

S-215-2007  
(Consultant)

**MEMORANDUM**

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**DATE:** April 25, 2007

**SUBJECT:** **Jefferson County**  
**1200 056**  
**Mars # 6554101D**  
**Ramp 42 3RD Street to I-65 SB (S0180, Bridge B3RD-1)**  
**Ohio River Bridges Project**  
**Kennedy Interchange Reconstruction – Section 1**  
**Item No. 5-118.18 & .19**

The geotechnical engineering report for this structure has been completed by Barr & Prevost. We have reviewed and concur with the recommendations as presented in this report.

A copy of the report is attached. If you have any questions, please contact this office at 502-564-2374.

Attachment:

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Stuart Edwards (Barr & Prevost) (w/o attachment)

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**FINAL REPORT  
GEOTECHNICAL INVESTIGATION  
KENNEDY INTERCHANGE RECONSTRUCTION  
RAMP 42 3RD STREET to I-65 SB  
(S0180, BRIDGE B3RD-1)  
JEFFERSON COUNTY, KENTUCKY**

**For:**

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**April 18, 2007**

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**FINAL REPORT**  
**GEOTECHNICAL INVESTIGATION**  
**KENNEDY INTERCHANGE RECONSTRUCTION – SECTION 1**  
**RAMP 42 3RD STREET to I-65 SB**  
**(S0180, BRIDGE B3RD-1)**  
**JEFFERSON COUNTY, KENTUCKY**  
**FOR**  
**KENTUCKY TRANSPORTATION ASSOCIATES**

**1. LOCATION AND DESCRIPTION**

This report presents the results of our geotechnical investigation for part of the proposed new ramp 42 carrying traffic from the 3<sup>RD</sup> Street on-ramp to Southbound (SB) I-65, north of downtown Louisville, Kentucky. This structure is part of the Kennedy Interchange Reconstruction, in turn a component of the Louisville-Southern Indiana Ohio River Bridges Project. The project area is north of the Louisville central business district in Jefferson County. The bridge begins at the north end of 3<sup>RD</sup> Street and ends at the beginning of Bridge B3RD-8 (Structure S0660). A Site Location Map is presented as Exhibit 1.

Working documents and drawings germane to the Kennedy Interchange Reconstruction are stored on ProjectWise, an internet-based data management system, utilized by Kentucky Transportation Associates' (KTA) team members and subcontractors. Subsequent reference to this system will take the form "ProjectWise 061014" referencing a document available on October 14, 2006. The bridge is being designed using the Load and Resistance Factor Design (LRFD) method as set forth in the American Association of State Highway and Transportation Officials (AASHTO) Publication "LRFD Bridge Design Specifications 4th Edition". Where this document is referenced subsequently, the term 'LRFD BDS' will be used.

According to the plans prepared by WMB, (ProjectWise 070216), the four-span bridge has a total length of 578.09 feet. The bridge section consists of an 8-foot inside shoulder, 15-foot lane width and a 6-foot outside shoulder. The initial part of Ramp 42 is built on an embankment that increases in height from ground level to about 14 feet at Abutment 1 of B3RD-1. The bridge then rises at a 5 percent (%) grade up to its junction with the Great Lawn Structures at Pier 4. The proposed start/finish substructure centerlines are shown in Table 1. The bridge plan is presented in Exhibit 2.

Table 1: Proposed Substructures Stationing<sup>1</sup>

Structure	Substructure	Centerline Station
B3RD-1	Abutment 1	423+81.92
B3RD-1	Pier 1	424+97.25
B3RD-1	Pier 2	426+09.58
B3RD-1	Pier 3	427+79.25
B3RD-1	Pier 4	429+59.92

<sup>(1)</sup> Preferred Layout, ProjectWise 070316.

The purpose of this investigation is to characterize the subsurface conditions of the site, perform geotechnical engineering analyses, and provide recommendations for the design and construction of the foundations of the bridge piers and abutment.

## 2. SITE TOPOGRAPHY AND GEOLOGIC CONDITIONS

### 2.1 General

The proposed ramp alignment crosses both West River Road and Bingham Way. The existing terrain is relatively flat varying in elevation from about 438 – 440 feet mean sea level (msl) through Pier 3 and about 444 feet msl under the final span.

Buried utilities have been mapped in the area as shown in ProjectWise 070216; those relevant to foundation construction appear to include:

- 12-inch combined sewer directly beneath Abutment 1
- 15-inch sanitary sewer directly beneath Pier 2
- 12-inch sanitary sewer beneath possible pile cap for Pier 3
- 12-inch sanitary sewer immediately adjacent and north of possible Pier 4 pile cap

Each of these will be potentially impacted by the installation of deep foundations.

### 2.2 Geology

The interchange system will be constructed in close proximity to the Ohio River, and generally within its left bank flood plain. Subsurface conditions are, therefore, dominated by the morphology of the river valley and more recent depositional history.

In the Louisville area, the Ohio River flows through a broad, relatively flat valley, carved into bedrock during post-glacial periods. Bedrock typically consists of Devonian and Silurian-age limestone, dolomite

and shale; the valley sides southeast of the site area are composed of Sellersburg and Jeffersonville limestone, the base of which occurs at about elevation 480-490 feet msl. The geologic sequence below this is reported as follows:

Table 2: Geologic Sequence – Louisville, Kentucky

<b>Formation</b>	<b>Rock Type</b>	<b>Thickness (ft.)</b>	<b>Base Elevation<sup>(1)</sup> (Ft. – msl)</b>
Jefferson	limestone	18-30	480-490
Louisville	limestone	45-75	405-445
Waldren	shale	9-13	392-436
Laurel	dolomite	50	346-386
Osgood	shale/dolomite	17	329-369
Brassfield	dolomite	10-30	200-359
Drake	dolomite, limestone, shale and mud stone	55-150	149-304

<sup>(1)</sup> Calculated based on mapped contact elevation at base of Jefferson Limestone.

Bedrock in the floor of the buried valley is recorded as being between 334 feet and 337 feet at more than a dozen well sites in the 2-mile stretch of the aquifer upstream from the current I-71/I-64 interchange (Unthank & Nelson, 2006). It is likely that the valley floor continues downstream, with some variation, at about that elevation into the area covered by this investigation. Based on the geological information presented above, the valley floor could be in the Osgood, Brassfield or Drake formations. The composition is therefore uncertain, as these tend to be composed of a variety of rock types, although dominantly of dolomite.

Bedrock is masked in the broad flood plain by Quaternary sediments that are frequently in excess of 100-foot thick. These consist of glacial outwash and river alluvium. Locally, man-made fill is present. The alluvium is generally less than 35 feet in thickness and may consist of clay, sand or gravel. Outwash deposits typically comprise sand and gravel, coarsening with depth and generally extending to bedrock. A geologic map is presented as Exhibit 3.

The subsurface investigation results are consistent with this geologic setting. The typical observed stratigraphy consists of man-made fill (~10 feet) overlying alluvial clay (~20 feet) followed by outwash deposits (sand and gravel mixtures) to the total depth explored (80 – 116 feet).

Groundwater conditions in the flood plain are influenced heavily by the stage of the Ohio River. In the extreme, flood levels up to elevation 450 feet (100-year flood) may be experienced, but lesser floods can be expected annually that are capable of inundating the project area. The river stage is regulated and for much of the year the normal stage for McAlpine Upper Pool is elevation 420 feet. Groundwater levels near the river may be expected at a similar or slightly higher elevation.

## 2.3 Seismicity

The seismic acceleration coefficient ( $A$ ) for the site is derived from contour maps prepared by the U.S. Geological Survey (USGS) (LRFD BDS Figures 3.10.2-1-3) for the design earthquake. The Louisville area coefficient is 6% (corresponding to an acceleration of 0.06 times gravitational acceleration). This places the structure in Seismic Zone 1 ( $A \leq 0.09$ ). There is a 90% probability that this acceleration will not be exceeded during a 50-year period. The return frequency for such an event is about 475 years.

Design specifications for seismic design are currently under review at AASHTO and are expected to undergo major changes, perhaps as soon as 2007 (Imbsen, 2006). The Designer should be aware of the status of these changes as the design moves forward and request ongoing assistance from a geotechnical engineer, as appropriate.

## 3. FIELD INVESTIGATION PROGRAM

Drilling and sampling were performed by Fuller, Mossbarger, Scott and May Engineers, Inc. (FMSM) of Lexington, Kentucky between May 22 and 31, 2006. The subsurface exploration consisted of three structure borings, designated as Borings 4B-248, 4B-249, and 4B-250. Boring 4W-270 was drilled for an adjacent structure and serves to increase the level of detail for both. Borings 4B-248 and 4B-249 were drilled to a depth of 80 feet, while Boring 4B-250 was terminated at a depth of 116.5 feet bgs. Boring 4W-270 was drilled to a depth of 80 feet. The borings were marked in the field by a Qk4 survey crew in accordance with an approved plan.

As-drilled boring locations are shown on Exhibit 2, and summaries of the drilling information are presented in Table 3. The Coordinate Data Submission Form, provided by the Kentucky Transportation Cabinet (KYTC) Geotechnical Branch, is completed and included in Appendix A. The logs of three borings drilled in 1962 for construction of the original interchange (Holes #DD, #E, #EE) were consulted for general stratigraphic correlation.

Table 3: Boring Summary for the Recommended Structure

Sub-Structure	Boring Number	Boring Location <sup>(1)</sup> (Station, Offset)	Surface Elevation (NGDV-Feet)	Bottom of Hole Elevation (Feet)
Abutment 1	4W-270	335+17, 133 RT	445.05	365.05
Abutment 1	4B-248	336+93, 136 RT	443.69	363.69
Pier 1	4B-248	336+93, 136 RT	443.69	363.69
Pier 2	4B-249	338+75, 136 RT	440.58	360.58
Pier 3	4B-249	338+75, 136 RT	440.58	360.58
Pier 4	4B-250	341+95, 101 RT	444.07	327.57

<sup>(1)</sup> Feet along I-64 Centerline.

The borings were drilled by FMSM using either an all-terrain-vehicle-mounted drill rig, Model CME 55, or a truck-mounted CME 85 equipped with 3.25-inch hollow stem augers. The first 6.5 feet of Boring 4W-270 was drilled by vacuum extraction to address concern for sensitive utilities in the area. Water was used to reduce the amount of “blow-back” of the granular materials once the borings advanced below the surficial clay.

Soil sampling was performed using an auto hammer according to the AASHTO T-206 “Standard Method for Penetration Test and Split Barrel Sampling of Soils.” FMSM indicated that the auto hammer efficiency of this hammer system is 80 percent. Undisturbed samples were obtained from the cohesive soil layers according to AASHTO T-207 “Standard Method for Thin-Walled Tube Sampling of Soils.” The soil samples were obtained at 5-foot intervals starting 2 feet below the existing ground surface. Upon completion, borings were backfilled with soil cuttings.

An engineer monitored the drilling activities and maintained field boring logs. The field logs included results of Standard Penetration Tests (SPT) recorded as blows per 6 inches. These data were used to determine the values shown on the Subsurface Data Sheet (Exhibit 4) as N values representing blows per 12 inches. In the field, the soils were classified according to American Society for Testing and Materials (ASTM) D-2488 “Description and Identification of Soils: Visual – Manual Procedure”.

All split-spoon samples collected in the field were placed in sealed, glass jars and transported to Wang Engineering Inc. (WEI’s) geotechnical laboratory in Lombard, Illinois. At the time the field investigation was conducted, WEI was the responsible geotechnical engineering firm for this structure. All field visual classifications were reviewed in the laboratory.

#### 4. LABORATORY TESTING AND RESULTS

Selected soil specimens recovered from split-spoon and Shelby tube samples were tested for natural moisture content, particle size distribution, Atterberg limits, and unconfined compressive strength after

the KYTC approved the laboratory test request. One-dimensional consolidation, and consolidated-undrained triaxial resistance were measured using some of the Boring 4B-248 samples primarily to support analysis of the Abutment 1 foundation. The laboratory tests were performed according to applicable AASHTO or Kentucky Methods.

The test results are shown on the Subsurface Data Sheet (Exhibit 4), and the results of consolidation and triaxial tests are presented in Appendix B. Soils were classified using indices based on ASTM D-2487 "Classification of Soils for Engineering Purposes (Unified Soil Classification System)" and AASHTO M-145 "Classification of Soils and Soil-Aggregate Mixtures for Highway Construction Purposes." The test results were used to establish material properties, and in subsequent engineering analyses to evaluate foundation design and construction alternatives.

#### **4.1 Undisturbed (Shelby) Tube Samples**

Undisturbed (Shelby) tube samples of the clay layers were obtained from all borings. The samples were extruded from the tubes, visually described, and trimmed into 6-inch specimens. Unit weight (dry and moist) and natural moisture contents were determined for each specimen. Selected specimens were subjected to engineering classification, unconfined compressive strength, one-dimensional consolidation and consolidated-undrained triaxial tests.

##### **4.1.1 Engineering Classification Testing**

Classification testing was performed on selected Shelby tube samples. In most cases, one soil classification test was completed per Shelby tube. Soils from the Shelby tube samples were classified mostly as clay (CL), according to the Unified Soil Classification System, and as A-6(15) and A-6(16), based on the AASHTO classification system.

##### **4.1.2 Unconfined Compressive Strength Tests**

The results of unconfined compressive strength tests are presented on the Subsurface Data Sheet (Exhibit 4) and are summarized in Table 4. The undrained shear strength (commonly referred as "undrained cohesion") is one-half of the unconfined compressive strength.

Table 4: Unconfined Compressive Strength Test Results

Boring Number	Location	Sample Interval Depth (ft)	Dry Unit Weight (pcf)	Moist Unit Weight (pcf)	Moisture Content (%)	Unconfined Compressive Strength (tsf)
4B-248	336+93, 136 RT	30 – 32	104.46	128.91	23.41	0.76
4B-249	338+75, 136 RT	20 – 22	103.05	126.64	22.89	1.29
		25 – 27	97.74	121.49	24.29	0.23
4B-250	341+95, 101 RT	15 – 17	96.82	123.92	27.99	0.81
		20 – 22	98.89	125.53	26.94	0.65
		25 – 27	101.49	125.53	23.69	1.84
		30 – 32	127.26	102.10	24.65	0.52

#### 4.1.3 One-Dimensional Consolidation Tests

Two one-dimensional consolidation tests were performed on clay samples from 20 and 30 feet below the ground surface (bgs). The results are summarized in Table 5, as follows:

Table 5: One-Dimensional Consolidation Test Results

Boring/Sample Number	Depth (bgs) (ft)	$C_c^{(1)}$	OCR <sup>(2)</sup>	$C_v^{(3)}$ (ft <sup>2</sup> /day)
4B-248 #4	20-22	0.189	1.0	0.15
4W-270 #7	30-32	0.298	1.0	0.10

<sup>(1)</sup>  $C_c$ -field corrected compression index

<sup>(2)</sup> OCR-over consolidation ratio

<sup>(3)</sup>  $C_v$ -consolidation coefficient

Laboratory data are provided in Appendix B.

#### 4.1.4 CU Triaxial Test

A consolidated-undrained triaxial test was performed on a sample of clay from Boring 4B-248 at 20-27 feet bgs. The effective friction angle ( $\phi'$ ) was 32.1 degrees and the effective cohesion ( $c'$ ) was zero. Laboratory data are provided in Appendix B.

## 4.2 Standard Penetration Test Samples

Standard Penetration Tests (SPT) and split-barrel (commonly known as split-spoon) sampling of soils were performed in granular materials or where drilling resulted in poor Shelby tube recovery. SPT N-

values ranged from 0 (push) to high values typically in the mid-thirties [blows per foot (bpf)]. The SPT N-values and laboratory test results, which include natural moisture contents, fine (silt and clay) contents, and engineering classifications, are presented on the Subsurface Data Sheet (Exhibit 4). SPT values used for calculation of internal friction angles were adjusted in a two-step process. First, the value was modified to account for the high efficiency (80%) hammer used in the test since N- $\phi$  correlations are based on older 60% efficiency hammer systems. Then the values were normalized to a constant effective stress of 1 atmosphere, creating a N1(60) equivalent. These results are shown in Appendix C.

## 5. SUBSURFACE CONDITIONS

Detailed information describing the soil conditions encountered during the subsurface investigation are presented on the Subsurface Data Sheet (Exhibit 4). In general, the borings encountered similar soil conditions. Specifically, below either asphalt pavement or a few inches of black, silty-clay topsoil, there was 8 to 10 feet of fill consisting of silty clay, silty sand and gravel. Non-cohesive fill soils tended to be loose to medium dense and the fine-grained soils soft to medium stiff. Below the fill is soft to hard, brown and gray, lean clay that extends to about 35 feet bgs. The clay is usually underlain by medium-dense sand with gravel, followed by medium-dense to dense, poorly graded, sand and gravel, which extend to the termination depth. The top of sand elevation was plotted at the current and 1962 boring locations. Contouring was used to estimate the clay/sand interface surface configuration, as shown in Table 6.

Table 6: Clay/Sand Interface

Substructure	Top of Sand Surface Elevation
Abutment 1	407.5
Pier 1	409.0
Pier 2	410.5
Pier 3	410.0
Pier 4	409.0

The borings produced samples at discrete locations within the subsurface environment beneath the site, and test data, whether insitu or exsitu, are representative only of those locations. To evaluate the general characteristics of the subsurface units, the data collected during field and laboratory testing were examined by observation, supported by statistical methods.

### 5.1 Man-made Fill

Fill consists of a variety of cohesive and non-cohesive soils. The unconfined compressive strength of near surface clay fill tends to be relatively high, as indicated by penetrometer testing in the field, suggesting that the material has been subjected to some compactive effort. The properties of the fill will have only minor impact on the design of pier foundations, which are usually placed 5 feet or so below existing grade. Fill

properties below about 5 feet are characterized as loose, silty sand for purpose of pier and shaft design, with an estimated effective internal friction angle of 31 degrees ( $^{\circ}$ ), based on the SPT derived values presented in Appendix C.

The abutment design is more sensitive to the presence of fill because of the potential for slope instability, and for settlement with associated downdrag on piling. For assessment of these effects and the design of abutment foundations, the fill is assumed to be soft to medium-stiff, silty clay (Boring 4B-248).

## 5.2 Flood Plain Deposits (Clay)

This stratum is a significant member of the overall soil profile, capping the deeper outwash sands with about 25 feet of near-normally consolidated clay. The material tends to be silty clay, and unconfined compression testing suggests a highly variable strength profile. However, it is possible that this variability is more a reflection of the silt content than the actual strength of the material; silty samples with low cohesion tend to yield low apparent unconfined, compressive strength.

An analysis of the unconfined, compressive strength test results shows no significant correlation of strength with depth but the mean shear strength is similar from boring to boring. The mean shear strength (based on 50% of the unconfined, compressive strength) is calculated to be 871 psf, however the standard deviation is high (534 psf). Using a  $t'$  test produces 90% confidence that the mean is greater than 377 psf. By eliminating two high outliers the standard deviation is reduced and the 90% confidence limit mean increases to 490 psf.

Theoretically, the shear strength of normally consolidated clay should increase in a generally linear fashion with depth, proportional to the effective stress under which it has consolidated. Using this approach, a shear strength profile of 262 psf at a depth of 10 feet (base of the fill) increasing to 814 psf at the base of the clay layer would be predicted. The mean shear strength would be 538 psf, generally consistent with the means of the measured values.

As indicated above, the silt content of the soil is likely to play a large role in determining the unconfined compressive strength. This effect is minimized in the consolidated-undrained triaxial test and in fact, the siltier the sample the higher  $\phi'$  is likely to be. Results of such a test conducted on a sample from Boring 4B-248 and tests performed on generally similar material from other borings yielded the following results:

Table 7: CU Triaxial Test Results

Boring	Classification	Effective Friction Angle	Effective Cohesion
1W-66	A-6(8)	$\phi' = 31.2^{\circ}$	$c' = 190$ psf
2B-106	A-6(15)	$\phi' = 31.4^{\circ}$	$c' = 240$ psf
4B-248	A-6(9)	$\phi' = 32.1^{\circ}$	$c' = 0$ psf

### 5.3 Sands

Sands extend beneath the floodplain clays to the total depth investigated. They comprise a variety of gradations and silt content with silty sands at one end of the spectrum and gravel (less frequently) at the other.

Effective stress parameters were estimated using corrected SPT blow counts – N1(60) values as presented in Appendix C. The values of  $\phi'$  (effective internal friction angle) ranges from a high of  $49^\circ$  to a low of  $30^\circ$  throughout the area investigated. Since these values are all derived from N values, the variability must be examined in that context. For example, the highest  $\phi'$  value is in a gravel (GP-GM) sample as may be expected ( $\phi' = 49^\circ$ ). More surprisingly, the second to highest is a silty sand (SM) sample  $\phi' = 45^\circ$ . This result suggests that the presence of occasional large gravel particles could be having a significant impact on the raw N values, which is then reflected, in the calculated friction angles. Conversely, variations in the water level in the borings change effective stress conditions and may lead to under reporting of N, as appears to happen in the deeper part of Boring 4B-250.

An analysis of the friction angle data set for the four borings showed no significant correlation of  $\phi'$  with depth and indicated that the results as a whole were near-normally distributed. The overall mean value of  $\phi'$  is  $34.6^\circ$ , although this is higher than the individual mean in Boring 4W-270, suggesting that it would be a non-conservative value to use in design. The 90% confidence level mean is  $33.7^\circ$ , the value recommended for use in design.

The base of the outwash deposits was not determined with certainty, but nearby Boring B-247 was terminated at refusal at a depth of 93.6 feet (elevation 345.46 feet), which may be the top of bedrock. Other borings in the area (e.g. Boring 4B-250) were extended to elevations as low as 327.57 feet without encountering refusal. It should be noted that regional geological analysis suggests a valley floor at about elevation 335 feet. However, elevations could be substantially higher depending on proximity to the valley wall, or deeper where there are local irregularities.

### 5.4 Rock

Rock compressive strength testing ( $q_u$ ) was conducted on samples from a nearby boring 4B-252 to support possible design of drilled shafts founded on rock. The rock appears to be dolomite based on its position in the geological column, and samples were tested at elevations 348.5 feet, 344.0 feet and 336.5 feet. The results varied from 8660 psi to 14730 psi with a mean of 11067 psi. Using a  $t'$  test, the mean is estimated to be greater than 5632 psi at a 90% confidence level.

## 5.5 Groundwater

FMSM installed groundwater observation wells in Borings 3B-177 and 3B-190 drilled for other bridges, and provided groundwater fluctuations during an established period. Groundwater levels fluctuated between elevations 420 to 423 feet. Groundwater was encountered in the borings at depths between 27 and 40 feet, corresponding to elevations of 413 and 405 feet. These elevations correspond roughly with the base of the lean, clay layer, which serves as a confining layer. Stabilized readings were not taken after completion of drilling. It should be noted that groundwater levels fluctuate significantly throughout the year with springtime increases that may be measured in tens of feet. The Federal Emergency Management Agency (FEMA) flood map for Louisville shows a 100-year flood elevation of 450 feet msl, which would inundate the bridge site to a depth of about 6 feet.

## 5.6 Selection of Design Parameters

Examination of the soil conditions along the bridge alignment suggests that the westerly end of the project (Boring 4W-270) is underlain by slightly better conditions (less fill and floodplain deposits) than are present elsewhere. Accordingly, for purposes of design, the substructures have been divided into two groups: Abutment 1 (A1) and Piers 1-4 (P1P4).

The soil strength profiles for these two design cases have been generalized in Table 8. Depths are below ground surfaces.

Table 8: Soil Strength Profiles

Profile	Depth (feet)	Soil	
Abutment 1 (A1)	0-7	silty clay (fill)	$\bar{S}_u=500$ psf
	7-29.5	silty clay	$\bar{S}_u=500$ psf
	29.5-116	fine-medium sand	$\bar{\phi}'=33.7^\circ$
Piers 1-4 (P1P4)	0-10	silty sand (fill)	$\bar{\phi}'=31.0^\circ$
	10-36	silty clay	$\bar{S}_u=500$ psf
	36-116	fine-medium sand	$\bar{\phi}'=33.7^\circ$

Deep foundations may encounter bedrock at depths in excess of about 80 feet.

Design of foundations should consider that groundwater levels are likely to be at, or near, normal (elevation ~ 421 feet) during construction, but that seasonal increases will occur such that groundwater levels will be at the ground surface for extended periods. This provides the worst-case scenario for axial resistance analysis of deep foundations.

## 6. SETTLEMENT AND GLOBAL STABILITY EVALUATIONS

The west end of the planned bridge is supported on an abutment behind which there will be approximately 14 feet of new embankment. Settlement of the foundation soils under embankment loads

may create differential movement at the transition from bridge to embankment, and may also create downdrag forces on deep foundation elements. Analyses were conducted to estimate the magnitude of settlement-related design constraints at the abutment.

The upper 29 feet of foundation soil at the abutment is fine grained and much of it (the lower 22 feet) is normally consolidated. The upper 7 feet is man-made fill and the stress history is not known. It is likely that some compaction occurred during placement and that subsequent land use imposed stresses in excess of those that would be expected in normally consolidated material. The deeper soils are sands and are subject to elastic compression under increased load.

Settlement estimates for the fine-grained soils were made using the computer program EMBANK (FHWA 1992) and the elastic settlement of non-cohesive soil was estimated primarily using the Hough method (LRFD BSD), discussed below. Results of these analyses are presented in Table 9.

Table 9: Abutment 1 Settlement Estimates

Stratum	Settlement (inches)	Notes
fill and floodplain silty clay	14.1	fill is normally consolidated
	9.0	fill over-consolidated 1,000 psf
outwash sands	0.6	Hough's method
	1.1	Schmertmann (1970)
	1.2	simple elastic calculation with Boussinesq stress distribution

Clearly, the fine-grained soils are the source of the major portion of the settlement, representing more than 90% of the movement that may be expected, regardless of the stress history of the existing fill. However, the estimated settlement of the sands could be critical in this design because of the potential for downdrag loads on deep foundations. Hence, the use of multiple techniques to assess whether the Hough method produces a conservative result (as it is often believed to) or not. Alternative methods suggest that the compression of the sand, while still small compared to the cohesive soils, could be greater than that predicted by Hough's method. According to LRFD BDS, the frictional resistance of materials in the zone above a point on the pile where settlement of the foundation soils is equal to 0.4 inches or greater must be ignored, and computed as a downdrag load. It is therefore important to locate the depth at which 0.4 inches of settlement occurs as accurately as is practicable. The maximum depths at which 0.4 inches of settlement is calculated to occur using the above three methods are as follows:

- Hough 37 feet
- Schmertmann 73 feet
- Boussinesq 70 feet

The significance of these values is that, first, there is a wide spread in the results and fine-tuning the

analyses is not likely to reduce it much. Second, if the larger values (i.e., greater depths) are to be believed, then deep foundations that rely on skin friction for geotechnical resistance may not be feasible. Even the Hough-derived value of 37 feet will carry a substantial downdrag load/lost capacity penalty. This topic is pursued further in the section on deep foundation design.

The rate of settlement for the clay will be quite slow: 90% of the movement is expected to take several years. The sand will compress quickly with movement largely complete soon after construction of the embankment. This points to perhaps the simplest method for reducing downdrag-related loads and resistance penalties associated with the sand. If the embankment can be constructed before the deep foundations are installed, subsequent settlement associated with the sands can be virtually eliminated and with it the portion of the downdrag and capacity reduction associated with the sands. This may require temporarily building the embankment and then removing a portion of it in order to construct the abutment foundation, but there would be no need for an extended 'preload' period as there would if consolidation of the cohesive soils were the objective.

There is no simple way to mitigate the relatively large total settlement that is anticipated for the embankment and the associated differential movement between the fill and the abutment. Preloading is unlikely to be feasible because of the extended time that would be required. Lightweight fill may be too buoyant during anticipated floods. Wick drains may be a possible means for accelerating settlement but are still unlikely to produce the desired results within an acceptable time frame. Stone columns may be a feasible method of providing more stable embankment support. Alternatively, moving Abutment 1 to the west about 90 feet down the ramp to station 423+00 (i.e., add a span to the structure) would eliminate the need for new fill and solve both the downdrag and settlement problems.

The 12-inch, combined sewer located behind Abutment 1 and under the fill will likely suffer significant vertical movement, the actual magnitude of which will depend on its depth. Relocation of the sewer may be the best option, although moving the abutment, as suggested above, would eliminate the need for mitigating the vertical movement of the sewer.

The embankment height is less than 20 feet and does not trigger the requirement for global stability analysis (KYTC Geotechnical Manual GT-601-5); it is considered stable.

## **7. FOUNDATION ANALYSES AND RECOMMENDATIONS**

### **7.1 General**

Two alternate four-span structures are under consideration for the final design of Bridge B3RD-1, but foundation selection is unlikely to be affected by this choice. One end of the structure connects to an elevated span of Bridge 3RD-8, so only one abutment is required near the ground level terminus of Ramp 42. The bridge plan for the recommended structure is presented in Exhibit 2.

The presence of a substantial thickness of compressible clay at the site effectively eliminates consideration of shallow foundations for the abutment or the piers because of concern for excessive settlement. Of the deep foundation alternatives, driven steel friction H-piles would normally be the most likely choice. However, the need for the abutment foundations to resist quite severe downdrag loads requires that the possibility of using end bearing H-piles be considered too, as discussed below.

In this section of the report the axial geotechnical resistance of friction H-piles is addressed first. This sets the stage for discussion of pile load conditions, especially the impact of downdrag loads on the abutment foundations caused by the construction of the approach ramp embankment. Possible means for mitigation of these conditions are then presented, and the process is repeated for a drilled shaft alternative design. Three sizes of friction H-piles (12x53, 14x73, and 14x89) were analyzed for static and dynamic axial-bearing capacity, for pullout axial capacity and for pile drivability. Geotechnical resistances of four diameters of drilled shafts (30", 36", 42" and 48") were estimated.

The LRFD design method requires independent consideration of the reliability of loads and resistances, and the assessment of structure performance under various limit states (e.g., service, strength). For geotechnical resistance [or 'load bearing capacity' as it is known under the Allowable Stress Design (ASD) system], this is accomplished through the use of resistance factors that are applied to the nominal (or ultimate) resistance estimates to yield 'factored resistance'. These should not be confused with 'allowable load-bearing capacities'. They must only be used by the Designer in the context of an LRFD design and with appropriately factored loads.

### **7.2 Pile Axial Capacity**

#### **7.2.1 Friction H-Piles**

Driven pile axial capacity evaluations were performed using the Tomlinson method for clays and Nordlund/Thurman method for sands to derive the nominal side resistance and nominal end bearing. Both methods are incorporated in the computer software APILE plus 4.0 (ENSOFT Inc., 2004). The total, nominal, geotechnical axial resistance at a particular depth is the sum of the nominal side resistance and nominal end bearing at that depth. The total, factored, geotechnical axial compressive resistance for static analysis was derived by multiplying the corresponding total nominal resistance by a resistance factor of 0.35 for clays and 0.45 for sands. The total, factored, geotechnical axial compressive resistance

for dynamic analysis was derived by multiplying the corresponding nominal resistance by a unique resistance factor of 0.65 for both clays and sands. The total, factored, geotechnical axial uplift resistance was obtained from the nominal side resistance multiplied by a resistance factor of 0.25 for clays and 0.35 for sands. These factors are specified in LRFD BDS Table 10.5.5.2.3-1.

Axial capacity evaluations for three selected sizes of steel H-piles (12x53, 14x73, and 14x89) were performed for the project based on the assessment of soil conditions along the bridge alignment. The results of pile evaluations for each foot of depth below the bottom of the pile cap are presented in the tables included in Appendix D. It is to be noted that axial compressive resistance values in Appendix D are based on the assumption that positive skin friction develops along the entire pile length. Pile capacities have been calculated for Piers 1-4 as a group based on generalized soil conditions that were described in Section 5.

According to KYTC policy (August 18, 2006), maximum factored geotechnical resistance is limited as follows for 50 kips per square inch (ksi) steel H-piles:

HP 12x53 - 200 kips (100 tons)  
HP 14x73 - 280 kips (140 tons)  
HP 14x89 - 340 kips (170 tons)

Table 10 summarizes the required pile lengths and pile tip elevations as estimated for the maximum-factored, geotechnical resistance indicated above for each of the three pile sizes. Should more (or less) capacity be required, refer to the pile capacity tables in Appendix D. H-pile lengths at each substructure location should be adjusted accordingly, based on the actual pile driving conditions encountered in the field and the designed pile cap elevations.

It should be noted that nearby drilling encountered refusal at an elevation above some of the predicted pile tip elevations presented in Table 10 (elevation 345.46 feet); regional analysis of the bedrock conditions suggest that the valley floor may be encountered at about elevation 335 feet. Therefore the possibility of meeting refusal while driving piles to the stated depths cannot be ruled out. Where this occurs, it may be assumed that the design geotechnical axial compressive resistance of the pile has been attained.

Table 10: Estimated H-Pile Lengths for Maximum Factored Geotechnical Resistance,  
Based on Static Analysis Method.

Substructure	Pile Size	Maximum Total Factored Geotechnical Axial Compressive Resistance		Estimated Pile Length <sup>(1)</sup> (feet)	Estimated Pile Tip Elevation (feet)	Total Factored Geotechnical Axial Uplift Resistance <sup>(2)</sup>	
		(kips)	(tons)			(kips)	(tons)
Abutment 1	HP 12x53	200	100	see	discussion	below	
	HP 14x73	280	140				
	HP 14x89	340	170				
Piers 1-4	HP 12x53	200	100	85	354	148	74
	HP 14x73	280	140	95	344	208	104
	HP 14x89	340	170	108	331	252	126

<sup>(1)</sup> Below the bottom of the pile cap.

<sup>(2)</sup> Estimated uplift resistance for the corresponding estimated pile length.

## 7.2.2 Abutment Foundations

### *Standard Driven H-Piles*

Absent downdrag loading, a friction H-pile foundation design for the abutment would be generally similar to that shown above for Piers 1-4. The piles would be a little shorter because the soil profile is slightly more favorable, but for the purpose of this discussion it is assumed that they are the same.

As described in Section 6, the soils beneath the abutment are compressible and subject to an increase in effective stress caused by the placement of 14 feet of fill. The resulting movement is likely to result in additional loading to deep foundations and these downdrag loads must be accommodated in the design of substructure components.

LRFD BDS requires that the downdrag load be calculated as the frictional resistance existing above a point on the pile where the soil settlement is equal to or greater than 0.4 inches. Further, the geotechnical resistance of this part of the pile is to be ignored. The approach discussed in Section 6 indicates that the 0.4-inch settlement point is estimated to occur at a depth between 37 and 73 feet.

Analysis of the 85-foot long HP 12x53 pile described in Table 10 using the neutral plane method, suggests that the neutral plane is 50 feet below the ground surface – somewhere near the mid-point of depths predicted by the settlement analyses. The neutral plane is located at a depth where the downdrag load approaches zero, and geotechnical resistance can begin to take effect. The analysis assumes that the maximum factored resistance of 100 tons is imposed, 90% of which is dead load and 10% live. Using a

load factor of 1.25, the actual dead load would be 72 tons (or 144 kips).

With a neutral plane at a depth of 50 feet, the nominal geotechnical resistance for the pile between 50 feet and 85 feet is 150 tons, and the downdrag is 90 tons. The factored geotechnical resistance (using 0.45 for sand) is 67.5 tons, considerably less than the unfactored downdrag. Analysis of the HP 14x89 pile produces similar results: a factored geotechnical resistance of 110 tons that is less than the unfactored downdrag of 143 tons.

It must therefore be concluded that, based on LRFD BDS requirements for accommodating downdrag, standard friction H-piles cannot be used to support the abutment. It should be noted that if the abutments are not supported on friction H-piles, their possible use in the pier foundations must be reviewed carefully to ensure compatibility with the system that is ultimately selected for the abutment.

#### *Alternative Foundation Systems*

Relocation As indicated in Section 6.0, the most effective mitigation from a geotechnical standpoint is to relocate the abutment about 90 feet to the west in order to eliminate the need for fill. This in turn would eliminate the downdrag forces thereby making feasible the use of standard friction H-piles of a similar size and length as those used for the piers.

Ground Improvement Another approach to reducing settlement is to improve the soil conditions beneath the fill. This could be accomplished by using stone columns, controlled modulus columns, or one of the other hybrid systems that essentially provides a dense grid of support throughout the foundation footprint. The columns work by effectively supporting the embankment on the underlying sands, thereby relieving the stress in the compressible clays. These techniques are increasing in popularity as their effectiveness is demonstrated on similar projects. Specialty contractors typically specify and build the systems based on their proprietary designs and equipment. Lower costs may be achieved if there are multiple applications for this technique throughout the interchange and economies of scale can be realized. Through use of this technology, the need for a deep driven foundation might even be eliminated.

Precompression In Section 6, the possibility of reducing downdrag by accelerating elastic compression of the non-cohesive soils was raised. This involves construction of the embankment in advance of deep foundation installation. In this scenario, the outwash sands become the equivalent of an incompressible layer, and the neutral plane can be assumed (for simplicity) to be at the base of the overlying compressible soil. As a result, two improvements in pile performance can be anticipated. First, the length of pile subject to downdrag and for which no geotechnical resistance can be claimed is reduced, and second the magnitude of the downdrag itself is reduced. The results are illustrated in the following table.

Table 11: Downdrag Effects – Precompression of Sands

Pile Size	Table 10 Maximum Factored Geotechnical Resistance (kips)	Length (Ft.)	Downdrag Load (factored) (kips)	New Factored Geotechnical Resistance (kips)	“Capacity” Reduction <sup>(1)</sup> (%)
HP 12x53	200	85	69	113	43.5
HP 14x73	280	95	81	182	35.0
HP 14x89	340	108	83	236	30.6

<sup>(1)</sup> Considers downdrag load and reduction in frictional resistance.

These values are not intended to be used in design, but are illustrative of the reduced impact of downdrag after precompressing the sands. The % reduction is the impact on overall load bearing capacity caused by downdrag effects. Note that the reduction without precompression is more than 100% as discussed above, so this represents a significant improvement. On this basis, it appears that driven friction H-piles are a feasible foundation type subject to appropriate construction sequencing and compensation for the reduced efficiency of each pile. If this approach is adopted, individual piles may be designed as described below.

Static Analysis Method

$$RC = (RF_z - RDD + RT_z) (0.45)$$

and

$$RUL = (RULF_z - RUL_{DD})$$

where RC is the maximum factored geotechnical compressive resistance for a pile of length z ft., (kips), RUL is the maximum factored geotechnical uplift resistance for the same pile,  $RF_z$  is the nominal side (friction) resistance at depth z from the appropriate Appendix D Table, RDD is the nominal downdrag from Table 12 below (kips),  $RT_z$  is the nominal tip resistance at depth z from Appendix D,  $RULF_z$  is the factored uplift resistance at depth z feet from Appendix D,  $RUL_{DD}$  is the factored uplift reduction due to downdrag\* from Table 12, and 0.45 is the axial compressive resistance factor for sand. [\*Note that this reduction is added back, in part at least, as a factored load in the downward direction during calculation of uplift resistance.]

Table 12: Downdrag Parameters

<b>Pile Size</b>	<b>Nominal Downdrag RDD (kips)</b>	<b>Factored Uplift Reduction RUL<sub>DD</sub> (kips)</b>
HP 12x53	49	13
HP 14x73	58	15
HP 14x89	59	17

*Dynamic Analysis Method*

Compressive resistance computed using the dynamic analysis method produces a higher level of confidence in the predicted capacity, but must still be discounted for the effect of downdrag. Values presented in Appendix D may be adjusted as follows:

$$RC_{DYN} = (R_{DYNz} - RDD_{DYN})$$

where  $R_{DYN}$  is the maximum factored axial geotechnical resistance (dynamic analysis method) for a pile of length  $z$  ft.,  $R_{DYNz}$  is total factored geotechnical axial compressive resistance (dynamic analysis method) at depth  $z$  from appropriate Appendix D table, and  $RDD_{DYN}$  is the factored downdrag from Table 13 below.

Table 13: Dynamic Method Downdrag Parameters

<b>Pile Size</b>	<b>Factored Downdrag RDD<sub>DYN</sub> (kips)</b>	<b>Unfactored Downdrag Load (kips)</b>
HP 12x53	37	57
HP 14x73	44	68
HP 14x89	46	71

The unfactored downdrag loads should be appropriately factored and added to other loads applied to the pile.

Bituminous Coatings In the class of solution that deals with the drag load, there are two approaches to minimizing its impact. First, this load can be significantly reduced by coating the pile with bitumen in the downdrag zone. This has been known to reduce downdrag by anywhere from 50% to 90% in tests. Bitumen creates a low-shear strength zone that is estimated to be on the order of 200 psf (FHWA, 1998). The clays primarily responsible for creating the shallow downdrag forces have an average shear strength of 500 psf. Downdrag would therefore be effectively reduced by about 60%. The reduction in the sands

theoretically could be greater, but it may be difficult to guarantee the integrity of the bitumen coating during driving through sand; it would be prudent to just accept the more conservative reduction value in the clay. However, there is still a large loss of geotechnical resistance because only that below the 0.4-inch soil settlement mark (or the neutral plane) may be considered. The net result is estimated to be, for example, a 108-foot long HP-14x89 with a factored geotechnical resistance of 110 tons and an unfactored downdrag of 57 tons. With a load factor of 1.25 this becomes an offset of 71 tons leaving fewer than 40 tons of 'useful' capacity in a pile that, as shown in Table 10, started life with 170 tons of factored axial resistance. If precompression of the sands is accomplished, as described above, 60% reduction of the remaining downdrag forces would substantially eliminate this capacity reduction.

Founding on Bedrock The second approach to downdrag mitigation involves extending the piles to bedrock. If as suspected, the bedrock horizon is at about 335 feet, it may be possible to drive the piles to bedrock and eliminate the reduction of capacity caused by the '0.4-inch settlement' rule. In other words, the example above would still retain a 170-ton factored resistance because support is not derived from frictional resistance above or below the 0.4-inch settlement mark. The downdrag magnitude would not be affected; if the pile were untreated, the downdrag would be on the order of 50 tons (unfactored). If bitumen coating were to be used, the downdrag would likely be on the order of 20 tons (unfactored) leaving more than 140 tons of useful resistance. If the abutment piles are to be driven to bedrock, consideration also should be given to driving all the piles to bedrock (piers as well) where differential vertical movement might be unacceptable.

### 7.2.3 Drilled Shafts

As an alternate to H-pile foundation support, a drilled shaft option was analyzed. Axial capacity evaluations were performed using the alpha-method for clays and beta-method for sands to derive the nominal side resistance, and the method developed by O'Neill and Reese (FHWA, 1999) was used to derive the nominal end bearing capacity. These analyses result in a shaft that derives the bulk of its support from frictional resistance in the soil column.

The following factors were applied to the nominal resistance in order to obtain the factored resistance: 0.45 for side resistance in clays, 0.40 for end bearing in clays, 0.55 for side resistance in sands, and 0.50 for end bearing in sands. The total factored geotechnical axial compressive resistance at a particular depth is the sum of the factored side resistance and factored end bearing at the corresponding depth. The total factored geotechnical axial uplift resistance was obtained from the nominal side resistance multiplied by a resistance factor of 0.35 for clays and 0.45 for sands (LRFD BDS Table 10.5.5.2.4-1). Where the upper 5 feet of a shaft was surrounded by cohesive soils, no frictional resistance was assumed to exist in this zone (FHWA, 1999).

Axial capacity evaluations for four drilled shaft diameters (30", 36", 42", and 48") were conducted for the Pier 1-4 soil profile. The results are presented in the tables included in Appendix E. It is to be noted that

axial compressive resistance values in Appendix E are based on the assumption that positive skin friction develops along the entire shaft length unless the upper 5 feet of the shaft is in cohesive soil.

Design of a shaft-supported foundation for Abutment 1 involves consideration of embankment-induced downdrag, as was the case for the driven pile designs discussed above. For the purpose of this assessment, it is assumed that the geotechnical characteristics of the subsurface profile at the abutment are sufficiently similar to those at the piers for the same shaft design to be applicable (i.e. as presented in Appendix E).

A 30-inch diameter shaft is evaluated to determine the likely impact of downdrag. A 114-foot deep shaft (assumed to be founded in soil, not on bedrock) has a factored, geotechnical resistance of a little over 1200 kips if there is no downdrag effect. This means that the maximum factored load cannot exceed 1200 kips of which 90% is assumed to be dead load. If the load factor is 1.25, then the actual dead load is 864 kips.

Constructing a neutral plane diagram shows the plane to be 65 feet below ground surface. The associated downdrag is 320 kips (unfactored) and the nominal geotechnical resistance has now been reduced to 700 kips, or 316 kips factored. The shaft is therefore so inefficient that its use cannot be recommended for support of the structure. An analysis with similar results can be made for the other shaft diameters.

Solutions of the same general classes as for the driven pile system can be implemented, i.e., relocation of the abutment, ground improvement and end bearing on bedrock. Bituminous coating is not feasible, but other methods for creating a low-shear strength zone may be possible. Of these options, however, only end bearing on bedrock requires further elaboration.

The axial resistance of a shaft founded on (or socketed into) bedrock is a function of the compressive strength and structure of the rock formation providing support. In this case the rock type is not known with certainty. At nearby Boring 4B-252, was logged starting at an elevation of 354.7 feet msl (RQD 60 – 70%). Another boring in the area, (4B-247), encountered refusal at 345.5 feet msl. Near Pier 4, however, Boring 4B-250 was drilled to 327.6 feet msl without encountering refusal. Based on these elevations, the most likely rock type to be encountered would be dolomite.

Assuming a compressive strength of 5600 psi, the nominal geotechnical resistance of a 30-inch diameter shaft in a 45-inch deep rock socket will be about 3100 kips. With a resistance factor of 0.50, the factored resistance becomes 1550 kips. As indicated above, downdrag on this size of shaft is estimated to be on the order of 320 kips (unfactored), or 400 kips factored. This demonstrates the geotechnical feasibility of designing a reasonably efficient end-bearing shaft. The geotechnical resistance of shafts of various diameters is shown below, along with a general indication of the downdrag to be considered in design. Final values can only be provided once actual dead loads have been allocated to the shafts.

Table 14: Shaft-Resistance and Downdrag  
 (all in kips except shaft diameter)

Shaft Diameter (inches)	Nominal End Bearing Resistance	Unfactored Downdrag (no precompression)	Unfactored Downdrag (with precompression)	Axial Factored End Bearing Resistance
30	3088	282	61	1544
36	4445	339	74	2222
42	6050	395	86	3025
48	7903	452	98	3452

(Subject to confirmation that q

### 7.3 Pile Driveability

A driveability assessment was made to provide guidance on the size of hammer likely to be required for installation of H-piles to the specified resistances without over-stressing the pile on the one hand, or using excessive blow counts, on the other.

For purposes of the analysis, it was assumed that the Contractor will be installing piles with maximum-factored geotechnical resistance, as shown in Table 10. The soil profile representing Abutment 1 and Piers 1-4 was selected for analysis. The Delmag family of hammers was selected as representative of diesel hammer performance for the applicable energy levels.

Analyses were conducted using Wave Equation software (GRLWEAP) developed by Pile Dynamics, Inc. for 12x53, 14x73 and 14x89 steel H-Piles and using hammers with a range of driving energy from 20 ft-kips to 80 ft-kips.

The soils contributing to the driving resistance include the upper clay layer and the underlying sand layers. In order to account for soil remolding, pore water pressure increase and corresponding shear strength decrease, the following skin friction reduction percentages proposed by KYTC were used to calculate driving resistance: 50% for clay and 25% for sand. The driving resistances corresponding to the pile lengths in Table 10 can be found in Appendix D. For each pile section/hammer combination, the relationship between static soil resistance (kips) and dynamic driving resistance [blows per foot (bpf)] was developed and plotted graphically.

The hammer energy to adequately drive the piles without delivering excessive blow counts or overstressing the piles has been set on criteria established by FHWA (Soil and Foundation Workshop Manual) as blow counts between 30 and 144 blows per foot and maximum stress in steel less than 0.9 times the yield (45 ksi for Grade 50 steel). The results are shown in Exhibit 5. It appears that the 60 ft-kip and 80 ft-kip hammers are capable of driving all three sections to the maximum factored axial resistance without excessive blow counts and /or stress to the pile. The 60 ft-kip hammer appears suitable

for 12x53 and 14x73 sections, but a larger hammer (at least 80 ft-kip) should be considered for 14x89 section H-piles.

The Designer may use the driving resistance table in Appendix D and Exhibit 5 to correlate minimum driving resistance required to achieve a desired factored geotechnical axial resistance for other pile lengths, as needed.

It should be noted that the information presented in Exhibit 5 is general in nature, and is not intended to replace independent analyses by pile driving contractors using actual hammer and driving accessory combinations. Further, the results of wave equation analyses should be calibrated based on dynamic field testing prior to production driving and reliance on the predictions of static axial load capacity.

#### **7.4 Lateral Squeeze**

Bridge abutments supported on piles driven through soft compressible cohesive soils may, under some circumstances, tilt because of lateral movement of the foundation soil associated with settlement strains. This phenomenon, known as lateral squeeze, is accentuated if there is a significant unbalanced load behind the abutment. The analysis of this potential condition is made using two rules of thumb for purposes of preliminary foundation evaluation.

First, lateral squeeze is not likely to occur if the vertical pressure exerted by the embankment behind the abutment is less than three times the undrained shear strength. At this structure, the embankment load is estimated to be 1820 psf. This suggests that if the foundation soil shear strength is less than 600 psf, lateral squeeze should be considered in design. In fact, the design shear strength for pile and shaft design is 500 psf. This is considered to be a conservatively low mean value but should, in any case, be cause for concern.

The magnitude of lateral squeeze displacement effects is assessed by the second rule of thumb, which suggests that horizontal movement may be as much as 0.25 times the vertical settlement. This could amount to 2 – 3 inches, with the top of the abutment tending to move away from the embankment.

There are various solutions to this problem. First, lateral squeeze is unlikely to occur once the settlement of the embankment is substantially complete, at which time the foundations could be installed. This may not be a remedy that can be employed here because of timing issues. Eliminating the problem through the use of lightweight fill, whereby the embankment load is reduced below 1500 psf, is likely to be feasible, provided the fill is still sufficiently dense to resist buoyancy forces. Relocation of the abutment, as discussed earlier, provides a positive and guaranteed solution.

Engineering solutions that cope with the problem include provision of large expansion shoes that can accommodate increased movement, and the use of steel H-piles (as opposed to concrete) that have high

tensile strength in flexure. The latter point merely assists in preserving the load bearing ability of the foundation once movement has occurred.

## **7.5 Lateral Load Analysis Parameters**

Deep foundation elements subjected to horizontal loads should be analyzed for maximum bending moments and lateral deflections. The required lateral load capacity can be obtained by increasing the diameter or the embedment depth of the foundation element. The Site Specific Idealized Soil Profile and corresponding recommended lateral soil modulus, and soil strain to be used to analyze the laterally loaded pile by the p-y curve method are shown in Appendix F. The p-y curve method for laterally loaded pile analyses is routinely performed using either the L-Pile Plus or Com 624 computer software.

## **7.6 Seismic Load Evaluation and Liquefaction Potential**

The soil profile at the site corresponds most closely to Type 3 (LRFD BDS – Table 3.10.5.1-1) resulting in a Site Coefficient (S) of 1.5. (A profile with soft to medium stiff clays and sands characterized by 30 feet or more of soft to medium stiff clays with or without intervening layers of sand or other cohesionless soils.)

Liquefaction of saturated, loose cohesionless soil may occur in response to vibration such as occurs during earthquakes. The attempted movement of individual soil particles is constrained by interstitial water, the pressure of which rises for the duration of the shaking. This increase in pore pressure corresponds to a decrease in effective stress and a reduction of associated strength parameters critical to foundation stability. Because saturated, cohesionless soils are present beneath the B3RD-1 bridge site and USGS has established seismic design parameters for the area, a preliminary assessment of liquefaction potential was made.

The soils below a depth of about 35 feet bgs are predominantly silty sands, sands and gravelly sand mixtures. At this depth the soils are permanently saturated. The shallow surficial fill in the area is not dominantly cohesionless, but pockets of silty sand are present (e.g., Boring 4B-250). These soils are typically not saturated, but may become so during the annual increase in stage of the Ohio River. Accordingly, an assessment was made considering the soils at a depth of 6 feet with a groundwater table at the surface, and at 35 feet, which is below the permanent water table.

The analysis was conducted in accordance with the methodology proposed by Idriss and Boulanger (2004). In this method the cyclic shear ratio (CSR) is calculated and compared with the cyclic resistance ratio (CRR) computed based on corrected SPT 'N' values. The process is simplified by the use of graphical solutions to the underlying equations.

Results of the assessment are presented in Exhibit 6, and show that the potential for liquefaction of the deep soils is remote. While that of the shallow fill material is higher, it too is very low.

As indicated in Section 2.2, seismic design guidelines are currently under review by AASHTO; this section should be updated accordingly, if changes occur.

## **8. SUMMARY OF FOUNDATION RECOMMENDATIONS**

### **8.1 General**

1. The bridge foundation design is complicated by anticipated settlement of the proposed approach embankment that is expected to be on the order of 1 foot. This creates an interface problem where the embankment and abutment meet; differential settlement at this location should generally be kept to less than 1 inch. Settlement of the foundation soils also creates the potential for significant downdrag loads on deep foundation components.

Mitigation of the differential settlement problem can be accomplished by relocation of the abutment about 90 feet west to eliminate the need for embankment fill. Alternatively, ground improvement in the form of stone columns or similar proprietary systems (e.g., constant modulus columns) could be undertaken in order to substantially reduce settlement and associated downdrag loads.

Downdrag loads on both driven friction H-piles and drilled friction shafts are potentially severe and can render those foundation types infeasible unless special precautions are taken. These include:

- Precompression of the sands through early construction of the embankment
- The use of bitumen coating to reduce friction forces in the downdrag zone
- Founding support members on bedrock

These precautions each provide a benefit, as described in earlier sections, and can be employed either individually or collectively.

2. No embankments are planned in the vicinity of the piers and so settlement induced down drag is not a design consideration, however compatibility with the foundation at the abutment should be considered particularly if it is decided to found the support members on bedrock.

3. Based on the geotechnical information developed during this investigation, end-bearing H-piles appear to represent the best candidate for supporting the proposed bridge substructures. Shorter friction H-piles may be used for the piers, provided this is compatible with the abutment foundations.

4. Foundation analyses were performed according to the provisions of LRFD BDS.

## 8.2 Steel H-piles

1. Three sizes of Grade 50 steel H-friction piles (12x53, 14x73, 14x89) were analyzed for static and dynamic axial bearing capacity, for pullout axial capacity and for pile drivability.
2. Pile capacities were developed for two soil profiles, one representative of conditions at Abutment 1, and the other, at the piers. Axial capacity estimates for single, steel H-piles are provided in Appendix D. Upon determination of the final H-pile locations, arrangement and loads, the designer should use the capacity estimates to determine the H-pile size and length. The factored geotechnical axial resistance estimates provided in Appendix D are based on the LRFD BDS resistance factors.
3. If load testing and/or dynamic analysis of driven piles in soil is conducted, the LRFD resistance factors used to determine the factored axial capacity for design purposes may be revised as outlined in LRFD BDS Table 10.5.5.2.3-1, based on site variability and the number and type of tests performed. The Designer should note that lateral capacity requirements will need to be revisited if the shaft lengths are revised based on load testing and/or dynamic analysis.
4. All pile axial capacities presented in Appendix D are for single piles. In addition to applying appropriate resistance factors, individual capacities for piles in-group configurations may be further reduced depending upon soil type, bearing condition of the pile cap, or center-to-center spacing as recommended in LRFD BDS. The following criteria should be observed:

Table 15: Group Efficiency Factors

CTC Spacing	Group Efficiency Factor		
	Cohesive Soils		Cohesionless Soils (Cap in or not in firm Contact with Ground)
	Cap not in firm Contact with Ground	Cap in firm Contact with Ground	
6B	1.00	1.00	1.00
2.5B	0.65	1.00	1.00

The notation "B" is the shaft diameter and the percent reduction may be linearly interpolated between the values and spacing provided.

5. LRFD BDS include a resistance factor for horizontal geotechnical resistance of a single pile or pile group of 1.0 for lateral capacity analyses. Appendix F provides the recommended soil parameters for use in lateral load pile analyses.
6. For the initial group of piles installed, the piles should be left for a minimum of one day so excessive

pore pressure caused by driving operations can dissipate and the soils can set-up. After the one-day waiting period, the piles should be re-struck to see if the required bearing capacity was achieved. If the set-up effect is not significant, subsequent groups of piles will not require re-striking.

7. Hammer selection was based on the ultimate driving resistance that 12x53, 14x73 and 14x89 steel H-piles would experience during the installation process. The results of these calculations are presented in Exhibit 5.

8. Upon selecting the pile size and length required to support the applied loads, the Designer should select the appropriate hammer required to drive the piles to the specified depths from Exhibit 5. The Designer should place a note on the drawings that states: "A hammer system capable of delivering a minimum energy of \_\_\_ foot-kips will be necessary to drive the piles to the maximum total factored geotechnical axial resistance without encountering excessive blow counts and over-stressing the piles. The Contractor should submit appropriate pile driving systems to the Kentucky Transportation Cabinet for approval prior to the installation of the first pile. Approval of the pile driving system by the Engineer will be subject to satisfactory field performance of the pile driving procedures."

9. Drivability studies were performed assuming continuous driving. If interruptions in driving individual piles should occur, difficulties in continuing the installation process will likely occur due to the pile "set-up" characteristics.

10. LRFD BDS (Section 6.5.4.2) recommends the following resistance factors for determining the structural capacity of steel H-piles:

Table 16: Resistance Factors – H-Pile Structural Capacity

Loading Condition	Resistance	
	Piles Subjected to Damage From Severe Driving Conditions <sup>(1)</sup>	Good Driving Conditions
Axial Resistance In Compression	$\phi_c=0.50$	$\phi_c=0.60$
Combined Axial and Flexural Resistance	N/A	$\phi_c=0.70$ $\phi_f=1.00$

<sup>(1)</sup>Apply these values only to the section of the pile likely to be damaged during driving (LRFD BDS Section 6.15.2).

### 8.3 Drilled Shafts

1. This option has been evaluated for comparison with H-pile supported foundations.

2. Axial capacity estimates for drilled shafts are provided in Appendix E and were derived utilizing the LRFD resistance factors in LRFD BDS. The capacities presented are for single shafts and do not include group reduction factors. Upon determination of the final shaft locations and shaft loads, the Designer should use the capacity estimates to determine shaft size and length for each shaft.
3. If load testing of drilled shafts in soil is conducted, the LRFD resistance factors used to determine the factored axial capacity for design purposes may be revised as outlined in LRFD BDS Table 10.5.5.2.4-1.
4. All shaft capacities presented in Appendix E are for single shafts. In addition to applying appropriate resistance factors, individual capacities for shafts in group configurations should be further reduced depending upon center-to-center spacing as specified in the LRFD BDS. The following criteria should be observed:

Table 17: Drilled Shaft Group Efficiency Factors

CTC Spacing	Group Efficiency Factor for Cohesive Soils	Group Efficiency Factor for Non-Cohesive Soils
6B	1.00	N/A
4B	0.80	1.00
2.5B	0.65	0.65

The notation "B" is the shaft diameter and the percent reduction may be linearly interpolated between the values and spacing provided.

5. LRFD BDS calls for a resistance factor for horizontal geotechnical resistance of a single shaft or shaft group of 1.0 for lateral capacity analyses. Appendix F provides the recommended soil parameters for use in lateral load analyses.
6. The Contractor should embed the drilled shafts to the plan tip elevation or to an elevation as directed by the Engineer.
7. If a temporary casing for drilled shafts is used during construction, the Contractor should either wait until concrete has been placed for the entire length of the shaft before pulling the casing, or the level of the concrete being placed should be maintained several feet above the hydrostatic head as the casing is retrieved. These measures should be implemented by the Contractor to reduce the likelihood of soils collapsing into the shaft excavation and detrimentally affecting the structural integrity of the drilled shafts.

8. It is recommended that Class A Modified concrete, which is in accordance with the current KYTC Special Note for Drilled Shafts, be used in construction of the drilled shaft. The concrete should also exhibit good workability, i.e., high slump. Once an excavation is complete and the steel reinforcing cage has been placed, concrete should be tremmied to the bottom of the shaft and should replace/displace any water or slurry remaining after drilling operations.

9. If drilling slurry is to be used during shaft installations, the slurry should be capable of suspending the soil particles encountered and not leave a thick coating of slurry, or "mud", on the excavation sides or bottom. In accordance with the current "Special Note for Drilled Shafts", the Contractor shall submit a detailed plan for its use and disposal along with a drilled shaft installation plan to the Geotechnical Branch of the KYTC for approval prior to implementation. The Contractor shall supply all equipment and construction techniques involving slurry that are necessary to maintain environmental standards.

10. Drilled Shaft Integrity Testing will be required for each drilled shaft. An appropriate number of Crosshole Sonic Logging (CSL) access tubes (approximately 3), consisting of 2-inch nominal diameter schedule 40 steel pipes, will be required. These tubes should be shown on the drilled shaft details with the following note on the Drilled Shaft Detail Sheet:

*Perform non-destructive Drilled Shaft Integrity Testing on the Drilled Shafts using Crosshole Sonic Logging (CSL) in accordance with the "Special Note for Non-Destructive Testing of Drilled Shafts". The Department will pay for this testing and associated cost at the contract unit bid price for "CSL Testing". This includes CSL Testing Mobilization and CSL Testing. The access tubes are incidental to the shaft.*

11. Unless otherwise specified, all construction methods and materials used for drilled shaft installations shall be in accordance with the current "Special Note for Drilled Shafts".

## 9. QUALIFICATIONS

This investigation was performed in accordance with accepted geotechnical engineering practice for the purpose of determining bridge foundation recommendations only. Verification of the subsurface conditions for purposes of determining contamination, difficulty of excavation, the effect of excavation on slope stability or existing structures, and trafficability is beyond the scope of this study. The analyses and recommendations submitted in this report are based upon the data obtained from the borings drilled at the locations shown on Exhibit 2, and on the Subsurface Data Sheet (Exhibit 4). This report does not reflect any variations that may occur between the borings or elsewhere on the site, and variations whose nature and extent may not become evident until a later stage of construction. In the event that any changes in the nature, design or location of the proposed bridges are made, the conclusions and recommendations contained in this report should not be considered valid until the changes are reviewed,

Kennedy Interchange  
S0180, B3RD-1  
April 18, 2007

and the conclusions and recommendations in this report have been modified or verified in writing by a geotechnical engineer.

It has been a pleasure to be of service to Kentucky Transportation Associates in performing this geotechnical investigation for Structure 0180, Bridge B3RD-1.

Respectfully Submitted,

**BARR & PREVOST**

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Gary R. Simmons, P.E.  
Principal

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Stuart Edwards, P.E.  
Senior Project Manager

## ***REFERENCES***

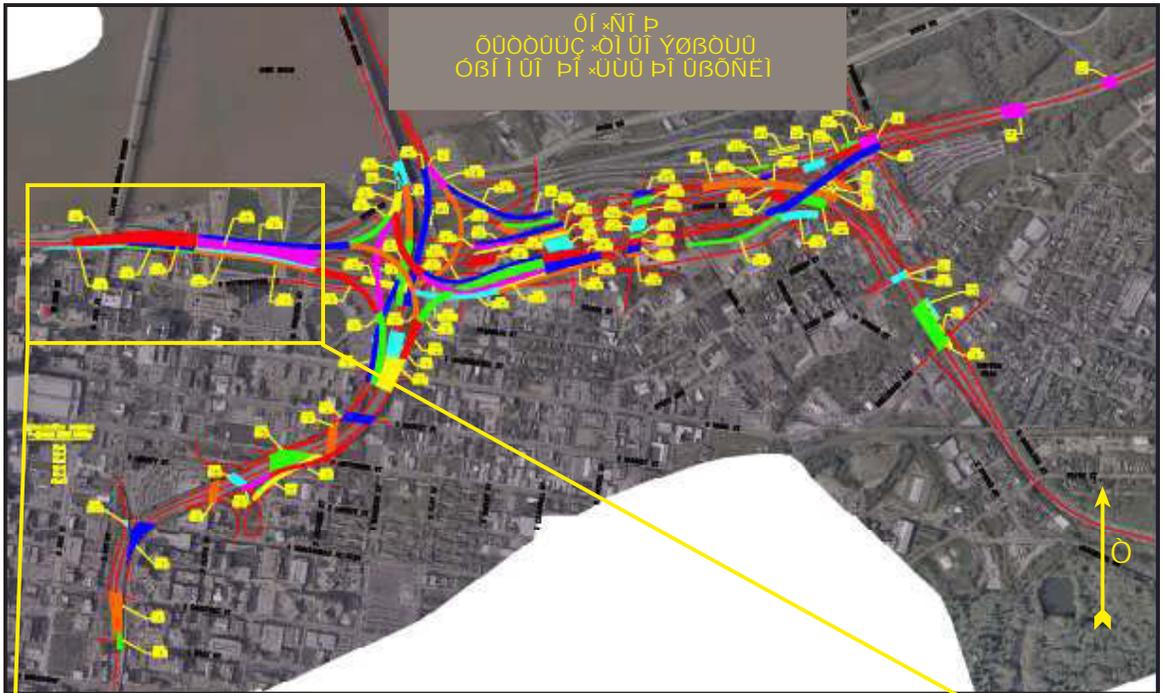
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**EXHIBITS 1-6**



$\hat{I} \cdot r \gg \hat{O} \pm \frac{1}{2} \zeta \cdot r \pm^2$	
$\hat{U} \cdot \cdot \cdot \frac{3}{4} \cdot r \cdot \hat{i}$	$\hat{Y} \pm^3 \pm^2 \gg \hat{\zeta} \cdot r \pm^0 \hat{O} \gg^2 \cdot \frac{1}{2} \mu \hat{S}$ $\hat{U} \gg^0 \hat{\zeta} \cdot r \pm^3 \gg^2 \cdot r \pm^0 \hat{O} \cdot 1 \cdot \hat{\zeta} \hat{S}$
 	$\hat{Y} \text{NEOIC}$ $\hat{O} \hat{O} \hat{U} \hat{U} \hat{O} \hat{T} \hat{I} \hat{N} \hat{O}$
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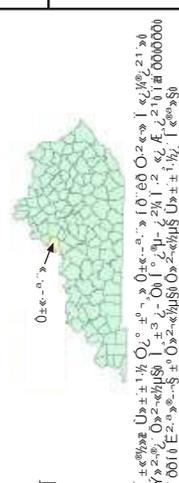


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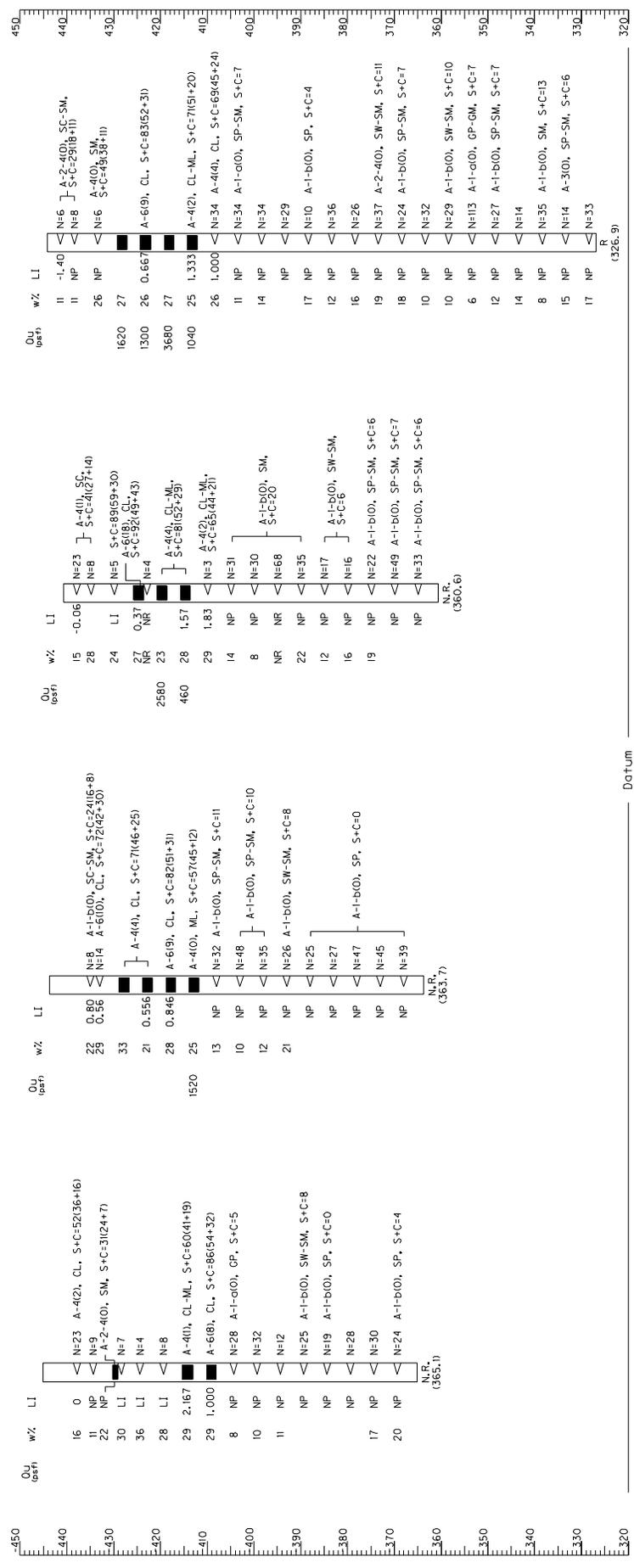
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Kentucky Transportation Associates	YNEOIC
P. 1/4 » P. 1/4	

Hole No. 4W-270 (1-64) 335+17 133.0' RT. 445.05  
 Location Offset Elev.  
 Hole No. 4B-248 (1-64) 336+93 136.0' RT. 442.69  
 Location Offset Elev.  
 Hole No. 4B-249 (1-64) 338+75 136.0' RT. 440.58  
 Location Offset Elev.  
 Hole No. 4B-250 (1-64) 341+95 101.0' RT. 444.07  
 Location Offset Elev.



Datum

DATE:	DESIGNED BY:	REVISION	CHECKED BY:
DETAILED BY:	Commonwealth of Kentucky DEPARTMENT OF HIGHWAYS COUNTY: JEFFERSON		
ROUTE: I-64	CORRECTING: RIVER ROAD	PROJECTED BY: PREFERRED LAYOUT BRIDGE B3RD-1 DRAWN BY: Division of Structural Design GEOTECHNICAL BRANCH	

ITEM NUMBER  
**5-118.18 & 5-118.19**  
**EXHIBIT 4**





**APPENDICES**  
**A - F**

**APPENDIX A**  
**COORDINATE DATA SUBMISSION FORM**



**APPENDIX B**  
**CONSOLIDATION AND STRENGTH TESTING**  
**LABORATORY RESULTS**



Wang Engineering, INC.  
Geo-Environmental Engineers  
1146 N Main Street  
Lombard, IL 60148  
Phone : 630 953-9928  
Fax: 630 953-9938

Project Name Kennedy Interchange - Phase 4 - S0160  
Source 4B-248 #4, 20'-22'  
Cv computation Method: Cassagrande

Initial Void Ratio = 0.579  
In-Situ Vertical Effective Stress = 2400 psf

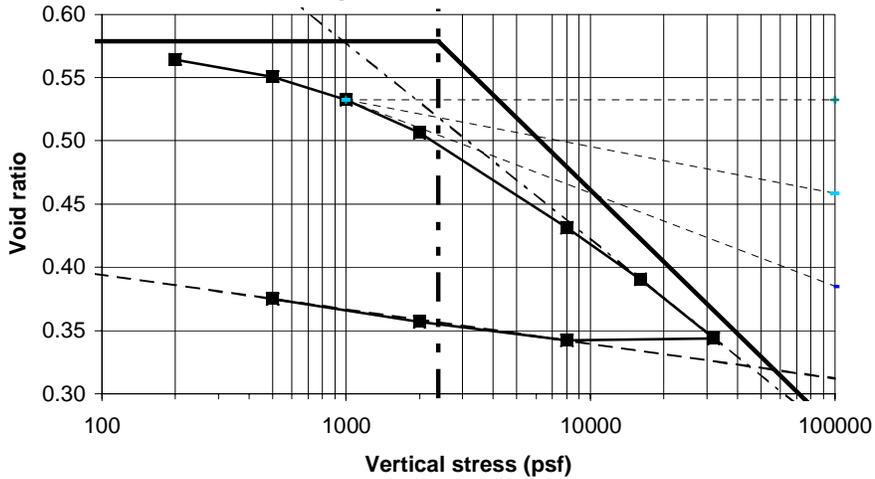
**Compression and Swelling Indices**

Compression index  $C_c = 0.155$   
Field corrected  $C_c = 0.189$   
Swelling index  $C_s = 0.027$

**Preconsolidation pressure,  $\sigma_c$**   
Casagrande Method = 2382 psf  
Over-Consolidation Ratio (OCR) = 1.0

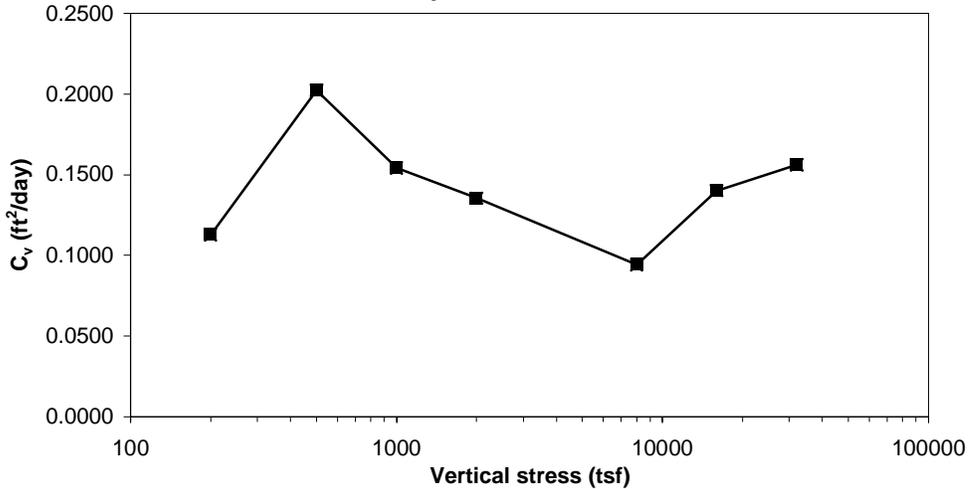
**CONSOLIDATION CURVE**

Sample 4B-248 #4, 20'-22'



**CONSOLIDATION COEFFICIENT ( $C_v$ ) vs. VERTICAL STRESS**

Sample 4B-248 #4, 20'-22'





Wang Engineering, INC.  
Geo-Environmental Engineers

1146 N Main Street  
Lombard, IL 60148  
Phone : 630 953-9928  
Fax: 630 953-9938

Project Name Kennedy Interchange - Phase 4 - S9070  
Source 4W-270 #7 (30'-32')  
Cv computation Method: Cassagrande

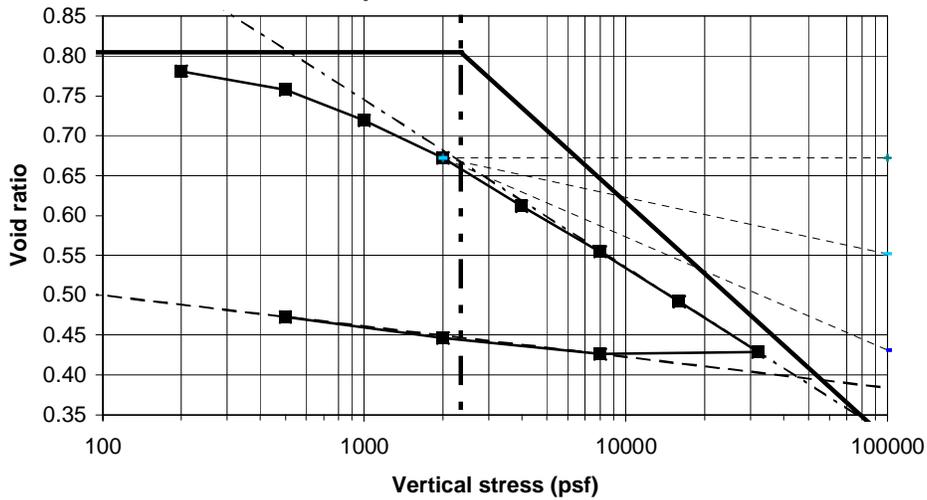
Initial Void Ratio = 0.805  
In-Situ Vertical Effective Stress = 2364 psf

**Compression and Swelling Indices**

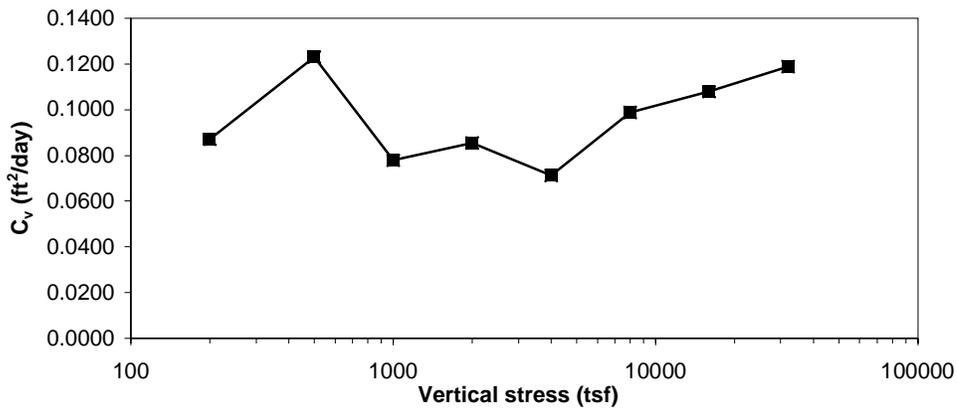
Compression index  $C_c = 0.210$   
Field corrected  $C_c = 0.298$   
Swelling index  $C_s = 0.039$

**Preconsolidation pressure,  $\sigma_c$**   
Casagrande Method = 2340 psf  
Over-Consolidation Ratio (OCR) = 1.0

**CONSOLIDATION CURVE**  
Sample 4W-270 #7, 30'-32'



**CONSOLIDATION COEFFICIENT ( $C_v$ ) vs. VERTICAL STRESS**  
Sample 4W-270 #7 (30'-32')

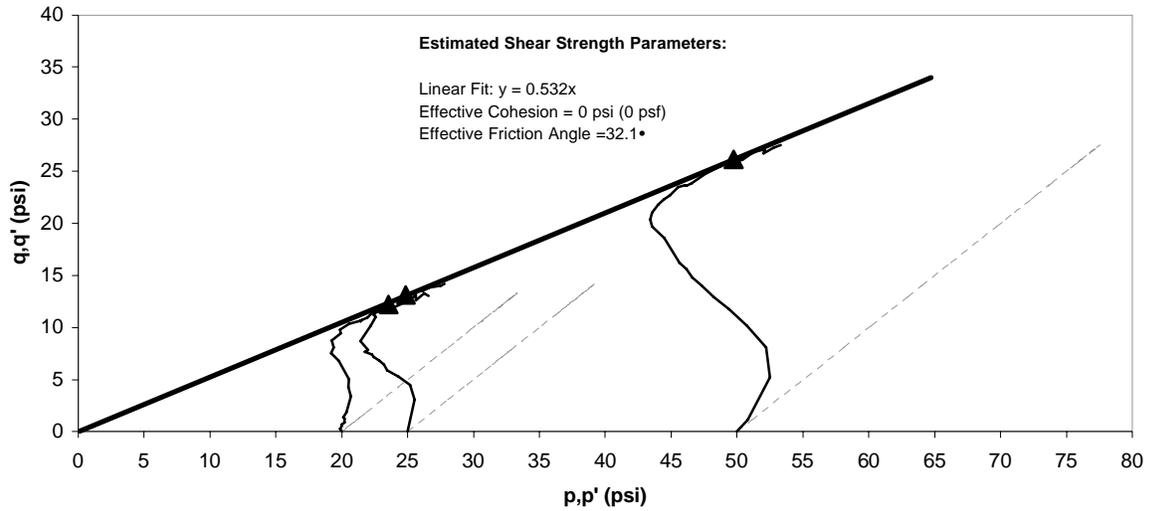




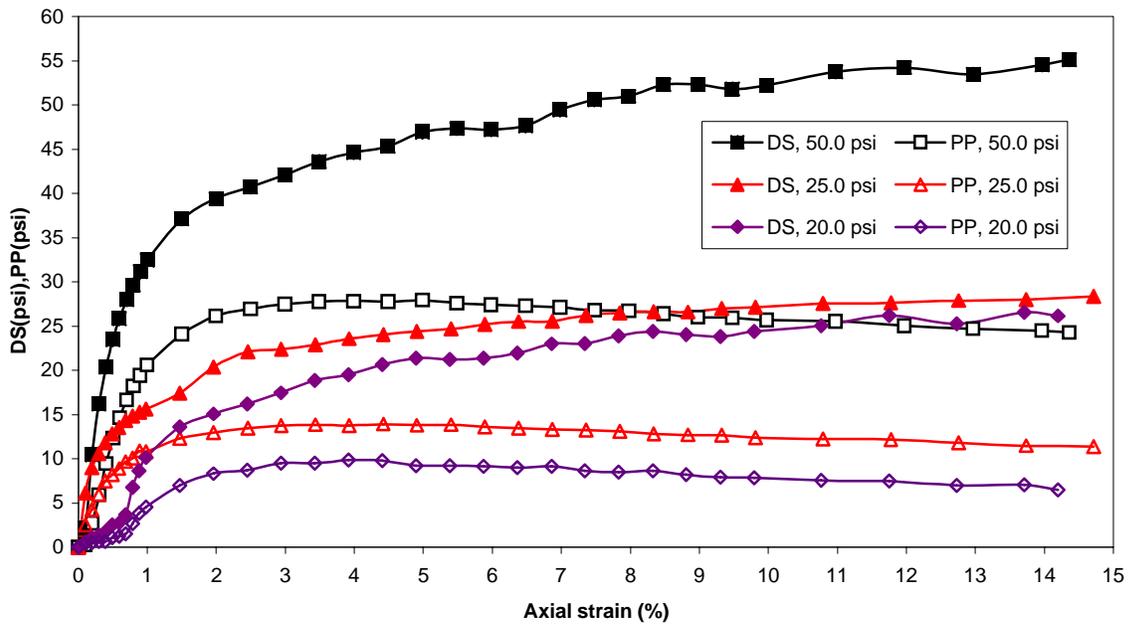
Project Kennedy Interchange - Phase 4 - S0160: Bridge 64-2  
Sample ID 4B-248336+ 93,1 S Rt., 20'to 27'

Failure Criterion : Maximum Effective Principal Stress Ratio

### Total and Effective Stress Paths and Effective Stress Strength Envelope



### Deviatoric Stress (DS) and Pore Water Pressure (PP) VS. Axial Strain



**APPENDIX C**  
**CORRECTED SPT N-VALUES AND SOIL PARAMETERS**

Corrected SPT N-Values  
and Soil Parameters  
4B-248

Boring 4B-248

Kennedy Interchange Reconstruction - Phase 4

Ramp 42 (3rd ST to I-65 SB)

GWL (extreme) = 0 ft bgs (flood stage)

GWL (normal) = 22.69 ft bgs (Elevation 421 ft) estimated based on Ohio River Pool elevation

Soil Description	Classification (USCS)	Top Elevation (ft)	Bottom Elevation (ft)	Undrained Shear Strength (psf)	Raw Field SPT $N_{field}$ (Blow Count)	Hammer Corrected SPT $N_{60}$ (Blow Count)	Final Corrected SPT $N_{160}$ (Blow Count)	Internal Friction Angle (degree)
Clayey sandy silt	SC-SM	443.69		125	8	11	13	31
Soft silty clay	CL		434.2	500	14	19	21	
	CL	434.2		250				
	CL			500				
	CL			1000				
	ML		410.7	760				
Silty sand with gravel	SP-SM	410.7			32	43	35	37
	SP-SM				48	64	50	41
	SP-SM				35	47	35	37
	SW-SM		390.7		26	35	25	34
Medium sand	SP	390.7			25	33	24	34
	SP				27	36	25	34
	SP				47	63	42	39
	SP				45	60	39	38
	SP		363.7		39	52	33	36

Equations:  $N_1 = C_N N_{field}$ ;  $C_N = [0.77 \log_{10}(40/\sigma'_v)]$ , and  $C_N < 2.0$ ; <sup>1)</sup>  
 $N_{60} = (80/60) N_{field}$ ; [corresponding to 80% hammer efficiency] <sup>1)</sup>  
 $N_{160} = C_N N_{60}$ ; <sup>1)</sup>  
 $\phi(\text{deg}) = 27.1 + 0.3 N_{cor} - 0.00054 N_{cor}^2$  <sup>2)</sup>

References: 1) AASHTO LRFD Bridge Design Specifications, Interim 2006, Section 10.4.6.2.4  
 2) Wolff, T. F. (1989). "Pile Capacity Prediction Using Parameter Functions," in *Predicted and Observed Axial Behavior of Piles, Results of a Pile Prediction Symposium* sponsored by the Geotechnical Engineering Division, ASCE, Evanston, IL, June, 1989, ASCE Geotechnical Special Publication No. 23, pp. 96-106.

Corrected SPT N-Values  
and Soil Parameters  
4B-249

Boring 4B-249

Kennedy Interchange Reconstruction - Phase 4

Ramp 42 (3rd ST to I-65 SB)

GWL (extreme) = 0 ft bgs (flood stage)

GWL (normal) = 19.58 ft bgs (Elevation 421 ft) estimated based on Ohio River Pool elevation

Soil Description	Classification (USCS)	Top Elevation (ft)	Bottom Elevation (ft)	Undrained Shear Strength (psf)	Raw Field SPT $N_{field}$ (Blow Count)	Hammer Corrected SPT $N_{60}$ (Blow Count)	Final Corrected SPT $N_{1,60}$ (Blow Count)	Internal Friction Angle (degree)
Stiff silty clay	SC	440.58		4500	23	31	48	
	CL		432.1	1000	8	11	14	
	CL	432.1		1000	5	7	8	
	CL			750	4	5	5	
	CL-ML			1290				
	CL-ML			230				
	CL-ML		407.6	1250	3	4	3	
Silty sand with gravel	SM	407.6			31	41	34	37
	SM				30	40	32	36
	SM				68	91	70	45
	SM		387.6		35	47	35	37
Medium silty sand	SW-SM	387.6			17	23	16	32
	SW-SM				16	21	15	31
	SP-SM				22	29	20	33
	SP-SM				49	65	43	39
	SP-SM		360.6		33	44	28	35

Equations:  $N_1 = C_N N_{field}$ ;  $C_N = [0.77 \log_{10}(40/\sigma'_v)]$ , and  $C_N < 2.0$ ; <sup>1)</sup>  
 $N_{60} = (80/60) N_{field}$ ; [corresponding to 80% hammer efficiency] <sup>1)</sup>  
 $N_{1,60} = C_N N_{60}$ ; <sup>1)</sup>  
 $\phi(\text{deg}) = 27.1 + 0.3 N_{cor} - 0.00054 N_{cor}^2$  <sup>2)</sup>

- References:
- 1) AASHTO LRFD Bridge Design Specifications, Interim 2006, Section 10.4.6.2.4
  - 2) Wolff, T. F. (1989). "Pile Capacity Prediction Using Parameter Functions," in *Predicted and Observed Axial Behavior of Piles, Results of a Pile Prediction Symposium* sponsored by the Geotechnical Engineering Division, ASCE, Evanston, IL, June, 1989, ASCE Geotechnical Special Publication No. 23, pp. 96-106.

Corrected SPT N-Values  
and Soil Parameters  
4B-250

Boring 4B-250

Kennedy Interchange Reconstruction - Phase 4

Ramp 42 (3rd ST to I-65 SB)

GWL (extreme) = 0 ft bgs (flood stage)

GWL (normal) = 23.07 ft bgs (Elevation 421 ft) estimated based on Ohio River Pool elevation

Soil Description	Classification (USCS)	Top Elevation (ft)	Bottom Elevation (ft)	Undrained Shear Strength (psf)	Raw Field SPT $N_{field}$ (Blow Count)	Hammer Corrected SPT $N_{60}$ (Blow Count)	Final Corrected SPT $N_{160}$ (Blow Count)	Internal Friction Angle (degree)
Silty sand with clay	SC-SM	444.1			6	8	12	31
	SC-SM				8	11	14	31
	SM		433.6		6	8	9	30
Silty clay	CL	433.6		810				
	CL			650				
	CL			1840				
	CL-ML			520				
Silty sand with gravel	CL		408.1	250	34	45	37	
	SP-SM	408.1			34	45	35	37
	SP-SM				34	45	34	37
Medium sand	SP-SM		391.1		29	39	28	35
	SP	391.1			10	13	9	30
	SP				36	48	33	36
	SP		376.1		26	35	23	34
Silty sand with sandy gravel lenses	SW-SM	376.1			37	49	32	36
	SP-SM		see sheet 2 of 2		24	32	20	33

Equations:  $N_1 = C_N N_{field}$ ;  $C_N = [0.77 \log_{10}(40/\sigma_v)]$ , and  $C_N < 2.0$ ; <sup>1)</sup>  
 $N_{60} = (80/60) N_{field}$ ; [corresponding to 80% hammer efficiency] <sup>1)</sup>  
 $N_{160} = C_N N_{60}$ ; <sup>1)</sup>  
 $\phi(\text{deg}) = 27.1 + 0.3 N_{cor} - 0.00054 N_{cor}^2$  <sup>2)</sup>

- References:
- 1) AASHTO LRFD Bridge Design Specifications, Interim 2006, Section 10.4.6.2.4
  - 2) Wolff, T. F. (1989). "Pile Capacity Prediction Using Parameter Functions," in *Predicted and Observed Axial Behavior of Piles, Results of a Pile Prediction Symposium*, sponsored by the Geotechnical Engineering Division, ASCE, Evanston, IL, June, 1989, ASCE Geotechnical Special Publication No. 23, pp. 96-106.

Corrected SPT N-Values  
and Soil Parameters  
4B-250

Boring 4B-250

Kennedy Interchange Reconstruction - Phase 4

Ramp 42 (3rd ST to I-65 SB)

GWL (extreme) = 0 ft bgs (flood stage)

GWL (normal) = 20.07 ft bgs (Elevation 421 ft) estimated based on Ohio River Pool elevation

Soil Description	Classification (USCS)	Top Elevation (ft)	Bottom Elevation (ft)	Undrained Shear Strength (psf)	Raw Field SPT $N_{field}$ (Blow Count)	Hammer Corrected SPT $N_{60}$ (Blow Count)	Final Corrected SPT $N_{160}$ (Blow Count)	Internal Friction Angle (degree)
Silty sand with sandy gravel lenses	SP-SM				32	43	26	35
	SW-SM				29	39	23	34
	GP-GM				113	151	88	49
	SP-SM				27	36	20	33
	SP-SM				14	19	10	30
	SM				35	47	25	34
	SP-SM				14	19	10	30
	SP-SM		327.6		33	44	23	34

Equations:  $N_1 = C_N N_{field}$ ;  $C_N = [0.77 \log_{10}(40/\sigma_v)]$ , and  $C_N < 2.0$ ; <sup>1)</sup>  
 $N_{60} = (80/60) N_{field}$ ; [corresponding to 80% hammer efficiency] <sup>1)</sup>  
 $N_{160} = C_N N_{60}$ ; <sup>1)</sup>  
 $\phi(\text{deg}) = 27.1 + 0.3 N_{cor} - 0.00054 N_{cor}^2$  <sup>2)</sup>

- References:
- 1) AASHTO LRFD Bridge Design Specifications, Interim 2006, Section 10.4.6.2.4
  - 2) Wolff, T. F. (1989). "Pile Capacity Prediction Using Parameter Functions," in *Predicted and Observed Axial Behavior of Piles, Results of a Pile Prediction Symposium*, sponsored by the Geotechnical Engineering Division, ASCE, Evanston, IL, June, 1989, ASCE Geotechnical Special Publication No. 23, pp. 96-106.

Corrected SPT N-Values  
and Soil Parameters  
4W-270

Boring 4W-270

Kennedy Interchange Reconstructio

Ramp 42 (3rd ST to I-65 SB)

GWL (extreme) = 0 ft bgs (flood stage)

GWL (normal) = 24.05 ft bgs (Elevation 421 ft) estimated based on Ohio River Pool elevation

Soil Description	Classification (USCS)	Top Elevation (ft)	Bottom Elevation (ft)	Undrained Shear Strength (psf)	Raw Field SPT $N_{field}$ (Blow Count)	Hammer Corrected SPT $N_{60}$ (Blow Count)	Final Corrected SPT $N_{160}$ (Blow Count)	Internal Friction Angle (degree)
Silty clay	CL	445.05	436.1	3500	23	31	41	
Silty sand w/ gravel	SM	436.1	429.1		9	12	14	31
Silty clay	CL	429.1		500	7	9	9	
	CL			125	4	5	5	
	CL			750	8	11	9	
	CL-ML			125				
	CL		405.1	250				
Sandy gravel	GP	405.1			28	37	29	35
	GP				32	43	32	36
	GP		389.6		12	16	12	31
Medium sand	SW-SM	389.6			25	33	23	34
	SP				19	25	17	32
	SP				28	37	25	34
	SP				30	40	26	34
	SP		365.1		24	32	20	33

Equations:  $N_1 = C_N N_{field}$ ;  $C_N = [0.77 \log_{10}(40/\alpha_v)]$ , and  $C_N < 2.0$ ;  
 $N_{60} = (80/60) N_{field}$ ; [corresponding to 80% hammer efficiency] <sup>1)</sup>  
 $N_{160} = C_N N_{60}$ ; <sup>1)</sup>  
 $\phi(\text{deg}) = 27.1 + 0.3 N_{cor} - 0.00054 N_{cor}^2$

- References: 1) AASHTO LRFD Bridge Design Specifications, Interim 2006, Section 7.7.2  
 2) Wolff, T. F. (1989). "Pile Capacity Prediction Using Parameter Functions," in *Predicted and Observed Axial Behavior of Pile* a *Pile Prediction Symposium*, sponsored by the Geotechnical Engineering Division, ASCE, Evanston, IL, June, 1989, ASCE (Special Publication No. 23, pp. 96-

**APPENDIX D**  
**SINGLE STEEL H-PILE CAPACITY**  
**EVALUATIONS**

## SINGLE STEEL H-PILE CAPACITY EVALUATIONS

<u>Steel H-Pile Capacities</u>		Page No.
Piers 1-4	12x53	D-1
	14x73	D-5
	14x89	D-8

<u>Steel H-Pile Driving Resistance</u>		
Piers 1-4	12x53	D-11
	14x73	D-13
	14x89	D-15

### Resistance Factors for LRFD – Driven Piles

Axial Capacity		
Skin Friction and End Bearing In Clays – Alpha Method		0.35
Skin Friction and End Bearing in Sands – Nordlund/Thurman Method		0.45
Uplift Resistance		
Clays	Alpha Method	0.25
Sands	Nordlund Method	0.35











Γ → 0: D → Y<sub>2</sub>° 2½ →  
 Π Γ Uoi D → 1 oi  
 ι ι' ef 0: D → 2 eo μ →

U → 3 2½ → D → 2½ → Y<sub>2</sub>° U → 2½ → 2 a i f c i o 0  
 E 2½ → 2½ → 2½ → U → 2½ → 2 a i f c i o 0

		U → 2½ D → 2½ Y <sub>2</sub> °	O ± 3 2½ Γ → 2½	O ± 3 2½ D → 2½	Γ ± 2½ U → 2½ 2½ 2½ Γ → 2½	Φ <sub>1</sub> Γ ± 2½ U <sub>2</sub> ± 2½ U → 2½ 2½ 2½ B <sub>2</sub> 2½ Y ± 3 2½ 2½ 2½ Γ → 2½ 2½ Γ → 2½ B <sub>2</sub> 2½ 2½ O → 2½	Φ <sub>2</sub> Γ ± 2½ U <sub>2</sub> ± 2½ U → 2½ 2½ 2½ B <sub>2</sub> 2½ Y ± 3 2½ 2½ 2½ Γ → 2½ 2½	Φ <sub>3</sub> Γ ± 2½ U <sub>2</sub> ± 2½ U → 2½ 2½ 2½ B <sub>2</sub> 2½ E <sub>2</sub> 2½ Γ → 2½ 2½ 2½	Φ <sub>4</sub> Γ ± 2½ U <sub>2</sub> ± 2½ U → 2½ 2½ 2½ B <sub>2</sub> 2½	
0°	0°	0°	0°	0°	0°	0°	0°	0°	0°	
1 0 i 0 0	1 1 e e 0	1 1 e e 0	1 1 e e 0	e i a i 0	e i a e e	1 e a i 0	e i a e c	1 1 e e 0	f e a i l	1 e a i f
1 1 i 0 0	1 1 f a i 0	1 1 f a i 0	1 1 f a i 0	e e a i e	e e a i e	1 e a i f	e e a e 0	1 1 a i 0	f e a 0 e	1 c a i 0
1 1 i 0 0	1 1 f a i 0	1 1 f a i 0	1 1 f a i 0	e i a i 0	e c a e 0	1 c a c 0	c i f e i f	1 e a e e	1 0 e a f	1 0 a i e
1 1 i 0 0	1 1 e e 0	1 1 e e 0	1 1 e e 0	e e a i 0	e i a i 0	1 1 e e 0	c e a c i f	1 c a i e	1 1 a i i	1 1 e a i
1 1 i 0 0	1 1 e e 0	1 1 e e 0	1 1 e e 0	e e a i 0	e e a i 0	1 1 e e 0	1 0 i a i e	e i a i f	1 e a i c	1 1 i 0 c
1 e i 0 0	1 e i a e 0	1 e i a e 0	1 e e a e 0	e i a i e	e 0 a e f	1 e a i f	1 0 c a e e	e i a e f	1 c a i 0	1 1 e a i
1 e i 0 0	1 e i a c 0	1 1 a i 0	1 e e a i 0	e e a e e	e i a e 0	1 e a i e	1 1 e a i e	e e a e i	e i a c f	1 e a c e
1 e i 0 0	1 e i a i 0	1 1 e e 0	1 e e a i 0	c i a i 0	e e a e i	1 c a i i	1 1 0 c 0	e 0 a i e	e i a e e	1 e a i f
1 e a 0 0	1 e a c 0	1 1 a c 0	1 c i a e 0	c e a i 0	e i a e e	1 a i f c	1 1 e a e i	e i a i f	e e a e e	1 e a c i
1 c a 0 0	1 e e a e 0	1 e a i 0	1 0 i a e 0	1 0 i a c 0	e e a e i	1 f a i f	1 1 a i e	e e a i l	e 0 a c f	1 0 a i e
e 0 i 0 0	1 c e a e 0	1 e a e 0	1 1 i 0 0	1 0 e a e 0	c 0 a e e	1 e a i e	1 1 e a i e	e c a i f	e i a 0 i	1 1 i 0 f
e i 0 0 0	1 0 e a e 0	1 e a e 0	1 1 i a i 0	1 1 i a i 0	c i a c c	1 e a e 0	1 1 i a e e	e i a i e	e e a i f	1 1 e a i
e i 0 0 0	1 1 e a e 0	1 e a e 0	1 1 i a e 0	1 1 e a e	c c a i e	1 c a e c	1 e 0 a e i	e e a i 0	e 0 a i f	1 e a i 0
e i 0 0 0	1 1 f a c 0	1 e a e 0	1 e 0 a e 0	1 1 e a i e	1 0 e a e i	e i a e f	1 e i a e i	e i a i i	e e a e e	1 e a i c
e e a 0 0	1 1 f a 0 0	1 e a c 0	1 e a c 0	1 1 c a c e	1 1 i a e e	e e a c f	1 e e a c i	e i a i e	e c a c e	1 c a c e
e e a 0 0	1 e i a i 0	1 e a i 0	1 e c a i 0	1 1 i a e 0	1 1 e a 0 e	e e a 0 f	1 e i a c e	e e a i c	e i a i e	1 1 e a e
e e a 0 0	1 e i a i 0	1 e a i 0	1 e e a e 0	1 1 c a i e	1 1 0 a i f	e 0 a i e	1 e i a i e	c 0 a e e	e e a i f	1 1 a i e
e e a 0 0	1 e 0 a i 0	1 e a e 0	1 e e a i 0	1 1 i a e e	1 1 i a e e	e i a i e	1 e e a i e	c i a e i	e c a e e	1 1 e e e
e c a 0 0	1 e c a e 0	1 e a 0 0	1 c e a e 0	1 1 e a e e	1 1 e a e c	e i a i c	1 c i a i e	c e a e c	c i a e i	1 e a i f
e 0 i 0 0	1 e e a e 0	1 e a i 0	1 0 e a e 0	1 e i a i 0	1 1 f a c e	e e a i c	1 c c a i f	c c a e i	1 e a c e	1 e a c e
e i 0 0 0	1 c e a e 0	1 e a e 0	1 e a i 0	1 e a i 0	1 1 e a i 0	e e a e 0	1 0 e a e i	1 0 i a e e	c c a i f	1 c a e e
e i 0 0 0	1 0 e a e 0	1 e a e 0	1 1 e a e 0	1 e i a e 0	1 1 a i f	e 0 a e i	1 1 i a e i	1 0 e a e i	1 0 a i c	e i a i e
e i 0 0 0	1 1 e a 0 0	1 c a i 0	1 f e a i 0	1 e e a e e	1 1 e a i	e i a e e	1 1 e a e i	1 0 e a c i	1 0 e a e i	e i a e e
e i 0 0 0	1 1 e a i 0	1 c a i 0	1 1 i a i 0	1 e i a i 0	1 1 c a e c	e i a c e	1 1 f a e e	1 1 i a c c	1 0 e a e 0	e i a i e
e e a 0 0	1 1 f a i 0	1 c a e 0	1 e i a e 0	1 e e a c 0	1 e i a i f	e e a 0 e	1 1 c a c e	1 1 i a c c	1 1 i a e e	e e a c f
e e a 0 0	1 1 a i 0	1 c a c 0	1 e i a i 0	1 e i a e 0	1 e a i e	e c a i e	1 1 e a 0 e	1 1 e a 0 i	1 1 e a 0 e	e e a e f
e e a 0 0	1 e i a i 0	1 0 a i 0	1 e i a e 0	1 e a i 0	1 e i a e e	e i a i c	1 1 i a i c	1 1 i a i 0	1 1 e a i e	e c a i f
e e a 0 0	1 e i a e 0	1 0 a i 0	1 e i a c 0	1 c 0 a c e	1 e e a e e	e i a i e	1 1 e a i l	1 1 a i f	1 1 i a i l	e 0 a e i
e c a 0 0	1 e 0 a e 0	1 0 a e 0	1 c a i 0	1 c e a e 0	1 e i a 0 i	e e a e i	1 e i a i f	1 1 e a i f	1 1 i a e e	e i a i f
e 0 i 0 0	1 e c a e 0	1 1 i 0 0	1 0 0 a e 0	1 0 0 a i 0	1 e e a i e	e e a e i	1 e 0 a e i	1 1 0 a i e	1 1 e a e i	e i a c f
e i 0 0 0	1 e e a c 0	1 1 a i 0	1 i 0 a i 0	1 0 e a i 0	1 e c a e 0	e c a e e	1 e e a e i	1 1 f a i f	1 1 i a i f	e e a e i
e i 0 0 0	1 c e a 0 0	1 1 e e 0	1 i c a e 0	1 0 c a e e	1 e i a e c	c i a e i	1 e i a e e	1 1 e a i f	1 1 i a i f	e e a i f
e i 0 0 0	1 0 e a i 0	1 1 e e 0	1 1 e a c 0	1 1 i a i e	1 e e a c f	c i a c e	1 e e a e c	1 f c a i c	1 1 e a i 0	e e a e 0
e i 0 0 0	1 1 e a i 0	1 1 a i 0	1 f e a i 0	1 1 c a i e	1 c i a i e	c e a 0 e	1 1 a i 0 e	1 1 i a i e	1 1 0 a e e	e 0 a c
e e a 0 0	1 1 e a i 0	1 1 a i 0	1 1 e a e 0	1 1 f a c 0	1 c e a i f	c e a i f	1 c i a 0 e	1 1 e a e i	1 1 f a e 0	e i a c 0
e e a 0 0	1 1 i a e 0	1 1 e e 0	1 e e a i 0	1 1 e a e e	1 0 0 a e i	1 0 0 a i 0	1 c e a i f	1 1 e a e e	1 1 e a c c	e i a i c
e e a 0 0	1 1 f a e 0	1 1 a c 0	1 e e a e 0	1 1 f a i e	1 0 i a e i	1 0 i a i f	1 0 i a i f	1 e i a e i	1 e 0 a i e	e e a 0 c
e e a 0 0	1 e i a e 0	1 1 a i 0	1 e e a c 0	1 f e a c e	1 0 i a e f	1 0 i a e f	1 0 c a i f	1 e i a e e	1 e i a i e	e e a e e











[->> QfD: > U<sup>0,21</sup> ] >---γ2½>  
 Pí Í Uúí D>® í úí  
 Yí "éí QfD: > úed μ--

U-~3 γ>¼ P<-> ±° D> > Y<° U>° <-±2 á í çúð °ú  
 E<~>¼ |<¼> O>±> < D±± U>° <-±2á í í úð °ú

U>° γ< P>±° D:> Y<°	Q±3 .2< Í .¼> Í >---γ2½>	Q±3 .2< U>²¼ P>γ<21	QfD: > U <sup>0,21</sup> Í >---γ2½>	
θ<~>	θμ<° ->	θμ<° ->	θμ<° ->	θ±2->
éí úð	í ðí úed	éí çð	í í í ú í	í í í ú í
ééúð	í í í ú ed	éí çð	í í í ú í	í í éú é
ééúð	í í í ú çð	éí çð	í í éú í	í í çú í
ééúð	í í í ú í ð	éí çð	í í éú è	í í í ú éí
ééúð	í í í ú ed	éí çð	í éí ú í ð	í í éú í ð
éçúð	í éí ú í ð	éí çð	í éçú éí	í í çú é
éúúð	í éí ú ed	éí çð	í ééú ed	í í í ú í ð
éí úð	í éð ú ed	éí çð	í éí ú ed	í í éú ðí
éí úð	í éð ú ed	éí çð	í éí ú ed	í í éú é
éí úð	í çð ú ed	éí çð	í éú çú í	í í í ú é
éí úð	í ðð ú ed	éí çð	í çéú ed	í í éú í
ééúð	í í ú ed	éí çð	í ðí ú ed	í éí ú ðí
ééúð	í í í ú ed	éí çð	í í í ú ed	í ééú çð
ééúð	í í í ú í ð	éí çð	í í çú éí	í éçú é
ééúð	í í í ú ed	éí çð	í í éú í ð	í éí ú ed
éçúð	í éí ú í ð	éí çð	í í éú è	í ééú éí
éðúð	í éí ú çð	éí çð	í í í ú í	í éí ú éí
éí úð	í éí ú ed	éí çð	í éí ú í	í ééú é
éí úð	í éí ú ed	éí çð	í éçú í	í éçú éí
éí úð	í çéú í ð	éí çð	í ééú ed	í éí ú ed
éí úð	éðéú ed	éí çð	í ééú çú í	í ééú çé
ééúð	éí éú ed	éí çð	í éí ú í	í çú í ú í
ééúð	éí éú ed	éí çð	í çú í éé	í çéú í í
ééúð	éí ðú í ð	éí çð	í ðú í ð	í ðð ú ed
ééúð	ééí ú ed	éí çð	í ðçú éé	í ðí ú éé
éçúð	ééí ú í ð	éí çð	í í éú è	í ðçú í ç
çúð	ééí ú ed	éí çð	í í éú ed	í í í ú éí
çí úð	éééú í ð	éí çð	í í éú éé	í í éú çú í
çí úð	éçéú í ð	éí çð	í í í ú éé	í í í ú í ç
çí úð	éí ðú í ð	éí çð	í éí ú ed	í í éú é
çí úð	éí í ú í ð	éí çð	í éí ú ed	í í í ú í é
çéúð	éí í ú í ð	éí çð	í éí ú éé	í í éú ç

Í <²¼

U>° γ< P>±° D:> Y<°	Q±3 .2< Í .¼> Í >---γ2½>	Q±3 .2< U>²¼ P>γ<21	QfD: > U <sup>0,21</sup> Í >---γ2½>	
θ<~>	θμ<° ->	θμ<° ->	θμ<° ->	θ±2->
çéúð	éí éú ed	éí çð	í éðú çú í	í í ðú í é
çéúð	éééú ed	éí çð	í çú ú í é	í í éú ed
çéúð	ééí ú í ð	éí çð	í çú ú í é	í í çú éí
ççúð	ééí ú ed	éí çð	éðéú éí	í éí ú í
í ðú ed	éçéú í ð	éí çð	éí éú è	í éçú í
í ðí ed	éðçú ed	éí çð	éí éú ed	í éí ú çð
í ðí ed	éí í ú ed	éí çð	éí éú í ð	í ééú ed
í ðí ed	éí í ú ed	éí çð	éí éú ed	í éí ú ed
í ðí ed	éí éú ed	éí çð	éééú éí	í ééú í
í ðéú ed	ééú ed	éí çð	éééú éé	í ééú í
í ðéú ed	ééí ú ed	éí çð	éééú éé	í ééú è
í ðéú ed	éééú í ð	éí çð	éééú éí	í çú í é
í ðéú ed	éðð ú ed	éí çð	éçéú ed	í çéú í ç
í ðçú ed	éí í ú í ð	éí çð	éðéú ed	í ðí ú í é
í í ðú ed	éí éú ed	éí çð	éí éú éí	í ðéú í
í í í ú ed	éí í ú í ð	éí çð	éí éú í ð	í í í ú éé
í í í ú ed	éééú í ð	éí çð	éí éú è	í í éú éí
í í í ú ed	ééçú í ð	éí çð	éí éú éé	í í í ú çú í
í í í ú ed	ééí ú í ð	éí çð	éééú è	í í çú í ç
í í éú ed	éçéú í ð	éí çð	éééú éé	í í í ú í

Í <²¼



[->> OñD: > U<sup>o.a.21</sup> ] >---γ2½>  
 Pí Í Uñi D> > Ì ñi  
 ÿ ÿ "éí OñD: > ñeð μ-->

U>°γ<sub>1</sub> O±³.2γ<sub>1</sub> O±³.2γ<sub>1</sub> OñD: > U<sup>o.a.21</sup>  
 P>±°D: > Í.¼> U2¼ Í >---γ2½>  
 Y<sub>γ</sub>° Í >---γ2½> P>γ<sup>o.21</sup>

	U>°γ <sub>1</sub> P>±°D: > Y <sub>γ</sub> °	O±³.2γ <sub>1</sub> Í.¼> Í >---γ2½>	O±³.2γ <sub>1</sub> U2¼ P>γ <sup>o.21</sup>	OñD: > U <sup>o.a.21</sup> Í >---γ2½>	
	θ <sup>o</sup> -+	θμ <sup>o</sup> -+	θμ <sup>o</sup> -+	θμ <sup>o</sup> -+	θγ±2--+
γ <sup>2¼</sup>	éí ñð	í í çñð	í ðñçð	í éí ñí é	í í éñéé
	éí ñð	í í çñéð	í ðñçð	í é çñí ð	í í çñéé
	éí ñð	í éðñí ð	í ðñçð	í éññí ð	í í í ñéð
	ééñð	í éí ñí ð	í ðñçð	í éññí ð	í í éñéé
	ééñð	í éí ñí ð	í ðñçð	í éí ñéé	í í í ñéé
	ééñð	í çí ñí ð	í ðñçð	í çí ñéé	í í éñçí
	ééñð	í ðí ñéð	í ðñçð	í ðññí é	í éðñí í
	éçñð	í í éñéð	í ðñçð	í ðñééé	í éí ñí é
	éðñð	í í éñí ð	í ðñçð	í í éñí ð	í ééñéé
	éí ñð	í í éñéð	í ðñçð	í í éñçí	í éí ñçé
	éí ñð	í éðñí ð	í ðñçð	í í í ñéð	í ééñí é
	éí ñð	í éí ñí ð	í ðñçð	í í í ñéé	í éí ñéé
	éí ñð	í éí ñí ð	í ðñçð	í éí ñí é	í ééñí í
	ééñð	í ééñí ð	í ðñçð	í éí ñí é	í éðñéí
	ééñð	í çéñí ð	í ðñçð	í éðñéé	í ééñí é
	ééñð	éí ðñí ð	í ðñçð	í éçñéð	í éçñéé
	ééñð	éí í ñéð	í ðñçð	í éçñí ð	í çí ñéð
	éçñð	éí éñí ð	í ðñçð	í çéñí é	í ççñí ç
	éðñð	éí éñçð	í ðñçð	í ðñéñí	í ðí ñçí
	éí ñð	ééðñéð	í ðñçð	í í éñí é	í ðñééé
	éí ñð	ééí ñí ð	í ðñçð	í í éñçé	í í í ñí é
	éí ñð	ééññí ð	í ðñçð	í í éñéí	í í éñí í
	éí ñð	éççñí ð	í ðñçð	í í éñí é	í í í ñí í
	ééñð	éí í ñéð	í ðñçð	í éññí é	í í éñí í
	ééñð	éí éñéð	í ðñçð	í éññí é	í í í ñí í
	ééñð	éí çñí ð	í ðñçð	í ééñí ð	í í éñí é
	ééñð	ééí ñéð	í ðñçð	í éñéñð	í í í ñí é
	éçñð	éééñí ð	í ðñçð	í çéñéð	í í éñí é
	çðñð	ééðñí ð	í ðñçð	éðéñçé	í éí ñí ç
	çí ñð	éçí ñð	í ðñçð	éí éñí ð	í ééñéð
	çí ñð	éðéñð	í ðñçð	éí éñçð	í éí ñçé

	U>°γ <sub>1</sub> P>±°D: > Y <sub>γ</sub> °	O±³.2γ <sub>1</sub> Í.¼> Í >---γ2½>	O±³.2γ <sub>1</sub> U2¼ P>γ <sup>o.21</sup>	OñD: > U <sup>o.a.21</sup> Í >---γ2½>	
	θ <sup>o</sup> -+	θμ <sup>o</sup> -+	θμ <sup>o</sup> -+	θμ <sup>o</sup> -+	θγ±2--+
γ <sup>2¼</sup>	çí ñð	éí í ñí ð	í ðñçð	éí éñí é	í éçñí í
	çí ñð	éí éñí ð	í ðñçð	éí çñí í	í éí ñéé
	çéñð	ééðñéð	í ðñçð	ééçñéé	í éçñí çí
	çéñð	éééñí ð	í ðñçð	ééðñéí	í éññí é
	çéñð	ééçñéð	í ðñçð	ééí ñéé	í çðñéí
	çéñð	éçí ñí ð	í ðñçð	éçí ñéí	í çéñí í
	ççñð	éðçñí ð	í ðñçð	éðí ñéí	í ðí ñéé
	í ðññð	éí í ñð	í ðñçð	éí í ñçð	í ðéñí é
	í ðí ñð	éí çñí ð	í ðñçð	éí éñí í	í í í ñí í
	í ðí ñð	ééí ñí ð	í ðñçð	éí éñéé	í í éñéé
	í ðí ñð	ééçñéð	í ðñçð	éí çñí	í í í ñéí
	í ðí ñð	ééí ñçð	í ðñçð	ééðñéé	í í ðñí ç
	í ðéñð	çðñí ð	í ðñçð	ééí ñí ð	í í éñí ð
	í ðéñð	çí éñð	í ðñçð	ééí ñçð	í í í ñçé
	í ðéñð	çí í ñéð	í ðñçð	éçéñéé	í í éñéí
	í ðéñð	çí éñéð	í ðñçð	éðéñéí	í éí ñéé
	í ðçñð	çéí ñéð	í ðñçð	éí çñéí	í éçñéé
	í í ñð	çéçñéð	í ðñçð	éí í ñéð	í éñéñð
	í í í ñð	ççéñéð	í ðñçð	éí í ñéé	í éí ñéé
	í í í ñð	í ðí í ñí ð	í ðñçð	éééñçé	í ééñçç
í í í ñð	í ðí éñéð	í ðñçð	éééñí é	í éí ñí í	
í í í ñð	í ðí éñð	í ðñçð	ééðñéé	í çðñí í	
í í éñð	í ðéí ñéð	í ðñçð	éçí ñí é	í çéñéç	





**APPENDIX E**  
**SINGLE DRILLED SHAFT EVALUATIONS**

## SINGLE DRILLED SHAFT EVALUATIONS

### Resistance Factors for LRFD – Drilled Shafts

#### Axial Capacity

Side Resistance in Clays	Alpha Method	0.45
End Bearing in Clays	Total Stress	0.40
Side Resistance in Sands	Beta Method	0.55
End Bearing in Sands	SPT Method	0.50

#### Uplift Resistance

Clays	Alpha Method	0.35
Sands	Beta Method	0.45

















$\Gamma_{\pm 2} \text{B} \cdot \Gamma_{\pm 2} \text{I} \gg \dots \Gamma_{\pm 2} \gg$   
 $\text{P} \Gamma \text{I} \text{U} \text{O} \text{I} \text{D} \gg \dots \text{I} \text{O} \text{I}$   
 $\text{I} \Gamma \text{O} \text{I} \text{U} \text{I} \text{D} \gg \dots \text{I} \text{O} \text{I}$

$\text{U} \gg \dots \Gamma_{\pm 2} \text{B} \cdot \Gamma_{\pm 2} \text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$   
 $\text{E} \Gamma_{\pm 2} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$

$\text{U} \gg \dots \Gamma_{\pm 2} \text{B} \cdot \Gamma_{\pm 2} \text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$ $\text{D} \cdot \text{Y} \Gamma_{\pm 2} \text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$		$\text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$ $\text{U} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$ $\text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$		$\text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$ $\text{U} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$ $\text{Y} \Gamma_{\pm 2} \text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$		$\text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$ $\text{U} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$ $\text{I} \gg \dots \Gamma_{\pm 2} \text{I} \text{O} \text{I}$		
$\text{O} \mu \text{O}$	$\text{O} \mu \text{O}$	$\text{O} \mu \text{O}$	$\text{O} \mu \text{O}$	$\text{O} \mu \text{O}$	$\text{O} \mu \text{O}$	$\text{O} \mu \text{O}$	$\text{O} \mu \text{O}$	
éçúð	éçúçé	í ðeaf ð	éçeaf é	í í çú í í	í eéaçí	í í eéaçe	í ðí açe	í eí açe
eðúð	eí ðueí	í ðeaf ð	çí eafí	í eeaúí	í eeaéé	í í í eí	í í í açe	í eeaçe
eí úð	eí eaeé	í ðeaf ð	çí í açe	í eeaçç	í çeaeé	í í eaeé	í í ðuðí	í eðuðí
eí úð	eí eaeé	í ðeaf ð	çeí úí é	í eeaúé	eðeaeé	í eí eae	í í eaf í	í eí úí í
eí úð	eéaaf é	í ðeaf ð	çéúae	í eeaí é	eí eaeé	í eaeí	í í eaf ð	í eeaí é
eí úð	eéí aeé	í ðeaf ð	çeçúí é	í çí aeé	eí eafí	í eí açe	í í aeé	í eí úí í
eéúð	eðí aeé	í ðeaf ð	í ðeaeçe	eðí açe	eí eafí é	í eeaí í	í eí úí í	í eaeé
eéúð	eí í aeé	í ðeaf ð	í ðí eaeçe	eí í úí ç	eí eaeí	í eí í é	í eí aeç	í eaeçe
eéúð	eí ðúçí	í ðeaf ð	í ðí eaf í	eí í úí í	eéaaf ç	í eaeé	í eaeé	í eafí é
eéúð	eéðúí í	í ðeaf ð	í ðeaeí	eí í aeí	eéaeçe	í eí açe	í eafí ç	í eaeí
eçúð	eé.ðuð	í ðeaf ð	í ðeaf ð	eí í aeé	eéaeçe	í eafí ç	í eafí	í çí aeé
çúúð	eççaeí	í ðeaf ð	í í ðeaf í	eéí aeé	eéçaeð	í çí aeé	í çeafé	í çeafé
çí úð	çí çúç	í ðeaf ð	í í eaf ð	eéí aeð	eðeafí	í ðeaf é	í ðeaf ç	í ðí aeé
çí úð	çí ðuí é	í ðeaf ð	í í eaf é	eéí aeí	eí í aeç	í ðeaf í	í eaf í	í ðeafí
çí úð	çéðaeé	í ðeaf ð	í í eaeçe	eéí açe	eí í aeí é	í í í aeé	í í í í í	í í í í í
çí úð	çeí af í	í ðeaf ð	í í eaeí	eçí af í	eí í aeí	í í eaf é	í í í aeí	í í eaeé
çeúð	í ðúí af í	í ðeaf ð	í í ðeaeí	eðí aeé	eí eafí	í í í aeí	í í í af í	í í í aeé
çeúð	í ðí í af é	í ðeaf ð	í í í eaeé	eí í af í	eéaeafí	í í eafí	í eí aeé	í eaf í
çeúð	í ðí í aeé	í ðeaf ð	í í í çúçé	eí í açe	eéçaf é	í í í aeé	í eí af í	í í í af í
çeúð	í ðeaf ð	í ðeaf ð	í í eí aeé	eí eaeé	eéí af ð	í í ðeaf	í eí açe	í í eaeç
ççúð	í ðeaeçí	í ðeaf ð	í í çí af í	eí eaeí	eçí af í	í í eaeé	í eí aeð	í í ðeae
í ðuðúð	í í ðeaeé	í ðeaf ð	í í í eafí é	eéaeé	eðeaf í	í eí aeð	í çí aeé	í í eaeç
í ðí úð	í í í í aeí	í ðeaf ð	í í í eaf í	eéaeé	eí eaf ð	í eaeé	eðí aeé	í eaeé
í ðí úð	í í eí af é	í ðeaf ð	í í eaeé	eéçaeí	eí çaeð	í eí aeé	eí í aeí	í eaeé
í ðí úð	í í eaeçe	í ðeaf ð	í í eí af é	eçí af í	eí í af í	í eí aeé	eí í aeé	í eaeé
í ðí úð	í í ççaeí	í ðeaf ð	í í ðeafí	eðí aeí	eéí aeí	í eafí	eí í aeí	í eafí
í ðeafúð	í í í í aeí	í ðeaf ð	í í í aeí	eí í aeí	eéaf é	í eí aeí	eí í af é	í eí af é
í ðeafúð	í í í eafí	í ðeaf ð	í í eí af í	eí eaeí	eéafí	í çafí	eéí aeð	í eafí
í ðeafúð	í í eafí í	í ðeaf ð	í í eí aeí	eí eafí	eçí açe	í çeaf é	eéí af í	í eí aeé
í ðeafúð	í í çí açe	í ðeaf ð	í í çeaf í	eí çaf í	eaeaf ç	í ðí açe	eéí açe	í eaeçe
í ðçúð	í í í eaeé	í ðeaf ð	í eí í aeé	eéí aeí	eí çafí	í çaeé	eéí aeé	í çí af í
í í ðuð	í í í ðeaf	í ðeaf ð	í eí eafí	eéí aeé	eí í af í	í í eafí	eçeafé	í çeaeé
í í í úð	í í eafí	í ðeaf ð	í eafí í	eéaf é	eí eaeé	í í í aeé	eðeaf ð	í ðí af í
í í í úð	í í eafí é	í ðeaf ð	í eafí ç	eçeaf ç	eéçafí	í í çaeé	eí eafí	í ðeafé





$\Gamma_{\pm 2}^{\pm 2} \beta_{\pm 2}^{\pm 2} \Gamma_{\pm 2}^{\pm 2}$   
 $\Gamma_{\pm 2}^{\pm 2} \beta_{\pm 2}^{\pm 2} \Gamma_{\pm 2}^{\pm 2}$   
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$\Gamma_{\pm 2}^{\pm 2} \beta_{\pm 2}^{\pm 2} \Gamma_{\pm 2}^{\pm 2}$   
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	$\Gamma_{\pm 2}^{\pm 2}$							
éçïòð	éçïçï	í éëí é	í ðëðí ð	èí ðíð	éëðíé	í èðí í	í í èí ç	í èí èç
èðíðð	èí ííí	í éëí é	í ðëðí ð	èí ðíð	éëí íéí	í èëçí	í èëéí	í èëéí é
èí ððð	èí íéé	í éëí é	íí ððçí	èëðí é	èéí íí í	í çí íéé	í èëéé	í èí íéé
èí ððð	èéí éé	í éëí é	íí íí íéí	èëðíé	èçí íéé	í çéí é	í èëí í	í èëéé
èí ððð	èéí éé	í éëí é	íí íí íéí	èéí íé é	èðéí í	í ðí ðé	í èí íéè	í çí íí ç
èí ððð	èçééí	í éëí é	íí èí ðð	èéí ðð	èí èí éí	í ðéçð	í çí íí í	í çéðé
èëððð	çí èí í	í éëí é	íí èëéð	èçí íéé	èí çéí	í íí íéí	í ðí íéí	í ðí çí
èëððð	çí çíðé	í éëí é	íí ðéí í	èðí íéí	èí íéé	í í ðéç	í í íéç	í ðéèð
èëððð	çéí íðí	í éëí é	íí í çí ð	èí íéð	èéí íéé	í í ééí	í í íí é	í í íéí
èëððð	çéí íé é	í éëí é	íí èí íí	èí èéí	èëéèç	í í íçí	í í íí é	í í íéí
èçíðð	í ðéðéð	í éëí é	í í èí íéé	èí èí çí	èééí í	í í çí í	í í íéé	í í íéç
çéíðð	í ðí èí ð	í éëí é	í í çéí é	èí èí é	èçðéí	í í éí é	í èí èç	í í èí ç
çí ððð	í ðéí íí	í éëí é	í í í çí é	èéçéí	èðí íí í	í èí íéé	í èí íí	í í íðé
çí ððð	í ðéí íí é	í éëí é	í í í íéí	èéí íí í	èí èíð	í èëðí	í èí íéí	í í èí é
çí ððð	í ðéçéçð	í éëí é	í í èëíðé	èéí íðí	èí èí çí	í èí í é	í èëíðé	í í íéí
çí ððð	í í í íéí	í éëí é	í í èçí éð	èçí íéé	èí í çí	í èðíçé	í çééí	í í èëé
çéíðð	í í í éí í	í éëí é	í í í íéé	èðéèç	èééíðé	í èëéí	èðéí é	í èí íí
çéíðð	í í èçéí	í éëí é	í í í èíð	èí èí éé	èééí í	í èí í é	èí èí í	í èëéé
çéíðð	í í çí çð	í éëí é	í í èí ðé	èí íðí	èéí íéí	í çðíé	èí èí é	í èí íí
çéíðð	í í í íéð	í éëí é	í í èëéé	èí íí í	èçéí é	í çééí	èí çí í	í èçéé
ççíðð	í í í íí í	í éëí é	í èí íéð	èéééé	èðéèçí	í ðí í é	èéðéí	í èëí é
í ððíðð	í í èëí í	í éëí é	í èí íéé	èééí ç	èí í éí	í í íí é	èéí íéð	í èðèçð
í ðí ððð	í í çí íéí	í éëí é	í èéí íçð	èéðíçé	èí èëé	í í éí í	èéí íð	í èëéð
í ðí ððð	í í í çí ð	í éëí é	í èëéí é	èçí íéí	èéðéí	í í éí é	èéí íéð	í çí íí é
í ðí ððð	í í í èíðé	í éëí é	í èí íí í	èðééí	èéí íéç	í í íí í	èçéí ð	í çéí é
í ðí ððð	í í èí íí í	í éëí é	í èí çí é	èí çéí	èéçí í	í í çéí	èðéíðí	í ðí ðí
í ðéíðð	í í çéí ð	í éëí é	í èëééé	èí íéé	èçí íéé	í í ééí	èí çéí	í ðçéçí
í ðéíðð	í í í í çð	í éëí é	í èçí íðé	èí èíðí	çðéí í	í èí íí	èí íéé	í í èëé
í ðéíðð	í í èðíé	í éëí é	í èí èéí	èéçí í	çí í íçé	í èí í é	èí íéí	í í í çí
í ðéíðð	í í èééí	í éëí é	í èí èíð	èéí íçð	çí èéð	í èéíçð	èééèçé	í í èéçé
í ðçíðð	í èðí íéé	í éëí é	í èéí íðí	èéééí	çéí íéé	í èëí é	èééí ð	í í íí ð
í í ðíðð	í èí íí í	í éëí é	í èðíéð	çðéí é	çéééé	í èí çí	èéðíéé	í í ðí é
í í ííðð	í èéðíðí	í éëí é	í èí èíð	çí í íí ð	çéí íí í	í çí íéé	èçí íðí	í í èéí

**APPENDIX F**  
**RECOMMENDED SOIL PARAMETERS FOR**  
**LATERAL LOAD ANALYSIS**

Estimated Soil Parameters for Lateral Load Pile Analysis  
Kennedy Interchanges Project – Bridge B3RD-1  
Structure S0180

Ground Water Level = 444 ft.

**Abutment 1**

Idealized Soil Profile	Approximate Soil Elevations (ft)	Average Saturated Unit Weight (pcf)	Estimated Corrected Friction Angle (degrees)	Estimated Undrained Shear Strength (psf)	Estimated Lateral Soil Modulus Parameter, k (pci)	Estimated Soil Strain Parameter, E <sub>50</sub> (%)
silty clay – fill	444-437	125	-	500	215	0.01
clay - soft to stiff	437-414.5	125	-	500	215	0.01
fine to medium sand – medium dense - dense	414.5-328	130	33.7	-	130	-

**Piers 1-4**

Idealized Soil Profile	Approximate Soil Elevations (ft)	Average Saturated Unit Weight (pcf)	Estimated Corrected Friction Angle (degrees)	Estimated Undrained Shear Strength (psf)	Estimated Lateral Soil Modulus Parameter, k (pci)	Estimated Soil Strain Parameter, E <sub>50</sub> (%)
silty sand (fill)	444-434	125	31	-	40	-
clay - soft to stiff	434-408	125	-	500	215	0.01
fine to medium sand – medium dense to dense	408-328	130	33.7	-	130	-