THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	"N" VALUE	STD. PEN. TEST	LABORATORY DATA				SAMPLE TYPE	USCS SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION
				MOISTURE (% of dry wt.)	DRY DENSITY (lb <sub>s</sub> /cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT				
6386	4-5	30	10	NA	27	16	D	SC		Clayey sand 0/77/23	
6381	9-10	36	8	NA	36	18	D	SC		Clayey sand 0/60/40	

BORING: SSC-DH-5  
 DATE DRILLED: 9/25/86  
 EQUIPMENT USED: SIMCO 2800 HS  
 LOCATION: SW<sup>1/4</sup>, SW<sup>1/4</sup>, Sec 28, T11N, R8E  
 ELEVATION: 6390  
 TOTAL DEPTH: 15 feet

NOTES:

NV = no value  
 NP = non-plastic

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

BORING: SSC-DH-6  
 DATE DRILLED: 9/26/86  
 EQUIPMENT USED: SIMCO 2800 HS  
 LOCATION: SW $\frac{1}{4}$ , SE $\frac{1}{4}$ , Sec 26, T11N, R9E  
 ELEVATION: 6345  
 TOTAL DEPTH: 24 feet

ELEVATION (ft.)	DEPTH (ft.)	'N' VALUE	STD. PEN. TEST	MOISTURE (% of dry wt.)	LABORATORY DATA				USCS SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION
					DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	SAMPLE TYPE <small>U-Undisturbed D-Disturbed</small>			
6341	4-5	27	7	NA	27	20	D	SH-SC		Silty sand-clayey sand	3/67/30
6336	9-9 $\frac{1}{2}$	REF	12	NA	42	23	D	CL		Lean clay with sand	0/20/80
6326	19-19 $\frac{1}{2}$	REF	16	NA	51	22	D	CH		Fat clay	0/13/87

NOTES:

NV = no value  
 NP = non-plastic

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	"N" VALUE	STD. PEN. TEST	LABORATORY DATA				SAMPLE TYPE	USCS SYMBOL	GEOLOGIC SYMBOL	TOTAL DEPTH: 30 feet
				MOISTURE (% of dry wt.)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT				
6396	4-5	41	6	NA	23	19	D	SM-SC		SOIL DESCRIPTION	
6391	9-10	13	5	NA	22	21	D	SM		Silty sand-clayey sand 0/57/43	
6381	19-20	REF	7	NA	29	23	D	SM		Silty sand 0/73/27	
6371	29-30	REF	7	NA	28	26	D	SM		Silty sand 0/61/39	
										Silty sand 0/79/21	

BORING: SSC-DH-7  
 DATE DRILLED: 9/29/86  
 EQUIPMENT USED: SIMCO 2800 HS  
 LOCATION: NW¼, NW¼, Sec 32, T11N, R10E  
 ELEVATION: 6400  
 TOTAL DEPTH: 30 feet

NOTES:  
  
 NV = no value  
 NP = non-plastic

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	"N" VALUE	STANDARD PEN. TEST				LABORATORY DATA			USCS SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION Gravel/Sand/Fines (%)
			MOISTURE (%)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	SAMPLE TYPE					
6406	4-5 1/2	25	5	NA	21	16	D	SM-SC			SOIL DESCRIPTION	
6401	9-10	23	7	NA	27	22	D	CL			Silty sand-clayey sand 0/66/34	
6391	19-20	28	5	NA	20	18	D	SM			Sandy lean clay 0/45/55	
6381	29-30	34	2	NA	NP	NP	D	SP-SM			Silty sand 0/54/46	
											Poorly graded sand with silt and gravel 23/69/8	

BORING: SSC-DH-8  
 DATE DRILLED: 9/30/86  
 EQUIPMENT USED: SIMCO 2800 HS  
 LOCATION: SE 1/4, NE 1/4, Sec 32, T11N, R10.  
 ELEVATION: 6410  
 TOTAL DEPTH: 30 feet

NOTES:  
 NV = no value  
 NP = non-plastic





















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BORING: SSC-DH-18  
 DATE DRILLED: 10/24/86  
 EQUIPMENT USED: SIMCO 2800 HS  
 LOCATION: NE $\frac{1}{4}$ , SE $\frac{1}{4}$ , Sec 32, T8N, R9E  
 ELEVATION: 6135  
 TOTAL DEPTH: 30 feet

ELEVATION (ft.)	DEPTH (ft.)	SPT VALUE	STANDARD PEN. TEST	MOISTURE (% of dry wt.)	LABORATORY DATA						USCS SYMBOL	GEOLOGIC SYMBOL
					DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	SAMPLE TYPE	SOIL DESCRIPTION	Gravel/Sand/Fines (%)		
6131	4-5 $\frac{1}{2}$	29	7	NA	25	21	D	SM-SC				Silty sand-clayey sand 0/53/47
6122	9-9 $\frac{3}{4}$	21	11	NA	28	17	D	CL		Lean clay with sand 4/22/74		
6121 $\frac{1}{2}$	9 $\frac{3}{4}$ -10 $\frac{1}{2}$	21	26	NA	30	16	D	CL		Lean clay 0/9/91		
6112	19-20	39	4	NA	NP	NP	D	SP-SM		Poorly graded sand with silt and gravel 21/70/9		
6111	20-20 $\frac{1}{2}$	39	10	NA	32	19	D	CL		Sandy lean clay 0/35/65		
6102	29-30	REF	22	NA	63	23	D	CH		Fat clay 0/11/89		

NOTES:

NV = no value  
NP = non-plastic

BORING LOG  
NEW MEXICO SSC PROPOSAL JULY 31, 1987

THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	"N" VALUE	STD. PEN. TEST	LABORATORY DATA				SAMPLE TYPE U-Undisturbed D-Diameter	USCG SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION						
				MOISTURE (% of dry wt.)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT					TOTAL DEPTH: 30.5 feet	BORING: SSC-DH-20	DATE DRILLED: 10/28/86	EQUIPMENT USED: SIMCO 2800 HS	LOCATION: NW <sup>1</sup> / <sub>4</sub> , NW <sup>1</sup> / <sub>4</sub> , Sec 29, T8N, R9E	ELEVATION: 6146
6146	0-1½	9	16	NA	34	18	D	CL		Sandy lean clay 0/42/58							
6144½	1½-3	14	8	NA	32	17	D	CL		Lean clay with sand 0/37/63							
6143	3-4½	14	14	NA	32	25	D	ML		Silt 2/12/86							
6141½	4½-6	17	15	NA	35	21	D	CL		Lean clay with sand 2/19/79							
6140	6-7½	18	15	NA	34	15	D	CL		Lean clay with sand 2/23/75							
6138½	7½-9	15	17	NA	29	17	D	CL		Lean clay with sand 0/20/80							
6137	9-10½	18	18	NA	37	20	D	CL		Lean clay with sand 0/15/85							
6135½	10½-12	15	16	NA	29	17	D	CL		Sandy lean clay 1/44/55							
6134	12-13	13	11	NA	24	NP	D	ML		Silt 0/38/62							
6133	13-13½	--	--	NA	32	15	D	CL		Lean clay with sand 0/19/81							
6132½	13½-15	19	16	NA	41	15	D	CL		Lean clay with sand 0/15/85							
6131	15-16½	21	26	NA	37	17	D	CL		Lean clay with sand 0/26/74							
6129½	16-18	32	12	NA	31	14	D	CL		Lean clay with sand 3/18/79							
6128	18-19½	14	30	NA	57	22	D	CH		Fat clay 1/11/88							
6126½	19½-21	17	21	NA	43	16	D	CL		Lean clay 0/6/94							
6125	21-22½	14	29	NA	40	19	D	CL		Lean clay 0/9/91							
6117	29-30½	40	15	NA	34	20	D	CL		Lean clay 6/3/91							

NOTES:  
NV = no value  
NP = non-plastic

BORING LOG  
NEW MEXICO SSC PROPOSAL JULY 31, 1987



THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	'N' VALUE	STD. PEN. TEST	MOISTURE (% of dry wt.)	LABORATORY DATA				USCS SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION Gravel/Sand/Fines (%)
					DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	SAMPLE TYPE (Disturbed/Undisturbed)			
6291	4-5½	14	11	NA	25	21	D	CL-ML		Silty clay 0/22/78	
6286	9-10½	24	9	NA	24	17	D	CL-ML		Silty clay 0/32/68	
6276	19-20½	31	5	NA	25	19	D	SM-SC		Silty sand-clayey sand 0/65/35	
6266	29-30½	29	7	NA	22	19	D	SM		Silty sand 0/61/39	

NOTES:

NV = no value  
 NP = non-plastic

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

BORING: SSC-DH-22  
 DATE DRILLED: 10/29/86  
 EQUIPMENT USED: SIMCO 2800 HS  
 LOCATION: SW¼, NE¼, Sec 6, T8N, R10E  
 ELEVATION: 6295  
 TOTAL DEPTH: 30.5 feet

THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	'N' VALUE	STD. PEN. TEST	LABORATORY DATA					USCS SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION Gravel/Sand/Fines (%)
				MOISTURE (% of dry wt.)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	SAMPLE TYPE U=Undisturbed D=Disturbed			
6221	5-5½	25	12	NA	44	20	D	CL		Lean clay with sand 0/23/77	
6216	9-10½	23	7	NA	26	18	D	CL		Sandy lean clay 0/49/51	
6206	19-20½	32	22	NA	47	23	D	CL		Lean clay 0/14/86	
6196	29-30½	29	18	NA	38	19	D	CL		Lean clay with sand 0/15/85	

BORING: SSC-DH-23  
 DATE DRILLED: 10/30/86  
 EQUIPMENT USED: SIMCO 2800 HS  
 LOCATION: NW¼, NE¼, Sec 13, T9N, R8E  
 ELEVATION: 6225  
 TOTAL DEPTH: 30.5 feet

NOTES:

NV = no value  
 NP = non-plastic

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

JULY 31, 1987

THIS LOG IS REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	"N" VALUE	STD. PEN. TEST	LABORATORY DATA						USCS SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION
				MOISTURE (% of dry wt.)	DRY DENSITY (lbs./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	SAMPLE TYPE U=Undisturbed D=Disturbed	GRAVEL/SAND/FINES (%)			
6251	4-5 1/2	23	9	NA	32	20	D	CL			Lean clay 0/14/86	
6246	9-10 1/2	20	7	NA	29	19	D	CL			Lean clay 0/12/88	
6236	19-20 1/2	23	13	NA	24	21	D	ML			Silt with sand 7/21/72	
6226	29-30 1/2	7	22	NA	29	12	D	CL			Sandy lean clay 0/34/66	

BORING: SSC-DH-24  
 DATE DRILLED: 10/30/86  
 EQUIPMENT USED: SINCO 2800 HS  
 LOCATION: SW 1/4, NW 1/4, Sec 35, T10N, R8E  
 ELEVATION: 6255  
 TOTAL DEPTH: 30.5 feet

NOTES:

NV = no value  
 NP = non-plastic

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

BORING: SSC-DH-25  
 DATE DRILLED: 10/31/86  
 EQUIPMENT USED: SIMCO 2800 HS  
 LOCATION: NW¼, NW¼, Sec 23, T10N, R8E  
 ELEVATION: 6275  
 TOTAL DEPTH: 30.5 feet

ELEVATION (ft.)	DEPTH (ft.)	"N" VALUE	STD. PEN. TEST	MOISTURE (% or dry wt.)	LABORATORY DATA				SAMPLE TYPE	USCS SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION
					DRY DENSITY (lbs./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT					
6271	4-5½	7	4	NA	NP	NP	D	SP-SM			Gravel/Sand/Fines (%) 17/72/11	
6266	9-9½	29	4	NA	NP	NP	D	SP			Poorly graded sand with silt and gravel	
6265½	9½-10½	29	9	NA	31	18	D	CL			Coarse sand * 18/47/35	
6256	19-19½	REF	8	NA	31	25	D	CL			Sandy lean clay 1/37/62	
6255	20-20½	REF	8	NA	28	19	D	CL			Sandy lean clay 5/43/52	
6246	29-30½	28	7	NA	28	16	D	SC			Sandy lean clay 0/43/57	
											Clayey sand with gravel 49/20/31	

NOTES: REF = refusal  
 \*Field description

NV = no value  
 NP = non-plastic

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

SOURCE: SSM

NOTED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	"N" VALUE	STD. PEN. TEST	LABORATORY DATA				SAMPLE TYPE U-Undisturbed D-Disturbed	USCS SYMBOL	GEOLOGIC SYMBOL	SOIL DESCRIPTION
				MOISTURE (% of dry wt.)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT				
6301	4-5½	14	8	NA	30	19	D	CL			
6296	9-9½	20	4	NA	23	19	D	CL-ML		Sandy lean clay 0/38/62	
6286	19-20½	39	16	NA	46	23	D	CL		Lean clay-silt 4/46/50	
6276	29-30½	19	21	NA	31	19	D	CL		Lean clay 0/12/88	
											Lean clay with sand 1/27/72

NOTES:

NV = no value  
 NP = non-plastic

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

BORING: SSC-DH-26  
 DATE DRILLED: 10/31/86  
 EQUIPMENT USED: SIMCO 2800 HS  
 LOCATION: NW¼, NE¼, Sec 21, T10N, R8E  
 ELEVATION: 6305  
 TOTAL DEPTH: 30.5 feet

6001174



CORRECTION TO BE MADE TO THE DATA OBTAINED FROM THE BOREHOLE AT OTHER LOCATIONS AND TIMES.

BORING: SSC-DH-27  
 DATE DRILLED: 4/20/87  
 EQUIPMENT USED: CME-55  
 LOCATION: NW $\frac{1}{4}$ , SE $\frac{1}{4}$ , Sec. 31, T10N, R10E  
 ELEVATION: 6380  
 TOTAL DEPTH: 149 feet

ELEVATION (ft.)	DEPTH (ft.)	WATER VALUE	MOISTURE (% of dry wt.)	LABORATORY DATA				SAMPLE TYPE U.D.C.P.	USCS, AGI SYMBOL	GEOLOGIC SYMBOL OR FORMATION	DESCRIPTION OF ROCK AND SOIL
				DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	PLASTICITY INDEX				
6380	0-3	NA	NV	NV	NA	NA	NA	NA	NA		
6377	3-9	NA	NV	150	NA	NA	C	SS	Pg		not sampled
6371	9-19	NA	NV	NV	NA	NA	C	SS	Pg		clean quartz sandstone
6361	19-29	NA	NV	NV	NA	NA	C	SS	Pg		clean quartz sandstone with iron staining
6351	29-39	NA	NV	NV	NA	NA	C	SS	Pg		at 26 feet hit brecciated zone
6341	39-49	NA	NV	NV	NA	NA	C	SS	Pg		very fractured zone
6331	49-59	NA	NV	150	NA	NA	C	SS	Pg		brecciated zone
6321	59-69	NA	NV	NV	NA	NA	C	SS	Pg		minor limonitic seams
6311	69-79	NA	NV	NV	NA	NA	C	SS	Pg		limonitic partings
6301	79-89	NA	NV	NV	NA	NA	C	SS	Pg		very fractured
6291	89-99	NA	NV	NV	NA	NA	C	SS	Pg		very weathered, soft
6281	99-109	NA	NV	162	NA	NA	C	SS	Pg		some minor heavy metals
6271	109-119	NA	NV	NV	NA	NA	C	SS	Pg		clean quartz sandstone
6261	119-129	NA	NV	NV	NA	NA	C	SS	Pg		very fractured
6251	129-139	NA	NV	156	NA	NA	C	SS	Pg		copper and manganese staining
6241	139-149	NA	NV	150	NA	NA	C	SS	Pg		cross-bedded sandstone
											iron-stained sandstone

NOTES: Located at Highway Department rock quarry approximately one and one-half mile south-east of Montoya Ranch.

**SAMPLE TYPES**  
 U = undisturbed soil  
 D = disturbed soil  
 C = rock core  
 P = plug (cuttings recovered)

NA = Not applicable  
 NV = no value  
 NP = non-plastic

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	"w" VALUE	MOISTURE % of dry wt.	LABORATORY DATA				SAMPLE TYPE U.O.C.P.	USCS, AGI SYMBOL	GEOLOGIC SYMBOL OR FORMATION	TOTAL DEPTH: 53 feet
				DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT					
											DESCRIPTION OF ROCK AND SOIL
6576	0-5	NA	NA	NV	NA	NA	P	ML-CL			sandy silt-silty clay
6571	5-10	NA	NA	NV	NA	NA	P	ML			Sandy silt
6566	10-15	NA	NA	NV	NA	NA	P	SM-ML			Silty sand-sandy silt
6561	15-20	NA	NA	NV	NA	NA	P	SM-SP			Silty sand-gravelly sand
6556	20-25	NA	NA	NV	NA	NA	P	SP			gravelly sand
6551	25-30	NA	NA	NV	NA	NA	P	SP			coarse sand
6546	30-35	NA	NA	NV	NA	NA	P	SP			medium sand
6541	35-40	NA	NA	NV	NA	NA	P	SP			gravelly sand
6536	40-45	NA	NA	NV	NA	NA	P	SP			gravelly sand
6531	45-50	NA	NA	NV	NA	NA	P	SP			medium sand with gravel
6526	50-53	NA	NA	NV	NA	NA	P	SP			medium sand with gravel

BORING: SSC-DH-28  
 DATE DRILLED: 4/20/87  
 EQUIPMENT USED: Failing 1500W  
 LOCATION: NW<sup>1</sup>/<sub>4</sub>, NW<sup>1</sup>/<sub>4</sub>, Sec. 4, T11N, R8E  
 ELEVATION: 6576

**NOTES:** Drilling mud was used in hole after 20 feet. Drilled on King's land, 1 mile east of Hyer.  
 NA = Not applicable  
 NV = no value  
 NP = non-plastic

**SAMPLE TYPES**  
 U = undisturbed soil  
 D = disturbed soil  
 C = rock core  
 P = plug (cuttings recovered)

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

NOTED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	LABORATORY DATA							USCS, AGI SYMBOL	GEOLOGIC SYMBOL OR FORMATION	DESCRIPTION OF ROCK AND SOIL
		NP VALUE	MOISTURE (% of dry wt.)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	SAMPLE TYPE				
										BORING: SSC-DH-29	
										DATE DRILLED: 4/21-22/87	
										EQUIPMENT USED: Failing 1500W	
										LOCATION: NW 1/4, NW 1/4, Sec. 36, T10N, R9E	
										ELEVATION: 6228	
										TOTAL DEPTH: 355 feet	
6228	0-5	NA	NV	NA	NA	NA	P	SM			
6223	5-10	NA	NV	NA	NA	NA	P	SM		silty sand 70/25/5	
6218	10-15	NA	NV	NA	NA	NA	P	ML-CL		silty sand, 60/30/10	
6213	15-20	NA	NV	NA	NA	NA	P	SM-ML		sandy silt-silty clay, 30/50/20	
6208	20-35	NA	NA	NA	NA	NA	C	SS	Pg	silty sand-sandy silt, 50/40/10	
6193	35-50	NA	NA	150	NA	NA	C	SS	Pg	clean quartz sandstone	
6178	50-100	NA	NA	NV	NA	NA	P	SS	Pg	clean quartz sandstone	
6128	100-120	NA	NA	NV	NA	NA	P	SS	Pg	clean quartz sandstone, water at 98 ft	
6108	120-135	NA	NA	150	NA	NA	C	SS	Pg	iron staining	
6093	135-160	NA	NA	NV	NA	NA	P	SS	Pg	well-cemented, some cross-bedding	
6068	160-175	NA	NA	156	NA	NA	C	SS	Pg	iron staining	
6053	175-235	NA	NA	NV	NA	NA	P	SS	Pg	some weathered zones	
5993	235-280	NA	NA	NV	NA	NA	P	SS	Py	color change to orange-red at 235 ft	
										orange and red fragments of siltstone and claystone	
5948	280-295	NA	NA	137	NA	NA	C	SS	Py	silty sandstone, easily scratched, some weathered clay zones, abundant bioturbation	
5928	300-340	NA	NA	NV	NA	NA	P	SS/Sh	Py	brick-red to gray sandstones and claystones	
5888	340-355	NA	NA	137	NA	NA	C	Sh	Py	4 feet of recovery, soft sandy claystone with drusy quartz growths	

NOTES: Permian Glorieta sandstone from 18 feet to 235 feet = 217 feet thick  
Permian Yeso Formation from 235 feet

NA = Not applicable  
NV = no value  
NP = non-plastic

SAMPLE TYPES

U = undisturbed soil  
D = disturbed soil  
C = rock core  
P = plug (cuttings recovered)

BORING LOG  
NEW MEXICO SSC PROPOSAL JULY 31, 1987

INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

BORING: SSC-DH-30  
 DATE DRILLED: 4/21/87  
 EQUIPMENT USED: CME-55  
 LOCATION: NW<sub>4</sub>, NE<sub>4</sub>, Sec. 32, T11N, R10E  
 ELEVATION: 6400  
 TOTAL DEPTH: 89 feet

ELEVATION (ft.)	DEPTH (ft.)	LABORATORY DATA										DESCRIPTION OF ROCK AND SOIL
		WATER VALUE		MOISTURE (C% of dry wt.)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	SAMPLE TYPE	USCS, AGI SYMBOL	GEOLOGIC SYMBOL OR FORMATION		
6400	0-3	NA	NA	NA	NA	NA	NA	NA	NA	NA	NA	
6397	3-5	NA	NV	NV	NV	NV	U	CL-ML				not sampled
6395	5-8	15	14	NV	NV	NV	D	SC				silty clay
6392	8-10	NA	NV	NV	NV	NV	U	ML				clayey sand
6390	10-13	14	14	NV	NV	NV	D	SM				sandy silt
6387	13-15	NA	NV	NV	NV	NV	U	SM				silty sand
6385	15-18	22	7	NV	NV	NV	D	SM				silty sand
6382	18-20	NV	NV	NV	NV	NV	U	SM				silty sand
6380	20-23	17	18	NV	NV	NV	D	SM-SC				silty sand
6377	23-25	NA	NV	NV	NV	NV	U	SM				silty sand-clayey sand
6375	25-28	34	8	NV	NV	NV	D	SM				silty sand with gravel
6372	28-33	>50	12	NV	NV	NV	D	SS/Sh	Rd			silty sand
6367	33-38	>50	8	NV	NA	NA	D	SS/Sh	Rd			dark red-top of Chinle
6362	38-43	>50	7	NV	NA	NA	D	SS/Sh	Rd			maroon sandstone and claystone
6357	43-48	>50	2	NV	NA	NA	D	SS/Sh	Rd			fissile sandstone and claystone
6352	48-53.5	>50	8	NV	NA	NA	D	SS/Sh	Rd			green reduction spots
6342	58-59	NA	8	NV	NA	NA	C	SS/Sh	Rd			red fissile sandstone
6341	59-69	NA	3	144	NA	NA	C	SS/Sh	Rd			mottled green sandstone and claystone
6331	69-79	NA	NV	NV	NA	NA	C	SS/Sh	Rd			red sandstone and claystone
6321	79-89	NA	NV	147	NA	NA	C	SS/Sh	Rd			red sandstone and claystone

**NOTE 3:** Located just northeast of Marshal Rowley Ranch  
 At 28 feet contact of Triassic Chinle Formation.  
 Rock from 33-61 feet very soft and weathered.

NA = Not applicable  
 NV = no value  
 NP = non-plastic

**SAMPLE TYPES**  
 U = undisturbed soil  
 D = disturbed soil  
 C = rock core  
 P = plug (cuttings recovered)

**BORING LOG**  
**NEW MEXICO SSC PROPOSAL JULY 31, 1987**

BORE HOLE CONDITIONS AT OTHER LOCATIONS AND TIMES.

BORING: SSC-DH-31  
 DATE DRILLED: 4/22/87  
 EQUIPMENT USED: CME-55  
 LOCATION: NE $\frac{1}{4}$ , NE $\frac{1}{4}$ , Sec. 6, T11N, R9E  
 ELEVATION: 6421  
 TOTAL DEPTH: 61 $\frac{1}{2}$  feet

ELEVATION (ft.)	DEPTH (ft.)	LABORATORY DATA							SAMPLE TYPE	USCS, AGI SYMBOL	GEOLOGIC SYMBOL OR FORMATION	DESCRIPTION OF ROCK AND SOIL
		MP VALUE	MOISTURE (%)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	USCS, AGI SYMBOL	USCS, AGI SYMBOL				
6421	0-2	>50	NA	NA	NA	NA	NA	NA	NA	NA		
6419	2-7	33	7.47	NV	NA	NA	D	SM			not sampled	
6414	7-12	>50	11.37	NV	NA	NA	D	SM			silty sand with gravel	
6409	12-17	>50	10.27	NV	NA	NA	D	SM-SC			silty sand	
6404	17-22	>50	1.86	NV	NA	NA	D	SP			silty-clayey sand	
6399	22-28	>50	5.65	NV	NA	NA	D	SM			gravelly-fine sand	
6393	28-34	>50	NV	NV	NA	NA	D	SM			silty sand	
6387	34-40	>50	NV	NV	NA	NA	D	SM-ML			silty sand with caliche	
6381	40-45	>50	NV	NV	NA	NA	D	SM			silty sand-clayey silt	
6376	45-51	>50	NV	NV	NA	NA	D	SM			silty sand	
6370	51-55	>50	NV	NV	NA	NA	D	SP			silty sand	
6366	55-61	>50	NV	NV	NA	NA	D	SM			sand with gravel	
											silty sand with gravel	

NOTE: On Hyer Road 3 miles west of NM-41.

NA = Not applicable  
 NV = no value  
 NP = non-plastic

**SAMPLE TYPES**  
 U = undisturbed soil  
 D = disturbed soil  
 C = rock core  
 P = plug (cuttings recovered)

**BORING LOG**  
**NEW MEXICO SSC PROPOSAL JULY 31, 1987**

RECORD THESE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	W <sub>m</sub> VALUE	MOISTURE (% of dry wt.)	LABORATORY DATA					SAMPLE TYPE U.D.C.A.	USCS, AGI SYMBOL	GEOLOGIC SYMBOL OR FORMATION	DESCRIPTION OF ROCK AND SOIL
				DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT						
												BORING: SSC-DH-32 DATE DRILLED: 4/23/87 EQUIPMENT USED: Falling 1500W LOCATION: SW <sub>4</sub> , SW <sub>4</sub> , Sec. 35, T12N, R9E ELEVATION: 6540 TOTAL DEPTH: 155 feet
6540	0-20	NA	NA	NA	NA	NA	NA	P	MI-CL			
6520	20-45	NA	NA	NA	NA	NA	NA	P	CL-ML			sandy silt-silty clay
												silty clay-sandy silt (some caliche zones)
6495	45-55	NA	NA	NA	NA	NA	NA	P	SP-GP			coarse sand-fine gravel
6485	55-75	NA	NA	NA	NA	NA	NA	P	SP-GP			coarse sand-fine gravel (minor clay present)
6465	75-95	NA	NA	NA	NA	NA	NA	P	ML-CL			sandy silt-lean clay (minor gravel)
6445	95-100	NA	NA	NA	NA	NA	NA	P	SP-GP			coarse sand-fine gravel
6440	100-120	NA	NA	NA	NA	NA	NA	P	ML-CL			sandy silt-lean clay (about 10% gravel)
6420	120-140	NA	NA	NV	NA	NA	NA	P	Sh	Km		light-yellowish brown shale
6400	140-155	NA	NA	134	NA	NA	NA	C	Sh	Km		blue-black sandy mudstone, breaks easily along bedding planes, beds dipping at 25-45°

NOTES: Located at intersection of Hyer Road and Highway 41. At 120 feet contact of Cretaceous Mancos Shale.

NA = Not applicable  
NV = no value  
NP = non-plastic

SAMPLE TYPES  
U = undisturbed soil  
D = disturbed soil  
C = rock core  
P = plug (cuttings recovered)

**BORING LOG**  
NEW MEXICO SSC PROPOSAL JULY 31, 1987

THE LOG OF SUBSURFACE CONDITIONS SHOWN HEREON APPLIES ONLY AT THE SPECIFIC BORING LOCATION AND AT THE DATE INDICATED. IT IS NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	NP VALUE	MOISTURE (% of dry wt.)	LABORATORY DATA				SAMPLE TYPE U.D.C.P.	USCS, AGI SYMBOL	GEOLOGIC SYMBOL OR FORMATION	DESCRIPTION OF ROCK AND SOIL
				DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT					
											TOTAL DEPTH: 38.5 feet
6576	0-3	NA	NA	NA	NA	NA	NA	NA	NA	not sampled	
6573	3-5	NA	NV	NV	NV	NV	U	ML		sandy silt	
6571	5-8	23	8.38	NV	NV	NV	D	ML		sandy silt with gravel	
6568	8-13	48	8.71	NV	NV	NV	D	ML		sandy silt with 0.5 inch carbonate nodules	
6563	13-18	26	7.25	NV	NV	NV	D	SM		silty sand with interbedded sand	
6558	18-23	>50	4.31	NV	NV	NV	D	SM		silty sand with gravel	
6553	23-28	32	6.16	NV	NV	NV	D	SM-ML		silty sand-sandy silt with carbonate nodules	
6548	28-32	>50	NA	NA	NA	NA	NA	NA	NA	no sample recovered	
6544	32-37	>50	9.36	NV	NV	NV	D	SM-ML		silty sand-sandy silt with gravel and carbonate	
6539	37-38.5	>50	6.82	NV	NV	NV	D	ML		sandy silt with gravel	

NOTE 3: Located on Hyer Road, same location as DH-28 and 34.  
Hit hard caliche at approximately 10 feet.

NA = Not applicable  
NV = no value  
NP = no plastic

**SAMPLE TYPES**

U = undisturbed soil  
D = disturbed soil  
C = rock core  
P = plug (cuttings recovered)

**BORING LOG**

NEW MEXICO SSC PROPOSAL JULY 31, 1987

BORING: SSC-DH-33  
DATE DRILLED: 4/23-24/87  
EQUIPMENT USED: CME-55  
LOCATION: NW 1/4, NW 1/4, Sec. 4, T11N, R8E  
ELEVATION: 6576

... REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

ELEVATION (ft.)	DEPTH (ft.)	LABORATORY DATA								USCS, AGI SYMBOL	GEOLOGIC SYMBOL OR FORMATION	DESCRIPTION OF ROCK AND SOIL
		W/W VALUE	MOISTURE (% of dry wt.)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	SAMPLE TYPE	U.D.C.P.				
6576	0-25	NV	NA	NA	NA	NA	P	ML-CL				
6551	25-65	NV	NA	NA	NA	NA	P	SP-SM				Sandy silt-silty clay
6541	65-105	NV	NA	NA	NA	NA	P	SP				Coarse sand with silt
6471	105-135	NV	NA	NA	NA	NA	P	SM-GM				Coarse sand with fine gravel
6441	135-160	NV	NA	NA	NA	NA	P	SM-GM				Medium sand-medium gravel with clay
												Medium sand-medium gravel with dark red clay
6416	160-180	NV	NA	NV	NA	NA	P	Sh	R d			dark-red sandy claystone
6496	180-195	NV	NA	156	NA	NA	C	SS/Sh	R d			dark-red sandy siltstone with claystone
												beds dip approximately 30°. Expansive claystone forced open sample barrel.

BORING: SSC-DH-34  
 DATE DRILLED: 4/24/87  
 EQUIPMENT USED: Falling 1500W  
 LOCATION: NW<sup>1</sup>/<sub>4</sub>, NW<sup>1</sup>/<sub>4</sub>, Sec. 4, T11N, R8E  
 ELEVATION: 6576  
 TOTAL DEPTH: 195 feet

NOTE: Located on Hyer Road, same location as DH-28 and 32.  
 At 160 feet, contact of Triassic Chinle claystone.  
 Some green claystone beds in Chinle.

NA = Not applicable  
 NV = no value  
 NP = non-plastic

SAMPLE TYPES  
 U = undisturbed soil  
 D = disturbed soil  
 C = rock core  
 P = plug (cuttings recovered)

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987



RESULTS TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TINES.

ELEVATION (ft)	DEPTH (ft)	LABORATORY DATA								USCS, AGI SYMBOL	GEOLOGIC SYMBOL OR FORMATION	DESCRIPTION OF ROCK AND SOIL
		WATER VALUE	MOISTURE (% of dry wt.)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	SAMPLE TYPE U.D.C.P.					
6183	0-3	NA	NA	NA	NA	NA	NA	NA	NA	NA	DESCRIPTION OF ROCK AND SOIL	
6180	3-4.75	NA	NV	NV	NV	NV	U	CL			not sampled	
6178	4.75-8	12	18	NA	NV	NV	D	CL			green lean clay with gypsum	
6175	8-9.75	NA	8	NV	NV	NV	U	ML			green clay with gypsum	
6173	9.75-13	15	13	NA	NV	NV	D	SM			sandy silt	
											fine clean sand with some silts and clays	
6170	13-15	NA	NV	NV	NV	NV	U	SM			fine sand with mottled clay	
6168	15-18.5	7	18	NA	NV	NV	D	ML-SP			clayey silt with fine sand at top	
6164	18.5-20.25	NA	NV	NV	NV	NV	U	CL-CH			lean clay-plastic clay	
6163	20.25-23.25	5	32	NA	NV	NV	D	CH			plastic clay with carbon filaments	
6160	23.25-25	NA	16	NV	NV	NV	U	SP			fine clean sand	
6158	25-28.25	30	NV	NA	NV	NV	D	SP			fine clean sand-poorly graded sand	
6155	28.25-28.5	NA	14	NV	NV	NV	U	SP			saturated gravelly sand	
6154	28.5-33	NA	NA	NA	NA	NA	NA	NA	NA	NA	no sample recovered	
6150	33-35	NA	NV	NV	NV	NV	U	CL			yellowish-red hard clay	
6148	35-38	15	30	NA	NV	NV	D	CH			very stiff clay with carbonate nodules	
6144	38.5	NV	NA	NA	NA	NA	NA	NA	NA	NA	no sample recovered, running sand, saturated	
6144	38.5-43.5	NV	NA	NA	NA	NA	NA	NA	NA	NA	no sample recovered, running sand, saturated	
6139	43.5	NV	NA	NA	NA	NA	NA	NA	NA	NA	no sample recovered, running sand, saturated	
6139	43.5-45	22	17	NA	NV	NV	D	SM			silty sand with clay at top	
6138	45-51	23	28	NA	NV	NV	D	CL-SM			stiff clay with silty sand at top	
6132	51-53	NA	NA	NA	NA	NA	NA	NA	NA	NA	no sample recovered	
6130	53-54.75	NA	25	104	NV	NV	U	CH-SM			yellow clay with running sand at top	
6128	54.75-58	NA	NA	NA	NA	NA	NA	NA	NA	NA	no sample recovered	
6125	58-59.25	NA	NV	103	NV	NV	U	CL-ML			silty clay with carbonate nodules	

BORING: SSC-DH-35  
 DATE DRILLED: 4/24/87  
 EQUIPMENT USED: CME-55  
 LOCATION: SE4, SE4, Sec. 17, T9N, R9E  
 ELEVATION: 6183  
 TOTAL DEPTH: 65 feet

NOTES: Located on Martinez Road just east of Salt Draw.  
 Water at 50.5 ft depth after 2 days.

NA = Not applicable  
 NV = no value  
 NP = non-plastic

SAMPLE TYPES  
 U = undisturbed soil  
 D = disturbed soil  
 C = rock core  
 P = plug (cuttings recovered)

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987











RESULTS ARE NOT WARRANTED TO BE REPRESENTATIVE OF SUBSURFACE CONDITIONS AT OTHER LOCATIONS AND TIMES.

BORING: SSC DH-40  
 DATE DRILLED: 4/27/87  
 EQUIPMENT USED: CME-55  
 LOCATION: SE4, SE4, Sec. 19, T9N, R9E  
 ELEVATION: 6198  
 TOTAL DEPTH: 69.5 feet

ELEVATION (ft.)	DEPTH (ft.)	LABORATORY DATA										GEOLOGIC SYMBOL OR FORMATION	DESCRIPTION OF ROCK AND SOIL
		WATER VALUE	MOISTURE (% of dry wt.)	DRY DENSITY (lb./cu. ft.)	LIQUID LIMIT	PLASTIC LIMIT	SAMPLE TYPE	U.S.C.S. AGI SYMBOL					
6198	0-3	NA	NA	NA	NA	NA	NA	NA	NA	NA	NA	NA	not sampled
6195	3.25-5	NA	NV	NV	NV	NV	U	ML					sandy silt
6193	5-8	9	9	NV	NV	NV	D	ML					sandy silt with rock fragments
6190	8-10	NA	NV	NV	NV	NV	U	SM					silty sand
6188	10-13	28	11	NV	NV	NV	D	ML/SM					sandy silt-silty sand with caliche nodules
6185	13-14.25	NA	NV	NV	NV	NV	U	CL-ML					sandy silty clay
6184	14.25-18	18	16	NV	NV	NV	D	CL-ML					sandy, silty clay
6180	18-18.5	NA	NV	NV	NV	NV	U	SM					silty sand with gravel
6179	18.5-23	28	10	NV	NV	NV	D	ML/CL					sandy silt-silty clay, siltier at top
6175	23-28	22	20	NV	NV	NV	D	CL					gravel at top, white lake clay bottom
6170	28-33	41	14	NV	NV	NV	D	SP-SM					poorly graded sand with silt
6165	33-38	41	11	NV	NV	NV	D	SP					fine sand, iron staining
6160	38-43	29	21	NV	NV	NV	D	CH					clay with some gravelly sand stringers
6155	43-48	23	19	NV	NV	NV	D	CL-ML					silty clay-clayey silt with carbonate stringers
6150	48-53	22	21	NV	NV	NV	D	SP-ML					fine sand-sandy silt
6145	53-55	NA	NV	NV	NV	NV	U	SC					clayey sand
6143	55-58	26	18	NV	NV	NV	D	SM-SC					silty, clayey sand with carbonate nodules
6140	58-63	27	18	NV	NV	NV	D	SM/ML					silty sand-sandy silt with marl clasts
6135	63-68	31	24	NV	NV	NV	D	ML/CL					sandy silt-stiff clay with gravel
6130	68-69.5	>50	8	NV	NV	NV	D	GW					sandy gravel, mostly Pm clasts

NOTE: Located on west end of airport.

NA = Not applicable  
 NV = no value  
 NP = non-plastic

SAMPLE TYPES  
 U = undisturbed soil  
 D = disturbed soil  
 C = rock core  
 P = plug (cuttings recovered)

BORING LOG  
 NEW MEXICO SSC PROPOSAL JULY 31, 1987

APPENDIX II

Backhoe trench logs



APPENDIX III

Laboratory test data for soil

Sample No.	Depth of Sample (ft)	Eng. Geo. Unit	Moisture Content (%)	Atterberg Limits		Sieve Analysis (% Passing)											USCS SOIL DESCRIPTION	AASHTO SOIL DESCRIPTION
				LL	PI	No. No.												
						1/2"	3/8"	1/4"	4	20	40	60	80	100	140	200		
S-1	2	Qld		NP	NP	100	100	100	99	97	94	88	77	41	12	SP	sand	A-2-4(0)
S-2	2	Qld		24	5	100	100	100	95	92	87	83	73	65	50	CL-ML	silty clay with sand	A-4(0)
S-3	2	Qao		33	13	100	100	100	98	97	96	96	95	88	75	CL	clay	A-6(8)
S-4	2	Qao		21	4	89	83	80	77	69	63	48	39	34	26	SM-SC	clayey sand with silt	A-2-4(0)
S-5	2	Qao		29	8	100	100	100	81	65	64	63	62	58	44	SM	silty sand with gravel	A-4(1)
S-6	2	Qay		NP	NP	90	81	71	63	25	14	9	8	7	5	SW	sand with gravel	A-1-a(0)
S-7	2	Qld		27	10	100	100	100	100	100	99	99	95	88	68	CL	clay with sand	A-4(4)
S-10	2	Qay		29	10	100	100	100	100	99	98	96	92	83	66	CL	sandy clay	A-4(4)
S-11	2	Qao		25	6	100	100	100	99	98	97	94	89	75	56	CL-ML	silty clay	A-4(1)
S-12	2	Qay		NP	NP	100	100	100	99	97	94	88	76	39	12	SP-SM	sand with silt	A-2-4(0)
S-13	2	Qay		NP	NP	96	91	77	60	4	3	2	2	2	2	SP	sand with gravel	A-1-a(0)
S-14	2	Qld		NP	NP	87	85	82	78	27	13	7	6	5	4	SW	sand with gravel	A-1-b(0)
S-15	2	Qld		34	14	100	100	100	83	79	75	73	70	61	46	SC	clayey sand	A-6(3)
S-16	2	QTa		28	11	100	100	100	97	97	96	95	94	87	74	CL	sandy clay	A-6(6)
S-17	2	QTa		NP	NP	28	24	20	18	10	8	7	6	5	4	GP	gravel	A-1-a(0)
S-18	2	QTa		30	13	100	100	100	98	98	99	97	96	94	88	CL	clay	A-6(10)
S-19A	2	Qay		25	6	100	100	100	99	99	98	96	92	78	53	CL-ML	silty clay with sand	A-4(1)
S-19B	2	Qay		NP	NP	53	42	35	31	16	13	11	10	9	7	GP	gravel with sand	A-1-a(0)
S-20	2	Qay		43	18	100	100	100	99	99	98	97	96	87	59	CL	sandy clay	A-7-6(9)
DH-1	2	Qay	6	26	7	100	100	100	96	88	79	75	71	63	54	CL-ML	silty clay	A-4(1)
DH-1	5	QTa	6	19	3	100	100	100	99	97	93	88	85	76	72	ML	silt with sand	A-4(0)
DH-1	8	QTa	2	NP	NP	100	100	100	99	97	93	88	85	76	72	ML	silt with sand	A-1-b(0)
DH-1	14	QTa	10	26	7	100	100	100	98	96	94	90	87	74	60	CL-ML	silty clay	A-4(2)
DH-2	4-5.5	QTa	7	21	NP	100	100	100	100	95	82	71	63	40	21	SM	silty sand	A-2-4(0)
DH-2	9-10.5	QTa	16	53	31	100	100	100	100	100	100	100	100	99	97	CH	clay	A-7-6(34)
DH-2	19-20.5	QTa	12	25	6	100	100	100	96	83	74	68	63	50	36	SM-SC	clayey sand with silt	A-4(0)
DH-2	29-30.5	QTa	9	30	11	100	100	100	98	90	83	77	72	52	29	SC	clayey sand	A-2-6(0)
DH-2	39-40.5	QTa	13	28	12	100	100	100	97	95	91	88	83	74	59	CL	clay	A-6(4)

Laboratory Test Data for Soil Units. NP = nonplastic

Sample No.	Depth of Sample (ft)	Eng. Geo. Unit	Moisture Content (%)	Atterberg Limits		Sieve Analysis (% Passing)											USCS SOIL DESCRIPTION GROUP SYMBOL	AASHTO SOIL DESCRIPTION	
				LL	PI	No. No.													
						1/2"	3/8"	1/4"	4	20	40	60	80	100	140	200			
DH-3	9-10.5	Qao	12	38	NP	100	100	100	100	91	75	63	56	52	43	33	SM	silty sand	A-2-4(O)
DH-3	19	Qao	4	NP	NP	100	99	86	76	38	22	12	9	4	1	0	SP	sand with gravel	A-1-b(O)
DH-3	30	Qao	9	29	14	100	100	100	100	99	87	84	83	80	73	61	CL	sandy clay	A-6(6)
DH-3	45	Qao	12	32	15	100	100	100	100	78	67	60	55	52	44	34	SC	clayey sand	A-2-6(1)
DH-4	4-5	QTa	10	26	9	100	100	99	99	95	83	72	66	60	46	30	SC	clayey sand	A-2-4(O)
DH-4	9-10	QTa	11	23	NP	100	100	100	100	99	93	83	76	70	53	29	SM	silty sand	A-2-4(O)
DH-4	19-20	QTa	7	25	4	100	100	100	100	97	90	83	77	72	55	31	SM-SC	clayey sand with silt	A-2-4(O)
DH-4	29-30	QTa	13	32	15	100	100	100	100	97	88	78	72	67	56	40	SC	clayey sand	A-6(2)
DH-5	4-5	QTa	10	27	11	100	100	100	100	95	84	75	68	60	40	23	SC	clayey sand	A-3(O)
DH-5	9-10	QTa	8	36	18	100	100	100	100	97	89	81	76	72	59	40	SC	clayey sand	A-6(3)
DH-6	4-5	QTa	7	27	7	100	100	98	97	90	83	76	70	64	47	30	SM-SC	clayey sand with silt	A-2-4(O)
DH-6	9-9.5	QTa	12	42	19	100	100	100	100	98	96	94	92	90	86	80	CL	clay with sand	A-7-6(15)
DH-6	19-19.5	QTa	16	51	29	100	100	100	100	100	100	99	97	96	93	87	CH	clay	A-7-6(27)
DH-7	4-5	QTa	6	23	4	100	100	100	96	92	90	84	81	76	60	43	SM-SC	clayey sand with silt	A-4(O)
DH-7	9-10	QTa	5	22	1	100	100	100	100	97	91	84	78	72	51	27	SM	silty sand	A-2-4(O)
DH-7	19-20	QTa	7	29	6	100	100	100	100	98	91	77	69	63	53	39	SM	silty sand	A-4(O)
DH-7	29-30	QTa	7	28	2	100	100	100	99	88	76	65	56	49	34	21	SM	silty sand	A-2-4(O)
DH-8	4-5	QTa	5	21	5	100	100	100	100	99	92	82	76	71	55	34	SM-SC	clayey sand with silt	A-2-4(O)
DH-8	9-10	QTa	7	27	9	100	100	100	100	100	96	90	86	83	75	55	CL	sandy clay	A-4(2)
DH-8	19-20	QTa	5	20	2	100	100	100	100	98	91	84	78	65	46	31	SM	silty sand	A-4(O)
DH-8	29-30	QTa	2	NP	NP	100	93	82	77	56	47	29	22	17	8	3	SP-SM	sand with silt	A-1-b(O)
DH-9	4-5	Qld	13	24	4	100	100	100	100	100	99	96	94	91	78	57	CL-ML	silty clay	A-4(O)
DH-9	9-10	Qld	10	25	6	100	100	100	100	99	93	87	82	77	59	35	SM-SC	clayey sand with silt	A-2-4(O)
DH-9	19-20	Qld	19	42	18	100	100	100	100	99	92	83	76	72	62	47	SC	clayey sand	A-7-6(5)
DH-9	29-30	Qld	14	36	13	100	100	100	100	100	99	95	90	86	83	75	CL	clay	A-6(9)

Laboratory Test Data for Soil Units. NP = nonplastic

Sample No.	Depth of Sample (ft)	Eng. Geo. Unit	Moisture Content (%)	Atterberg Limits		Sieve Analysis (% Passing)											USCS SOIL DESCRIPTION		AASHTO SOIL DESCRIPTION
				LL	PI	1/2"	3/8"	1/4"	4	20	40	60	80	100	140	200	GROUP SYMBOL	GROUP NAME	
DH-10	4-5	Qld	13	20	NP	100	100	100	99	95	88	81	75	57	33	SM	silty sand	A-2-4(0)	
DH-10	9-10	Qld	6	NP	NP	100	100	100	98	70	46	33	24	19	14	SM	silty sand	A-1-b(0)	
DH-10	19-20	Qld	15	20	4	100	100	100	99	99	92	83	78	71	58	CL-ML	silty clay	A-4(0)	
DH-10	29-30	Qld	13	21	9	100	100	100	100	100	100	97	95	83	63	CL	clay	A-4(2)	
DH-11	4.5-5	QTa	7	23	NP	100	100	100	100	99	95	89	74	52	38	SM	silty sand	A-4(0)	
DH-11	9-9.5	QTa	7	24	3	100	100	100	100	99	96	94	89	69	50	ML	sandy silt	A-4(0)	
DH-11	19-20.5	QTa	11	26	4	100	100	100	100	99	98	98	97	94	80	CL-ML	silty clay	A-4(2)	
DH-11	29-29.5	QTa	12	35	12	100	100	100	100	99	96	93	90	81	70	CL	sandy clay	A-6(7)	
DH-12	4-5.5	QTa	7	25	5	100	100	100	99	88	78	72	68	54	36	SM-SC	clayey sand with silt	A-4(0)	
DH-12	9-10.5	QTa	6	24	6	100	100	100	99	92	81	73	68	55	40	SM-SC	clayey sand with silt	A-4(0)	
DH-12	19-20.5	QTa	8	29	10	100	100	100	99	91	82	75	70	57	40	SC	clayey sand	A-4(1)	
DH-13	4-5.5	QTa	5	23	4	100	100	100	94	89	78	68	49	40	30	SM-SC	clayey sand with silt	A-2-4(0)	
DH-13	9-10.5	QTa	11	39	16	100	100	100	100	100	99	99	99	98	96	CL	clay	A-6(17)	
DH-13	19-20.5	QTa	7	23	4	100	100	100	100	100	99	98	96	89	65	CL-ML	silty clay with sand	A-4(0)	
DH-14	4-5.5	QTa	11	39	16	100	100	100	97	94	91	88	85	76	64	CL	sandy clay	A-6(9)	
DH-14	9-10.5	QTa	10	39	20	100	100	100	99	99	98	97	96	89	79	CL	clay	A-6(15)	
DH-14	19-19.5	QTa	9	36	20	100	100	100	99	98	96	96	95	91	81	CL	clay with sand	A-6(15)	
DH-15	4-5.5	Qao	13	51	25	100	100	100	94	90	86	84	82	73	60	CH	sandy clay	A-7-6(13)	
DH-15	9-9.5	Qao	11	57	31	100	100	100	97	96	95	94	93	91	86	CH	clay	A-7-6(29)	
DH-15	19-20.25	Qao	13	60	39	100	100	100	98	97	96	96	95	94	89	CH	clay	A-7-6(38)	
DH-16	4-4.5	Qld	7	28	8	100	100	100	96	96	96	95	95	95	93	CL	clay	A-4(7)	
DH-16	9-9.83	Qld	12	42	22	100	100	100	96	96	96	96	95	95	92	CL	clay	A-7-6(21)	

Laboratory Test Data for Soil Units. NP = nonplastic

Sample No.	Depth of Sample (ft)	Eng. Geo. Unit	Moisture Content (%)	Atterberg Limits		Sieve Analysis (% Passing)										USCS SOIL DESCRIPTION			AASHTO SOIL DESCRIPTION		
				LL	PI	1/2"	3/8"	1/4"	No.					GROUP SYMBOL	GROUP NAME						
									40	60	80	100	140			200					
DH-17	4-4.5	Q1d	15			100	100	100	100	100	100	100	99	99	99	99	99	99	CL-MI	silty clay	A-4(1)
DH-17	4.5-5.5	Q1d	6	24	3	100	100	100	100	99	98	97	94	81					ML	silt with sand	A-6(18)
DH-17	9-10.5	Q1d	13	38	18	100	100	100	100	100	99	99	99	97					CL	clay	A-7-6(54)
DH-17	19.25-20.5	Q1d	40	79	52	100	100	100	100	100	99	99	98	91					CH	clay	A-6(3)
DH-17	29-30.5	Q1d	14	27	12	100	100	100	99	94	89	85	82	70	51				CL	sandy clay	A-6(13)
DH-17	40	Q1d	17	36	18	100	100	93	88	87	87	86	86	84	80				CL	clay	
DH-18	4-5.5	Q1d	7	25	4	100	100	100	100	99	98	96	92	71	47			SM-SC	clayey sand with silt	A-4(0)	
DH-18	9-9.75	Q1d	11	28	11	100	100	100	96	91	89	88	88	87	74			CL	clay with sand	A-6(6)	
DH-18	9.75-10.5	Q1d	26	30	14	100	100	100	100	99	99	98	98	94	91			CL	clay	A-6(11)	
DH-18	19-20	Q1d	4	NP	NP	79	79	79	64	49	31	25	21	15	9			SP-SM	sand with silt and gravel	A-1-b(0)	
DH-18	20-20.5	Q1d	10	32	13	100	100	100	98	97	96	94	92	81	65			CL	sandy clay	A-6(6)	
DH-18	29-30	Q1d	22	63	40	100	100	100	96	95	94	93	93	91	89			CH	clay	A-7-6(39)	
DH-19	4-4.5	Q1d	9	22	6	100	100	100	97	97	96	96	94	93	87			CL-ML	silty clay	A-4(3)	
DH-19	9-9.5	Q1d	11	27	13	100	100	100	88	86	83	82	79	74	63			CL	sandy clay	A-6(5)	
DH-19	19.5	Q1d	15	22	7	100	100	100	91	88	84	77	73	67	58			CL-ML	silty clay	A-4(1)	

Laboratory Test Data for Soil Units. NP = nonplastic

Sample No.	Depth of Sample (ft)	Eng. Geo. Unit	Moisture Content (%)	Atterberg Limits		Sieve Analysis (% Passing)													USCS SOIL DESCRIPTION		AASHTO SOIL DESCRIPTION
				LL	PI	No. No. No. No. No. No. No. No. No. No. No. No. No. No.													GROUP SYMBOL	GROUP NAME	
						1/2"	3/8"	1/4"	4	20	40	60	80	100	100	140	200				
DH-20	0-1.5	Q1d	16	34	16	100	100	100	99	96	95	92	83	58	CL	sandy clay	A-6(6)				
DH-20	1.5-3	Q1d	8	32	15	100	100	100	100	100	99	99	95	85	CL	clay with sand	A-6(12)				
DH-20	3-4.5	Q1d	14	32	7	100	100	100	97	94	92	91	89	86	ML	silt	A-4(6)				
DH-20	4.5-6	Q1d	15	35	14	100	100	98	96	95	94	93	92	86	CL	clay with sand	A-6(10)				
DH-20	6-7.5	Q1d	15	34	19	100	100	98	97	96	95	94	93	87	CL	clay with sand	A-6(12)				
DH-20	7.5-9	Q1d	17	29	12	100	100	100	99	99	98	97	96	90	CL	clay with sand	A-6(8)				
DH-20	9-10.5	Q1d	18	37	17	100	100	100	99	99	98	97	96	92	CL	clay with sand	A-6(14)				
DH-20	10.5-12	Q1d	16	29	12	100	100	100	99	94	89	86	84	75	CL	sandy clay	A-6(4)				
DH-20	12-13	Q1d	11	24	NP	100	100	100	98	95	89	87	84	74	ML	silt	A-4(0)				
DH-20	13-13.5	Q1d		32	17	100	100	100	98	98	97	97	96	92	CL	clay with sand	A-6(12)				
DH-20	13.5-15	Q1d	16	41	26	100	100	100	100	99	99	98	95	85	CL	clay with sand	A-7-6(21)				
DH-20	15-16.5	Q1d	26	37	20	100	100	100	100	99	99	98	92	74	CL	clay with sand	A-6(13)				
DH-20	16.5-18	Q1d	12	31	17	100	100	97	93	92	91	91	90	87	CL	clay with sand	A-6(11)				
DH-20	18-19.5	Q1d	30	57	35	100	100	99	98	97	97	97	95	88	CH	clay	A-7-6(33)				
DH-20	19.5-21	Q1d	21	43	27	100	100	100	99	99	99	98	98	94	CL	clay	A-7-6(26)				
DH-20	21-22.5	Q1d	29	40	21	100	100	100	99	99	99	99	99	98	CL	clay	A-6(20)				
DH-20	29-30.5	Q1d	15	34	14	100	100	100	94	93	93	93	93	91	CL	clay	A-6(13)				
DH-21	4-5.5	Q3o	5	19	3	100	100	100	84	80	70	57	48	31	SM	silty sand	A-2-4(0)				
DH-21	9-10.5	Q3o	7	29	12	100	100	100	91	88	81	78	75	47	SC	clayey sand	A-6(2)				
DH-21	19-20.5	Q3o	5	22	8	100	100	100	83	78	67	61	54	26	SC	clayey sand	A-2-4(0)				
DH-21	29-30.5	Q3o	9	35	15	100	100	100	97	95	87	81	77	55	CL	sandy clay	A-6(6)				
DH-22	4-5.5	Q3o	11	25	4	100	100	100	99	99	97	94	92	86	CL-ML	silty clay	A-4(2)				
DH-22	9-10.5	Q3o	9	24	7	100	100	100	100	100	97	92	87	79	CL-ML	silty clay	A-4(2)				
DH-22	19-20.5	Q3o	5	25	6	100	100	100	98	97	86	73	65	35	SM-SC	clayey sand with silt	A-4(0)				
DH-22	29-30.5	Q3o	7	22	3	100	100	100	97	95	88	83	77	65	SM	silty sand	A-4(0)				

Laboratory Test Data for Soil Units. NP = nonplastic

Sample No.	Depth of Sample (ft)	Eng. Geo. Unit	Moisture Content (%)	Atterberg Limits		Sieve Analysis (% Passing)												USCS SOIL DESCRIPTION		ASHTO SOIL DESCRIPTION
				LL	PI	1/2"	3/8"	1/4"	4	20	40	60	80	100	140	200	GROUP SYMBOL	GROUP NAME		
																			No.	
DH-23	4-5.5	Qao	12	44	24	100	100	100	99	98	97	95	92	85	77	CL	clay with sand	A-7-6(18)		
DH-23	9-10.5	Qao	7	26	8	100	100	100	96	94	94	90	78	51	CL	sandy clay	A-4(1)			
DH-23	19-20.5	Qao	22	47	24	100	100	100	100	99	99	98	95	86	CL	clay	A-7-6(22)			
DH-23	29-30.5	Qao	18	38	19	100	100	100	99	99	98	97	94	85	CL	clay with sand	A-6(16)			
DH-24	4-5.5	Qao	9	32	12	100	100	100	100	100	99	99	96	86	CL	clay	A-6(10)			
DH-24	9-10.5	Qao	7	29	10	100	100	100	100	100	100	99	98	88	CL	clay	A-4(8)			
DH-24	19-20.5	Qao	13	24	3	100	100	93	92	92	91	89	87	72	ML	silt with sand	A-4(0)			
DH-24	29-30.5	Qao	22	29	17	100	100	100	99	98	96	95	92	83	CL	sandy clay	A-6(8)			
DH-25	4-5.5	Qao	4	NP	NP	96	94	89	83	65	52	37	29	24	17	11	SP-SM	sand with silt and gravel	A-2-4(0)	
DH-25	9-9.5	Qao	4			100	100	85	82	66	58	54	53	51	47	35	SM-SC	clayey sand with silt		
DH-25	9.5-10.5	Qao	9	31	13	100	100	100	99	93	90	88	86	84	77	62	CL	sandy clay	A-6(6)	
DH-25	19-19.5	Qao	8	31	8	100	100	100	95	85	80	75	72	69	61	52	CL	sandy clay	A-4(2)	
DH-25	20-20.5	Qao	8	28	9	100	100	100	98	95	92	89	85	74	57	CL	sandy clay	A-4(3)		
DH-25	29-30.5	Qao	7	28	12	51	51	51	51	50	49	47	47	39	31	9C	clayey sand with gravel	A-6(0)		
DH-26	4-5.5	QTa	8	30	11	100	100	100	100	100	99	98	96	84	62	CL	sandy clay	A-6(5)		
DH-26	9-9.5	QTa	4	23	4	100	100	100	96	85	81	78	77	75	65	50	CL-ML	silty clay with sand	A-4(0)	
DH-26	9.5-10.5	QTa	10	41	23	100	100	100	100	100	99	99	98	92	80	CL	clay with sand	A-7-6(18)		
DH-26	19-20.5	QTa	16	46	23	100	100	100	100	100	100	100	99	96	88	CL	clay	A-7-6(22)		
DH-26	29-30.5	QTa	21	31	12	100	100	100	99	98	97	96	95	87	72	CL	clay with sand	A-6(7)		
DH-30	5-6.5	Qay	14																	
DH-30	10-11.5	Qay	14																	
DH-30	15-16.5	Qay	7			100	100	95	92	72	63	56	51	48	42	34	SM*	silty sand		
DH-30	20-21.5	Qay	18																	
DH-30	25-26.5	Qay	8			100	97	87	84	63	58	54	49	46	39	29	SM*	silty sand		
DH-30	28-29.5	Qay	12																	

\*USCS designation based on estimated plasticity.

Laboratory Test Data for Soil Units. NP = nonplastic

Sample No.	Depth of Sample (ft)	Eng. Geo. Unit	Moisture Content (%)	Atterberg Limits		Sieve Analysis (% Passing)													USCS SOIL DESCRIPTION		AASHTO SOIL DESCRIPTION
				LL	PI	No.													GROUP SYMBOL	GROUP NAME	
						1/2"	3/8"	1/4"	4	20	40	60	80	100	140	200	No.	No.			
DH-31	2-3.5	QTa	7			95	88	72	64	43	37	33	30	28	25	20	SM*	silty sand			
DH-31	7-8.5	QTa	11			100	100	100	99	95	90	76	69	63	45	34	SM-SC*	clayey sand with silt			
DH-31	12-13.5	QTa	10			100	79	73	70	37	32	27	24	21	6		SM*	silty sand			
DH-31	16-19	QTa	2																		
DH-31	22.5-24	QTa	6			92	92	92	92	91	90	87	85	81	66	47	SM*	silty sand			
DH-33	5-6.5	QTa	8			100	100	100	100	100	99	98	96	91	71	46	SM*	silty sand			
DH-33	8-9.5	QTa	9			100	100	100	100	100	99	98	96	91	71	46	SM*	silty sand			
DH-33	13-14.5	QTa	7			100	100	100	100	100	99	98	96	91	71	46	SM*	silty sand			
DH-33	18-18.5	QTa	4			100	100	98	97	94	85	69	61	55	49	40	SM*	silty sand			
DH-33	23-24.5	QTa	6			100	100	100	99	93	86	75	67	63	55	43	SM*	silty sand			
DH-33	32-33.5	QTa	9			100	100	100	100	98	97	95	94	93	91	86	CL*	clay			
DH-33	37-38.5	QTa	7			100	100	100	100	98	97	95	94	93	91	86	CL*	clay			
DH-35	4.5-6	Qld	25			100	97	97	96	95	95	94	94	91	79	57	ML*	sandy silt			
DH-35	5-6.5	Qld	10			100	100	100	100	100	100	100	99	94	69	27	SP*	sand			
DH-35	8-9.5	Qld	8			100	100	100	100	96	94	93	93	93	90	84	CL*	clay with sand			
DH-35	9.5-11	Qld	13			100	100	100	100	96	94	93	93	93	90	84	CL*	clay with sand			
DH-35	15-16.5	Qld	18			100	100	100	100	96	94	93	93	93	90	84	CL*	clay with sand			
DH-35	20-21.5	Qld	32			100	100	100	100	99	99	98	97	97	95	90	CL*	clay			
DH-35	25-26.5	Qld	16			100	100	100	100	99	99	98	97	97	95	90	CL*	clay			
DH-35	28-28.5	Qld	14			100	100	100	100	99	99	98	97	97	95	90	CL*	clay			
DH-35	35-36.5	Qld	30			100	100	100	100	96	94	93	93	93	90	84	CL*	clay			
DH-35	43.5-45	Qld	17			100	100	100	100	99	99	98	97	97	95	90	CL*	clay			
DH-35	48.5-50	Qld	28			100	100	100	100	99	99	98	97	97	95	90	CL*	clay			
DH-35	53	Qld	25			100	100	100	100	99	99	97	95	93	86	69	CL*	sandy clay			
DH-35	59.5-61	Qld	22			100	100	100	100	99	99	97	95	93	86	69	CL*	sandy clay			

\*USCS designation based on estimated plasticity.



Laboratory Test Data for Soil Units. NP = nonplastic

Sample No.	Depth of Sample (ft)	Eng. Geo. Unit	Moisture Content (%)	Atterberg Limits		Sieve Analysis (% Passing)											USCS SOIL DESCRIPTION	ASHTO SOIL DESCRIPTION	
				LL	PI	No. No.													
						1/2"	3/8"	1/4"	4	20	40	60	80	100	140	200			GROUP SYMBOL
DH-36	17.5-19	Qao	7			100	100	100	100	100	99	98	97	92	76	CL*	clay with sand		
DH-36	20-21.5	Qao	6			100	100	100	92	86	79	71	64	48	SM*	silty sand			
DH-36	25.5-27	Qao	9																
DH-36	30-31.5	Qao	11			99	83	83	82	81	81	81	75	62	CL*	sandy clay			
DH-36	33-34.5	Qao	11																
DH-36	38-38.5	Qao	5			100	100	95	52	32	22	21	19	12	SP-SM*	sand with silt			
DH-36	43-44.5	Qao	30																
DH-36	53-54.5	Qao	15			100	100	100	99	99	99	98	98	92	CL*	clay			
DH-38	3-4.5	QTa	12			100	100	92	90	82	77	64	54	43	35	SM*	silty sand		
DH-38	8-9.5	QTa	4																
DH-38	13-14.5	QTa	6			100	100	100	96	95	94	82	69	62	53	SM*	silty sand		
DH-38	18-19.5	QTa	5																
DH-38	23-24.5	QTa	7			100	98	88	82	64	59	50	43	39	26	SM*	silty sand		
DH-38	28-29.5	QTa	7																
DH-38	33-34.5	QTa	9			100	98	92	89	83	81	70	57	49	38	SM*	silty sand		
DH-38	43-44.5	QTa	9																
DH-38	73-74.5	QTa	3			100	87	81	78	66	64	47	35	30	18	SC*	clayey sand		
DH-38	98-99.5	QTa	6																
DH-38	108-109.5	QTa	6			100	100	100	99	98	98	92	85	78	62	SM*	silty sand		
DH-39	3-4.5	QTa	19			100	100	100	98	97	97	96	96	96	93	CL-ML*	silty clay		
DH-39	8-8.5	QTa	8																
DH-39	13-13.5	QTa	2																

\*USCS designation based on estimated plasticity.

Laboratory Test Data for Soil Units. NP = nonplastic

Sample No.	Depth of Sample (ft)	Eng. Geo. Unit	Moisture Content (%)	Atterberg Limits		Sieve Analysis (% Passing)										USCS SOIL DESCRIPTION	AASHTO SOIL DESCRIPTION								
				LL	PI	1/2"		3/8"		1/4"		No. 40		No. 60				No. 80		No. 100		No. 140		No. 200	
						No.	No.	No.	No.	No.	No.	No.	No.	No.	No.			No.	No.	No.	No.	No.	No.	No.	No.
DH-40	5-6.5	Q1d	9			100	100	100	99	99	97	96	94	87	70								ML*	sandy silt	
DH-40	10-11.5	Q1d	11			100	100	100	100	99	97	96	95	89	78								CL-ML*	silty clay	
DH-40	14.5-15.5	Q1d	16			94	89	80	79	75	74	74	74	74	72								CL*	gravelly clay	
DH-40	18-18.5	Q1d	13			100	100	100	100	98	93	87	83	79	70								ML*	sandy silt	
DH-40	18.5-20	Q1d	10			100	100	100	100	100	100	100	99	99	95								CL-ML*	silty clay	
DH-40	23-24.5	Q1d	20			100	100	100	97	69	66	61	57	53	40								SC*	clayey sand	
DH-40	28-29.5	Q1d	14			100	100	96	94	90	88	86	85	84	80								CL*	clay with sand	
DH-40	33-34.5	Q1d	11			100	100	100	100	100	100	100	99	99	98								CL-ML*	silty clay	
DH-40	38.5	Q1d	12			100	100	100	100	99	97	95	93	91	86								ML	silt with sand	A-4(7)
DH-40	38-39.5	Q1d	21			100	100	94	91	89	77	74	72	71	68								SM	silty sand	A-4(0)
DH-40	43-44.5	Q1d	19			100	100	100	100	100	99	97	96	92	69								SM	silty sand	A-4(0)
DH-40	48-49.5	Q1d	21			100	100	100	100	99	98	97	96	91	75								CL-ML	silty clay	A-4(0)
DH-40	55-56.5	Q1d	18			100	100	100	100	100	100	100	99	99	98								SM-SC	clayey sand with silt	A-4(0)
DH-40	58-59.5	Q1d	18			100	100	100	100	99	96	92	84	74	62								CL	clay	A-6(12)
DH-40	63-64.5	Q1d	24			100	100	100	100	99	96	92	84	74	62								ML	silt with sand	A-4(7)
BH-1	4	QTa	9	35	10	100	100	100	100	99	97	95	93	91	86								SM	silt with sand	A-4(0)
BH-1	6.6	QTa	6	NP	NP	100	94	91	89	77	74	72	71	68	58								SM	silty sand	A-4(0)
BH-1	8.2	QTa	10	27	5	100	100	100	100	99	97	96	92	69	47								SM	silty sand	A-4(0)
BH-1	9.4	QTa	9	26	4	100	100	100	100	99	98	97	96	91	75								CL-ML	silty clay	A-4(0)
BH-2	1.2	QTa	19	26	6	100	100	100	100	99	96	92	84	74	62								SM-SC	clayey sand with silt	A-4(0)
BH-2	3	QTa	11	37	13	100	100	100	100	100	99	98	98	96	93								CL	clay	A-6(12)
BH-2	5.5	QTa	7	36	9	100	100	100	100	99	99	98	96	92	87								ML	silt with sand	A-4(7)
BH-2	6.9	QTa	10	30	8	100	100	100	100	100	100	89	79	70	53								SC	clayey sand	A-4(0)
BH-2	7	QTa	10	26	6	100	100	100	100	99	98	96	94	91	85								CL-ML	silty clay	A-4(2)
BH-2	11.5	QTa	7	21	NP	72	72	70	70	68	65	61	55	49	39								SM	silty sand with gravel	A-6(0)

\*USCS designation based on estimated plasticity.

Laboratory Test Data for Soil Units. NP = nonplastic

Sample No.	Depth of Sample (ft)	Eng. Geo. Unit	Moisture Content (%)	Atterberg Limits		Sieve Analysis (% Passing)											USCS SOIL DESCRIPTION	AASHTO SOIL DESCRIPTION
				LL	PI	No. No.												
						1/2"	3/8"	1/4"	4	20	40	60	80	100	140	200		
BH-3	0.4	Qld	10	20	NP	100	100	100	98	88	73	62	52	34	19	SM	silty sand	A-2-4(0)
BH-3	1	Qld	12	19	NP	100	100	100	99	96	85	78	70	55	37	SM	silty sand	A-4(0)
BH-3	3	Qld	10	19	NP	100	100	100	92	83	71	58	41	32	23	SM	silty sand	A-2-4(0)
BH-3	5	Qld	3	NP	NP	100	100	100	71	28	10	7	6	4	2	SP	sand	A-1-b(0)
BH-3	7.8	Qld	13	34	NP	100	100	100	100	99	99	98	97	91	81	CL	clay with sand	A-6(1)
BH-4	1	Qay	12	NP	NP	100	100	100	97	94	84	78	71	59	44	SM	silty sand	A-4(0)
BH-4	3	Qay	3	NP	NP	100	100	100	97	94	85	71	63	54	36	SM	silty sand	A-4(0)
BH-4	4.4	Qay	5	NP	NP	61	60	49	42	24	19	14	13	9	8	GW-GM	gravel with silt	A-1-a(0)
BH-4	4.6	Qay	6	NP	NP	100	100	100	100	100	96	86	77	53	18	SM	silty sand	A-2-4(0)
BH-4	5	Qay	12	28	4	100	100	100	90	88	85	84	82	69	51	ML	sandy silt	A-4(0)
BH-4	6.6	Qay	12	24	8	100	100	100	98	96	94	92	90	83	71	CL	clay with sand	A-4(3)
BH-5	1	Qld	21	26	6	100	100	100	96	95	94	92	89	80	69	CL-ML	silty clay with sand	A-4(2)
BH-5	4	Qld	11	21	4	100	100	100	98	98	95	94	93	89	75	CL-ML	silty clay with sand	A-4(1)
BH-5	6.8	Qld	8	23	3	100	100	100	98	97	97	96	96	93	87	ML	silt	A-4(1)
BH-6	0.2	Qld	12	27	8	100	100	100	99	92	84	81	79	75	65	CL	sandy clay	A-4(3)
BH-6	1.8	Qld	21	NP	NP	100	100	100	99	98	96	94	87	43	19	SM	silty sand	A-2-4(0)
BH-6	3.2	Qld	4	26	6	100	100	100	100	100	100	99	96	90	71	CL-ML	silty clay with sand	A-4(2)
BH-6	4.6	Qld	5	24	0	100	100	100	95	90	87	84	82	74	48	SM	silty sand	A-4(0)
BH-6	5.4	Qld	4	29	11	100	100	100	100	99	98	98	93	59	28	SC	clayey sand	A-2-6(0)
BH-6	6	Qld	4	29	11	100	100	100	100	100	100	100	100	98	88	ML*	silt	
BH-6	6.5	Qld	10	NP	NP	100	100	100	100	99	98	97	94	72	30	SM	silty sand	A-2-4(0)
BH-6	7.3	Qld	8	24	6	100	100	100	100	99	98	97	95	90	57	CL-ML	silty clay with sand	A-4(1)
BH-6	7.9	Qld	11	NP	NP	100	100	100	100	100	100	99	96	75	32	SM	silty sand	A-2-4(0)
BH-6	8.4	Qld	5	30	12	100	100	100	100	99	97	95	75	71	54	CL	sandy clay	A-6(4)
BH-6	9	Qld	15	22	NP	100	100	100	100	98	95	89	76	48	32	SM	silty sand	A-2-4(0)

\*USCS designation based on estimated plasticity.

Laboratory Test Data for Soil Units. NP = nonplastic

Sample No.	Depth of Sample (ft)	Eng. Geo. Unit	Moisture Content (%)	Atterberg Limits		Sieve Analysis (% Passing)											USCS SOIL DESCRIPTION	GROUP SYMBOL	GROUP NAME	AASHTO SOIL DESCRIPTION
				LL	PI	1/2"	3/8"	1/4"	4	20	40	60	80	100	140	200				
				No.	No.	No.	No.	No.	No.	No.	No.	No.	No.	No.	No.	No.				
BH-7	0.4	Qld	12	25	NP	100	100	100	100	96	93	91	86	44	SM	silty sand	A-4(0)			
BH-7	2.9	Qld	14	24	NP	100	100	100	100	97	93	89	77	50	ML	sandy silt	A-4(0)			
BH-7	3.9	Qld	6	26	NP	100	100	100	100	100	98	95	67	25	SM	silty sand	A-2-4(0)			
BH-7	5.2	Qld	21	21	NP	100	100	100	100	98	94	84	72	50	ML	sandy silt	A-4(0)			
BH-7	6.3	Qld	8	27	NP	100	100	100	100	98	95	92	86	85	ML	silt with sand	A-4(0)			
BH-7	6.7	Qld	17	39	16	100	100	100	99	98	97	96	96	93	CL	clay	A-4(16)			
BH-7	7.3	Qld	3	23	NP	100	100	100	100	98	94	80	62	45	SM	silty sand	A-4(0)			
BH-7	7.7	Qld	8	35	16	100	100	100	100	100	99	99	97	86	CL	clay	A-6(13)			
BH-7	8.1	Qld	8	22	NP	100	100	100	97	90	76	56	37	19	SM	silty sand	A-3(0)			
BH-8	1.2	Qao	20	27	5	100	100	100	100	97	91	85	66	65	ML	sandy silt	A-4(2)			
BH-8	4.6	Qao	16	38	18	100	100	100	100	89	78	68	48	33	SC	clayey sand	A-2-6(1)			
BH-8	6	Qao	16	33	8	100	100	100	100	91	82	74	55	54	ML	sandy silt	A-4(2)			
BH-8	8.6	Qao	12	31	8	100	100	100	100	90	80	72	56	51	ML	sandy silt	A-4(2)			
BH-9	0.6	Qao	60	29	9	67	67	67	51	49	48	47	45	41	SC	clayey sand with gravel	A-2-4(0)			
BH-9	2.4	Qao	6	33	4	74	72	66	62	39	35	30	28	22	SM	silty sand with gravel	A-1-b(0)			
BH-9	7.2	Qao	2	NP	NP	55	47	41	39	14	5	3	2	2	GP	gravel with sand	A-1-a(0)			
BH-9	9.2	Qao	9	30	4	100	98	96	95	87	73	60	49	46	SM	silty sand	A-4(0)			

Laboratory Test Data for Soil Units. NP = nonplastic

Sample No.	Depth of Sample (ft)	Eng. Moisture Content (%)	Atterberg Limits		Sieve Analysis (% Passing)								USCS SOIL DESCRIPTION		AASHTO SOIL DESCRIPTION					
			LL	PI	1/2"	3/8"	1/4"	4	20	40	60	80	100	140		200	GROUP SYMBOL	GROUP NAME		
																			GROUP SYMBOL	GROUP NAME
BH-10	0.5	QTa	12														SM**	silty sand		
BH-10	1.8	QTa	14															CL-ML**	silty clay	
BH-10	2.5	QTa	12															CL-ML**	silty clay	
BH-10	4a*	QTa	17															ML**	silt	
BH-10	4b*	QTa	13															ML**	silt	
BH-10	6.5	QTa	4															SP-SM**	silty sand	
BH-10	7	QTa	10															ML/SM**	silt/silty sand	
BH-10	7.25	QTa	4															SP**	sand	
BH-10	8.5	QTa	4															GP-GW**	gravel	
BH-10	8.5	QTa	7															CL**	clay	
BH-10	10	QTa	5															fault gouge		
BH-10	9.25-10.75	QTa	15															CL**	clay	
BH-10	10.75	QTa	4															SP**	sand	

\*BH-10 4a located at station 12; BH-10 4b located at station 37.

\*\*USCS designation assigned in the field.

APPENDIX IV

Standard penetration test (SPT) results

Drill Hole Number	Depth of Sample (ft)	Eng. Geo. Unit	USCS GROUP SYMBOL	USCS GROUP NAME	N-Value (blows/ft)	Consistency (cohesive soils) # / Compactness (cohesionless soils) *	
						Stiff / Very Stiff	Loose / Very Stiff
DH-1	2	Qay	CL-ML	silty clay	15		
DH-1	5	QTa	ML	silt with sand	8		Loose
DH-1	8	QTa	SW-SM	sand with silt	25		Medium Dense
DH-1	14	QTa	CL-ML	silty clay	20		Very Stiff
DH-2	4-5.5	QTa	SM	silty sand	6		Loose
DH-2	9-10.5	QTa	CH	clay	27		Very Stiff
DH-2	19-20.5	QTa	SM-SC	clayey sand with silt	26		Medium Dense
DH-2	29-30.5	QTa	SC	clayey sand	27		Medium Dense
DH-2	39-40.5	QTa	CL	clay	25		Very Stiff
DH-3	9-10.5	Qao	SM	silty sand	21		Medium Dense
DH-3	19-20.5	Qao	SP	sand with gravel	>50		Very Dense
DH-3	29-30.5	Qao	CL	sandy clay	30		Very Stiff / Hard
DH-3	44-45.5	Qao	SC	clayey sand	--		--
DH-4	4-5	QTa	SC	clayey sand	14		Medium Dense
DH-4	9-10	QTa	SM	silty sand	9		Loose
DH-4	19-20	QTa	SM-SC	clayey sand with silt	29		Medium Dense
DH-4	29-30	QTa	SC	clayey sand	39		Dense
DH-5	4-5	QTa	SC	clayey sand	30		Medium Dense / Dense
DH-5	9-10	QTa	SC	clayey sand	36		Dense
DH-6	4-5	QTa	SM-SC	clayey sand with silt	27		Medium Dense
DH-6	9-9.5	QTa	CL	clay with sand	>50		Hard
DH-6	19-19.5	QTa	CH	clay	>50		Hard

\* After Low and Zaccaro, 1975.

Standard Penetration Data. TSF = tons/sq ft

Drill Hole Number	Depth of Sample (ft)	Eng. Geo. Unit	USCS GROUP SYMBOL	USCS GROUP NAME	N-Value (blows/ft)	Consistency (cohesive soils) #/ Compactness ( cohesionless soils) #
DH-7	4-5	QTa	SM-SC	clayey sand with silt	41	Dense
DH-7	9-10	QTa	SM	silty sand	13	Medium Dense
DH-7	19-20	QTa	SM	silty sand	>50	Very Dense
DH-7	29-30	QTa	SM	silty sand	>50	Very Dense
DH-8	4-5	QTa	SM-SC	clayey sand with silt	25	Medium Dense
DH-8	9-10	QTa	CL	sandy clay	23	Very Stiff
DH-8	19-20	QTa	SM	silty sand	28	Medium Dense
DH-8	29-30	QTa	SP-SM	sand with silt	34	Dense
DH-9	4-5	Q1d	CL-ML	silty clay	12	Stiff
DH-9	9-10	Q1d	SM-SC	clayey sand with silt	15	Medium Dense
DH-9	19-20	Q1d	SC	clayey sand	25	Medium Dense
DH-9	29-30	Q1d	CL	clay	45	Hard
DH-10	4-5	Q1d	SM	silty sand	21	Medium Dense
DH-10	9-10	Q1d	SM	silty sand	36	Dense
DH-10	19-20	Q1d	CL-ML	silty clay	12	Stiff
DH-10	29-30	Q1d	CL	clay	32	Hard
DH-11	4.5-5	QTa	SM	silty sand	22	Medium Dense
DH-11	9-9.5	QTa	ML	sandy silt	>50	Very Dense
DH-11	19-20.5	QTa	CL-ML	silty clay	59	Hard
DH-11	29-29.5	QTa	CL	sandy clay	>50	Hard
DH-12	4-5.5	QTa	SM-SC	clayey sand with silt	33	Dense
DH-12	9-10.5	QTa	SM-SC	clayey sand with silt	19	Medium Dense
DH-12	19-20.5	QTa	SC	clayey sand	20	Medium Dense

\* After Low and Zacco, 1975.



Standard Penetration Data. TSF = tons./sq ft

Drill Hole Number	Depth of Sample (ft)	Eng. Unit	USCS GROUP SYMBOL	USCS GROUP NAME	N-Value (blows/ft)	Consistency (cohesive soils) * / Compactness (cohesionless soils) *
DH-13	4-5.5	QTa	SM-SC	clayey sand with silt	22	Medium Dense
DH-13	9-10.5	QTa	CL	clay	22	Very Stiff
DH-13	19-20.5	QTa	CL-ML	stiffy clay	17	Very Stiff
DH-14	4-5.5	QTa	CL	sandy clay	23	Very Stiff
DH-14	9-10.5	QTa	CL	clay	33	Hard
DH-14	19-19.5	QTa	CL	clay with sand	>50	Hard
DH-15	4-5.5	Qao	CH	sandy clay	24	Very Stiff
DH-15	9-9.5	Qao	CH	clay	>50	Hard
DH-15	19-20.25	Qao	CH	clay	48	Hard
DH-15	29-29.5	--	--	--	>50	Hard
DH-16	4-4.5	Q1d	CL	clay	>50	Hard
DH-16	9-9.83	Q1d	CL	clay	>50	Hard
DH-17	4-4.5	Q1d	CL-ML	silty clay	56	Very Dense
DH-17	4.5-5.5	Q1d	ML	silt with sand	56	Very Stiff
DH-17	9-10.5	Q1d	CL	clay	24	--
DH-17	19-19.25	Q1d	--	--	22	--
DH-17	19.25-20.5	Q1d	CH	clay	22	Very Stiff
DH-17	29-30.5	Q1d	CL	sandy clay	26	Very Stiff
DH-17	40	Q1d	CL	clay	--	--

\* After Low and Zaccaro, 1975.

Standard Penetration Data. TSF = tons/sq ft

Drill Hole Number	Depth of Sample (ft)	Eng. Geo. Unit	USCS GROUP SYMBOL	USCS GROUP NAME	N-Value (blows/ft)	Consistency (cohesive soils) % / Compactness (cohesionless soils) *
DH-18	4-5.5	Q1d	SM-SC	clayey sand with silt	29	Medium Dense
DH-18	9-9.75	Q1d	CL	clay with sand	21	Very Stiff
DH-18	9.75-10.5	Q1d	CL	clay	21	Very Stiff
DH-18	19-20	Q1d	SP-SM	sand with silt and gravel	31	Dense
DH-18	20-20.5	Q1d	CL	sandy clay	31	Hard
DH-18	29-30	Q1d	CH	clay	>50	Hard
DH-19	4-4.5	Q1d	CL-ML	silty clay	>50	Hard
DH-19	9-9.5	Q1d	CL	sandy clay	>50	Hard
DH-19	19.0-19.5	Q1d	CL-ML	silty clay	>50	Hard
DH-20	0-1.5	Q1d	CL	sandy clay	9	Stiff
DH-20	1.5-3	Q1d	CL	clay with sand	14	Stiff
DH-20	3-4.5	Q1d	ML	silt	14	Medium Dense
DH-20	4.5-6	Q1d	CL	clay with sand	17	Very Stiff
DH-20	6-7.5	Q1d	CL	clay with sand	18	Very Stiff
DH-20	7.5-9	Q1d	CL	clay with sand	15	Stiff/Very Stiff
DH-20	9-10.5	Q1d	CL	clay with sand	18	Very Stiff
DH-20	10.5-12	Q1d	CL	sandy clay	15	Stiff/Very Stiff
DH-20	12-13	Q1d	ML	silt	13	Medium Dense
DH-20	13-13.5	Q1d	CL	clay with sand	13	Stiff
DH-20	13.5-15	Q1d	CL	clay with sand	19	Very Stiff
DH-20	15-16.5	Q1d	CL	clay with sand	21	Very Stiff
DH-20	16.5-18	Q1d	CL	clay with sand	32	Hard
DH-20	18-19.5	Q1d	CH	clay	14	Stiff
DH-20	19.5-21	Q1d	CL	clay	17	Very Stiff
DH-20	21-22.5	Q1d	CL	clay	14	Stiff
DH-20	29-30.5	Q1d	CL	clay	40	Hard

\* After Low and Zaccheo, 1975.

Standard Penetration Data. TSF = tons/sq ft

Drill Hole Number	Depth of Sample (ft)	Eng. Geo. Unit	USCS GROUP SYMBOL	USCS GROUP NAME	N-Value (blows/ft)	Consistency (cohesive soils) #/ Compactness (cohesionless soils) *
DH-21	4-5.5	Qao	SM	silty sand	18	Medium Dense
DH-21	9-10.5	Qao	SC	clayey sand	25	Medium Dense
DH-21	19-20.5	Qao	SC	clayey sand	26	Medium Dense
DH-21	29-30.5	Qao	CL	sandy clay	33	Hard
DH-22	4-5.5	Qao	CL-ML	silty clay	14	Stiff
DH-22	9-10.5	Qao	CL-ML	silty clay	24	Very Stiff
DH-22	19-20.5	Qao	SM-SC	clayey sand with silt	31	Dense
DH-22	29-30.5	Qao	SM	silty sand	29	Medium Dense
DH-23	4-5.5	Qao	CL	clay with sand	25	Very Stiff
DH-23	9-10.5	Qao	CL	sandy clay	23	Very Stiff
DH-23	19-20.5	Qao	CL	clay	32	Hard
DH-23	29-30.5	Qao	CL	clay with sand	29	Very Stiff
DH-24	4-5.5	Qao	CL	clay	23	Very Stiff
DH-24	9-10.5	Qao	CL	clay	20	Very Stiff
DH-24	19-20.5	Qao	ML	silt with sand	23	Medium Dense
DH-24	29-30.5	Qao	CL	sandy clay	7	Medium Stiff
DH-25	4-5.5	Qao	SP-SM	sand with silt and gravel	7	Loose
DH-25	9-9.5	Qao	SM-SC	clayey sand with silt	29	--
DH-25	9.5-10.5	Qao	CL	sandy clay	29	Very Stiff
DH-25	19-19.5	Qao	CL	sandy clay	>50	Hard
DH-25	20-20.5	Qao	CL	sandy clay	>50	Hard
DH-25	29-30.5	Qao	SC	clayey sand with gravel	28	Medium Dense

\* After Low and Zaccoho, 1975.

Standard Penetration Data. TSF = tons/sq ft

Drill Hole Number	Depth of Sample (ft)	Eng. Geo. Unit	USCS GROUP SYMBOL	USCS GROUP NAME	N-Value (blows/ft)	Consistency (cohesive soils) * / Compactness (cohesionless soils) *
DH-26	4-5.5	QTa	CL	sandy clay	14	Stiff
DH-26	9-9.5	QTa	CL-ML	silty clay with sand	20	Very Stiff
DH-26	9.5-10.5	QTa	CL	clay with sand	20	
DH-26	19-20.5	QTa	CL	clay	39	Hard
DH-26	29-30.5	QTa	CL	clay with sand	19	Very Stiff
DH-30	5-8	Qay	SC	clayey sand	15	Medium Dense
DH-30	10-13	Qay	SM	silty sand	14	Medium Dense
DH-30	15-18	Qay	SM	silty sand	22	Medium Dense
DH-30	20-23	Qay	SM-SC	clayey sand with silt	17	Medium Dense
DH-30	25-28	Qay	SM	silty sand	34	Dense
DH-31	0-2	QTa			>50	
DH-31	2-7	QTa	SM	silty sand with gravel	33	Very Dense
DH-31	7-12	QTa	SM	silty sand	>50	Very Dense
DH-31	12-17	QTa	SM-SC	clayey sand with silt	>50	Very Dense
DH-31	17-22	QTa	SP	sand with gravel	>50	Very Dense
DH-31	22-28	QTa	SM	silty sand	>50	Very Dense
DH-31	28-34	QTa	SM	silty sand	>50	Very Dense
DH-31	34-40	QTa	SM/ML	silty sand/silt	>50	Very Dense
DH-31	40-45	QTa	SM	silty sand	>50	Very Dense
DH-31	45-51	QTa	SM	silty sand	>50	Very Dense
DH-31	51-55	QTa	SP	sand with gravel	>50	Very Dense
DH-31	55-61	QTa	SM	silty sand with gravel	>50	Very Dense

\* After Low and Zaccaro, 1975.

Standard Penetration Data. TSF = tons/sq ft

Drift Hole Number	Depth of Sample (ft)	Eng. Unit	USCS GROUP SYMBOL	USCS GROUP NAME	N-Value (blows/ft)	Consistency (cohesive soils)*/ Compactness (cohesionless soils)*
DH-33	5-8	QTa	ML	sandy silt with gravel	23	Medium Dense
DH-33	8-13	QTa	ML	sandy silt	48	Dense
DH-33	13-18	QTa	SM	silty sand	26	Medium Dense
DH-33	18-23	QTa	SM	silty sand with gravel	>50	Very Dense
DH-33	23-28	QTa	SM/ML	silty sand	32	Dense
DH-33	28-32	QTa			>50	
DH-33	32-37	QTa	SM/ML	silty sand	>50	Very Dense
DH-33	37-38.5	QTa	ML	sandy silt with gravel	>50	Very Dense
DH-35	4.75-8	Q1d	CL	clay	12	stiff
DH-35	9.75-13	Q1d	SM	silty sand	15	Medium Dense
DH-35	15-18.5	Q1d	ML/SP	silt with sand at top	7	loose
DH-35	20.25-23.25	Q1d	CH	clay	5	Medium Stiff
DH-35	25-28.25	Q1d	SP	sand	30	Medium Dense/Dense
DH-35	35-38	Q1d	CH	clay	15	Stiff/Very Stiff
DH-35	43.5-45	Q1d	SM	silty sand	22	Medium Dense
DH-35	45-51	Q1d	CL/SM	clay	23	Very Stiff
DH-35	59.25-63	Q1d	CL	sandy clay	30	Very Stiff/Hard
DH-36	5-8	Qao	SM/CL	silty sand/lean clay	15	Stiff/Very Stiff
DH-36	9-9.5	Qao	SM	silty sand	14	Medium Dense
DH-36	17.5-19	Qao	CL/ML	lean clay / sandy silt	36	Hard
DH-36	19-22.5	Qao	SM	silty sand	>50	Hard
DH-36	22.5-28	Qao	CL/ML	lean clay / sandy silt	31	Hard
DH-36	28-33	Qao	CL/ML	lean clay / sandy silt	>50	Hard
DH-36	33-38	Qao	SP-SM	sand with silt	>50	Very Dense
DH-36	38-43	Qao	CL-ML	silty clay	>50	Hard
DH-36	43-48	Qao	CL	clay	39	Hard
DH-36	53-55	Qao	CL	clay	>50	Hard

\* After Low and Zaccaro, 1975.

Standard Penetration Data. TSF = tons/sq ft

Drill Hole Number	Depth of Sample (ft)	Eng. Geo. Unit	USCS GROUP SYMBOL	USCS GROUP NAME	N-Value (blows/ft)	Consistency (cohesive soils)*/ Compactness (cohesionless soils) #	
DH-38	3-8	QTa	SM	silty sand	9		Loose
DH-38	8-13	QTa	ML	sandy silt with gravel	38		Dense
DH-38	13-18	QTa	SM	silty sand	46		Dense
DH-38	18-23	QTa	SP	sand with gravel	>50		Very Dense
DH-38	23-28	QTa	SM	silty sand	>50		Very Dense
DH-38	28-33	QTa	SP-SM	sand with silt	>50		Very Dense
DH-38	33-38	QTa	ML/CL	sandy silt/silty clay	>50		Hard
DH-38	38-43	QTa			>50		
DH-38	43-73	QTa	SP	sand	>50		Very Dense
DH-38	73-98	QTa	SP	sand with gravel	>50		Very Dense
DH-38	98-108	QTa	SM	silty sand	44		Dense
DH-38	108-109.5	QTa	SM/SP	silty sand to sand	>50		Very Dense
DH-39	3-8	QTa	ML/CL	sandy silt/silty clay	11		Stiff
DH-39	8-13	QTa	CL-ML	silty clay	>50		Hard

\* After Low and Zaccheo, 1975.

Standard Penetration Data. TSF = tons/sq ft

Drill Hole Number	Depth of Sample (ft)	Eng. Geo. Unit	USCS GROUP SYMBOL	USCS GROUP NAME	N-Value (blows/ft)	Consistency (cohesive soils) * / Compactness (cohesionless soils) *
DH-40	5-8	Q1d	ML	sandy silt	9	Loose
DH-40	10-13	Q1d	ML/SM	sandy silt/silty sand	28	Medium Dense
DH-40	14.25-18	Q1d	CL-ML	silty clay with sand	18	Very Stiff
DH-40	18.5-23	Q1d	ML/CL	sandy silt/silty clay	28	Medium Dense/Very Stiff
DH-40	23-28	Q1d	CL	clay	22	Very Stiff
DH-40	28-33	Q1d	SP-SM	sand with silt	41	Dense
DH-40	33-38	Q1d	SP	sand	41	Dense
DH-40	38-43	Q1d	CH	clay	29	Very Stiff
DH-40	43-48	Q1d	CL-ML	silty clay	23	Very Stiff
DH-40	48-53	Q1d	SP/ML	sand	22	Medium Dense
DH-40	55-58	Q1d	SM-SC	clayey sand with silt	26	Medium Dense
DH-40	58-63	Q1d	SM/ML	silty sand/sandy silt	27	Medium Dense
DH-40	63-68	Q1d	ML/CL	sandy silt/clay	31	Dense/Hard
DH-40	68-69.5	Q1d	GW	gravel with sand	>50	Very Dense

\* After Low and Zaecheo, 1975.

APPENDIX V

Laboratory test data for rock



Drill Hole Number	Depth of Sample (ft)	Geologic Unit	Lithology	Diametral PLSI (MPa)*	Drill Hole Number	Depth of Sample (ft)	Geologic Unit	Lithology	Diametral PLSI (MPa)*
DH-27	6	Pg	sandstone	5.66	DH-27	92	Pg	sandstone	4.66
DH-27	6	Pg	sandstone	5.95	DH-27	92	Pg	sandstone	3.50
DH-27	6	Pg	sandstone	2.99	DH-27	92	Pg	sandstone	4.31
DH-27	6	Pg	sandstone	4.74	DH-27	101	Pg	sandstone	4.13
DH-27	6	Pg	sandstone	4.91	DH-27	101	Pg	sandstone	2.81
DH-27	13.5	Pg	sandstone	4.08	DH-27	112	Pg	sandstone	4.25
DH-27	13.5	Pg	sandstone	4.08	DH-27	112	Pg	sandstone	4.34
DH-27	13.5	Pg	sandstone	3.58	DH-27	112	Pg	sandstone	4.51
DH-27	19	Pg	sandstone	5.90	DH-27	116	Pg	sandstone	4.00
DH-27	32	Pg	sandstone	2.69	DH-27	116	Pg	sandstone	3.33
DH-27	32	Pg	sandstone	3.06	DH-27	124	Pg	sandstone	2.50
DH-27	44	Pg	sandstone	4.42	DH-27	124	Pg	sandstone	4.58
DH-27	47	Pg	sandstone	4.42	DH-27	124	Pg	sandstone	4.41
DH-27	47	Pg	sandstone	5.02	DH-27	128	Pg	sandstone	3.71
DH-27	47	Pg	sandstone	4.59	DH-27	128	Pg	sandstone	3.23
DH-27	47	Pg	sandstone	4.25	DH-27	135	Pg	sandstone	3.51
DH-27	55	Pg	sandstone	4.23	DH-27	145	Pg	sandstone	3.79
DH-27	55	Pg	sandstone	3.05	DH-27	145	Pg	sandstone	4.05
DH-27	55	Pg	sandstone	3.89	DH-27	145	Pg	sandstone	4.23
DH-27	62	Pg	sandstone	3.74	DH-29	41.5	Pg	sandstone	2.31
DH-27	62	Pg	sandstone	4.42	DH-29	45	Pg	sandstone	4.51
DH-27	69	Pg	sandstone	5.66	DH-29	45	Pg	sandstone	3.07
DH-27	79	Pg	sandstone	4.25	DH-29	121.5	Pg	sandstone	2.55
DH-27	79	Pg	sandstone	3.59	DH-29	122	Pg	sandstone	2.63
DH-27	79	Pg	sandstone	4.57	DH-29	133	Pg	sandstone	4.79
DH-27	85	Pg	sandstone	6.29	DH-29	173	Pg	sandstone	5.42
DH-27	85	Pg	sandstone	5.12					
DH-27	89	Pg	sandstone	7.28	DH-32	44.5	Km	shale	0.49
					DH-32	46.5	Km	shale	0.42
					DH-32	50.5	Km	shale	0.49

\*1MPa = 10.44 tons/sq ft

Number	Sample (ft)	Geologic Unit	Lithology	PLSI (MPa#)	Diameter (in)	Geologic Unit	Lithology	PLSI (MPa#)	Diameter (in)
DH-39	14.5	Pm	limestone	4.00					
DH-39	17	Pm	limestone	2.39					6.33
DH-39	22.5	Pm	limestone	3.83					5.93
DH-39	22.5	Pm	limestone	3.33					3.79
DH-39	32.5	Pm	limestone	6.60					5.76
DH-39	32.5	Pm	limestone	4.38					5.10
DH-39	39.5	Pm	limestone	7.83					5.12
DH-39	39.5	Pm	limestone	5.16					5.54
DH-39	39.5	Pm	limestone	5.33					4.84
DH-39	54	Pm	limestone	4.94					4.50
DH-39	54	Pm	limestone	3.90					
DH-39	54	Pm	limestone	4.78					1.64
DH-39	59	Pm	limestone	6.11					0.74
DH-39	59	Pm	limestone	3.16					0.93
DH-39	59.5	Pm	limestone	4.94					1.24
DH-39	59.5	Pm	limestone	3.50					1.24
DH-39	59.5	Pm	limestone	4.17					0.82
DH-39	62	Pm	limestone	6.25					1.81
DH-39	62	Pm	limestone	4.78					2.51
DH-39	66	Pm	limestone	5.76					1.97
DH-39	66	Pm	limestone	4.45					
DH-39	66	Pm	limestone	2.61					0.73
DH-39	72.5	Pm	limestone	4.80					0.72
DH-39	72.5	Pm	limestone	4.14					0.69
DH-39	79.5	Pm	limestone	6.58					0.98
DH-39	85.5	Pm	limestone	5.61					0.97
DH-39	85.5	Pm	limestone	3.72					1.08
DH-39	85.5	Pm	limestone	4.71					0.89
DH-29	281.5	Py	sandstone						1.97
DH-29	289	Py	sandstone						0.73
DH-29	354	Py	sandstone						0.72
DH-37	120	Py	sandstone						0.69
DH-37	120	Py	sandstone						0.98
DH-37	120	Py	sandstone						0.97
DH-37	127.5	Py	sandstone						1.08
DH-37	127.5	Py	sandstone						0.89
DH-37	127.5	Py	sandstone						0.92
DH-30	59.5	Trd	mudstone						0.73
DH-30	59.5	Trd	mudstone						0.72
DH-30	66.5	Trd	sandstone						0.69
DH-30	69	Trd	mudstone						0.98
DH-30	69	Trd	mudstone						0.97
DH-30	72	Trd	mudstone						1.08
DH-30	72	Trd	mudstone						0.89
DH-30	74	Trd	mudstone						0.92

\*1MPa = 10.44 tons/sq ft

Drill Hole Number	Depth of Sample (ft)	Geologic Unit	Lithology	Diametral	
				PLS (MPa)*	
DH-30	74	Trd	mudstone	0.75	
DH-30	74.5	Trd	mudstone	0.71	
DH-30	78	Trd	mudstone	1.07	
DH-30	78	Trd	mudstone	0.71	
DH-30	86.5	Trd	mudstone	1.04	
DH-30	86.5	Trd	mudstone	1.03	
DH-30	88.7	Trd	mudstone	0.83	
DH-34	186	Trd	sandstone	0.52	
DH-34	186	Trd	sandstone	0.52	
DH-34	193.5	Trd	sandstone	0.99	
DH-34	193.5	Trd	sandstone	0.79	
DH-34	193.5	Trd	sandstone	0.48	

\*1MPa = 10.44 tons/sq ft

Drill Hole Number	Depth of Sample (ft)	Geologic Unit	Lithology	Axial PLSI (MPa*)	Drill Hole Number	Depth of Sample (ft)	Geologic Unit	Lithology	Axial PLSI (MPa*)
DH-27	13.5	Pg	sandstone	3.55	DH-30	59.5	Trd	mudstone	1.70
DH-27	13.5	Pg	sandstone	4.51	DH-30	59.5	Trd	mudstone	2.23
DH-27	13.5	Pg	sandstone	4.35	DH-30	66.5	Trd	sandstone	1.32
DH-27	47	Pg	sandstone	4.15	DH-30	66.5	Trd	sandstone	1.36
DH-27	47	Pg	sandstone	4.86	DH-30	69	Trd	mudstone	1.30
DH-27	47	Pg	sandstone	4.24	DH-30	69	Trd	mudstone	2.19
DH-27	79	Pg	sandstone	3.90	DH-30	74	Trd	mudstone	2.05
DH-27	79	Pg	sandstone	4.70	DH-30	74	Trd	mudstone	1.61
DH-27	79	Pg	sandstone	3.72	DH-30	74	Trd	mudstone	1.55
DH-27	112	Pg	sandstone	4.02	DH-30	86.5	Trd	mudstone	2.87
DH-27	112	Pg	sandstone	4.75	DH-30	86.5	Trd	mudstone	1.24
DH-27	112	Pg	sandstone	4.88	DH-30	88.7	Trd	mudstone	1.55
DH-27	124	Pg	sandstone	5.89	DH-30	88.7	Trd	mudstone	1.30
DH-27	124	Pg	sandstone	5.38	DH-34	186	Trd	sandstone	3.60
DH-27	124	Pg	sandstone	5.38	DH-34	186	Trd	sandstone	5.69
DH-29	41.5	Pg	sandstone	2.92	DH-34	193.5	Trd	mudstone	1.14
DH-29	41.5	Pg	sandstone	3.81	DH-34	193.5	Trd	mudstone	1.68
DH-29	45	Pg	sandstone	2.88					
DH-29	45	Pg	sandstone	3.23	DH-32	146.5	Km	shale	1.38
DH-29	45	Pg	sandstone	2.97	DH-32	146.5	Km	shale	1.23
DH-29	133	Pg	sandstone	5.29	DH-32	150.5	Km	shale	0.81
DH-29	133	Pg	sandstone	4.62					
DH-29	173	Pg	sandstone	4.50	DH-29	281.5	Py	sandstone	2.11
DH-29	173	Pg	sandstone	5.71	DH-29	354	Py	sandstone	1.36
					DH-37	120	Py	sandstone	2.92
					DH-37	120	Py	sandstone	2.84
					DH-37	127.5	Py	sandstone	3.35
					DH-37	127.5	Py	sandstone	3.38

\*MPa = 10.44 tons/sq ft.

Drill Hole Number	Depth of Sample (ft)	Geologic Unit	Lithology	Axial PLSI (MPa*)
DH-39	32.5	Pm	limestone	3.49
DH-39	39.5	Pm	limestone	4.60
DH-39	39.5	Pm	limestone	4.43
DH-39	39.5	Pm	limestone	5.22
DH-39	54	Pm	limestone	5.60
DH-39	62	Pm	limestone	5.06
DH-39	62	Pm	limestone	4.91
DH-39	85.5	Pm	limestone	6.59
DH-39	85.5	Pm	limestone	8.52
DH-39	85.5	Pm	limestone	7.22

\*MPa = 10.44 tons/sq ft.

Sample Number	Block Number	Geologic Unit	Lithology/Description	Block PLSI (MP <sub>a</sub> )*	PLSI Test Type**	Dry Density (PCF)
R-3	1	Psa/Pg	tan sandstone, mod.-poorly indurated	2.89	BP	142
R-3	2	Psa/Pg	tan sandstone, mod.-poorly indurated	4.38	BP	
R-3	3	Psa/Pg	tan sandstone, mod.-poorly indurated	4.42	BP	
R-6	1	Trd	tan sandstone, mod. indurated, hematite stains	2.67	BP	132
R-6	2	Trd	tan sandstone, mod. indurated, hematite stains	3.73	BP	
R-6	3	Trd	tan sandstone, mod. indurated, hematite stains	4.27	BP	
R-6	4	Trd	tan sandstone, mod. indurated, hematite stains	4.29	BP	
R-6	5	Trd	tan sandstone, mod. indurated, hematite stains	2.42	BP	
R-6	6	Trd	tan sandstone, mod. indurated, hematite stains	2.47	BP	
R-7	1	Pg	purpleish-grey sandstone, hematite stains	9.76	BP	150
R-7	2	Pg	purpleish-grey sandstone, hematite stains	9.24	BP	
R-7	3	Pg	purpleish-grey sandstone, hematite stains	8.95	BP	
R-7	4	Pg	purpleish-grey sandstone, hematite stains	8.96	BP	
R-7	5	Pg	purpleish-grey sandstone, hematite stains	6.43	BP	
R-7	6	Pg	purpleish-grey sandstone, hematite stains	6.97	BP	
R-8	1	Pm	greyish brown limestone, v. well indurated	8.90	BP	168
R-8	2	Pm	greyish brown limestone, v. well indurated	8.13	BP	
R-8	3	Pm	greyish brown limestone, v. well indurated	6.75	BP	
R-8	4	Pm	greyish brown limestone, v. well indurated	7.33	BP	
R-8	5	Pm	greyish brown limestone, v. well indurated	6.42	BP	
R-8	6	Pm	greyish brown limestone, v. well indurated	5.98	BP	

\*1MP<sub>a</sub> = 10.44 tons/sq ft

\*\*BP = Block test; samples tested perpendicular to plane of greatest anisotropy (i.e., bedding, foliation).

Sample Number	Block Number	Geologic Unit	Lithology / Description	Block PLSI (MPa)*	PLSI Test Type**	Dry Density (PCF)
R-14	1	Pg	sandstone, well indurated, hematitic	5.75	BP	NP
R-14	2	Pg	sandstone, well indurated, hematitic	4.89	BP	
R-14	3	Pg	sandstone, well indurated, hematitic	5.38	BP	
R-14	4	Pg	sandstone, well indurated, hematitic	6.67	BP	
R-14	5	Pg	sandstone, well indurated, hematitic	5.40	BP	
R-14	6	Pg	sandstone, well indurated, hematitic	7.85	BP	
R-16	1	Pm	gray limestone with iron oxide solution bands	7.02	BP	167
R-16	2	Pm	gray limestone with iron oxide solution bands	3.30	BP	
R-16	3	Pm	gray limestone with iron oxide solution bands	9.38	BP	
R-16	4	Pm	gray limestone with iron oxide solution bands	4.86	BP	
R-17	1	Tm	monzonite, very altered, hematite stains	2.31	BP	149
R-17	2	Tm	monzonite, very altered, hematite stains	6.05	BP	
R-17	3	Tm	monzonite, very altered, hematite stains	7.71	BP	
R-17	4	Tm	monzonite, very altered, hematite stains	7.56	BP	
R-18	1	Tm	monzonite, altered	5.21	BP	164
R-18	2	Tm	monzonite, altered	7.14	BP	
R-19	1	Pa	light brown sandstone, v. well indurated	10.98	BP	154
R-19	2	Pa	light brown sandstone, v. well indurated	11.40	BP	
R-19	3	Pa	light brown sandstone, v. well indurated	10.64	BP	
R-19	4	Pa	light brown sandstone, v. well indurated	9.59	BP	
R-19	5	Pa	light brown sandstone, v. well indurated	10.42	BP	

\*1MPa = 10.44 tons/sq ft

\*\*BP = Block test; samples tested perpendicular to plane of greatest anisotropy (i.e., bedding, foliation).

Sample Number	Block Number	Geologic Unit	Lithology / Description	Block PLSI (MPa)*	PLSI Test Type**	Dry Density (PCF)
R-20	1	Tm	monzonite, fresh	17.05	BP	153
R-20	2	Tm	monzonite, fresh	11.53	BP	
R-20	3	Tm	monzonite, fresh	17.89	BP	
R-20	4	Tm	monzonite, fresh	17.08	BP	
R-20	5	Tm	monzonite, fresh	18.25	BP	
R-21	1	Pm	brown limestone, highly fractured, v. well indurated	6.38	BP	168
R-21	2	Pm	brown limestone, highly fractured, v. well indurated	4.67	BP	
R-21	3	Pm	brown limestone, highly fractured, v. well indurated	6.25	BP	
R-21	4	Pm	brown limestone, highly fractured, v. well indurated	7.58	BP	
R-21	5	Pm	brown limestone, highly fractured, v. well indurated	5.29	BP	
R-22	1	Pm	grayish-brown limestone	4.89	BP	NP
R-22	2	Pm	grayish-brown limestone	4.89	BP	
R-22	3	Pm	grayish-brown limestone	4.00	BP	
R-22	4	Pm	grayish-brown limestone	4.61	BP	
R-22	5	Pm	grayish-brown limestone	6.74	BP	
R-23	1	Tm	monzonite, slightly altered, hematite stains	7.73	BP	149
R-23	2	Tm	monzonite, slightly altered, hematite stains	2.85	BP	
R-23	3	Tm	monzonite, slightly altered, hematite stains	1.70	BP	
R-23	4	Tm	monzonite, slightly altered, hematite stains	2.18	BP	
R-24	1	Pm	tan limestone, v. well indurated	9.04	BP	154
R-24	2	Pm	tan limestone, v. well indurated	6.34	BP	
R-24	3	Pm	tan limestone, v. well indurated	7.03	BP	
R-24	4	Pm	tan limestone, v. well indurated	7.45	BP	

\*1MPa = 10.44 tons/sq ft

\*\*BP = Block test, samples tested perpendicular to plane of greatest anisotropy (i.e., bedding, foliation).



Sample Number	Block Number	Geologic Unit	Lithology/Description	Block PLSI (MPa)*	PLSI Test Type**	Dry Density (pcf)
R-25	1	Pb	pinkish-grey quartz sandstone, fractured, v. well indurated	5.85	BP	172
R-26	2	Pb	pinkish-grey quartz sandstone, fractured, v. well indurated	10.99	BP	
R-26	3	Pb	pinkish-grey quartz sandstone, fractured, v. well indurated	11.17	BP	
R-26	4	Pb	pinkish-grey quartz sandstone, fractured, v. well indurated	12.14	BP	
R-27	1	Pg	pink sandstone, cracks parallel to bedding, well indurated	4.28	BP	NP
R-27	2	Pg	pink sandstone, cracks parallel to bedding, well indurated	6.44	BP	
R-27	3	Pg	pink sandstone, cracks parallel to bedding, well indurated	7.38	BP	
R-27	4	Pg	pink sandstone, cracks parallel to bedding, well indurated	4.84	BP	
R-27	5	Pg	pink sandstone, cracks parallel to bedding, well indurated	3.07	BP	
R-28	1	Tr-d	sandstone, well indurated	3.98	BP	156
R-28	2	Tr-d	sandstone, well indurated	5.43	BP	
R-28	3	Tr-d	sandstone, well indurated	5.52	BP	
R-28	4	Tr-d	sandstone, well indurated	5.26	BP	
R-29	1	Pb-Psa	white sandstone, poorly indurated	2.71	BP	135
R-29	2	Pb-Psa	white sandstone, poorly indurated	1.15	BP	
R-29	3	Pb-Psa	white sandstone, poorly indurated	1.98	BP	
R-30	1	Pb	gray limestone, well indurated	12.22	BP	163
R-30	2	Pb	gray limestone, well indurated	11.23	BP	
R-30	3	Pb	gray limestone, well indurated	13.66	BP	
R-30	4	Pb	gray limestone, well indurated	15.54	BP	
R-30	5	Pb	gray limestone, well indurated	9.61	BP	
R-30	6	Pb	gray limestone, well indurated	12.36	BP	
R-32	1	pCs	schist, well-developed foliation	6.42	BP	161
R-32	2	pCs	schist, well-developed foliation	5.67	BP	
R-32	3	pCs	schist, well-developed foliation	5.82	BP	

\*1MPa = 10.44 tons/sq ft

\*\*BP = Block test; samples tested perpendicular to plane of greatest anisotropy (i.e., bedding, foliation).

Sample Number	Block Number	Geologic Unit	Lithology /Description	Block PLSI (MPa)*	PLSI Test Type**	Dry Density (PCF)
R-33	1	pCq	purpleish-gray metaquartzite, well indurated	12.89	BP	165
R-33	2	pCq	purpleish-gray metaquartzite, well indurated	13.40	BP	
R-33	3	pCq	purpleish-gray metaquartzite, well indurated	10.01	BP	
R-33	4	pCq	purpleish-gray metaquartzite, well indurated	10.31	BP	
R-33	5	pCq	purpleish-gray metaquartzite, well indurated	13.43	BP	
R-34	1	Pg	tan sandstone, moderately indurated	3.97	BP	140
R-34	2	Pg	tan sandstone, moderately indurated	3.88	BP	
R-34	3	Pg	tan sandstone, moderately indurated	3.55	BP	
R-34	4	Pg	tan sandstone, moderately indurated	4.43	BP	
R-34	5	Pg	tan sandstone, moderately indurated	3.94	BP	
R-35	1	Pg	purpleish-gray siltstone, v. well indurated	7.20	BP	148
R-35	2	Pg	purpleish-gray siltstone, v. well indurated	5.62	BP	
R-35	3	Pg	purpleish-gray siltstone, v. well indurated	5.27	BP	
R-35	4	Pg	purpleish-gray siltstone, v. well indurated	6.75	BP	
R-35	5	Pg	purpleish-gray siltstone, v. well indurated	6.69	BP	
R-36	1	QTa	pebble conglomerate, poorly indurated	1.24	BP	NP
R-36	2	QTa	pebble conglomerate, poorly indurated	1.63	BP	
R-36	3	QTa	pebble conglomerate, poorly indurated	2.85	BP	
R-36	4	QTa	pebble conglomerate, poorly indurated	3.28	BP	
R-37	1	Trd	tan sandstone, moderately indurated, hematite stains	2.75	BP	145
R-37	2	Trd	tan sandstone, moderately indurated, hematite stains	3.31	BP	
R-37	3	Trd	tan sandstone, moderately indurated, hematite stains	3.16	BP	
R-37	4	Trd	tan sandstone, moderately indurated, hematite stains	3.94	BP	
R-37	5	Trd	tan sandstone, moderately indurated, hematite stains	3.26	BP	

\*1MPa = 10.44 tons/sq ft

\*\*BP = Block test; samples tested perpendicular to plane of greatest anisotropy (i.e., bedding, foliation).

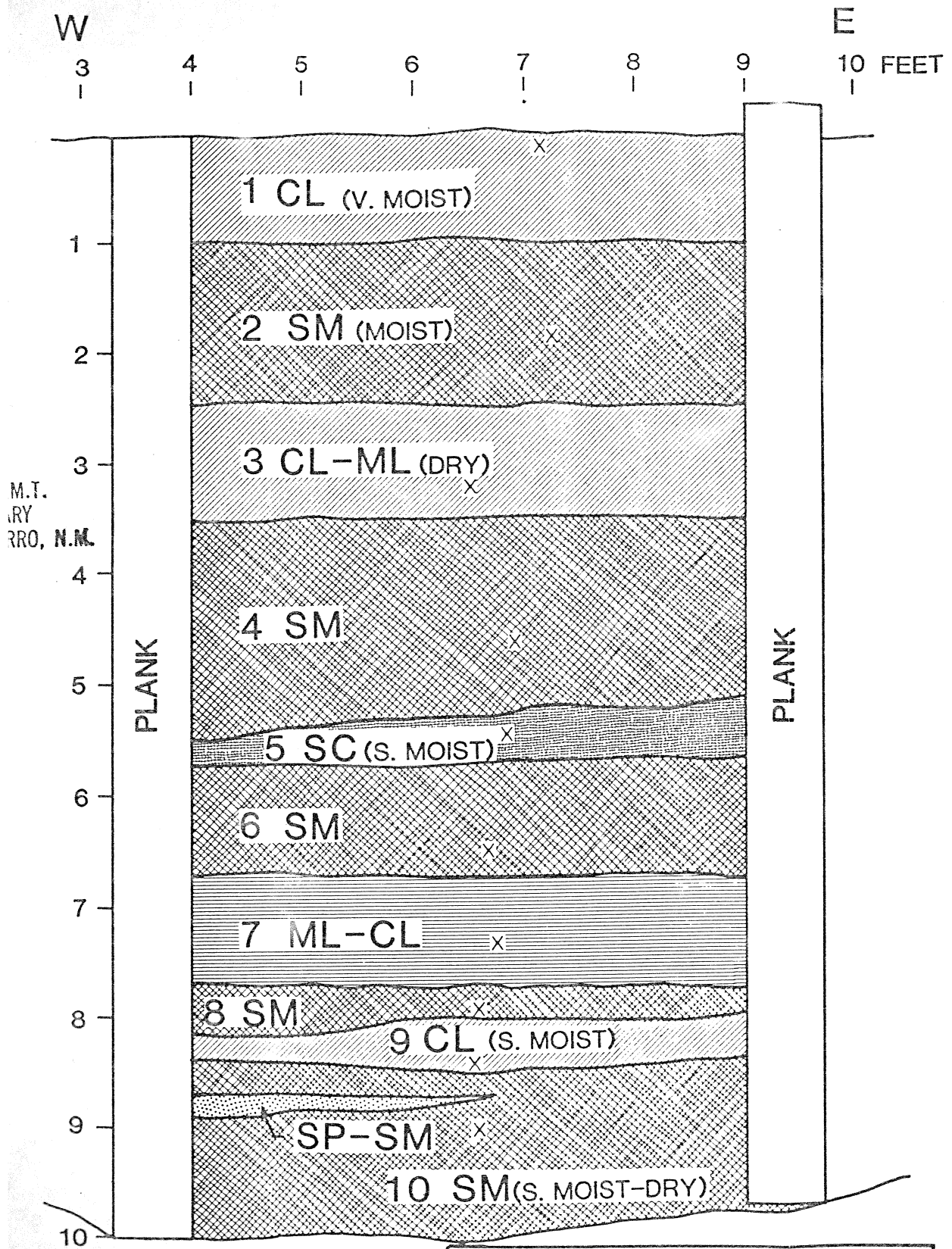
Sample Number	Block Number	Geologic Unit	Lithology / Description	Block PLSI (MPa)*	PLSI Test Type**	Dry Density (PCF)
R-38	2	Trd	grayish-brown sandstone, v. well indurated, iron oxide stains	7.54	BP	
R-38	3	Trd	grayish-brown sandstone, v. well indurated, iron oxide stains	5.50	BP	
R-38	4	Trd	grayish-brown sandstone, v. well indurated, iron oxide stains	7.62	BP	
R-38	5	Trd	grayish-brown sandstone, v. well indurated, iron oxide stains	8.72	BP	
R-38	6	Trd	grayish-brown sandstone, v. well indurated, iron oxide stains	6.20	BP	
R-39	1	Pm	gray limestone, v. well indurated	6.92	BP	171
R-39	2	Pm	gray limestone, v. well indurated	7.01	BP	
R-39	3	Pm	gray limestone, v. well indurated	7.35	BP	
R-39	4	Pm	gray limestone, v. well indurated	6.33	BP	
R-39	5	Pm	gray limestone, v. well indurated	7.01	BP	
R-39	6	Pm	gray limestone, v. well indurated	7.09	BP	
R-40	1	QTa	pebble conglomerate, poorly sorted, poorly indurated	0.90	BP	NP
R-40	2	QTa	pebble conglomerate, poorly sorted, poorly indurated	1.25	BP	
R-40	3	QTa	pebble conglomerate, poorly sorted, poorly indurated	1.90	BP	
R-40	4	QTa	pebble conglomerate, poorly sorted, poorly indurated	0.77	BP	

\*1MPa = 10.44 tons/sq ft

\*\*BP = Block test; samples tested perpendicular to plane of greatest anisotropy (i.e., bedding, foliation).

Laboratory Test Data for Rock Units. PCF = pounds per cubic foot

Drill Hole Number	Depth of Sample (ft)	Geologic Unit	Geologic Period	Lithology	Dry Density (PCF)
DH-27	6	Pg	Permian	clean quartz sandstone	148
DH-27	55	Pg	Permian	sandstone with minor limonitic seams	149
DH-27	101	Pg	Permian	clean quartz sandstone	160
DH-27	133.5	Pg	Permian	sandstone, crossbedded, very fractured	157
DH-27	142	Pg	Permian	sandstone, iron stained, fractured	151
DH-29	45	Pg	Permian	clean quartz sandstone, fractured	153
DH-29	120	Pg	Permian	sandstone, iron stained	151
DH-29	169	Pg	Permian	sandstone, weathered	156
DH-29	282	Py	Permian	silty sandstone	137
DH-29	354	Py	Permian	sandy mudstone	140
DH-30	62.3	Trd	Triassic	sandstone + mudstone	142
DH-30	74.5	Trd	Triassic	sandstone + mudstone	146
DH-30	88.5	Trd	Triassic	sandstone + mudstone	153
DH-32	144	Km	Cretaceous	sandy shale	131
DH-32	154	Km	Cretaceous	sandy shale	136
DH-34	182	Trd	Triassic	sandy siltstone	154
DH-37	124	Py	Permian	shaly fine sandstone, vuggy	135
DH-37	127.5	Py	Permian	shaly fine sandstone, vuggy	144
DH-39	25.5	Pm	Pennsylvanian	limestone, very fractured	167
DH-39	54	Pm	Pennsylvanian	limestone with shale partings	173
DH-39	78	Pm	Pennsylvanian	limestone, fractured	158
DH-39	85.5	Pm	Pennsylvanian	limestone, fractured	165
DH-39	90	Pm	Pennsylvanian	shaly limestone, fractured	176



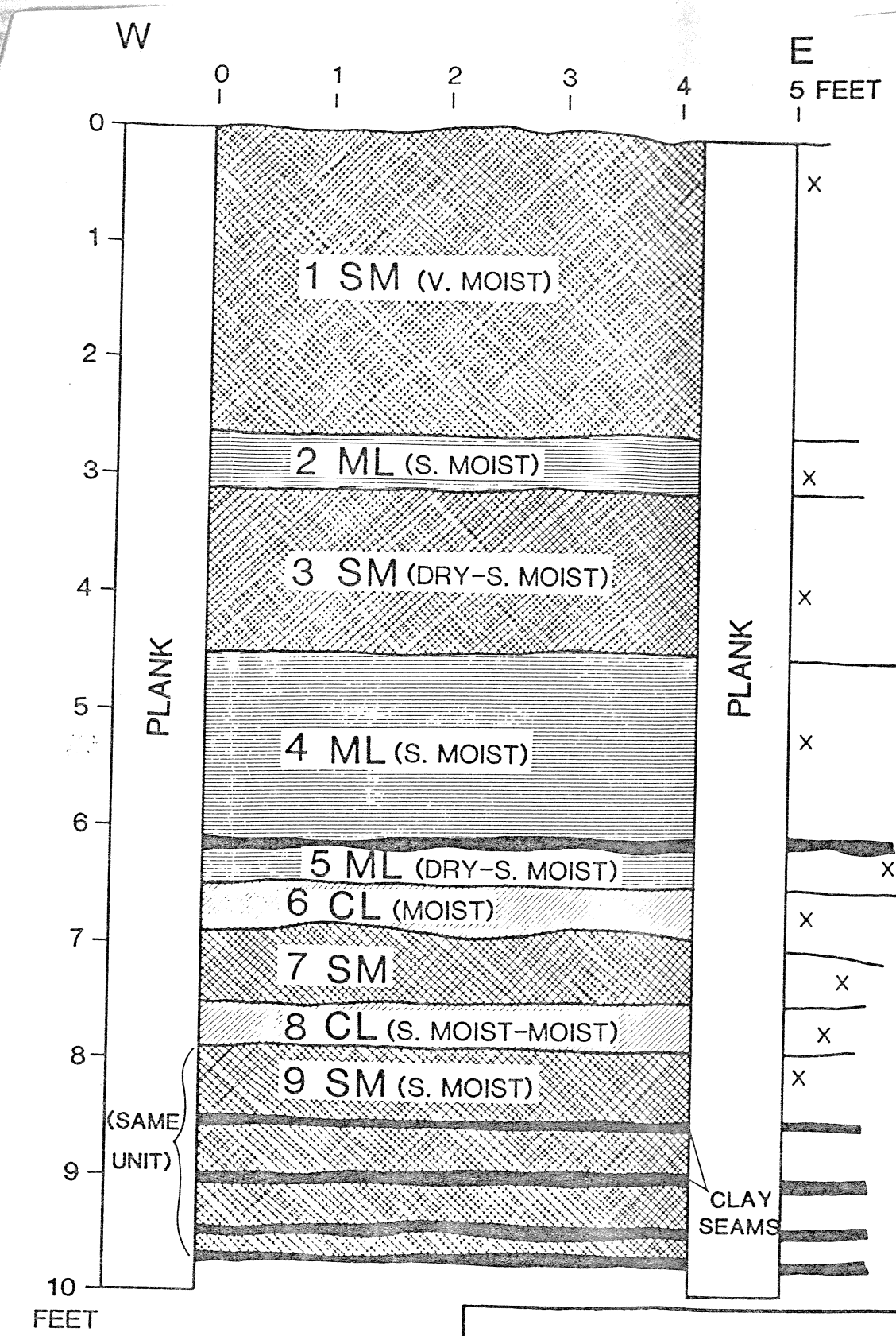
- 1 CL, SANDY, LEAN CLAY, V. MOIST, SOFT, PPV = 1.5kg/cm<sup>2</sup>, TSS = 1.75kg/cm<sup>2</sup>, VERY ORGANIC, WORM CASTINGS, ABUNDANT ROOTS, FEW CaCO<sub>3</sub> NODULES (LAKE), DARK BROWN.
- 2 SM, SILTY SAND, MOIST, SOFT, PPV = 1.5kg/cm<sup>2</sup>, TSS = 2.5kg/cm<sup>2</sup>, ABUNDANT BURROWS, GRUBS, SOFT CaCO<sub>3</sub> NODULES, PALE YELLOW.
- 3 CL-ML, SANDY SILTY CLAY, DRY, HARD, PPV > 4.5kg/cm<sup>2</sup>, TSS = 2kg/cm<sup>2</sup>, UPPER CONTACT GRADATIONAL, ABUNDANT ROOTS, FEW CLAY-FILLED BURROWS, ABUNDANT CaCO<sub>3</sub> NODULES, PALE YELLOW.
- 4 SM, SILTY SAND, HARD, PPV > 4.5kg/cm<sup>2</sup>, TSS = 1kg/cm<sup>2</sup>, MOTTLED AND STREAKED WITH RED/YELLOW IRON STAINING, YELLOW.
- 5 SC, CLAYEY SAND, S. MOIST, HARD, PPV > 4.5kg/cm<sup>2</sup>, TSS = 3kg/cm<sup>2</sup>, LOW PLASTICITY CLAY INTERBEDDED WITH SAND, FEW ROOTS, UPPER CONTACT WAVEY, V. PALE BROWN.
- 6 SM, SILTY FINE SAND, SMALL CLAYEY SILT NODULES THROUGHOUT, SCATTERED CaCO<sub>3</sub> NODULES, MOTTLED AND STREAKED WITH YELLOWISH RED IRON STREAKS, PALE BROWN.
- 7 ML-CL, INTERBEDDED SILTY CLAYS AND CROSSBEDDED FINE SAND, STIFF, PPV > 4.5kg/cm<sup>2</sup>, TSS = 2kg/cm<sup>2</sup>, YELLOWISH BROWN.
- 8 SM, SILTY SAND, CROSSBEDDED, HARD, PPV > 4.5kg/cm<sup>2</sup>, TSS = 1.0kg/cm<sup>2</sup>, FEW ROOTS, UPPER MUCH MORE CLAYEY, UPPER CONTACT IS WAVY, LIGHT YELLOWISH BROWN.
- 9 CL, SANDY LEAN CLAY, S. MOIST, MASSIVE, STIFF, PPV > 4.5kg/cm<sup>2</sup>, TSS = 3kg/cm<sup>2</sup>, YELLOWISH BROWN.
- 10 SM, SILTY SAND, CROSSBEDDED, S. MOIST-DRY, HARD, PPV > 4.5kg/cm<sup>2</sup>, TSS = 1.5kg/cm<sup>2</sup>, FEW ROOTS, DISCONTINUOUS LAYERS OF SP-SM, LIGHT YELLOWISH BROWN.

10 FEET

X = SAMPLE LOCATION  
 PPV = POCKET PENETROMETER VALUE  
 TSS = TORVANE SHEAR STRENGTH  
 S = SLIGHTLY

LOG OF TRENCH SSC-BH-6

JULY 31, 1987



- 1 SM, SILTY SAND, V. MOIST, SOFT, PPV=0.5kg/cm<sup>2</sup>, TSS=1.0kg/cm<sup>2</sup>, ABUNDANT ROOTS, A FEW SOFT CaCO<sub>3</sub> NODULES BELOW 1.5 FEET, NO STRUCTURE, DARK YELLOWISH BROWN.
- 2 ML, SANDY SILT, S. MOIST, PPV=4.0kg/cm<sup>2</sup>, TSS=2.5kg/cm<sup>2</sup>, NEARLY PLUGGED WITH CaCO<sub>3</sub>, NUMEROUS 0.5 INCH WORM BURROWS, GRADATIONAL CONTACT WITH UNIT 3 BELOW, WHITE.
- 3 SM, SILTY SAND, DRY TO SLIGHTLY MOIST, HARD, PPV=4.5kg/cm<sup>2</sup>, TSS=3.0kg/cm<sup>2</sup>, A FEW 0.5 INCH DIAMETER BURROWS, YELLOW MOTTLED WITH YELLOW IRON OXIDE STAINING.
- 4 ML, SANDY SILT, S. MOIST, HARD, PPV=4.5kg/cm<sup>2</sup>, TSS=1.0kg/cm<sup>2</sup>; POORLY CROSSBEDDED, SCATTERED CaCO<sub>3</sub> NODULES AND FEW ROOTS, UPPER CONTACT GRADATIONAL, V. PALE BROWN MOTTLED WITH ORANGE IRON OXIDE STAINING.
- 5 ML, SILT WITH SAND, DRY TO S. MOIST, FIRM, PPV=3.5kg/cm<sup>2</sup>, TSS=1.0kg/cm<sup>2</sup>, CROSSBEDDED, LIGHT BROWNISH GREY.
- 6 CL, LEAN CLAY, MOIST, HARD, PPV>5 4.5kg/cm<sup>2</sup>, TSS=5.0kg/cm<sup>2</sup>, SOME GYPSUM STRINGERS, ROOTS, V. PALE BROWN, IRON STAINED.
- 7 SM, SILTY SAND, SOFT, PPV=1.0kg/cm<sup>2</sup>, TSS=1.0kg/cm<sup>2</sup>, CROSSBEDDED, NUMEROUS CLAY CONCRETIONS UP TO .25 INCH DIAMETER, SHARP UPPER SURFACE, REWORKED CaCO<sub>3</sub> CONCRETIONS, YELLOW.
- 8 CL, LEAN CLAY, S. MOIST TO MOIST, STIFF, PPV=4.5kg/cm<sup>2</sup>, TSS=5.5kg/cm<sup>2</sup>, CONCRETIONS OF GYPSUM AT BASE, IRON OXIDE COLORATION ON PED SURFACES, UNIDENTIFIED RED ORGANIC FILAMENTS, LIGHT BROWNISH GREY.
- 9 SM, SILTY SAND, S. MOIST, HARD, PPV=4.5kg/cm<sup>2</sup>, TSS=2kg/cm<sup>2</sup>, STREAKED AND MOTTLED WITH ORANGE IRON OXIDE STAINING, FEW ROOTS, CaCO<sub>3</sub> LAMINAE AND CLAY SEAMS AT 9.0, 9.4, 8.5, AND 8.7 FEET, LIGHT BROWNISH GREY.

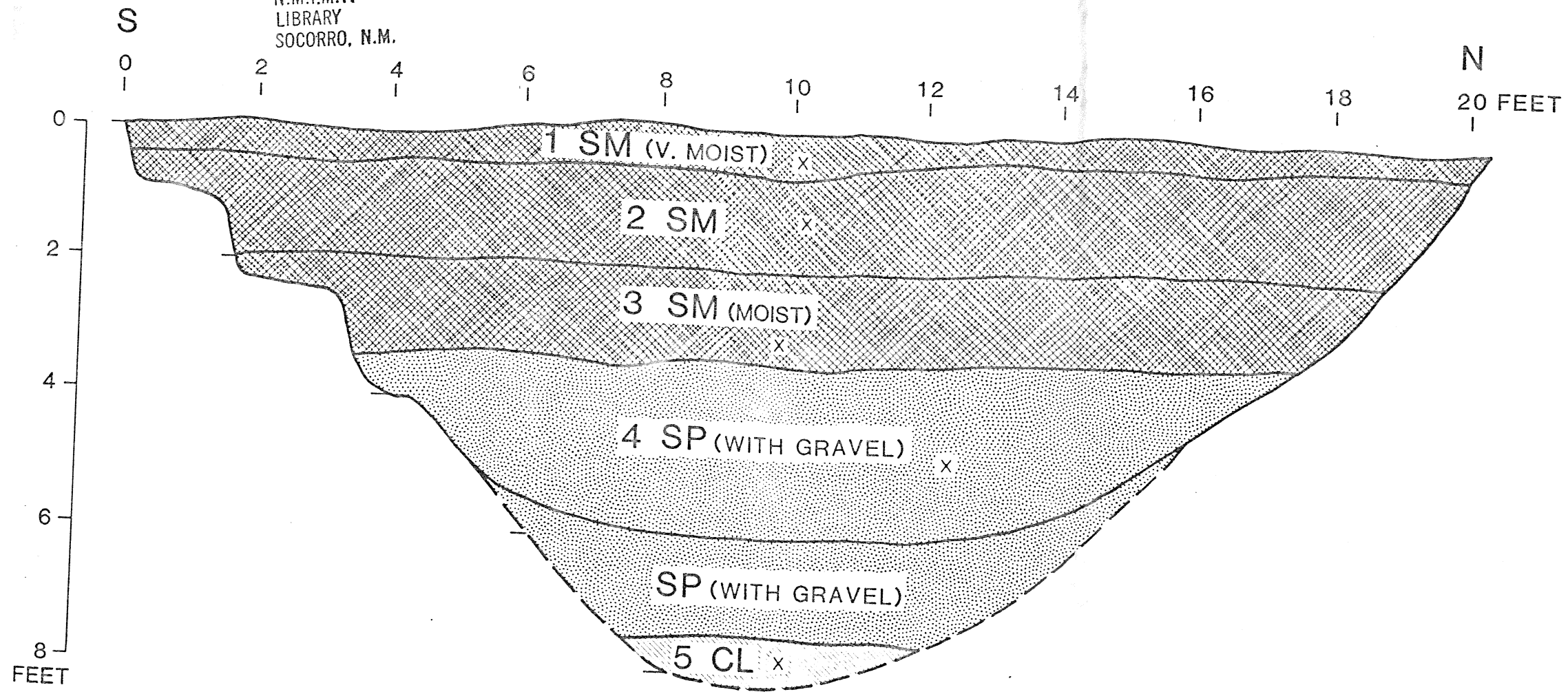
X= SAMPLE LOCATION  
 PPV=POCKET PENETROMETER VALUE  
 TSS=TORVANE SHEAR STRENGTH  
 S=SLIGHTLY

LOG OF TRENCH SSC-BH-7  
 JULY 31, 1987



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### VIEW WEST



- UNIT
- 1 SM, SILTY SAND, V. MOIST, LOOSE, CONTAINS ABUNDANT ROOTS AND ORGANIC MATTER.
  - 2 SM, SILTY SAND, LOOSE, ABUNDANT ROOTS, FEW CaCO<sub>3</sub> NODULES.
  - 3 SM, SILTY SAND, MOIST, LOOSE, ABUNDANT CaCO<sub>3</sub> NODULES, PEA-SIZED GRAVEL.
  - 4 SP, POORLY GRADED SAND WITH GRA LENSES, RUNNING SAND, GRAVEL CLASTS UP TO 0.5 INCH, LITHOLOGIES IGNEOUS ROCKS, QUARTZ, FELDSPAR
  - 5 CL, LEAN CLAY WITH SAND, S. MOIST STIFF, WHITE, BLOCKY TEXTURE (LAP DEPOSIT).

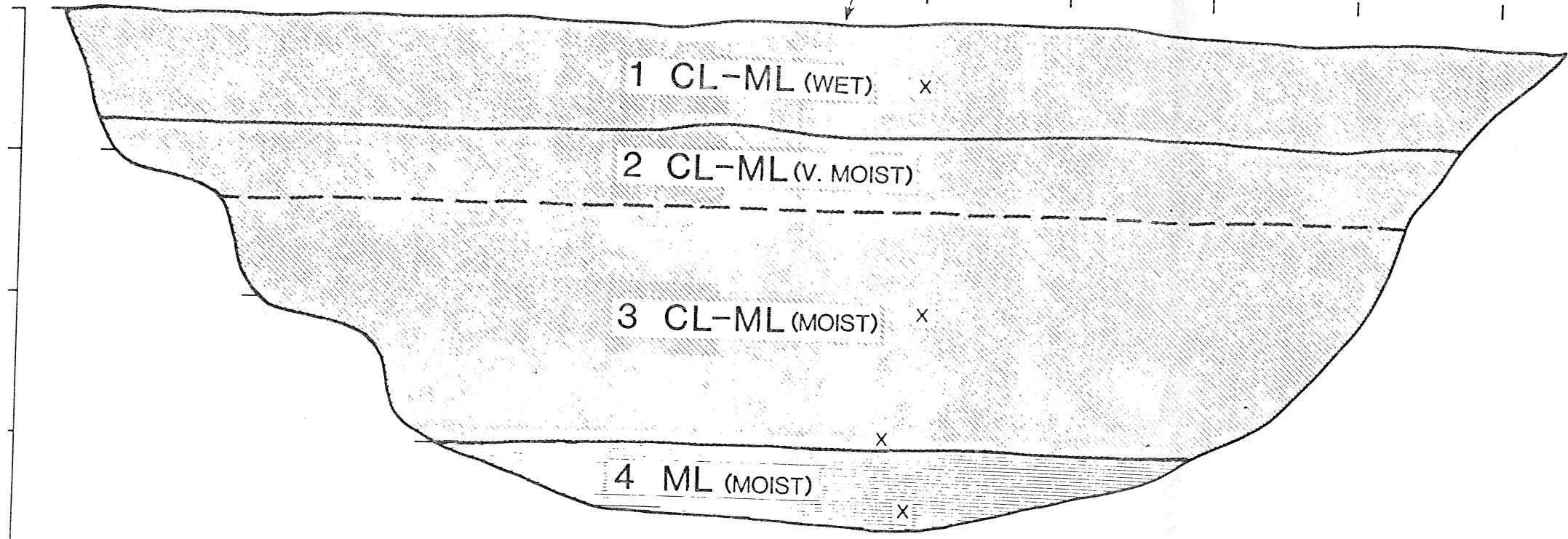
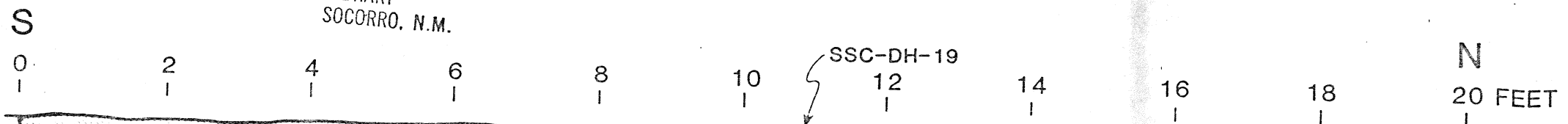
X = SAMPLE LOCATION  
PPV = POCKET PENETROMETER VALUE  
TSS = TORVANE SHEAR STRENGTH  
S = SLIGHTLY

LOG OF TRENCH SSC-BH-3  
JULY 31, 1987

NEW MEXICO SSC PROPOSAL

VIEW WEST

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UNIT

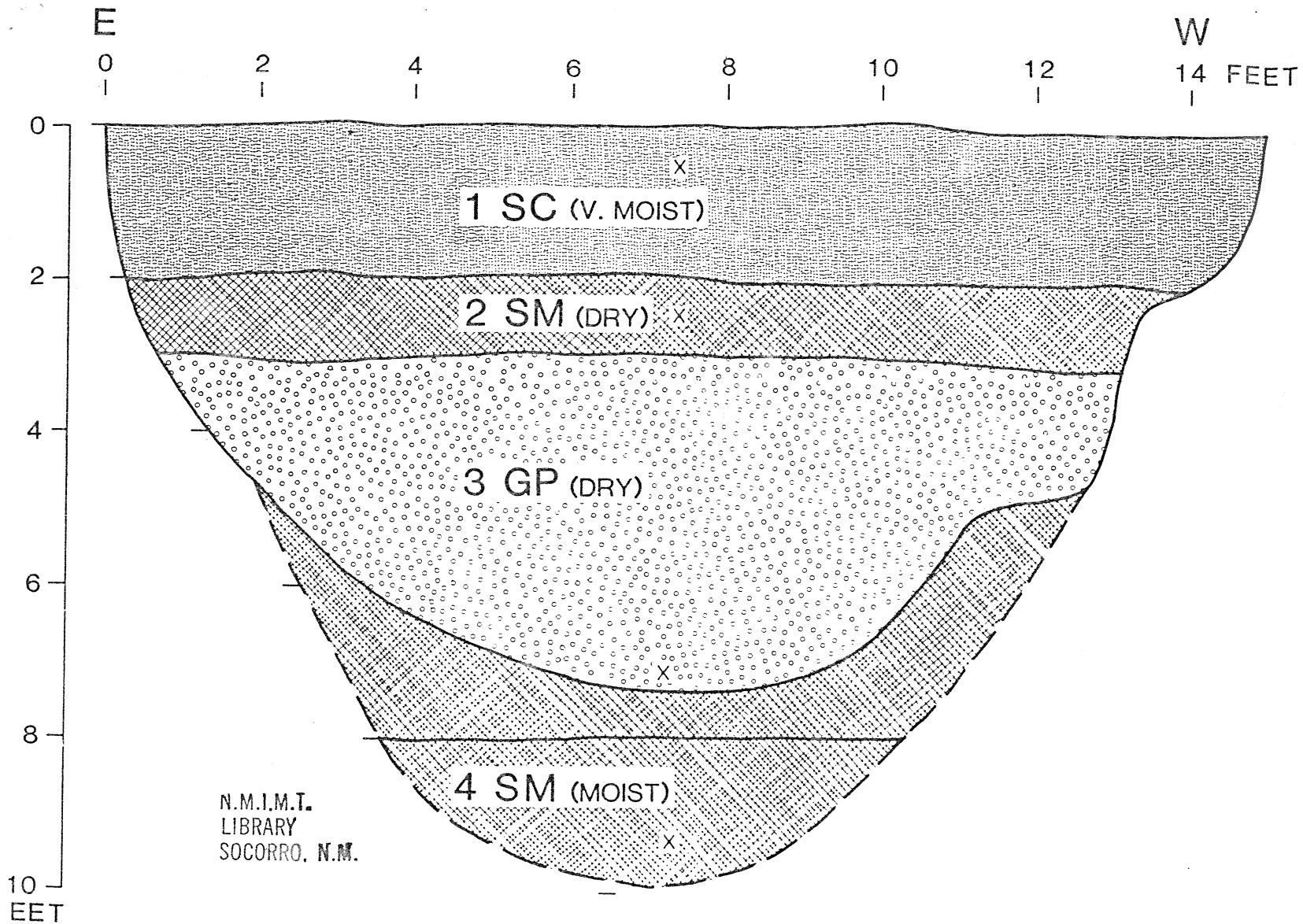
- 1 CL-ML, SANDY-SILTY CLAY, WET, SOFT, PPV < 0.5kg/cm<sup>2</sup>, GREEN/OLIVE (LAKE).
- 2 CL-ML, SANDY-SILTY CLAY, V. MOIST, SOFT-FIRM, PPV ≈ 2kg/cm<sup>2</sup>, WHITE.
- 3 CL-ML, SILTY CLAY, WITH SAND, MOIST, STIFF-HARD, PPV > 4.5kg/cm<sup>2</sup>, WHITE.
- 4 ML, SILT, S. MOIST, STIFF, PPV > 4.5kg/cm<sup>2</sup>, ROCK-LIKE IN CHARACTER, VUGGY, CaCO<sub>3</sub> CRYSTALS ON VUG SURFACES, WHITE (LAKE MARE).

X = SAMPLE LOCATION  
 PPV = POCKET PENETROMETER VALUE  
 TSS = TORVANE SHEAR STRENGTH  
 S = SLIGHTLY

LOG OF TRENCH SSC-BH-5  
 JULY 31, 1987



VIEW SOUTH



UNIT

- 1 SC, CLAYEY SAND WITH GRAVEL, V. MOIST, SOFT, VERY ORGANIC, ABUNDANT WORMS AND WORM BURROWS, SCATTERED GRAVEL (FINE TO MEDIUM).
- 2 SM, DRY, STAGE IV, CaCO<sub>3</sub>.
- 3 GP, POORLY-GRADED GRAVEL WITH SAND, AND SCATTERED BOULDERS AND COBBLES, DRY, CLASTS ARE MADERA LIMESTONE, SUBANGULAR TO SUBROUNDED, MOST ARE WELL COATED WITH CaCO<sub>3</sub>.
- 4 SM, SILTY SAND, MOIST, DENSE, CLASTS TO 0.5 INCH, SUBROUNDED, ALL MADERA LIMESTONE, BROWN TO DARK BROWN.

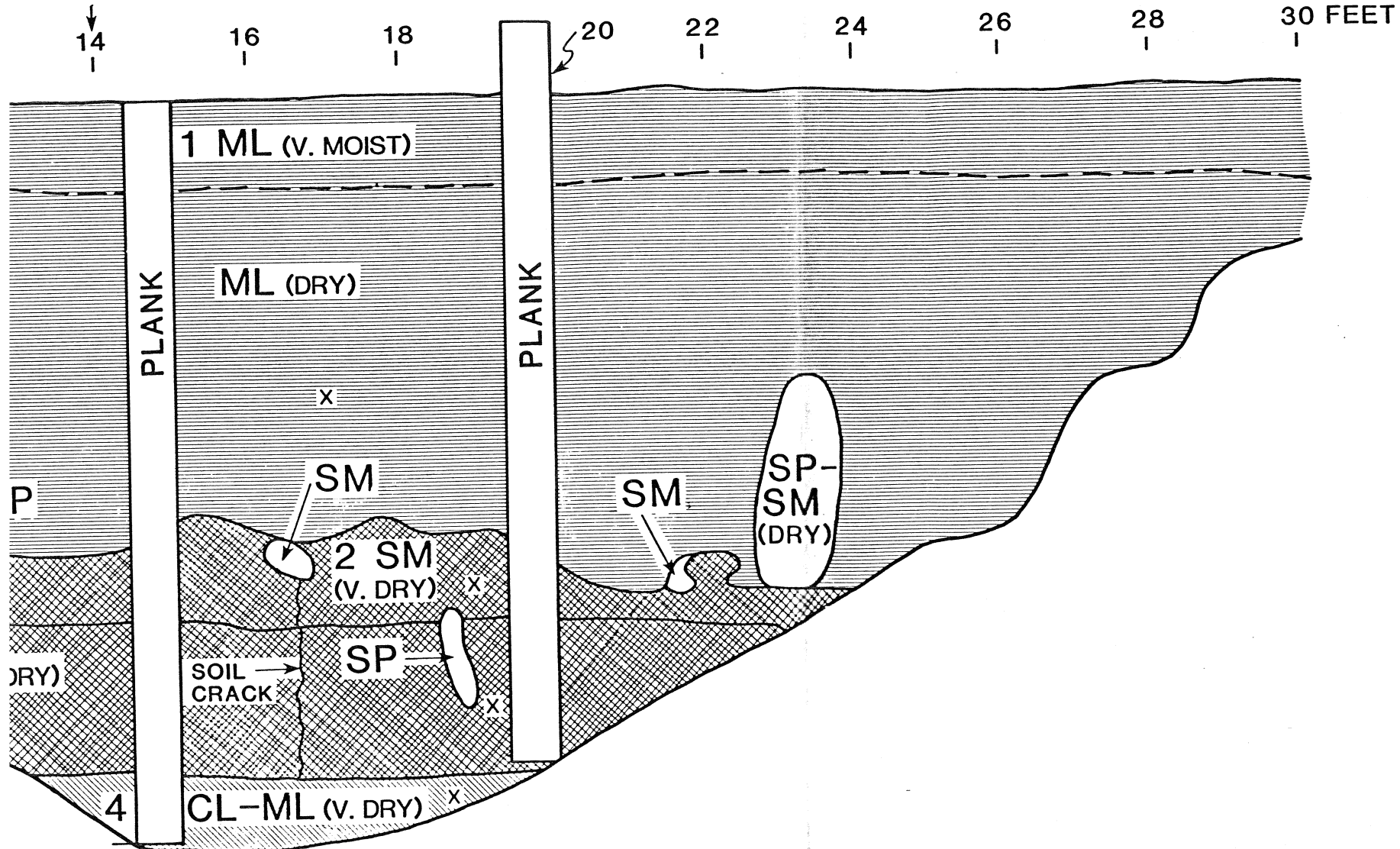
X = SAMPLE LOCATION  
 PPV = POCKET PENETROMETER VALUE  
 TSS = TORVANE SHEAR STRENGTH

LOG OF TRENCH SSC-BH-9  
 JULY 31, 1987

VIEW NORTH

SMIC LINE SR-1  
R-142

E



UNIT

- 1 ML, SANDY SILT, V. MOIST AT SURFACE, DRY BELOW 1 FOOT, SOFT AT SURFACE, STIFF BELOW 1 FOOT, CONTAINS GRAVEL SIZE CLASTS OF IGNEOUS AND LIMESTONE ROCK (SUBANGULAR),  $\text{CaCO}_3$  NODULES AND FILAMENTS.
- 2 SM, SILTY SAND, V. DRY, HARD,  $\text{CaCO}_3$  LAMINAR-SUBLAMINAR STAGE IV IN PLACES, BROKEN INTO GRAVEL SIZE PIECES THROUGHOUT. EXTENSIVELY BURROWED. BURROWS FILLED WITH FINE DARK COLORED SP-SM (GRAVELLY SAND-SILTY SAND), DRY, SOFT.
- 3 SM-SC, SILTY SAND-CLAYEY SAND, V. DRY, SOMEWHAT BURROWED.
- 4 CL-ML, SILTY CLAY, V. DRY, SCATTERED CARBON.

NOTES:

SOIL CRACK IS COATED WITH  $\text{CaCO}_3$ ,  
NO OFFSET NOTED.

SMALL ROOTS TO 6 FEET.

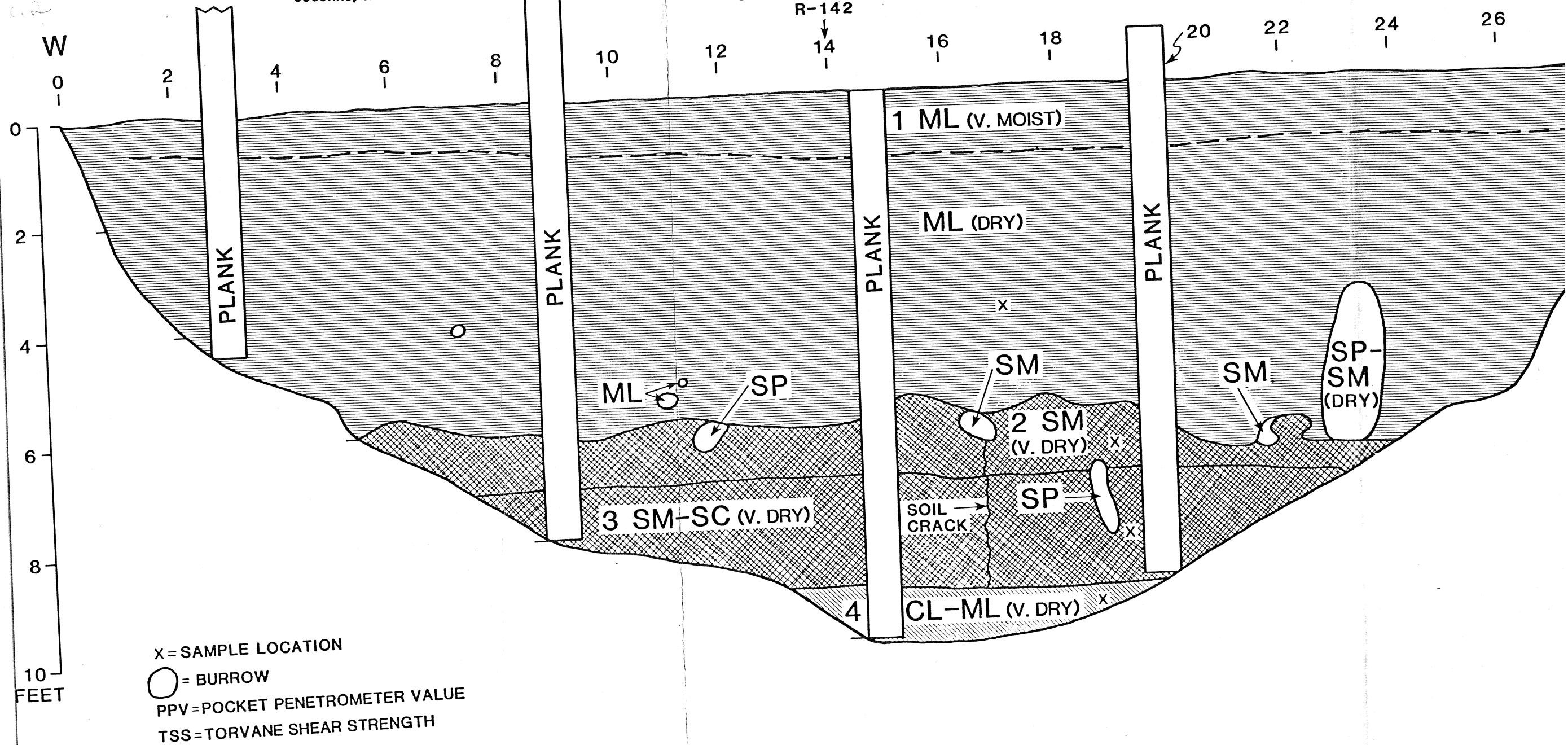
NO EVIDENCE FOR FAULTING IN TRENCH.

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SOCORRO, N.M.

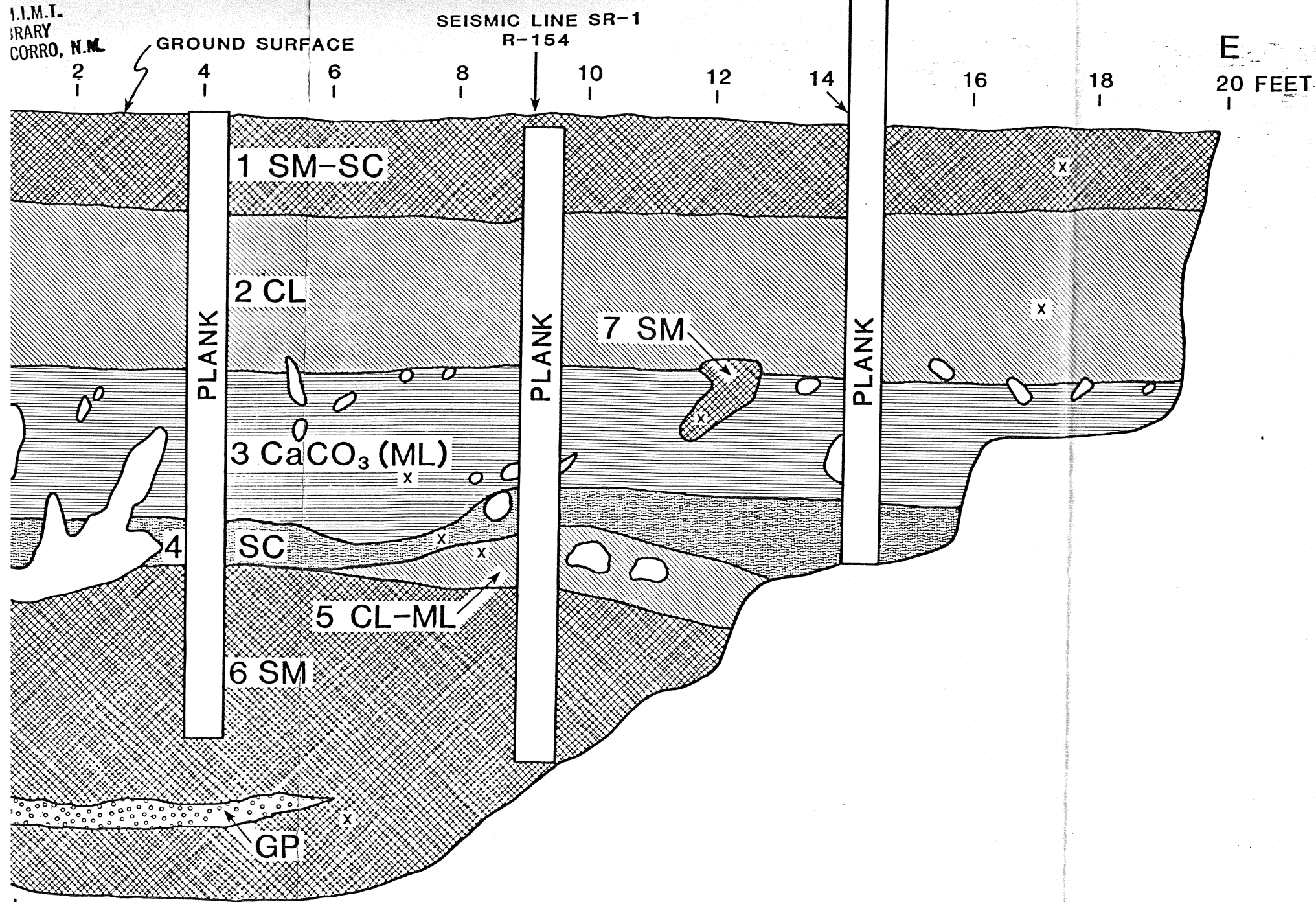
### VIEW NORTH

SEISMIC LINE SR-1  
R-142



LOG OF TRENCH SSC-BH-1
JULY 31, 1987

VIEW NORTH



UNIT

- 1 SM-SC, SILTY SAND-CLAYEY SAND, MOIST, SOFT, PPV = 1.0 kg/cm<sup>2</sup>, MANY ROOTS.
- 2 CL, LEAN CLAY, DRY, HARD, PPV > 4.5kg/cm<sup>2</sup>, CONTAINS FILAMENTS AND NODULES, MANY ROOTS.
- 3 ML, CaCO<sub>3</sub>, SILT WITH SAND, DRY, HARD-V. HARD, WHITE, STAGE III (NO LAMINAE), PPV > 4.5kg/cm<sup>2</sup>.
- 4 SC, CLAYEY SAND, HARD, PPV > 4.5kg/cm<sup>2</sup>.
- 5 CL-ML, SANDY SILTY CLAY, DRY, SOFT, FINELY LAMINATED FINE SAND, PPV < 0.5kg/cm<sup>2</sup>, CONTAINS .25 INCH SIZE INTRACLASTS OF CLAY, RODENT BONES.
- 6 SM, SILTY SAND, DRY, MED. DENSE, PPV > 4.5kg/cm<sup>2</sup>, DISCONTINUOUS GRAVEL LENSES.
- 7 SM, SILTY SAND, HARD-VERY HARD, FROM BURROW AT 5.5 FEET.

OMETER VALUE  
STRENGTH

1 SSC-BH-2

1987

C PROPOSAL

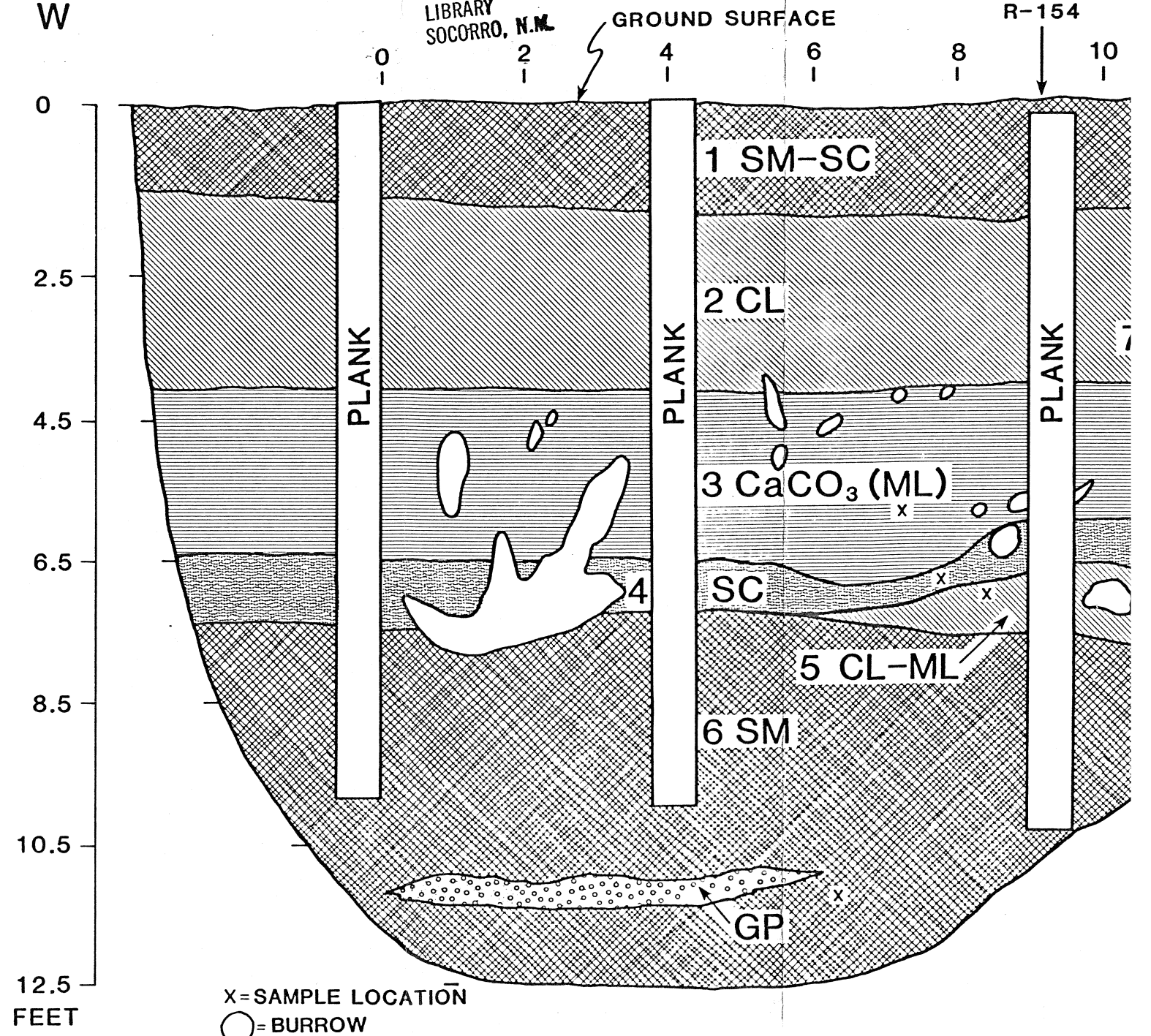


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VIEW NORTH

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SEISMIC LINE SR  
R-154



x = SAMPLE LOCATION  
○ = BURROW  
PPV = POCKET PENETROMETER VALUE  
TSS = TORVANE SHEAR STRENGTH

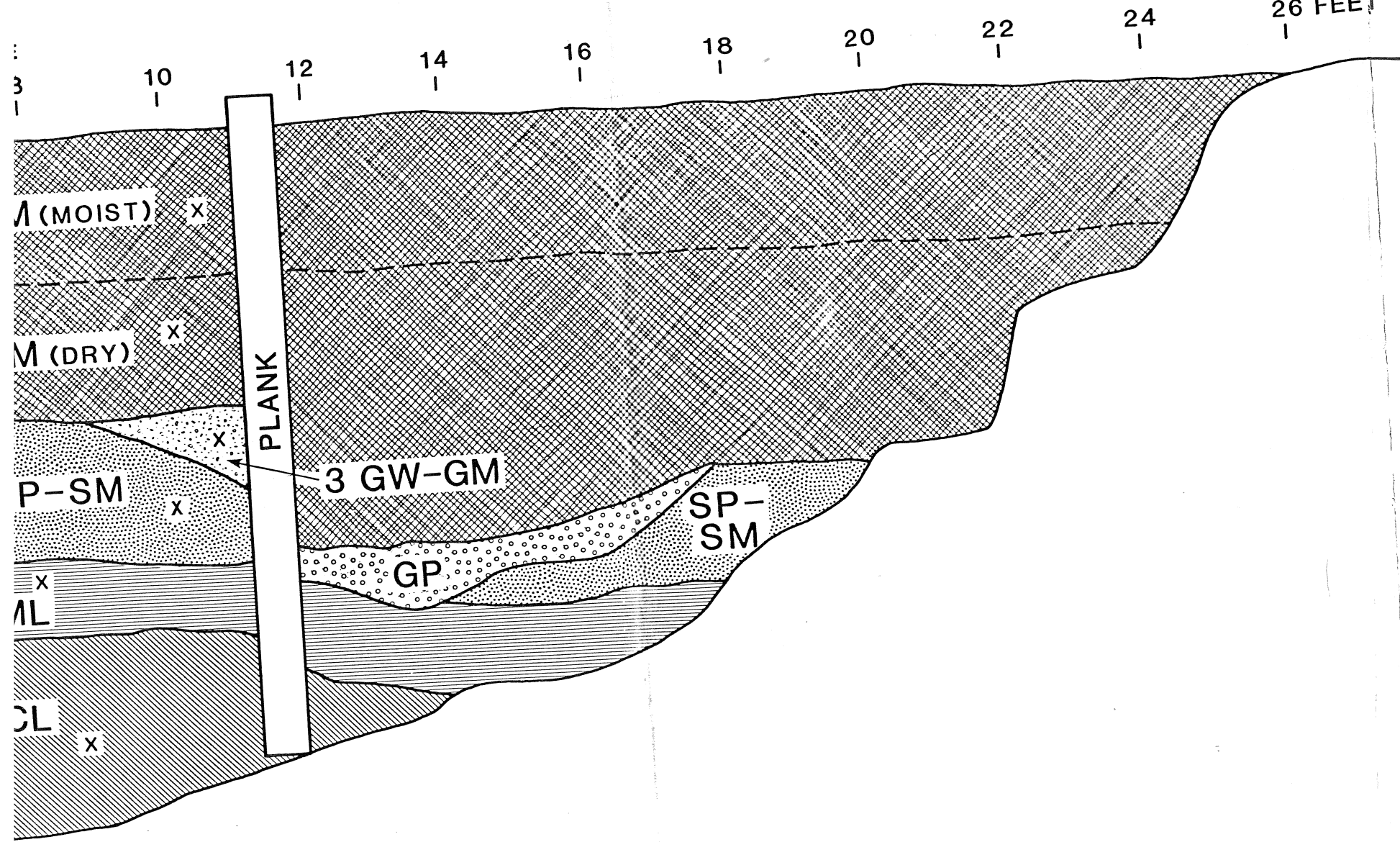
LOG OF TRENCH SSC-BH-2  
JULY 31, 1987

NEW MEXICO SSC PROPOSAL

VIEW NORTH

E

26 FEET



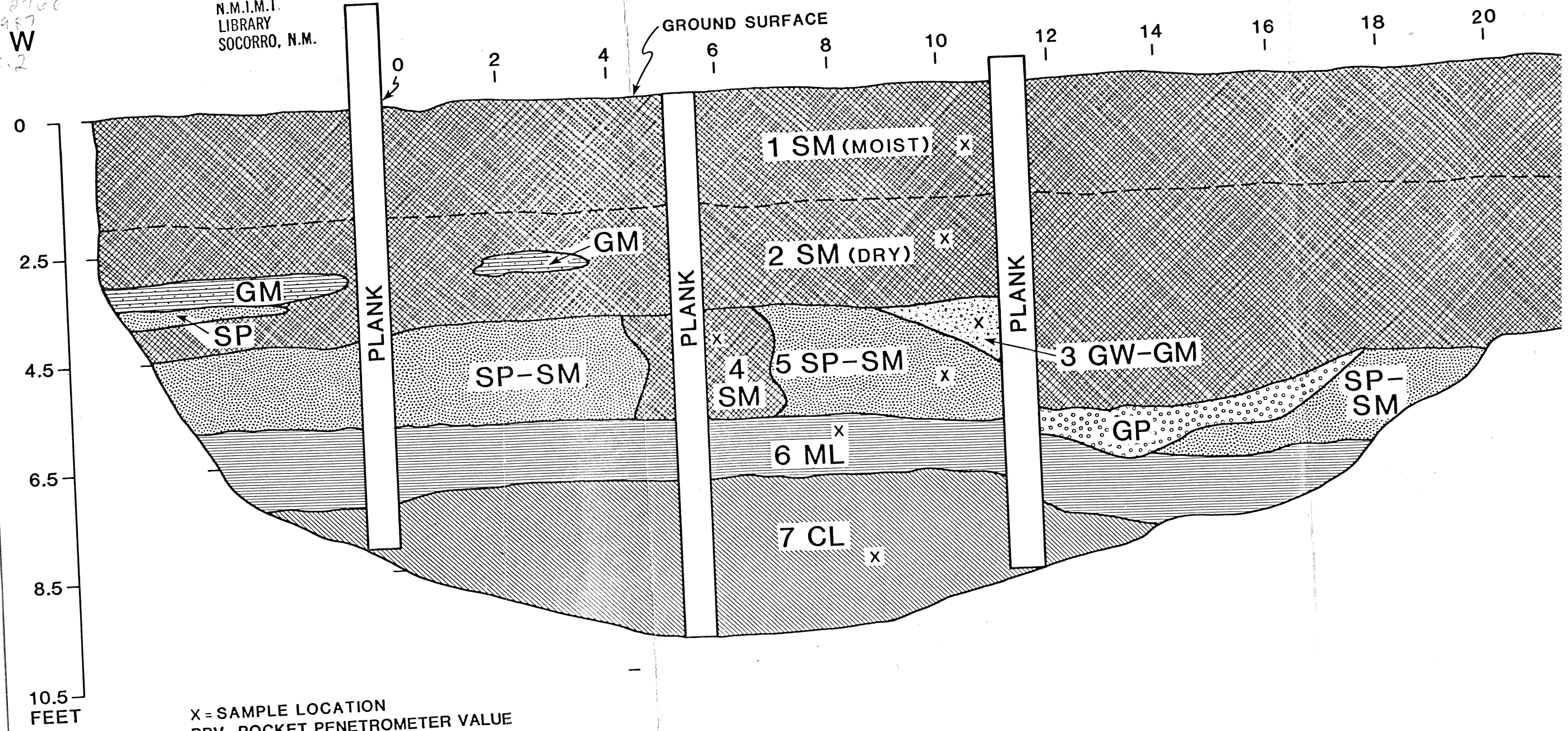
UNIT

- 1 SM, SILTY SAND, S. MOIST, LOOSE.
- 2 SM, SILTY SAND-GRAVELLY LENSES OF GM, (SILTY GRAVEL) AND SP, (CLEAN SAND). DRY, LOOSE, SNAIL SHELLS.
- 3 GW-GM, WELL GRADED GRAVEL WITH SILT.
- 4 SP, FINE SAND.
- 5 SP-SM, FINE SAND-SILTY SAND, DRY, MED. DENSE TO LOOSE, LAMINATED.
- 6 ML, SANDY SILT, DRY, STIFF,  $PPV > 4.5 \text{ kg/cm}^2$ .
- 7 CL, LEAN CLAY WITH SAND, DRY-S. MOIST, STIFF-HARD,  $PPV > 4.5 \text{ kg/cm}^2$ , WHITE.

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### VIEW NORTH

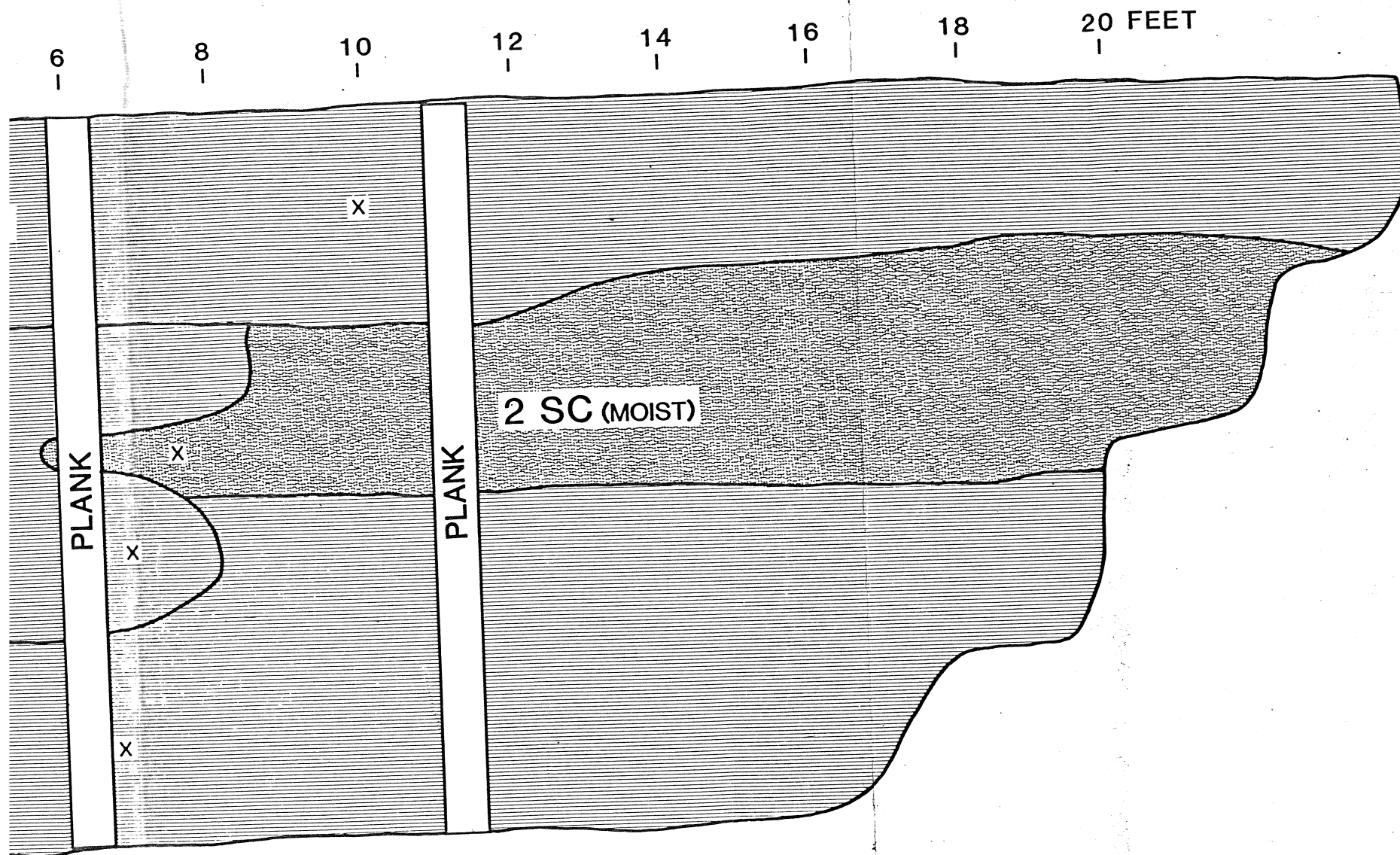


x = SAMPLE LOCATION  
PPV = POCKET PENETROMETER VALUE  
TSS = TORVANE SHEAR STRENGTH  
S = SLIGHTLY

LOG OF TRENCH SSC-BH-4  
JULY 31, 1987



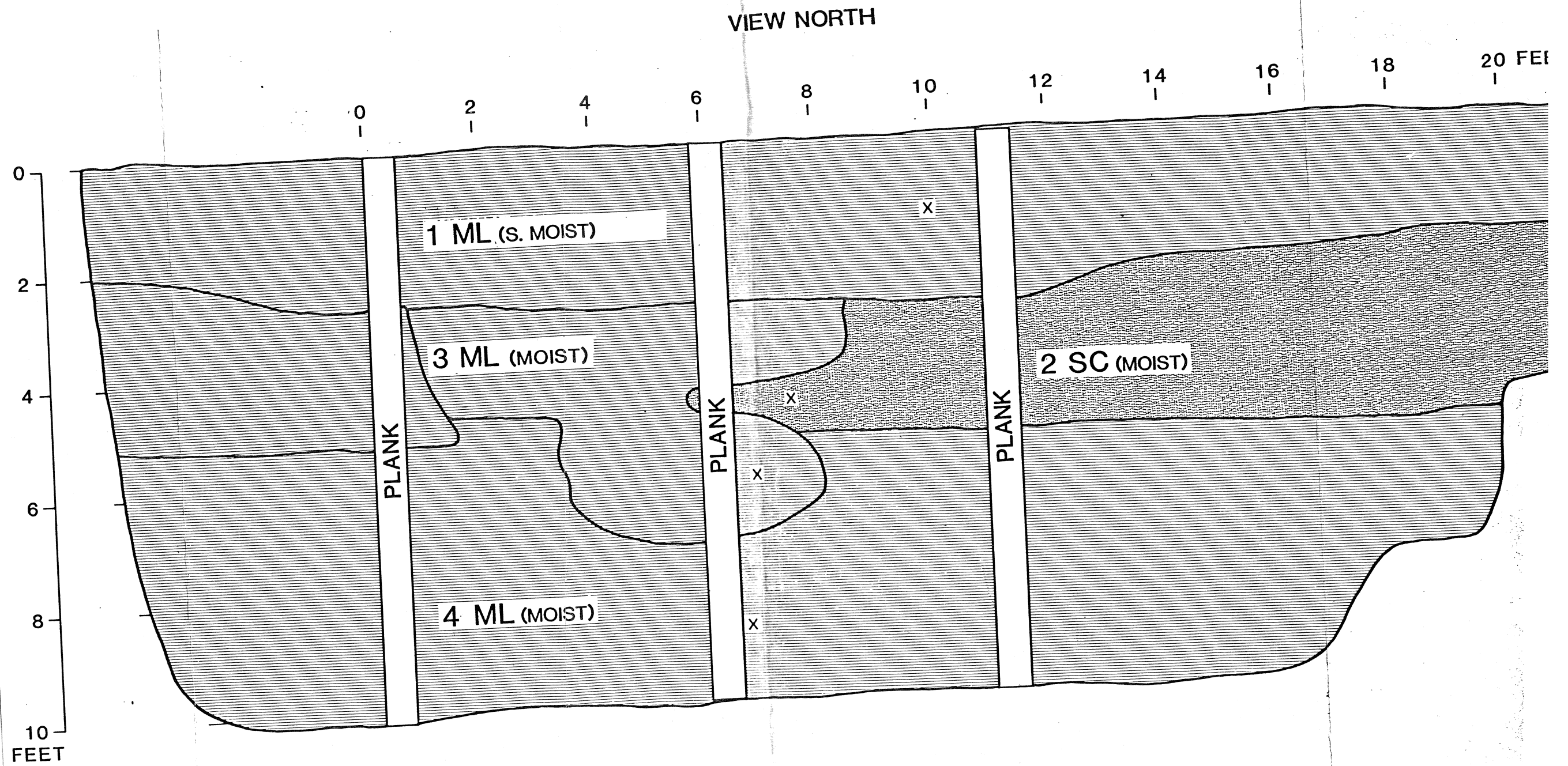
VIEW NORTH



UNIT

- 1 ML, SANDY SILT, MOIST,  $PPV = 2.0 \text{ kg/cm}^2$ ,  $TSS = 1.5 \text{ kg/cm}^2$ , ORGANIC RICH, HEAVILY BURROWED, MANY  $\text{CaCO}_3$  NODULES, SOME REWORKED CLASTS FROM UNIT 2 BELOW, DARK YELLOWISH BROWN (POSSIBLY ALLUVIUM).
- 2 SC, CLAYEY SAND, MOIST, HARD,  $PPV > 4.5 \text{ kg/cm}^2$ ,  $TSS = 9.5 \text{ kg/cm}^2$ , FINE ROOTS, CONTACTS BETWEEN UNITS ABOVE AND BELOW ARE SHARP, V. PALE BROWN (LAKE).
- 3 ML, SILT TO SANDY SILT, MOIST, SOFT,  $PPV = 1.5 \text{ kg/cm}^2$ ,  $TSS = 0.5 \text{ kg/cm}^2$ , MORE ORGANIC THAN OTHER UNITS, DISTINGUISHED FROM OTHER UNITS BY DARKER COLOR, STRONG REACTION TO HCl, FINE ROOTS, YELLOWISH BROWN.
- 4 ML, SANDY SILT, MOIST, HARD,  $PPV > 4.5 \text{ kg/cm}^2$ ,  $TSS = 3.0 \text{ kg/cm}^2$ , FEW ROOTS, STRONG REACTION TO HCl, GRADATIONAL CONTACT TO UNIT 3 ABOVE, STRONG BROWN.





X = SAMPLE LOCATION  
 PPV = POCKET PENETROMETER VALUE  
 TSS = TORVANE SHEAR STRENGTH

LOG OF TRENCH SSC-BH-8

JULY 31, 1987

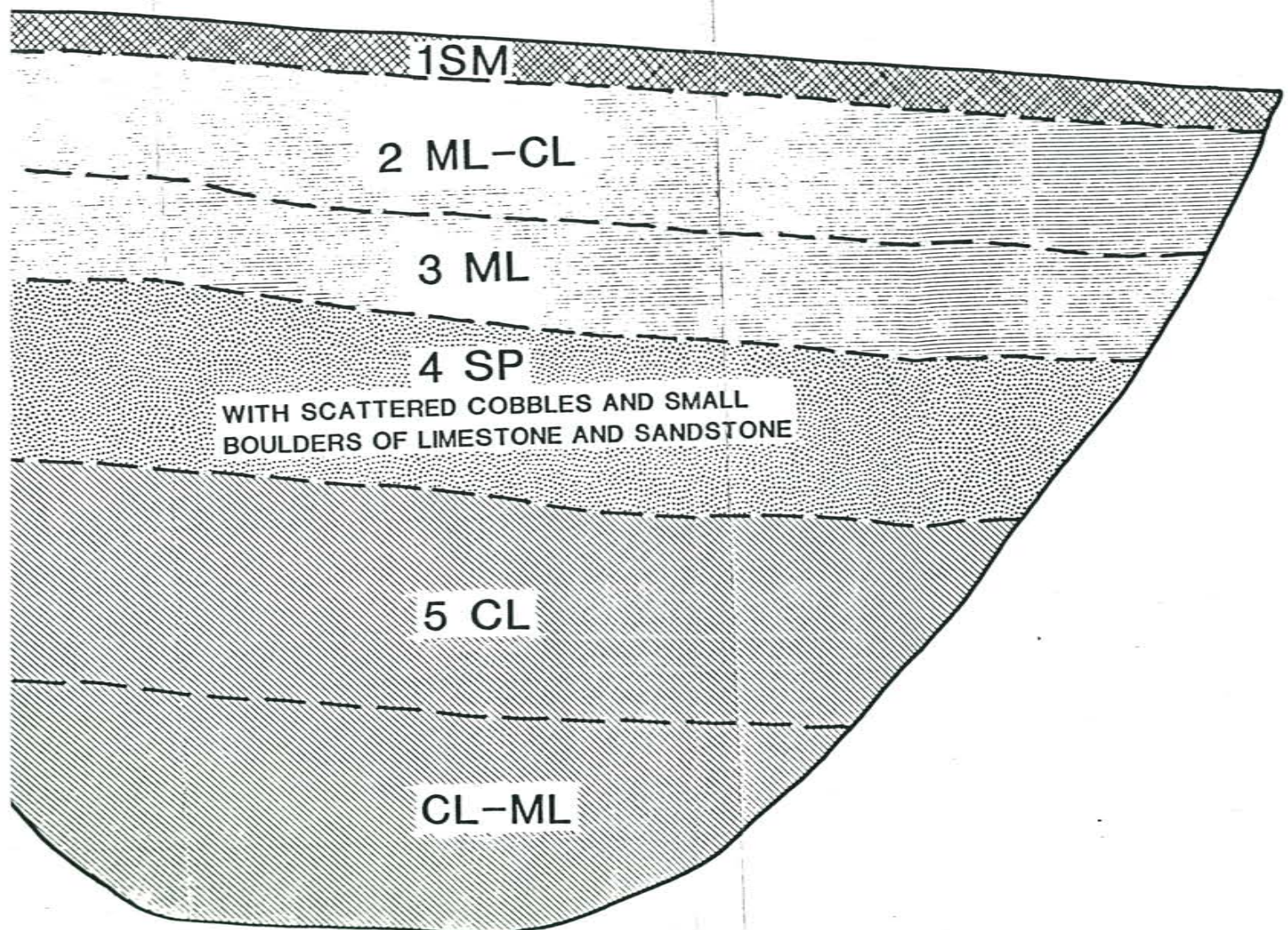


W

145 | 150 | 155 | 160 | 165 | 170 | 175 | 180 |

WEST TRENCH (PROJECTED)  
(NOT LOGGED IN DETAIL)

ACE



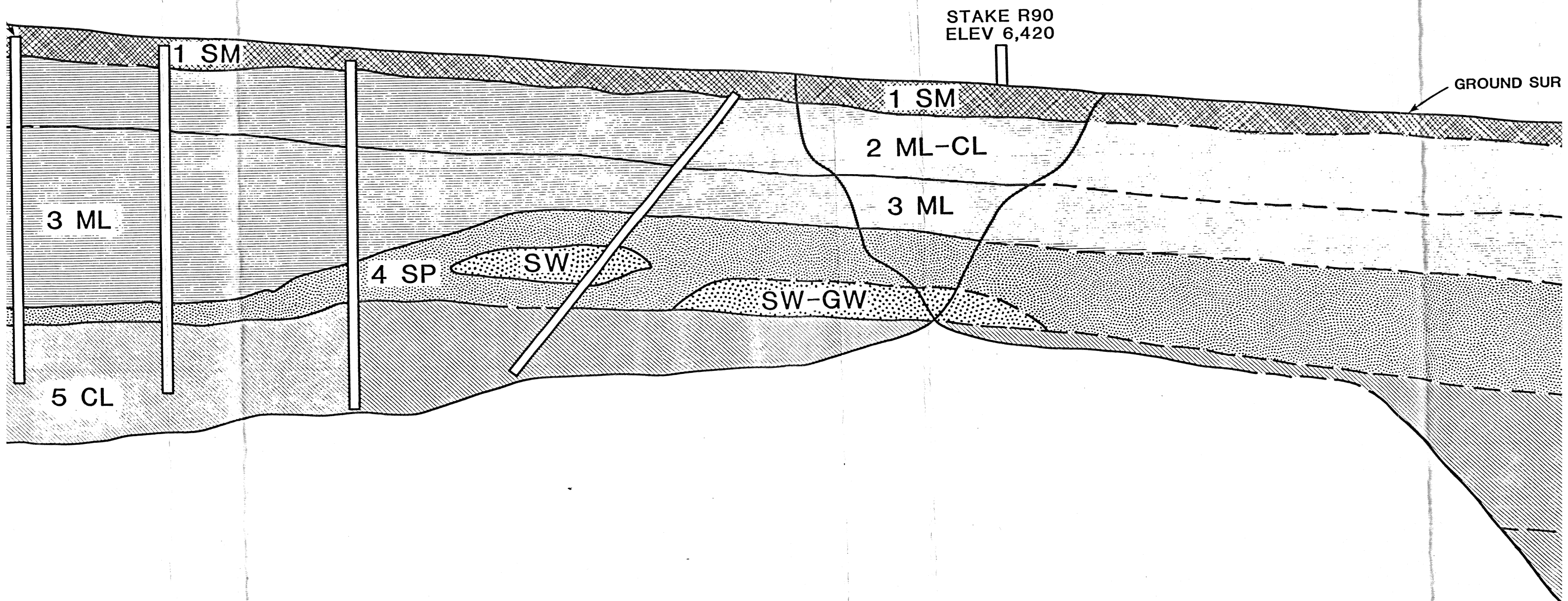
- UNIT
- 1 SM, FINE SAND, A HORIZON, MANY ROOTS, SCATTERED FINE GRAVEL, VERY MOIST-MOIST, 5YR 4/4 TO 4/6, REDDISH BROWN TO YELLOWISH RED.
  - 2 ML-CL, SANDY SILT TO SILTY CLAY, PEDOGENIC ARGILLIC HORIZON, SOFT CARBONATE NODULES, SCATTERED FINE GRAVEL (SOME COARSE GRAVEL), SLIGHTLY MOIST-MOIST, MANY SMALL ROOTS, 5 YR 5/6 TO 5/8, YELLOWISH RED.
  - 3 ML, SANDY SILT, PEDOGENIC CALCIC HORIZON IN UPPER PART, SCATTERED COARSE GRAVEL INCLUDING LIMESTONE AND VERY SOFT YELLOWISH SANDSTONE CLASTS, FEW ROOTS, 5YR 7/4 TO 6/6, PINK TO REDDISH YELLOW.
  - 4 SP, COARSE GRAVELLY SAND, 35% GRAVEL, MANY GRAVELS HAVE THICK CALCIUM CARBONATE RINDS, CONTAINS LENSES OF CLEAN SAND (SW), UP TO 1 INCH CARBONATE NODULES, 5 YR 5/4 TO 6/6, REDDISH BROWN TO REDDISH YELLOW.
  - 5 CL, SILTY CLAY WITH GRAVEL, BURIED STRONG PEDOGENIC CALIC HORIZON, VERY BLOCKY STRUCTURE; UP TO 2 INCH DIAMETER, HARD, WHITE, CARBONATE NODULES; BLACK STAINS ON PED SURFACES; CONTACTS WITH UNITS ABOVE AND BELOW ARE UNDULATORY; 7.5YR 8/2 TO 8/4 ; PINKISH WHITE TO PINK.
  - 6 GP-GW, POORLY TO WELL-GRADED GRAVEL, VERY HARD SANDY GRAVEL. 0.25 TO 0.5 INCH CLASTS, BACKHOE COULD NOT PENETRATE THIS BED.
  - 7 SP-GM, GRAVELLY SAND-SILTY GRAVEL, HARD, 7.5YR 7/6, REDDISH YELLOW.
  - 8 ML-SM, SANDY SILT-FINE SAND, CONTAINS RIP UP CLASTS OF UNIT 6, DRY, SCATTERED FINE GRAVEL, A FEW FINE ROOTS, GRADATIONAL CONTACT WITH UNIT 3, 5YR 4/6, YELLOWISH RED.

NOTE:  
SAMPLE 9 - FROM FAULT GOUGE ZONE, LAMINATIONS IN GOUGE ARE PARALLEL TO FAULT PLANE.

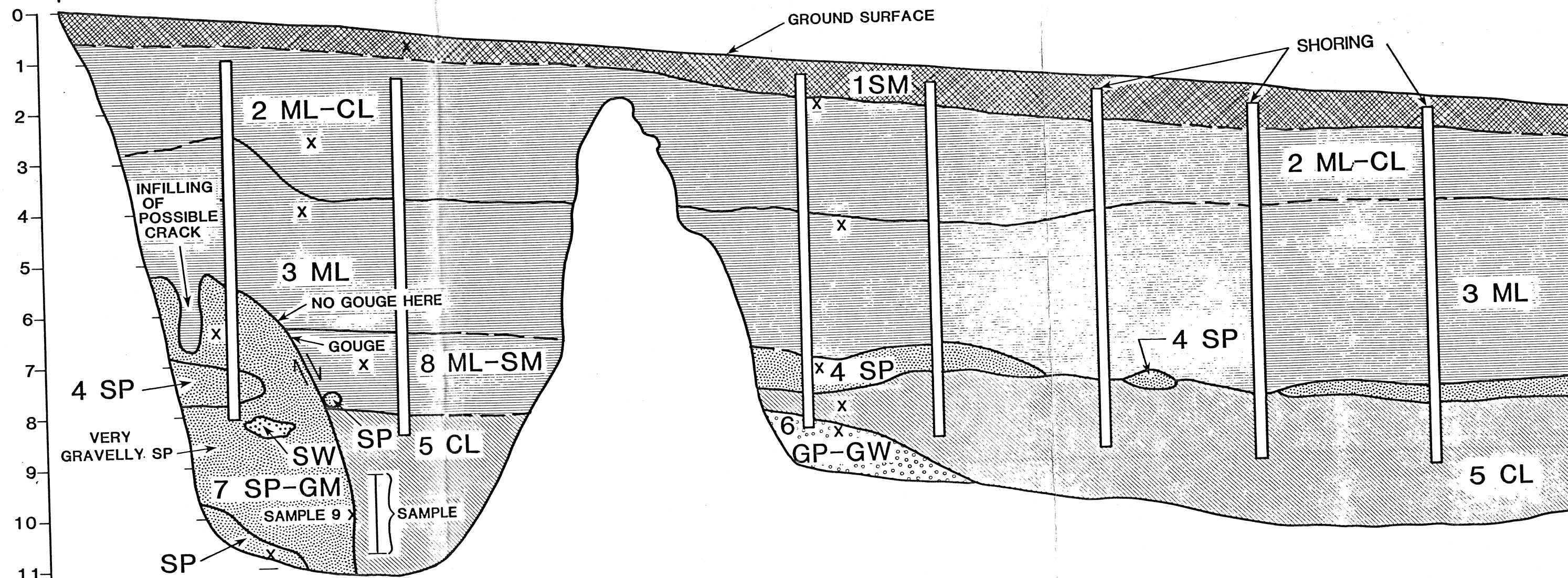


VIEW SOUTH

70 75 80 85 90 95 100 105 110 115 120 125 130 135 140



E 0 5 10 15 20 25 30 35 40 45 50 55 60 65 70  
 EAST TRENCH



X = SAMPLE LOCATION  
 ○ = BURROW (SCATTERED THROUGH-OUT TRENCH)  
 PPV = POCKET PENETROMETER VALUE  
 TSS = TORVANE SHEAR STRENGTH

LOG OF TRENCH SSC-BH-10  
 JULY 31, 1987