



DETROIT RIVER

Engineering Report

VOLUME 6: DETROIT RIVER BRIDGE STRUCTURE STUDY: APPENDIX D – GEOTECHNICAL REPORT

November 2008

Prepared by:



In association with:



CORRADINO

THE CORRADINO GROUP

Under agreement with:

Geotechnical Feasibility Evaluation Report for Proposed Foundation Elements

Proposed Detroit River International Crossing United States Shoreline

November 21, 2008

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EXECUTIVE SUMMARY

The Detroit River International Crossing (DRIC) Study is a bi-national effort to complete the environmental study processes for the border crossing between Detroit, Michigan and Windsor, Ontario for the United States, Michigan, Canada and Ontario governments. The study will identify solutions that support the region, state, provincial and national economies while addressing civil defense, national defense, and homeland security needs of the busiest trade corridor between the United States and Canada (Figure S-1).



Figure S-1: Detroit River International Crossing Study, Existing Detroit River International Crossings

Source: The Corradino Group of Michigan, Inc

The DRIC Study Draft Environmental Impact Statement (DEIS), previously released under separate cover, analyzes issues/impacts on the U.S. side of the crossing system over the Detroit River between Detroit, Michigan, and Windsor, Ontario, Canada. The alternatives are comprised of three components: the crossing, toll and customs plaza, and interchange connecting the plaza to I-75 (Figure S-2).



Figure S-2: Detroit River International Crossing Study, U.S. Area of Analysis for Crossing System

Source: The Corradino Group of Michigan, Inc.

Geotechnical Investigation Background

The initial geotechnical task performed by NTH for this project (Task 2330) was to study the Illustrative Alternative crossing locations from Belle Isle to the tip of Grosse Ile, including collecting the relevant available geotechnical data along the proposed project area and evaluating the data with respect to conceptual designs. The results of this task were presented in a report entitled Geotechnical Evaluation, Proposed Detroit River International Crossing, Task 2330, dated December 28, 2005.

After consideration of the available data, it became apparent that historical brine wells and associated cavities exist in the vicinity of the proposed bridge, but the exact locations of the historical solution mining were not known and the potential impacts to the proposed construction were not understood. As a result, the project team developed a Brine Well Cavity Investigation program (Brine Well Program) to delineate the size, locations, and shape of potential brine well cavities and to evaluate the possible impacts of such cavities. The Brine Well Program included the two proposed crossing corridors on the U.S. side of the river. An investigation plan was proposed and completed, utilizing a combined geophysical and geotechnical program including the drilling of multiple deep rock borings, performed in combination with cross-well seismic imaging. Forward modeling of geophysical methods in conjunction with preliminary historical rock mechanics analysis, borehole gravity, vertical seismic profiling (VSP), and downhole wireline logging were also included in the program. This report was entitled "Brine Well Cavity investigation Program Technical Report", and was presented as a portion of the DRIC Draft Environmental Impact Statement (DEIS) in February 2008.

In a report entitled "Preliminary Draft Geotechnical Investigation Report" revised September 21, 2006, NTH summarized the historical data specifically relevant to proposed crossings corridors X-10 and X-11, and provided a summary of expected engineering and construction issues related to the crossing corridors. This current report is an update of the September 21, 2006 report, and includes site specific information and analysis based on our field investigation. The field portion of this investigation was conducted between April 24 and June 1, 2008 and consisted of drilling a total of eight test borings, followed by laboratory testing of samples, and analysis of conditions with respect to the proposed construction.

Purpose of the Report

The purpose of the report is to present the details of our investigation, and to provide studylevel geotechnical analysis and recommendations with respect to the proposed construction concepts under consideration for the Detroit side of the proposed Detroit River International Crossing Bridge between Detroit, Michigan and Windsor, Ontario. This current report is not intended as a stand-alone document, and is intended for the use by the Corradino Group and Parsons Transportation to develop preliminary foundation concepts for the bridge design, as well as to evaluate probable construction costs and other impacts related to the project.

This report was developed to be placed in the Appendix of the complete Bridge Engineering Report, itself an attachment of the final DRIC Engineering report, which is part of the overall DRIC Environmental Impact Statement Report.

When this geotechnical investigation was undertaken (April 2008), two potential crossing corridors were under consideration, defined as Crossings X-10 and X-11 as shown on the attached Figure Nos. 1A and 1B, respectively, in Attachment A. The two subject crossings are in the same general vicinity, between the Ambassador Bridge, and Zug Island in southwest Detroit. The Crossing X-10 corridor generally consists of the area immediately north of Zug Island to historic Fort Wayne along the banks of the Detroit River. The Crossing X-11 corridor generally consists of the area along the banks of the Detroit River immediately north of historic Fort Wayne to the existing Mistersky Power Plant.

Crossing X-10 has now been selected as the preferred crossing alternative. While this report provides documentation of the investigation and data collected for both the crossing corridors, all evaluations and recommendations herein are made regarding Crossing X-10.

Executive Summary Conclusions

Based on the information gathered during this investigation, subsurface conditions appear variable in composition and thickness in the upper levels of the X-10 borings and become more consistent with depth as bedrock is approached. The subsoils generally consist of variable fill soils underlain by a thin fill layer of gravelly sand. Underlying the gravelly sand or fill is a relatively thick silty clay layer. The silty clay layer is underlain by clay or granular hardpan that extends to limestone and dolomitic limestone bedrock. The bedrock interface is generally characterized by a thin zone of low Rock Quality Designation (RQD) rock (<75%) that has intermittent layers of fragmented bedrock with gravel and silty clay. Underlying the low RQD bedrock at the interface, is competent (>75% RQD) limestone and dolomite bedrock extending to the explored depths.

Based on the results of the investigation, the existing fill deposits at both crossing locations are highly variable and are not considered suitable for support of any foundation elements.

The underlying silty clay or granular soils are not considered suitable for support of the heavy loading expected from primary or secondary bridge foundation elements, but may be sufficient for support of ancillary structures with light-to-moderate foundation loads. For the purposes of this document, primary foundation elements are defined as the main structural foundation for cable stay and suspension bridges and the anchorages for the suspension bridge. Secondary foundation elements are defined as foundation elements for the approach roadway piers. Ancillary structures include bridge approach elements such as retaining walls, signage foundations, etc.

Based on the overall evaluation of the subsurface data obtained during this investigation and consideration of the project background information, it is anticipated that deep foundation systems will be required to support primary and secondary bridge elements. Such systems may consist of large concrete elements cast within a deep shaft and bearing on sound bedrock, drilled straight-shaft concrete-filled caissons, or piles. The hardpan soils underlying both corridors are considered well suited for the heavy foundation loading anticipated from proposed secondary structural elements of the bridge using deep foundation elements. The list below summarizes the nominal pile driving resistance values (R_{NDR}) for pipe piles and H-piles recommended in the Michigan Department of Transportation (MDOT) Bridge Design Manual (BDM). The dynamic resistance factor (ϕ_{DYN}) presented by the MDOT BDM is equal to 0.4, and assumes that pile driving criteria will be developed by using the Federal Highway Administration (FHWA) modified Gates Dynamic formula. We recommend considering the use of dynamic testing in developing driving criteria. If American Association of State Highway and Transportation Officials (AASHTO) guidelines for dynamic testing are followed, a dynamic resistance factor of 0.65 may be used instead of 0.4.

Pile	R _{NDR} (tons)
12" O.D. (0.25" wall)	175
14" O.D. (0.312" wall)	200
14" O.D. (0.438" wall)	250
HP12x53	200
HP14x102	400

If a drilled pier bearing on the hardpan is used, a nominal resistance value of 40 tons per square foot (3.8 MPa) can be used if a settlement of approximately 5% of the shaft end diameter is acceptable. A resistance factor of 0.55 should be used with the drilled shaft geotechnical design.

The upper, highly weathered bedrock (<75% RQD) underlying the hardpan soils is generally considered suitable for the heavy foundation loading anticipated from primary and secondary foundation elements of the bridge, although bearing capacities any higher than for the hardpan (as discussed above) are not recommended.

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The competent bedrock (>75% RQD) underlying the hardpan soils and the weathered bedrock is well suited for the heavy foundation loading anticipated from primary and secondary foundation elements of the bridge.

The anticipated heavy foundation loading for proposed primary foundation elements may involve massive elements cast within circular concrete shafts or drilled concrete piers (also known as drilled caissons). Such foundation elements would be designed to extend through the upper fill, silty clay, granular soil layers, hardpan soils, and be founded on competent bedrock at least 5 feet (1.5 meters) into the competent limestone/dolomite bedrock, resulting in depths of approximately 100 to 120 feet (30 to 34 meters). Estimated load-settlement behavior is provided for drilled pier diameters of 2.5 meters (8.2 feet) and 3.3 meters (10.8 feet) at rock socket lengths of 5 feet (1.52 meters), 10 feet (3.05 meters), and 15 feet (4.57 meters). The ultimate nominal end resistance is approximately 300 tsf (28.7 MPa) while the ultimate nominal shaft side resistance in the bedrock is approximately 10.6 tsf (1.0 MPa). However, because the skin friction mobilizes at small strain, while the end resistance mobilizes at large strain, the ultimate values should not be summed to estimate the total resistance. Rather, a load resistance factor design (LRFD) procedure is outlined to estimate the total resistance that accounts for strain incompatibility. For the evaluation presented herein, an end resistance factor of 0.5 and a shaft side resistance factor of 0.65 are recommended, based on AASHTO and FHWA guidelines. If during final design, shaft side and end resistance values are obtained through the use of field load tests, the resistance factor for both end and shaft side resistance can be increased to 0.8.

Pipe piles to support the suspension bridge anchorage and/or main towers were also evaluated and could consist of 30-inch diameter reinforced concrete filled steel pipes. The pipe piles would be pre-drilled and driven to bear on or immediately above the bedrock, a reinforcing steel cage would then be placed within each pile, and then filled with concrete. For the concept design, it can be assumed that the bedrock end bearing resistance will be mobilized within a settlement of up to 5% of the pipe diameter, which will occur primarily as elastic settlement.

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The nominal pile driving resistance values for vertical and battered (3V:1H) 30-inch pipe piles is summarized below. The values assume plugged conditions at the pile tip. The MDOT BDM presents a dynamic resistance factor (ϕ_{DYN}) equal to 0.4, which assumes that pile driving criteria will be developed by using the FHWA-modified Gates Dynamic formula. We recommend considering the use of dynamic testing in developing driving criteria. If AASHTO guidelines for dynamic testing are followed, a dynamic resistance factor of 0.65 may be used instead of 0.4.

		R _{NDR} (tons)	
Pile	Axial	Vertical Comp.	Horizontal Comp.
30" O.D. (0.625")	990	939	312

This summary is general in nature and should not be considered apart from the entire text of the report with all the qualifications and considerations mentioned herein. All interpretations are for United States (US) side only and for Crossing X-10. It is also noted that the analysis and interpretations herein are with respect to the general feasibility and concept design for the bridge foundations. It is understood that the once the final design is undertaken, a more detailed geotechnical investigation and analysis will be conducted that will include additional test borings and laboratory testing. The additional investigation will consist primarily of additional soil and rock test borings (vertical and angled rock coreholes), in situ rock testing, in situ permeability testing for rock grouting design (if determined to be necessary), and additional rock core laboratory testing.

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1.0 INTRODUCTION

The Detroit River International Crossing (DRIC) Study is a bi-national effort to complete the environmental study processes for the border crossing between Detroit, Michigan and Windsor, Ontario for the United States, Michigan, Canada and Ontario governments. The study proposes solutions that support the region, state, provincial and national economies while addressing civil national defense and homeland security needs of the busiest trade corridor between the United States and Canada (Figure 1-1).



Figure 1-1: Detroit River International Crossing Study, Existing Detroit River International Crossings

Source: The Corradino Group of Michigan, Inc.

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The DRIC Draft Environmental Impact Statement (DEIS), previously released under separate cover, analyzes issues/impacts on the U.S. side of the border of the end-to-end crossing system over the Detroit River between Detroit, Michigan, and Windsor, Ontario, Canada. The alternatives are comprised of three components: The crossing, plaza (where tolls are collected and Customs inspections take place), and interchange connecting the plaza to I-75 (Figure 1-3). Figure 1-3 illustrates the approximate location of the Canadian sinkhole relative to the Detroit River International Crossing study.



Figure 1-3: Detroit River International Crossing Study, Canadian Sinkhole

Source: URS Canada

1.1 PURPOSE OF THE REPORT

The purpose of the report is to present the details of our site specific investigation, and to provide study-level geotechnical analysis and recommendations with respect to the proposed construction concepts under consideration for the Detroit side of the proposed Detroit River International Crossing Bridge between Detroit, Michigan and Windsor, Ontario. This current report is not intended as a stand-alone document, and is intended for the use by the Corradino Group and Parsons Transportation to develop preliminary foundation concepts for the bridge design, as well as to evaluate probable construction costs and other impacts related to the project.

This report was developed to be placed in the Appendix of the complete Bridge Engineering Report, itself an attachment of the final DRIC Engineering report, which is part of the overall DRIC Environmental Impact Statement Report.

When this geotechnical investigation was undertaken (April 2008), two potential crossing corridors were under consideration, defined as Crossings X-10 and X-11 as shown on the attached Figure Nos. 1A and 1B, respectively, in Attachment A. The two subject crossings are in the same general vicinity, between the Ambassador Bridge, and Zug Island in southwest Detroit. The Crossing X-10 corridor generally consists of the area immediately north of Zug Island to historic Fort Wayne along the banks of the Detroit River. The Crossing X-11 corridor generally consists of the area along the banks of the Detroit River immediately north of historic Fort Wayne to the existing Mistersky Power Plant.

Crossing X-10 has now been selected as the preferred crossing alternative. While this report provides documentation of the investigation and data collected for both the crossing corridors, all evaluations and recommendations herein are made regarding Crossing X-10.

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2.0 SITE CONDITIONS

The two subject crossings are in the same general vicinity, between the Ambassador Bridge and Zug Island in southwest Detroit, and are described as follows:

2.1 X-10 CROSSING CORRIDOR

The Crossing X-10 corridor generally consists of the area immediately north of Zug Island to historic Fort Wayne along the banks of the Detroit River. The area is generally flat with a slight drop in elevation at the river, with large vacated areas, parking lots, and paved/unpaved roads. Current land use includes light-to-moderate industrial areas, including a cement terminal, a major trucking terminal, truck ferry operation, and aggregate storage areas. Residential areas exist north of Jefferson Avenue, but are generally intermingled with light commercial and industrial uses. Historic land use includes light-to-heavy industrial areas, including a major chemical processing plant and power plant operations, along with two suspected solution well operations identified during the literature search portion of the Brine Well Program. Known solution mining wells exist outside the influence zone of the crossing X-10 primary and secondary bridge elements; adjacent to the Rouge River along the south portion of the corridor, as well as possible undocumented solution mining wells adjacent to the current Fort Wayne property. Historic maps also indicate the original shoreline of the Detroit River to be set back approximately 16 to 80 feet (5 to 24 meters) from its current position, with potentially abandoned and buried docks and former boat slips prevalent throughout.

The X-10 Crossing bridge foundation borings (TB-101 through TB-107) were located on the former Detroit Coke site, which was used for coke oven and coke oven gas byproducts operations from early 1900 until 1991. The project site has been the subject of environmental investigations performed by others not related to this project. The property is now subdivided between several owners including Lafarge, Inc., McCoig Concrete (Koenig/Michigan Foundation, Inc.), the Detroit Economic Growth Corporation (DEGC), and Yellow Trucking, Inc.

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Boreholes TB-102 through TB-104 are located on the Lafarge North America Detroit Cement Terminal facility located at 1301 Springwells Court along the Detroit River northeast of the McCoig aggregate facility. The Lafarge property consists primarily of open land with a storage silo and related structures adjacent to the Detroit River.

The remaining X-10 Crossing boreholes, TB-101 and TB-105 through TB-107, were drilled on property that is currently vacant land owned by the DEGC. This property is sparsely vegetated and mostly open, with several overhead and underground utilities.

2.2 X-11 CROSSING CORRIDOR

Although the X-11 corridor is no longer under consideration for this project, one of the test borings was drilled in this area, and as such, some site information is provided.

The Crossing X-11 corridor generally consists of the area along the banks of the Detroit River immediately northeast of historic Fort Wayne and southwest of the existing Mistersky Power Plant; between Jefferson Avenue and the river. The area is generally flat and vacant, with a slight drop in elevation at the river. Current land use in the immediate area includes light to moderate industrial regions, the Mistersky power generation facilities, and some residential use to the north. The residential areas exist north of Jefferson Avenue and are generally intermingled with light commercial and industrial areas. Historic land use includes light to heavy industrial areas, including a major copper and brass fabrication operation, along with two suspected solution well operations identified during the literature search portion of the Brine Well Program. Historic maps indicate that potential solution mining operations exist directly to the north of the historic copper and brass fabrication facility and the northern portion of the corridor, in what is now intermingled residential and commercial areas. Based on the results of the brine well investigation, it has been confirmed that no brine wells exist within the influence zone of the primary elements of the proposed bridge alignment. Historic maps also indicate the original shoreline of the Detroit River in the X-11 area to

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be set back approximately 10 to 50 feet (3 to 15 meters) from its current position, with possible docks and former boat slips prevalent throughout.

Test boring TB-108 was drilled on the former Revere Copper and Brass (Revere) site located at 5851 West Jefferson Avenue. The Revere site exists as vacant, unoccupied land with some lightly wooded areas south of West Jefferson, to the southeast of the Mistersky Power Plant. Previous environmental investigations conducted by others not related to this project have identified the Revere site as impacted with the byproducts of former industrial use.

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3.0 PROJECT BACKGROUND

3.1 REGIONAL SALT AND SOLUTION MINING ACTIVITIES

Salt (halite) has historically been solution mined in the area of the X-10 Crossing Corridor. As part of this solution mining process, fresh water was injected into the ground, natural salt beds were dissolved, and the resulting brine was brought to the surface and evaporated to make salt. The solution mining of salt layers ranging from approximately 900 to 1,600 feet (270 to 500 meters) below the ground surface was typically conducted in an uncontrolled method before standardized record keeping was common practice. This created underground cavities of unknown location, size, and shape. A solution mining cavity collapsed to the surface and formed a sinkhole on the Windsor side of the study area in 1954. At least two additional sinkholes occurred at Point Hennepin (on Grosse Ile) south of the DRIC crossing site on the U.S. side of the river. Also, settlement of several feet was observed near the Wyandotte, MI brinefield location to the south of the X-10 Corridor.

3.1.1 Brine Well Program

The Brine Well Program was developed to delineate the size, locations, and shape of potential brine well cavities in the X-10 and X-11 Crossing corridors on the U.S. side of the river. Approval was obtained from MDOT for a combined geophysical and geotechnical program, which included the drilling of 13 deep-rock borings in combination with cross-well seismic imaging. Forward modeling of geophysical methods in conjunction with preliminary historical rock mechanics analysis, borehole gravity, vertical seismic profiling (VSP), and downhole wireline logging were also included in the program.

Based on criteria established by MDOT in January 2006 and further defined at a June 2006 Geotechnical Advisory Group meeting, the proposed bridge in Corridors X-10 and X-11 requires: 1) foundations be located outside of the influence of any rock cavities that could have impact on the foundations, including those produced by solution mining

activities; and, 2) foundations be built on competent bedrock. The brine well investigation program was developed and implemented to define conditions in the corridors to determine if these criteria could be satisfied.

A total of 12 cross-well seismic imaging profiles were performed for the X-10 corridor. The processed cross-well images and other geophysical data show two anomalies of interest, neither of which are of significant concern. A report detailing the findings was presented under separate cover in the DRIC DEIS, as mentioned previously.

3.1.2 Rock Mechanics Investigation Results

A preliminary model of geotechnical rock mass characteristics was also completed as part of the Brine Well Program to evaluate the potential instability of possible solution cavities of similar shape and size of anomalies discovered during the geophysical investigation program. The evaluation was also based on review of the historical instability of existing solution cavities in the Detroit-Windsor vicinity and on the results of a three-dimensional, distinct-element (3DEC) analysis of suspected or potential solution cavity geometry.

3.1.3 Combined Geophysical Investigation and Rock Mechanics Results

Based on the observations made during the deep drilling and subsequent geophysical investigations, there is no evidence of cavities of concern in the X-10 Crossing Corridor, nor evidence of potential instability of the rock mass. In fact, the analysis shows that the observed anomalies have probably been filled by one or a combination of several mechanisms. In addition, even for the largest of the anomalies located, and assuming an unfilled cavity, the analysis shows the anomaly is stable and will not progress upward any significant distance to affect a nearby bridge foundation.

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3.2 SUMMARY OF ENVIRONMENTAL CONDITIONS IN X-10

Based on experience along the Detroit River shoreline and within the Detroit River sediments, some environmental issues will be present for any excavations along the United States shorelines and within the upper 5 to 10 feet (1.5 to 3 meters) of river sediment. Along the shoreline, fill soils to depths of 5 to 30 feet (1.5 to 9 meters) from previous activity are typically contaminated requiring disposal in Type II landfills. Within the river, sediments along the river bottom are also typically contaminated increasing in risk and contamination levels especially south of the downtown Detroit area.

The former Detroit Coke Site, originally owned by the Solvay Processing Company (Solvay), occupies most of the Crossing X-10 landing area between Jefferson Avenue and the Detroit River. The Detroit Coke Site was used for coke oven and coke oven gas by-products operations from early 1900 until 1991. Due to the presence of regulated deep underground injection wells in the western part of the property, it was also identified as a Resource Conservation and Recovery Act (RCRA) facility. Associated environmental impacts with the coke oven and coke oven gas by-products operations included tar, free phase hydrocarbons (free product), and soil and groundwater contamination. Almost the entire site has been impacted by the former industrial operations.

Previous investigations by others have indicated site soils are contaminated with volatile organic compounds (VOCs), semi volatile organic compounds (SVOCs), ammonia, cyanide, and metals at concentrations exceeding the Michigan Department of Environmental Quality (MDEQ) industrial criteria for indoor and ambient air, direct contact, particulate inhalation, and surface water protection. Site groundwater is contaminated with VOCs, SVOCs, ammonia, cyanide, and metals at concentrations exceeding the MDEQ industrial criteria for indoor air, direct contact, and surface water protection. Previous investigations have also indicated significant soil and groundwater contamination, including possible free phase coal tar. The underlying clay layer vertically confines the contamination (aquitard). The Michigan Department of Environmental

Quality (MDEQ) and the former site owner (who is under a consent decree to implement remedial actions) have both expressed concern of the potential for future construction to allow existing contamination to vertically migrate through the aquitard.

Honeywell, the current owner of the Detroit Coke Site and the RCRA primary responsible party, has installed a demarcation membrane in certain areas, and approximately 6 to 12 inches (15 to 30 centimeters) of clean fill material has been placed over the membrane to prevent contact with the impacted soil. However, this membrane and clean fill layer may not be present throughout the entire site as observed during field work activities. Honeywell has also installed groundwater collection trenches to limit impacted groundwater from discharging to the Rouge River and/or Detroit River.

The site may also have been used as a brine well processing facility, without any documented environmental impacts attributed to that operation.

3.2.1 X-10 Disposal Wells

Research for Brine Well Investigation Program uncovered the existence of three previously operated deep-injection disposal wells on the former Solvay/Honeywell (Crossing X-10) parcel. The wells were drilled from 1969 to 1978 to depths of greater than 4,000 feet (1,219 meters). The wells were used to inject hazardous waste into permeable formations (Munising Group) deep within the ground. Wells #1 and #3 were plugged and abandoned according to the Michigan Department of Environmental Quality (MDEQ) and court records, resolving how the operators of the hazardous waste injection operation were prosecuted for illegal activities. Well #2 was reportedly plugged during the winter of 2008, according to the Michigan Department of Environmental Quality (MDEQ). These former deep-injection disposal locations are presented on Figure No 1A in Attachment A.

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3.3 REGIONAL GEOLOGIC SUMMARY

The generalized subsurface geology for the area is summarized in this section of the report.

3.3.1 Overburden

The bedrock in the project corridors is overlain by glacial drift soils, which have been deposited either directly by glacial ice (till), glacial meltwater streams (glaciofluvial deposits), or impounded glacial lakes (lacustrine deposits). The upper soil formations along the alignment generally consist of a relatively thick mantle of Wisconsin-aged lacustrine clays (10,000 to 50,000 years ago) that, with the exception of the near-surface deposits are typically very soft to soft in consistency. The lacustrine soils were deposited as sediments from a series of glacial lakes impounded between the ice front and the Inner Defiance Moraine located near the northwest corner of Wayne County. The upper 10 to 20 feet (3 to 6 meters) of these deposits (where still present) have been desiccated during historical low-water periods, resulting in soils of very stiff to hard consistency near the surface. The clay soils frequently contain intermittent sand and gravel layers that were produced from glacial rivers carrying coarser sediments as lake levels fluctuated. Localized alluvial soils are present along existing rivers and streams that drain the inland areas. In some locations, lake shorelines are identified by relatively thick layers of sand and gravel.

The lacustrine deposits are typically underlain by a thin layer of highly over-consolidated glacial till, generally consisting of sand, silt, and gravel within a matrix of clay. This formation is locally termed "hardpan" and usually overlies the bedrock formation. Depending on the amount of clay binder contained in the hardpan, the material may range in nature from cohesive to granular. The hardpan is generally believed to be from the Illinoian Ice Age (200,000 years ago) and can also contain calcium carbonate producing a cemented condition. Given the glacial origins of the hardpan layer, occasional cobbles and large boulders are typically present in this layer. Methane and hydrogen sulfide gas may also be encountered in this layer.

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The total glacial drift thickness along the X-10 Crossing Corridor varies from approximately 94.5 to 99 feet (Elevations 483 to 494 feet; or 28.8 to 30.2 meters). The surface topography was formed during the Wisconsin stage (youngest) of Pleistocene Series glaciations of the Cenozoic Era, and has been somewhat modified by surface erosion since that time.

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3.3.2 Hydrogen Sulfide Occurrence and Project History

Hydrogen sulfide is a colorless gas that smells like rotten eggs and can paralyze the sense of smell (olfactory paralysis). Olfactory paralysis then prevents the recognition of its presence. Hydrogen sulfide in the atmosphere can blacken exposed materials and irritate the eyes, causing them to become swollen or very sensitive to light.

Hydrogen sulfide gas has a history of occurrence in the DRIC Crossing areas. On many recent and historical projects, the gas has caused toxic conditions during deep excavation and tunneling operations, even causing death to construction workers in some cases.

The Southwest Intake Rock Tunnel was constructed in 1957 as part of a water intake system for the southern portion of Wayne County. The tunnel was constructed within the bedrock from an intake structure located in the Detroit River at the middle portion of Fighting Island approximately 4.5 miles (7.2 kilometers) downriver from the X-10 bridge alignment. From the intake structure, the tunnel was mined to a shore shaft on the west bank of the Detroit River approximately 1 mile (1.6 km) away. The tunnel consists of a 12-foot (3.7 m) finished inside diameter rock tunnel which varies in depth from approximately 127 feet at the intake structure to 176 feet at the shore shaft, corresponding to Elevation 445 to 400 feet (38 m to 65 m, corresponding to Elevation 135 m to 122 m). The tunnel is located approximately 65 to 115 feet (20 to 35 m) below the rock/soil interface at the intake and shore shafts, respectively. The tunnel was constructed using drill-and-shoot methods to excavate the rough opening.

Based on discussions with an individual who worked on the project, the major problems encountered during construction were the presence of hydrogen sulfide gas together with large inflows of groundwater. During construction, the contractor's personnel were reportedly required to wear gas masks. Further, due to the highly fractured nature of the bedrock, inflows of groundwater and hydrogen sulfide required significant grouting activities during the construction. These efforts, although not entirely effective, provided sufficient control to complete the construction of the rock tunnel. It should be noted that

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during construction of the related shore tunnel to the west of the rock tunnel, a worker died after being overcome by hydrogen sulfide gas.

Several other projects that involved excavation into the bedrock have been constructed within 1 to 2 miles (1.6 to 3.2 km) of the project site, mostly related to the nearby Detroit Water and Sewerage Department's wastewater treatment plant. These projects include the Pump Station 2A project (1991), Pump Station 1 project (1960s), and the DRO2 project (construction terminated in 2004 due to catastrophic inflow of hydrogen sulfide laden groundwater into the unfinished tunnel). For each of these projects, dissolved sulfide levels within groundwater were reported to be 80 ppm or higher. It is valuable to note that contaminated water inflow was controlled effectively by pre-excavation grouting of the bedrock. In the case of the DRO2 project, the uncontrolled inflows occurred during tunneling in which grouting was performed during the excavation (i.e., pre-excavation grouting was not performed).

3.3.3 Bedrock

The proposed crossing corridor is located at the geologically-termed southeast margin of the Michigan Basin geomorphic province and within the Erie-Huron glacial lowland. The Michigan Basin is termed as such due to the structural basin shape of the bedrock, in which layers of Paleozoic era sedimentary rock that overlay the Precambrian Basement Complex, dip inwards to the center of the Lower Peninsula of Michigan from each direction as a series of bowls. The youngest layers of bedrock are first encountered in the center of the state, with older rock layers progressing outwards to the outer margins.

The Michigan Basin was initially formed during the early Cambrian Period, when the remnants of the mountains formed during the Cambrian-Penokean Orogeny remained in a belt extending from Ontario, Canada, across the central part of the Upper Peninsula to present-day Wisconsin. The erosion of these "northern highlands" began the series of depositions and erosions that constitute the modern basin. The later effects of the Appalachian Orogeny likely caused the structural deformation and localized downward movement in what had been a relatively stable interior continental region.

As a result, several intracratonic structural basins were formed throughout the central lowland areas of North America creating arches and domes. The Michigan Basin is bounded on the west by the Wisconsin Arch and Wisconsin Dome; on the north and northeast by the Canadian Shield; on the east and southeast by the Algonquin Arch in Ontario and the Findlay Arch in Ohio; and by the Kankakee Arch in northern Indiana and Illinois.

Based on the position of Detroit, Michigan, along the southeast rim of the Michigan Basin, the Paleozoic rocks that comprise the basin in this area typically dip to the northwest, with each formation being buried by successive younger formations in the direction of the dip. The regional dip is slight, and is estimated at approximately 30 to 50 feet per mile (6 to 10 meters per kilometer).

The topography of the bedrock surface within the area is somewhat variable and characterized by numerous irregular features in the bedrock surface. These features include many synclinal and anticlinal structures believed to have developed before the Pleistocene Epoch and subsequently modified by repetitive glacial action. The bedrock features also include the existence of ancient stream valleys and numerous healed faults that cut the bedrock surface. Based on historical information, the bedrock features are understood to be fairly broad, and become narrow as they reach the terminus of the Erie/Huron Lowlands.

Due to the movement of the earth's crust, these strata are seamed and fissured with vertical and horizontal joints that permit movement of ground water. Where carbon dioxide dissolved within the groundwater-filled cracks, solution cavities typically developed within the limestone, and to some degree within the dolomite. Both the limestone and dolomite formations are known to contain dissolved sulfides, which can produce hydrogen sulfide gas upon exposure to atmospheric conditions. The natural decay of organic compounds that also existed within the ancient seas became trapped within cavities formed in the limestone and dolomites and is evident today as petroleum,

carbon monoxide, and methane. Small amounts of petroleum found within the limestone and dolomite tend to cause discoloring, staining, and associative odors.

3.4 GROUNDWATER

The near surface granular deposits and fill layers in the Detroit area typically contain groundwater, which is perched above the underlying clay strata. This groundwater forms an intermittent unconfined aquifer, which varies seasonably in depth and extent. In addition, confined groundwater is often contained within relatively thin granular layers that are occasionally present within the thick cohesive deposits and / or hardpan present throughout the corridor areas. Such confined aquifers are usually limited in extent, and therefore, have limited recharge capabilities. However, surficial granular layers near the Detroit River shoreline can obtain hydraulic communication with the river, sometimes requiring extensive dewatering programs as discussed later during the restoration of the TB-108 location.

Groundwater in the X-10 Crossing Corridor can typically be distinguished according to its chemical constituency and can be sub-divided into fresh and mineral in the explored depths.

3.4.1 Fresh Groundwater

Fresh water is potable and is free of any deleterious, naturally-occurring chemicals or dissolved salts or solids. Fresh water aquifers generally exist in the upper glacial drift. In the project area, the fresh water aquifer is discontinuous, and often contaminated as a result of human activities. Where the fresh water aquifer is present in the glacial drift, groundwater generally flows toward the Detroit River, which generally behaves as a regional discharge feature. The deepest freshwater aquifer in the explored area is considered to be the base of the glacial drift.

3.4.2 Mineral Groundwater

Mineral water contains dissolved minerals or constituents that may alter to gas upon being exposed to the atmosphere. The dissolved compounds of interest expected for this investigation consisted of hydrogen sulfide, methane, and carbon monoxide which exist naturally in some mineral ground waters. Mineral waters are common in the lower glacial drift (hardpan) and upper Middle Devonian bedrock (Dundee Limestone and Detroit River Group - Lucas Formation). In the project area, mineral waters are found in the hardpan and bedrock, and typically exhibit flowing artesian conditions when encountered.

3.5 REGIONAL SEISMOLOGY

According to historical seismic risk maps published by the United States Geodetic Survey, Michigan is located within Seismic Risk Zone No. 1 and, as such, posses a relatively low risk for earthquake occurrence. While tremors from earthquakes with epicenters in other regions have been recorded in Michigan, only 34 earthquakes with epicenters in Michigan have been recorded since 1872. With the exception of two seismic events that occurred in the Keweenaw Peninsula at the turn of the 20th century, all recorded events had recorded intensities of less than IV on the modified Mercalli scale. This corresponds to approximately magnitude 4.7 on the Richter scale.

According the Geologic Survey Division of the Michigan Department of Environmental Quality, the majority of the above referenced seismic events resulted in slippage along deep-seated Pre-Cambrian Faults and is not believed to involve slippage along the faulting of the overlying Paleozoic units.

4.0 PROPOSED CONSTRUCTION

The geotechnical investigation was planned and carried out to provide a general understanding of the feasibility and concept-level design requirements for bridge concepts as presented in the Detroit River International Crossing, Bridge Conceptual Engineering Report, Revised February 2008 by Parsons Transportation.

4.1 BRIDGE CONCEPTS

The project team has developed the crossing concept as a three-lane each way (for a total of 6 lanes with shoulders and a central median) clear-span bridge with an anticipated clearance of approximately 133 feet (40.5 m) at the river's edge. The bridge is anticipated to be engineered for restriping from six to eight lanes with a center median. The overall structure is designed to achieve a 120-year structure life, with replaceable components being designed for those not able to achieve the intended lifespan.

Given the navigational requirements, the bridge is anticipated to be a suspension bridge or cable-stayed bridge with primary piers on or near the shoreline. For the purposes of this document, primary foundation elements are defined as the main structural foundation elements for cable stay and suspension bridge towers. Primary piers would most likely be supported on drilled concrete piers (also termed "drilled caissons").

Suspension bridge anchorages would be located approximately 1,000 to 1,500 feet (300 to 450 meters) behind the primary piers, which would be located at or near the river's edge. Foundations for suspension bridge anchorages would be on competent bedrock. These foundations could be constructed as large-diameter sunken caissons.

Secondary foundation elements would consist of concrete piers supported on deep foundations. Secondary foundations are defined as elements for the approach roadway piers to the bridge. The final spacing of the secondary foundation elements would be developed during the final design, but for the purposes of the concept evaluation, it is anticipated that these elements would be approximately 100 to 200 feet (30 to 60 meters) apart.

4.1.1 Suspension Bridge

For a suspension bridge, the main cables are constructed between two large towers or piers and are anchored to massive anchorage structures often founded on bedrock. These cables form the primary load-bearing structure for the bridge deck. The cables are under tension from only their own weight before the deck is placed. Suspender ropes support the deck from the main cables. The tension on the cables is then transferred to the earth via the anchorages.

A diagram of a typical suspension bridge can be found below in Figure No. 4-1 and also in the attached Figure No. 38 in Attachment A.



Figure No. 4-1: Typical Suspension Bridge Structure (Modified from Bridge Conceptual Engineering Report, Rev Feb 2008 by Parsons)

4.1.1.1 Currently Proposed Alignment X-10 Suspension Bridge

The proposed Detroit side pylon is approximately 463 feet (141 m) tall with respect to the top of footing and is located on land adjacent to the existing rail spur currently servicing the Lafarge Terminal. The main span consists of an approximate 2,800-foot (855 m) suspended deck, with a US-side approach backstay span of nearly 830 feet (253 m). The planned main anchorage is located to the north of Springwells Court, on the Lafarge property, and is proposed to resist the suspension cable pull through a combination of dead weight, passive soil resistance, and direct load transfer to underlying bedrock.

4.1.2 Cable-Stayed Bridge

In the cable-stayed structure, the main piers or pylons are considered the primary foundation elements. The bridge deck is supported by steel cables running directly from the deck structure to the towers. The pylons are currently envisioned in two forms, namely an A-Frame and Inverted Y shape to generally limit second order effects and to increase resistance to wind forces. A diagram of a typical cable-stayed bridge is presented below in Figure No. 4-2 and also in the attached Figure No. 37 in Attachment A.



Figure No. 4-2: Typical Cable-Stayed Bridge Structure (Modified from Bridge Conceptual Engineering Report, Rev Feb 2008 by Parsons)

4.1.2.1 Currently Proposed Alignment X-10 Cable-Stayed Bridge - This option consists of a 2,760-foot (840 m) main span with symmetric 1,050-foot (320 m) side spans. The heavier concrete box girder allows the side spans to be shorter than one half the main span length and act as counterweights when loaded with traffic, effectively eliminating uplift on the anchor piers. Two pylon configurations have been proposed, with both extending approximately 820 feet (250 m) above their footings. The proposed pylon is located on land adjacent to the existing rail spur currently servicing the Lafarge Terminal.

4.2 FOUNDATION CONCEPTS

Similar to the bridge concepts, the Bridge Conceptual Engineering Report developed general concepts for foundations for both the general bridge types being proposed.

4.2.1 Suspension Bridge Foundation Concepts

As discussed previously, the suspension bridge concept includes two major types of primary foundation elements; pier foundations supporting the main towers; and anchorages resisting the tensile forces from the main cables.

4.2.1.1 Suspension Bridge Towers

The proposed tower foundations generally consist of drilled piers and a concrete footing (or pier cap) at the base of each tower leg. A tie-beam then connects the two adjacent pier caps. The footings typically consist of regularly reinforced mass concrete and are generally poured in a single monolithic pour at each tower location. A diagram of a typical tower foundation is presented below in Figure No. 4-3.





4.2.1.2 Suspension Bridge Anchorages

The proposed suspension bridge anchorages for this project consist generally of a mass concrete anchorage block with a splay chamber for each cable where it is secured to anchor rods. These anchorages are gravity-type anchorages, which use the dead weight of the concrete to resist the pull of the main cables. The anchorages are generally assumed to be founded on bedrock with longitudinal resistance to the main cable provided by direct transfer to the bedrock and conservative estimates of passive soil and rock resistance. An example of a typical suspension bridge anchorage is presented below in Figure No. 4-4.


Figure No. 4-4: Typical Suspension Bridge Anchorage Structure (Modified from Bridge Conceptual Engineering Report, Rev Feb 2008 by Parsons)

4.2.2 Cable-Stayed Bridge Foundation Concepts

The Bridge Conceptual Engineering Report considered the following general foundation types for the cable stayed bridge option.

4.2.2.1 Pylon Foundations

The proposed pylon foundations generally consist of drilled piers and a footing (pier cap) at the base of each tower leg. A tie beam then connects the two adjacent pier caps. The footings typically consist of regularly reinforced mass concrete. A diagram of a typical pylon foundation is presented below in Figure No. 4-5.

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4.2.2.2 Anchor Piers

Conceptually, anchor piers for a cable-stayed bridge system consist of drilled shafts with a respective cast in place footing or pier cap. At this time, the footing is envisioned to be reinforced concrete and located entirely below grade. The subsequent pier is constructed using solid cast in place concrete columns with multiple lifts and splicing of column reinforcing steel. This type of anchorage is much smaller than for the suspension bridge concept.

4.2.3 Approach Piers

Approach piers for the roadway are independent of chosen span type and are currently envisioned to be constructed similar to the procedure described in Section 4.2.1.1.

Generally, the piers are founded on deep foundations consisting of drilled caissons (piers) resting on or socketed into bedrock (possibly hardpan soils depending on anticipated loading). Drilled shafts are conceptually 5 to 10 feet (1.5 to 3.0 m) in diameter depending on the anticipated loading.

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5.0 FIELD INVESTIGATION

The field portion of this investigation was conducted between April 24 and June 1, 2008 and consisted of drilling a total of eight test borings. Prior to drilling, clearances for underground utilities were obtained through the Miss Dig System. The test borings were designated as TB-101 through TB-108. The test borings were drilled to depths varying between 112.5 and 152.5 feet (34.3 and 46.5 meters), under the full time supervision of NTH field staff in liaison with the project engineer. The test boring locations were selected by Parsons Transportation Group and located in the field based on NTH's knowledge of the proposed and existing structures. The approximate as-drilled locations of the borings are shown in Figures 5-1 and 5-2, and again on the attached Test Boring Location Plans, Figure Nos.1A and 1B in Attachment A. Elevations of test borings were surveyed in the field after restoration activities, and as such, should be considered approximate.



Figure No. 5-1: Test Boring Location Plan for Crossing X-10.

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Figure No. 5-2: Test Boring Location Plan for Crossing X-11.

5.1 DRILLING PROCEDURES

The test borings were drilled with an ATV-mounted drilling rig using a combination of 3 ¹/₄-inch (8.3 centimeters) inside diameter (ID) hollow stem augers, 12 ¹/₄-inch (31.1 centimeters) wash rotary tri-cone rotary bits, NWJ rods, and NQ/NX wire-line diamond rock coring techniques. TB-101 through 107 and TB-108 were drilled in X-10 and X-11 Crossing Corridors, respectively, and were drilled for a total depth of 112.5 to 152.5 feet (34.3 to 46.5 meters). TB-102 was terminated at 112.5 feet (Elevation 468.5 feet) (34.3 meters, Elevation 142.8 meters) due to broken coring equipment blocking the borehole. TB-104 and TB-105 were offset several times from the originally drilled locations due to demolition debris encountered below the surface. At test boring location TB-104E, the surficial concrete obstruction was cored to advance the boring. The soil samples were

obtained by either the Standard Penetration Test Method or by advancing a thin walled Shelby tube into undisturbed soil using a standard Shelby tube sampler.

Once the underlying bedrock formation was encountered, wire-line rock coring techniques were used to extend the boring to its termination depth.

Soil and rock conditions encountered in each of the test borings have been evaluated and are presented in the Logs of Test Boring, Figure Nos. 3 through 10, attached in Attachment A. The boring logs present information relating to sample data, standard penetration test results, groundwater conditions observed in the borings, personnel involved, and other pertinent data. The logs included in this report have been prepared on the basis of laboratory testing, as well as visual classification of the soil and rock samples. Terminology used to classify subsurface conditions is presented as Figure No. 2A, General Notes, presented in Attachment A.

The stratification shown on the test boring logs represents the soil and rock conditions at the actual explored locations. Variations in subsoil conditions may occur between the borings. Additionally, the stratigraphic lines represent the approximate boundary between the soil or rock types; however, the transition may be more gradual than what is shown.

5.1.1 Standard Penetration Testing

The Standard Penetration Test (SPT) (ASTM D1586) consists of driving a 2.0-inch (5.1 centimeters) outside diameter split-barrel sampler into the soil with a 140-pound (63.5 kilograms) weight falling freely a distance of 30 inches (76.2 centimeters). The sampler is generally driven three successive 6-inch (15.2 centimeters) increments, with the number of blows for each increment being recorded. The number of blows required to advance the sampler the last 12 inches (30.5 centimeters) is termed the Standard Penetration Resistance (N) and is presented on the individual Logs of Test Boring. As added information, the blow counts for each 6-inch (15.2 centimeters) increment are also presented on the Logs of Test Boring.

The split barrel sampler generally contains a 1-3/8-inch (3.5 centimeters) inside diameter and 3-inch (7.6 centimeters) long liner insert. Soil samples recovered in these liners are designated as "LS" on the respective test boring logs, whereas samples recovered directly from the split barrel are designated as "S." All soil and rock samples obtained with the split-barrel sampler were sealed in jars and transported to NTH's laboratory for further classification and testing.

5.1.2 Thin-Walled Shelby Tube Sampling

Within two of the test borings, Shelby tube samples were obtained within the soft cohesive soil zones. The samples were obtained using a standard Shelby tube sampler and are considered relatively undisturbed. The sampling method consists of advancing a thin-walled steel Shelby tube into the soil using hydraulic pressure. The sampling was performed in accordance with the Standard Practice for Thin-Walled Tube Sampling of Soils (ASTM D1587). Standard Shelby tube samples are designated as "ST" on the Logs of Test Boring. Samples were sealed within the steel tubes and transported to NTH's laboratory for further classification and testing.

5.2 ROCK CORING AND TESTING PROCEDURES

The sampling procedures used within the rock portion of the test borings included continuous NQ wireline core sampling. Brief descriptions of these methods are presented in the following paragraphs.

Continuous rock core samples were obtained in general accordance with ASTM D2113. Diamond core drilling was accomplished at each test boring location with NQ/NX wire line core tooling. Double-tube, solid, swivel type core barrels with diamond tipped bottom discharge bits were used. Each core barrel was capable of obtaining a core run length of at least five feet with actual cores obtained in 0.7 to 10-foot (0.2 to 3 meters) coring runs.

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Upon removal of each core from the borehole, the cores were placed in boxes. All core boxes were provided with longitudinal separators and recovered cores were laid out from left to right and top to bottom. Spacer blocks or plugs were inserted into the core column to mark the beginning of each successive core run.

After placing the core into its respective box, the field engineer prepared selected samples for laboratory testing, then recorded the percent recovery, fractures per foot of run, prepared selected portions of the core for laboratory testing, lithology, and determined the Rock Quality Designation (RQD) value for each core run. The recovery is defined as the total length of core retrieved from the barrel divided by the total distance the barrel was advanced during coring. The RQD is defined as the total length of all intact rock core pieces greater than four inches in length divided by the total advanced distance. The fractures per foot were determined by the NTH field engineer immediately after sampling by evaluating fractures that appeared natural and were not obviously mechanical (caused by coring, handling of the core, or by intentional breakage).

A detailed Log of Core Boring for each core run was prepared based on visual classification and the logs are presented as Figure Nos. 11 through 18 in Attachment A. The Logs of Core Boring include information for each core run, including a detailed rock description, a description of fractures and mechanical breaks noted by the NTH field engineer at the time of sampling, results of sampling for hydrogen sulfide gas and methane during sampling.

The Logs of Core Boring present information relating to sample data, rock recovery, rock quality designation (RQD), personnel involved, and other pertinent data. Definitions related to the information presented on the Log of Core Boring are presented on the Summary of Rock Log Nomenclature, Figure No. 2B, attached in Attachment A.

The stratification shown on the Log of Core Boring represents the rock conditions at the actual explored locations. Variations in bedrock conditions may occur between the

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borings. Additionally, the stratigraphic lines represent the approximate boundary between the rock types; however, the transition may be more gradual than what is shown.

5.3 PIEZOMETER INSTALLATION

Following the completion of the rock coring in TB-101 and 102, pneumatic piezometers, designated as PZ-101 and PZ-102 were installed within the respective boreholes to provide continued groundwater information within the bedrock. The piezometers consist of two flexible tubes, separated by a porous stone and flexible rubber diaphragm exposed to the surrounding groundwater. The piezometers were installed in a sand pack within the bedrock at predetermined depths, sealed with a small section of bentonite plug, and then tremie grouted to the surface. Groundwater head was measured by applying a pressure across the flexible diaphragm through one of the tubes until air is observed escaping from the remaining tube. The pressure at which air returns to the surface was recorded as the equivalent of the groundwater pressure at that depth. The corresponding groundwater head was then calculated based on the tip elevation of the piezometer. Schematics containing the details of the piezometer installation and associated water level readings are attached as Figure Nos. 19 and 20 in Attachment A.

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6.0 LABORATORY TESTING

The details of laboratory testing for soil and rock are discussed as follows.

6.1 SOIL TESTING

Representative soil samples obtained from the test borings were subjected to laboratory testing to determine pertinent engineering characteristics. The testing program included the determination of the dry density and natural moisture content, unconfined compressive strength, Atterberg Limits, and particle size distributions of selected soil samples. All testing was performed in accordance with current ASTM standards. The dry density, moisture content, and unconfined compressive strength values are presented on the Logs of Test Boring, as well as on the Tabulation of Laboratory Test Data, attached as Figure No. 1 in Attachment A. The Atterberg Limits and particle-size distribution values are presented in the Tabulation of Laboratory Test Data. In addition, Grain Size Analyses are presented graphically as Grain Size Distribution Curves, Figure Nos. 2 through 15 in Attachment A.

6.2 ROCK TESTING

A total of 14 representative rock samples obtained during the field investigation were subjected to Uni-Axial Compressive Strength (UCS) testing. During these tests, on eight of the 14 samples, attempts were made to measure and record both the axial and lateral deformation of the sample. However, when the data was processed, it became apparent that the strain gauges had failed, apparently producing errant lateral deformation values. The purpose of this test was to measure the compressive strength and determine the Poisson's ratio of the rock (estimations of modulus of elasticity can be obtained from the testing), although this value was also determined as part of tri-axial testing and acoustic testing discussed below.

Two representative rock samples obtained during the field investigation were subjected to Tri-Axial Compressive Strength testing. This type of test simulates the behavior of rock underground, as significant confining pressure is applied to the sample during loading. Five tests are typically conducted for each test sample, each at a different confining pressure. The resulting sample strengths are then plotted on a stress difference versus axial strain curve, and on a Mohr circle, to compute the design strength parameters.

A total of 4 representative rock samples obtained during the field investigation were subjected to Indirect (Brazilian) Tensile Strength testing, which provide a measure of rock toughness, as well as tensile strength.

Two representative rock samples obtained during the field investigation, which were prepared for UCS testing, were subjected to testing where the velocities of compressive and shear ultrasonic waves passing through the core sample were measured. These values were used to calculate the elastic modulus and Poisson's ratio as an indication of the competency of the rock. From the wave velocities and the sample bulk density, the dynamic elastic modulus, and dynamic Poisson's ratio are then calculated.

The tests were performed at the Earth Mechanics Institute (EMI) of the Colorado School of Mines, Golden, Colorado, in general accordance with ASTM D7012, D 3967, and D 2845. All samples were tested at their "as-received" moisture content. The tests were performed on a number of selected rock core samples and were intended to determine the general nature of intact properties for the upper bedrock formations. The results of the rock tests are presented on EMI's rock testing summary and report, presented as Figure No. 36 in Attachment A. The results of testing are also presented on the individual Logs of Core Borings. Included within the EMI report are detailed test results, sample measurements, procedures, and photographs of the specimens after testing.

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7.0 SUBSURFACE CONDITIONS

Based on the information gathered during this investigation, it is determined that the subsurface conditions vary in composition and thickness in the upper soil horizons and become more consistent with depth. The subsoils generally consist of variable fill soils underlain by a thin fill layer of gravelly sand. Underlying the gravelly sand or fill is a relatively thick soft silty clay layer. The clay layer is underlain by clay hardpan that extends to bedrock. The bedrock interface is generally characterized by a thin zone of low RQD (RQD <75%) bedrock that has intermittent layers of fragmented bedrock with gravel and silty clay. Underlying the low RQD bedrock, is competent limestone/dolomite bedrock (RQD >75%) extending to the explored depths.

A generalized soil and rock profile is presented for illustration in Figure 7-1. This summary profile is intended for illustration only. For conceptual design or evaluation purposes, the reader should refer to the individual test borings.

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Figure No. 7-1: Generalized Soil and Rock Profile.

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7.1 SOIL CONDITIONS

As discussed in the Subsurface Investigation section of this report, soil conditions were investigated for both of the proposed crossing corridors.

7.1.1 X-10 Crossing Corridor

Test Boring TB-101 through TB-107 were drilled within the X-10 crossing corridor. Granular fill was encountered at each test boring location to depths of 14.5 to 21 feet (4.4 to 6.4 meters) below the existing ground surface, which corresponds to Elevations 560.2 to 573.1 feet (Elevations 170.7 to 174.7 meters). The fill typically consists of loose to very compact silty sand, gravelly sand, clayey sand, and sand. Coal fragments, coke, coke tar, ash, and construction debris (variable size brick and concrete fragments) were encountered in the fill along with cobbles and boulders. Soft organic clay was encountered in test boring TB-102 at depths between 17 and 22 feet (5.2 and 6.7 meters), and very soft organic clay was encountered in TB-104E at depths between 14.5 and 22 feet (4.4 and 6.7 meters)

Native granular soils were found to underlie the fill soils at test borings TB-101 through TB-104 and extended 27 to 56.5 feet (8.2 to 17.2 meters) below the existing ground surface, which corresponds to Elevations 560.6 to 525.7 feet (Elevations 170.9 to 160.2 meters). The native granular soils generally consist of loose to compact sand and silty sand. However, a layer of very compact silty sand was encountered in test boring TB-102 at a depth of 46 to 56.5 feet (14 to 17.2 meters), corresponding to Elevation 537.6 to 527.1 feet (Elevation 163.9 to 160.7 meters) and a layer of very loose silty sand was encountered in test boring TB-103 at a depth of 22 to 28 feet (6.7 to 8.5 meters), corresponding to Elevation 563.2 to 557.2 feet (Elevation 171.7 to 169.8 meters).

Native cohesive soils were encountered below the granular soils in test borings TB-101 through TB-104 as well as directly below the fill at test boring locations TB-105 through TB-107. Very soft to medium silty clay, sandy clay, and clay was encountered in each

test boring, extending to depths of 82 to 92 feet (25 to 28 meters) below the ground surface, corresponding to Elevations 499 to 490 feet (Elevations 152.1 to 149.4 meters). The very soft to medium clays are underlain by hardpan at test boring locations TB-101 through TB-106. The hardpan layer consists of very stiff to very hard silty, sandy, and gravelly clay and extends to bedrock, although a layer of very compact gravel and sand was encountered above the bedrock within TB-105A. Within TB-107, a layer of native granular soil consisting of medium compact gray clayey silt was encountered beneath the very soft to medium native cohesive soils.

7.1.2 X-11 Crossing Corridor

Test Boring TB-108 was drilled within the X-11 crossing corridor. Approximately 0.2 feet of topsoil was encountered at the surface in test boring TB-108. Fill was encountered below the topsoil and extended to a depth of 12 feet (3.7 meters) below the existing ground surface, corresponding to Elevation 565 feet (Elevation 171.6 meters). The fill consists of loose to medium compact silty sand and gravel, as well as very stiff sandy clay.

Below the fill in TB-108, native cohesive soils were encountered, consisting of soft to medium silty clay. This stratum extend to a depth of 87 feet (Elevation 490 feet), or (26.5 meters (Elevation 149.4 meters). The soft to medium clay is then underlain by a layer of stiff silty clay to 92 feet (Elevation 485 feet), or 28 meters (Elevation 147.8 meters). The cohesive soils are further underlain by very compact silty sand hardpan which extends to bedrock.

7.2 BEDROCK CONDITIONS

The bedrock consists of amorphous to fine-grained, fossiliferous limestone and dolomite of the Dundee Limestone and Detroit River Group (Lucas Formation). The bedrock extends to the explored depths of 113.5 to 152.5 feet (Elevations 470.2 to 429.2 feet) (34.6 to 46.5 meters, Elevations 143.3 to 130.8 meters). Within the test borings, bedrock consists of light gray amorphous limestone with some fossiliferous zones.

RQD values generally ranged from 50% to 100% with several notable exceptions, which are summarized in Table 7.1. Recoveries in the remaining core runs ranged between 80 and 100%, with the exception of Run 1, performed at TB-101 from 95.8 to 98 feet, which had a recovery of 56.8%.

Core Boring	Run	Depth	RQD (%)	Recovery (%)
TB-101	2	99-102 ft (30.2-31.1 m)	18	33
TB-102	1	97-101 ft (29.6-30.8 m)	48	61
12 102	4	111-113.5 ft (33.8-34.6 m)	49	49
TB-103	1	94.5-100 (28.8-30.5 m)	8	81
	2	100-105 (30.5-32 m)	36	36
TB-104E	1	99-102.5 ft (30.2-31.2 m)	33	28
TB-105A	1	95-100 ft (29-30.5 m)	12	75
TB-106	4	115-125 ft (35.1-38.1)	31	97
TB-108	2	108-109 ft (32.9 – 33.2 m)	0	28

Table No. 7-1: Core Runs With An RQD Value Less Than 50%.

Fractures per foot ranged from 0 to 2.6 for all eight borings except Run 1, performed at TB-103 from 94.5 to 100 feet (28.8 feet and 30.5 feet) and Run 2 performed at TB-107 from 100.7 to 105.7 feet (30.7 to 32.2 feet), which had 10 and 22 fractures per foot, respectively (intensely fractured).

7.3 CANADIAN SUBSURFACE CONDITIONS

For the purposes of comparison, the team has reviewed the generalized subsurface conditions that were determined from the Canadian Brine Well Program. It should be noted that the Canadian DRIC team has not yet conducted an investigation specifically for the purpose of evaluating bridge foundation support issues, and so this summary is based on information gathered during the Canadian Brine Well Program. Because the brine well investigation was focused on the bedrock, there is little information on the soil overburden.

7.3.1 Canadian Overburden Conditions

The subsurface conditions on the Canadian side of the proposed crossing corridors appear to be quite similar to those on the United States side. Conditions generally consist of bedrock overlain by glacial drift, which has been deposited either directly as till, glaciofluvial, or lacustrine deposits.

The area of Canadian investigation, as well as the area of the United States investigation, is generally located in what is commonly referred to as the St. Clair Clay Plain. Late Pleistocene era unconsolidated deposits, typically from the last major glaciations (Wisconsinan Stage, 10,000 to 14,000 years ago), overly the Dundee Limestone bedrock. The sediments typically consist of basal till and a sequence of lacustrine deposits. In some areas, the lacusterine deposits contain sand and gravel sequences overlying the bedrock, but generally consist of 65 to 100 feet (20 to 30 m) of fine-grained silt and clay materials.

7.3.2 Canadian Bedrock Conditions

The proposed crossing corridor on the Canadian side, as well as the area of the United States investigation, is located at the geologically-termed southeast margin of the Michigan Basin geomorphic province. Dundee Limestone (Dundee), a light gray, moderately hard, petroliferous, fossiliferous, laminated to thinly bedded, stylotic, pitted and vuggy limestone to dolomitic limestone is commonly encountered underlying the overburden. The Dundee in this area (from historical reports) typically exhibits unconfined compressive strengths on the order of 7 to 14.5 ksi (50 to 100 MPa), similar to values achieved on the Unites States side. RQD values for core obtained for the Canadian X-10 and X-11 Crossing locations ranged from 19 to 82%, with an average of approximately 44%, though the upper 6 feet (2 m) often exhibit values as low as 5 to 10%, which is consistent with observations made on the United States side.

7.3.3 Planned Additional Investigation

The Canadian DRIC team has indicated that to better understand surficial soil and bedrock conditions on the Canadian side of the proposed crossing corridors, an additional

investigation consisting of approximately 35 test borings and 35 Cone Penetration Test borings (CPT) are planned in the near future. This investigation will supplement the deep rock/brine well investigation that has already been performed by the Canadian team.

7.4 EVALUATIONS

Based on the information gathered during this United States investigation, subsurface conditions appear variable in composition and thickness in the upper levels of the borings and become more consistent with depth as bedrock is approached. The subsoils generally consist of variable fill soils underlain by a thin fill layer of gravelly sand. Underlying the gravelly sand or fill is a relatively thick silty clay layer. The clay layer is underlain by clay or granular hardpan that extends to limestone and dolomitic limestone bedrock. The bedrock interface is generally characterized by a thin zone of low RQD bedrock (RQD <75%) that has intermittent layers of fragmented bedrock with gravel and silty clay. Underlying the low RQD bedrock is more competent (RQD >75%) limestone and dolomite bedrock extending to the explored depths.

Based on the results of this investigation, the existing fill deposits at both crossing locations are highly variable and are not considered suitable for support of any foundation elements.

The underlying silty clay or granular soils are not considered suitable for support of the heavy loading expected from primary or secondary bridge foundation elements, but may be sufficient for support of ancillary structures with light to moderate foundation loads.

The hardpan soils underlying both corridors are considered well suited for the heavy foundation loading anticipated from proposed secondary structural elements of the bridge.

The upper highly weathered bedrock (RQD >75%) underlying the hardpan soils is considered suitable for the heavy foundation loading anticipated from primary and

secondary foundation elements of the bridge, although it is recommended that preliminary design bearing capacities are not any higher than for the hardpan.

The competent bedrock (RQD >75%) underlying the hardpan soils and the weathered bedrock is well suited for the heavy foundation loading anticipated from primary and secondary foundation elements of the bridge.

8.0 GROUND WATER CONDITIONS AND CONTROL

Groundwater was encountered during drilling at depths of 12.5 to 19.5 feet (Elevations 567.5 to 574 feet) (3.8 to 5.9 meters, Elevations 173 to 175 meters) in the test borings performed at the X-10 Crossing Corridor and at a depth of 6.5 feet (Elevation 570.5 feet) (2 meters, Elevation 173.9 meters) in TB-108 for the X-11 Crossing Corridor. Further groundwater level measurements during drilling and at completion were precluded due to the use of drilling fluids.

Pneumatic piezometers were installed in TB-101 (PZ-101) and TB-102 (PZ-102). PZ-101 was installed at Elevation 431 (131.4 meters) within the lower bedrock zone. PZ-102 was installed within the upper bedrock zone at approximately EL 470.5 (143.4 meters). Refer to Figure Nos. 19 and 20 in Attachment A for diagrams of the well installation. Based on the readings taken subsequent to June 17, 2008, hydrostatic artesian pressure head in the bedrock was recorded at 8.0 to 11.5 feet (2.4 to 3.5 meters) above ground surface at PZ-102 and PZ-101, respectively, corresponding to Elevation 592.7 feet (180.7 meters).

8.1 GROUNDWATER CONTROL

Due to the depth of the proposed excavations, groundwater control will be required to address groundwater conditions (and related gas conditions) within the granular soil portions of the soft ground profile above the hardpan, as well as artesian conditions within the hardpan and bedrock.

8.1.1 Groundwater in the Soil Horizon

Within the X-10 corridor (in particular TB-101, TB-102, and TB-103), relatively deep water bearing granular soils are present in close proximity to the Detroit River shoreline, which will act as a significant source of recharge to the groundwater within this granular aquifer. Based on analysis of the grain size testing for these soils, it appears that the permeability will be as high as approximately 0.01 centimeter per second (cm/sec). This

permeability is considered quite high and on the margin of what is considered feasible to dewater. In addition, it is likely that any dewatering effort within this aquifer would cause migration of existing known contamination on the former Detroit Coke site toward the river, and would require extensive investigation and evaluation in order to accommodate permitting by regulating agencies. Further, any dewatering discharge would require significant treatment to remove dissolved hydrogen sulfide prior to disposal, which has proven to be extremely expensive for nearby projects where this method was used.

For these reasons, it is anticipated that the only feasible method for groundwater control for large open excavations (such as for anchorages) within the granular soils present in the upper 50 to 60 feet (15 to 18 meters), will be to install a groundwater cutoff wall consisting of steel sheeting, a slurry wall, or possibly sinking caisson wall. Such a wall may be incorporated as part of the temporary earth retention system for installing the anchorage elements. In any case, a system will need to be designed that prevents cross contamination of upper aquifers (where present), and which is acceptable to regulating agencies.

For drilled or pre-drilled foundation elements (drilled piers or pre-drilled piles, respectively), such as for the main piers or for approach piers, it is expected that excavation can most likely be accomplished through the use of slurry during drilling (as discussed in later sections). It is unlikely that any other method (such as driving cutoff casing) will be successful without the use of slurry, due to the depth of the water bearing granular layers, and the depth to which the casing would need to be driven to provide cutoff. As discussed above for large excavations, any design for drilled piers or piles will need to consider the potential for cross contamination of upper aquifers. It is likely that for any design incorporating piles or drilled piers, the most practical approach would be to install a single perimeter cutoff wall around the pile group supporting the subject pier, rather than installing cutoff casing around each foundation element.

8.1.2 Groundwater and Gas in the Hardpan and Bedrock

The groundwater within the hardpan and bedrock typically contains dissolved sulfides, which can create hydrogen sulfide gas upon exposure to the atmosphere and groundwater discharge concerns if not addressed. Likewise, toxic or explosive gases may be potentially present in localized areas throughout the site.

Based on the highly fractured nature of the upper rock, as well as NTH's experience with this formation on a number of sites in the immediate vicinity of the proposed construction, it is estimated that the upper rock mass has permeability in the range of 0.01 to 0.001 cm/sec, which is considered relatively high. Given the anticipated relatively high rock mass permeability, it appears groundwater in the bedrock may require control during foundation construction. For applications that require exposure of large areas of the bedrock to atmospheric conditions (such as for suspension bridge anchorages), this is typically accomplished by either rock grouting, or sometimes by dewatering.

For foundation installation methods other than drilled piers (discussed in subsequent sections), groundwater control in the hardpan and bedrock units by dewatering is not considered to be feasible for this project. This is because the foundations will be relatively massive and would be expected to involve at least 12 months of underground construction and related dewatering. For such a prolonged dewatering effort, consolidation settlement would be induced in the thick cohesive strata overlying the hardpan, creating the potential for damage to surrounding infrastructure. In addition, the volume of water and the associated groundwater treatment to remove sulfides and other contaminants prior to disposal (as discussed above for the soil overburden) would be significant and very costly.

Rock grouting has proven to be a very cost-effective means of controlling groundwater and gas at the soil rock interface on several projects in close proximity to the DRIC, where construction involved excavation and exposure of the bedrock surface. Rock grouting of the Dundee Limestone would allow for exposing and cleaning the rock surface to confirm the primary and secondary foundation elements are founded on an adequate bearing surface. However, rock grouting of the upper weathered rock will require consideration for the presence of granular soils where encountered immediately over the bedrock. This was the encountered condition at TB-105A and TB-107 in X-10, and TB-108 in X-11. The granular soils have the potential to fill the upper rock fractures, which may inhibit grout penetration during a rock grouting program. This can be addressed by the use of low viscosity chemical grouting such as acrylimide, or hot bitumen in the upper rock.

For the purposes of the conceptual engineering, and if exposure of a large area of the rock surface is necessary to accommodate the construction, it is recommended that a rock grouting program be developed, with a concept design including a three stage, cementious, top down rock grouting program, with a secondary chemical grouting program of the upper rock to address granular infilling of fractures in the rock.

Even if such a rock grouting program is effectively implemented, some amount of groundwater will almost certainly be produced from the surface of the grouted rock. Any groundwater produced is expected to require treatment for dissolved sulfides and hydrogen sulfide prior to disposal. Additionally, odor control may be required for airborne hydrogen sulfide gas, although substantial efforts for odor control are not typically necessary or required for industrial areas such as the X-10 corridor. When necessary, such odor control typically involves collecting and treating the water before it has the opportunity to let gas escape, and/or masking the odor chemically.

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9.0 SITE PREPARATION

It is anticipated that final design grades will be close to existing site grades. However, due to the presence of fill on the site, some earthwork may be required to achieve final design grades. It is anticipated that some of the on-site fill can be reused for earthwork operations, but only in green belt areas.

The concept engineering should consider that at the start of earthwork operations, and after any demolition is complete, all existing pavement, vegetation, layers of topsoil, and any other exposed organic soils should be stripped and removed from within the proposed foundation footprint. In addition, any utilities present within the foundation area should be re-routed and then properly abandoned and removed following outside of the footprint.

A special consideration for the X-10 site will be the presence of the demarcation barrier and associated groundwater collection system, which will probably need to be maintained outside the immediate construction area. In addition, special handling and excavation methods will be required for any excavations through the upper granular soil layers to prevent exacerbation of contamination on the site. It is likely that installation of a contamination cutoff wall will be necessary, which could be coincident with the groundwater cutoff wall mentioned in Section 7.1.1.

9.1 TEMPORARY EXCAVATIONS

The concept engineering should required that all excavations deeper than 5 feet (1.5 meters) be properly sloped or otherwise structurally retained to provide stable and safe working conditions. In all cases, applicable regulations prescribed by the Michigan Department of Consumer and Industry Services (MDCIS), formerly known as MIOSHA, will need to be followed and adequate protection for workers, structures and utilities provided.

10.0 FOUNDATION RECOMMENDATIONS

Recommendations for support of bridge foundations have been developed on the basis of the currently proposed cable-stayed and suspension bridge concepts as developed in the Detroit River International Crossing, Bridge Conceptual Engineering Report, by Parsons Transportation.

Based on the concepts developed in the Bridge Conceptual Engineering Report, the main pier elements for both suspension and cable-stayed bridge options, along with secondary and approach pier foundation elements, may be supported on long slender deep foundation elements. For the purposes of this report, such foundation elements will consist of drilled piers, or pre-drilled or driven piles. In all cases, multiple pier or pile elements would be required to support a main pile cap, in turn supporting the pier or other structural element.

For conceptual design purposes, the Michigan Department of Transportation's (MDOT) Bridge Design Manual (BDM) is used to develop nominal pile driving resistance values (R_{NDR}) for the recommendations presented herein. The driving resistance values presented in the MDOT BDM assume that the FHWA-modified Gates dynamic formula is used to develop driving criteria. As such, both AASHTO (2007) LRFD Bridge Design Specifications and the MDOT BDM recommend a dynamic resistance factor (ϕ_{DYN}) equal to 0.4. It should be noted that the FHWA manual <u>Design and Construction of Driven Pile Foundations</u>, FHWA-HI-97-013 and -014, states the following:

",,,dynamic formulas do not provide information on pile driving stresses and, in many circumstances, have proven unreliable in determining pile capacity. Therefore, their continued use is not recommended on significant projects. Dynamic test methods using signal matching techniques can be used to: calculate pile installation stresses, determine pile integrity, estimate static pile capacity, and determine relative soil

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resistance distribution on the pile. This is also an appropriate means of establishing driving criteria."

If AASHTO guidelines are followed to establish driving criteria from dynamic testing with signal matching, a dynamic resistance factor of 0.65 may be used instead of 0.4. We recommend that consideration be given to establishing pile driving criteria using dynamic testing.

10.1 SUSPENSION BRIDGE ANCHORAGE OPTIONS

If the main structure is chosen to be a suspension bridge, multiple options have been proposed and evaluated for the anchorage foundation elements. Those options are discussed as follows.

10.1.1 Two Rectangular Caissons (Option I)

In this case, two large rectangular sinking caissons are designed and constructed to serve as both the temporary earth retention system and permanent anchorage structures for the proposed suspension bridge. The proposed caissons measure approximately 33 feet x 197 feet (10 meters x 60 meters) in plan dimension. A diagram has been prepared to illustrate Option I and is presented as Figure 10-1 below and Figure No. 39 in Attachment A.

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Figure No. 10-1: Schematic of Option I – Two Rectangular Caissons.

10.1.2 Single Circular Caisson (Option II)

In this case, one large diameter sinking caisson would be designed and constructed to serve as both the temporary earth retention system and permanent anchorage structure for the proposed suspension bridge. The proposed caisson has a diameter of approximately 165 feet (50 meters). A diagram has been prepared to illustrate Option II and is presented as Figure 10-2 below and Figure No. 40 in Attachment A.

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Figure No. 10-2: Schematic of Option II – Single Circular Caisson.

10.1.3 Drilled Shafts and Two Circular Caissons (Option III)

In this case, two circular sinking caissons, in combination with 12 drilled piers are designed and constructed to serve as the permanent anchorage structure for the proposed suspension bridge. The proposed sinking caissons measure approximately 65 feet (20 meters) in plan diameter and the drilled piers would be 10-feet (3 meters) in diameter. A diagram has been prepared to illustrate Option III and is presented as Figure 10-3 below and Figure No. 41 in Attachment A.

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Figure No. 10-3: Schematic of Option III – Drilled Shafts and Two Circular Caissons.

10.1.4 Drilled Shafts and Two Rectangular Caissons (Option IV)

In this case, two rectangular sinking caissons in combination with 12 drilled piers are designed and constructed to serve as the permanent anchorage structure for the proposed suspension bridge. The proposed caissons measure approximately 30 feet x 82 feet (9 meters x 25 meters) with nominal 10-feet (3 meters) diameter drilled shafts. A diagram has been prepared to illustrate Option IV and is presented as Figure 10-4 below and Figure No. 42 in Attachment A.

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Figure No. 10-4: Schematic of Option IV – Drilled Shafts and Two Rectangular Caissons.

10.1.5 Pipe Piles (Option V)

Pipe piles to support the suspension bridge anchorage could consist of reinforced concrete filled steel pipes, which are typically equipped with a driving boot. The pipe piles would be pre-drilled to the top or very near the top of bedrock, and mandrel driven to bear on top of bedrock. A reinforcing steel cage would then be placed within each pile and the pile filled with concrete. For the concept design, it can be assumed that the bedrock geotechnical end bearing resistance will be mobilized within a settlement of up to 5% of the pipe diameter, and that any settlement would occur primarily as elastic settlement.

Table 10-1 summarizes the recommended nominal and factored pile driving resistance values for 30-inch diameter pipe piles. The values presented in Table 10-1 assume that the tip of the pipe pile is plugged. The vertical and horizontal components are also provided for battered piles assuming a 3V:1H batter. The dynamic resistance factor

 (ϕ_{DYN}) presented by the MDOT BDM is equal to 0.4, and assumes that pile driving criteria will be developed by using the FHWA-modified Gates Dynamic formula.

Table No. 10-1: Conceptual Driving Resistance Values for Cast-in-place (C.I.P.Pipe Piles.

	Axial (tons)		Vert. Component (tons)		Horiz. Component (tons)	
Pile			3V:1H batter		3V:1H batter	
	R _{NDR}	$\phi_{DYN}R_{NDR}$	R _{NDR}	$\phi_{DYN}R_{NDR}$	R _{NDR}	$\phi_{DYN}R_{NDR}$
30" O.D.	000	396	030	376	31/	126
0.625" Wall	990	390	737	570	514	120
$\phi_{\text{DYN}} = 0.4$ based on using the FHWA-modified Gates dynamic formula to establish driving criteria.						

We recommend considering the use of dynamic testing in developing driving criteria. If AASHTO guidelines for dynamic testing are followed, a dynamic resistance factor of 0.65 may be used instead of 0.4.

The pipes should be spaced approximately 2 diameters apart or greater, to prevent significant reduction in capacity and increases in settlement due to grouping effects. The actual spacing could potentially be closer (or potentially further), based on final design analysis and in consideration of the final actual loading group configuration, pipe boot details, etc.

10.2 DEEP (SLENDER) FOUNDATION ELEMENTS FOR PRIMARY FOUNDATIONS

Drilled piers (also termed drilled caissons or drilled shafts) used for support of the primary foundations would extended through the upper fill, silty clay, granular soil layers, hardpan soils, and be founded at least 5 feet (1.5 meters) into the underlying competent limestone bedrock formation, resulting in depths of approximately 100 to 110 feet (30 to 34 meters). Extending these foundation elements will minimize uncertainties

in the concept design by providing a uniform and reliable bottom pier elevation bearing on competent bedrock.

The drilled shaft evaluation summarized herein was performed using LRFD methodology. AASHTO and FHWA (report number FHWA-IF-99-025) guidelines were followed. The following statement taken from FHWA-IF-99-025 summarizes the assumptions used for evaluation.

"Assume that the settlement of the drilled shaft (pier) above the rock socket is due only to elastic compression of the drilled shaft (pier) material and is negligible. It is also assumed that the load transferred in the overburden above the rock is minimal. That is, all of the load is transferred in the socket. This assumption will ordinarily result in overpredicted settlements, since some load is invariably transferred in the overburden."

Tables 10-2 and 10-3 summarize the nominal ultimate end and nominal ultimate side resistance values for the evaluated drilled piers. The nominal ultimate base resistance, R_{BN} , and nominal ultimate side resistance values, R_{SN} , are computed by the following equations.

$R_{\rm BN} = A_{\rm E}(q_{\rm max})$	[Eqn. 10.1]
$R_{SN} = A_S(f_{max})$	[Eqn. 10.2]

Where:

A_E

Drilled pier end area, and

A_S Drilled pier shaft area within bedrock.

	B = 2.5 m (8.2 ft)	B = 3.3 m (10.8 ft)	
Maximum End Resistance, q _{max} (tsf)	300		
Maximum Nominal End Resistance, R _{BN} (tons)	15,843	27,605	
Resistance Factor, ϕ	0.50		

Table No. 10-2: Nominal Ultimate End Resistance Values.

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	Socket Length (ft)	B = 2.5 m (8.2 ft)	B = 3.3 m (10.8 ft)
Maximum Side Resistance, f _{max} (tsf)		10.6	
Maximum Nominal Sida Desistance D	5	1,370	1,800
(tons)	10	2,730	3,600
(tons)	15	4,100	5,400
Resistance Factor, ϕ		0.65	
B = Pier diameter			

Table No. 10-3: Nominal Ultimate Side Resistance Values.

For final drilled pier design, if the end and side resistance values are determined using a field load test, both resistance factors can be increased to 0.8. The nominal total base resistance, R_T , is computed by summing the nominal end and side resistance values. However, the ultimate nominal values of end and side resistance should not be added as these ultimate resistance values will not be mobilized at the same strain, as illustrated in Figure 10-5. The ultimate side resistance (point A) is mobilized at small strain, while the ultimate end resistance (point B) is mobilized at large strain. Once the end bearing resistance is fully mobilized, the shaft resistance has usually reduced to residual strength conditions (near point C). The load-settlement evaluation presented herein accounts for this strain incompatibility.



Figure No. 10-5: Load-Settlement Behavior under Compression Loading (FHWA-IF-99-025).

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Figures 10-6 and 10-7 present the load-settlement behavior for pier diameters of 2.5 m (8.2 ft) and 3.3 m (10.8 ft), respectively. The load-settlement relation is modeled using three linear segments; although, the actual behavior is likely non-linear. The first segment presumes the behavior is elastic until side socket shear failure occurs. During the second segment, after side shear failure and before complete failure of base (plunging), it is assumed that the base behavior is elastic and the side resistance reduces to its residual strength value. The behavior in the third segment is plastic (plunging) where uncontrolled deformation occurs with very little to no additional loading.



Figure No. 10-6: Drilled Pier Results Summary for a Diameter of 2.5 m (8.2 ft).

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Figure No. 10-7: Drilled Pier Results Summary for a Diameter of 3.3 m (10.8 ft).

For Figures 10-6 and 10-7, the computed normalized skin friction transfer relationship is also provided. At small strain, the mobilized skin resistance is near 100% of its ultimate value. Allowing continued strain reduces the mobilized skin resistance to its residual strength value, which is approximately 17% of its ultimate value. Figures 10-6 and 10-7 are used to estimate the nominal and factored resistance values at a specified strain, considering strain incompatibility between skin resistance and end bearing resistance.

Tables 10-4 and 10-5 summarize in a matrix format the graphical results that are presented in Figures 10-6 and 10-7, for various rock socket lengths (L). The column δ /D represents the pier settlement normalized by the pier diameter found in the x-axis of Figures 10-6 and 10-7. In addition, the values presented in Tables 10-2 and 10-3 are used to develop factored resistance values that are listed in Tables 10-3 and 10-4. In Tables 10-4 and 10-5, R_N is the nominal total load, R_S is the nominal shaft skin friction load, R_{S(U)} is the nominal ultimate shaft skin friction load, and R_B is the nominal end bearing load.

δ/D	R _N	$\mathbf{R}_{\mathbf{S}}/\mathbf{R}_{\mathbf{S}(\mathbf{U})}$	R _S	R _B	φR _N	
(%)	(tons)		(tons)	(tons)	(tons)	
		$\mathbf{L} = \mathbf{L}$	5 feet			
0.6	4,700	0.84	1,147	3,553	2,522	
1.0	7,200	0.70	956	6,244	3,743	
1.4	9,600	0.56	765	8,835	4,915	
2.5	16,072	0.17	229	15,843	8,071	
L = 10 feet						
0.6	6,100	0.87	2,362	3,738	3,404	
1.0	8,800	0.69	1,884	6,916	4,683	
1.4	11,300	0.53	1,447	9,853	5,867	
2.2	16,313	0.17	470	15,843	8,227	
L = 15 feet						
0.6	7,400	0.89	3,645	3,755	4,247	
1.0	10,200	0.68	2,785	7,415	5,518	
1.4	13,000	0.48	1,966	11,034	6,795	
2.0	16,565	0.18	721	15,844	8,391	

Table No. 10-4: Summary of Nominal and Factored Resistance Values for a 2.5 m

(8.2 ft) Drilled Pier.

Table No. 10-5: Summary of Nominal and Factored Resistance Values for a 3.3 m

(10.8 ft) Drilled Pier.

δ/D	R _N	$R_S/R_{S(U)}$	R _S	R _B	φR _N	
(%)	(tons)		(tons)	(tons)	(tons)	
		$\mathbf{L} = \mathbf{S}$	5 feet			
0.6	7,500	0.83	1,133	6,367	3,920	
1.0	11,500	0.70	956	10,544	5,893	
1.4	15,700	0.57	778	14,922	7,967	
2.63	27,907	0.17	229	27,678	13,988	
		L = 1	0 feet			
0.6	9,400	0.86	2,335	7,065	5,050	
1.0	13,700	0.70	1,912	11,789	7,137	
1.4	18,000	0.54	1,475	16,525	9,221	
2.36	28,225	0.17	470	27,755	14,183	
L = 15 feet						
0.6	11,100	0.87	3,564	7,537	6,085	
1.0	15,700	0.69	2,826	12,874	8,274	
1.4	20,300	0.51	2,089	18,211	10,463	
2.15	28,558	0.18	721	27,837	14,387	
The evaluation presented herein is preliminary and is subject to verification based on performing additional soil borings and rock coring at the final locations of the bridge foundation units. Once the final foundation locations are selected, and loading information, settlement tolerance, and construction methods are known, this evaluation can be finalized to design the required drilled pier diameter and rock socket length for each foundation unit.

It is understood that a preliminary shaft diameter of approximately 98 to 168 inches (2.5 to 3.3 meters) is planned from a foundation load standpoint. For planning purposes, drilled piers should be spaced a minimum of 2 diameters apart (edge-to-edge, including any bells), although the final spacing should be confirmed on the basis of actual pier layout and geometry, loading, design depth, etc.

During the conceptual engineering of foundation systems expected to be subjected to lateral loading, preliminary values for the modulus of lateral subgrade reaction can be applied as follows:

	Lateral Subgrade Reaction Modulus
Soil Stratum	(pounds per cubic inch)
Granular Layers	60
Cohesive Layers	100
Hardpan	2,000
Bedrock (weathered)	2,500

If used in the conceptual engineering structural analyses, the above moduli (Welch and Reese, 1972, 1975) should be used in conjunction with caisson diameters, modeling method used, etc., to determine appropriate lateral caisson capacities.

Within test borings TB-105A, TB-107, and TB-108, layers of granular soils were encountered below the hardpan or below the cohesive soil layers extending to the bedrock. It is expected that the predominately sandy layer(s) will generally possess little

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to no standup time. In locations with this condition, it is expected that drilling slurry (as recommended above to control groundwater), will be necessary to preserve sidewall stability. Likewise, where this condition exists, the use of belled caissons in the soil will be difficult or impossible, since the bells cannot be relied on to be self supportive within predominantly granular layers.

In order to evaluate the feasibility of the drilled excavations in the soft to medium clay soil zones, probable overload factors (also termed stability numbers and defined as the ratio of overburden stress to soil shear strength), were calculated. Overload factors on the order of six to eight (6 to 8) typically indicate marginal sidewall stability, and values greater than eight (8) typically indicate squeezing conditions. Based on the soil data, it is estimated that the overload factors will approach twenty (20) as the excavation depth below the ground surface reaches the hardpan layers. This indicates that squeezing conditions will be present within the shaft excavations unless the shafts are drilled under slurry or fully cased. As such, the concept engineering should include provisions for the use of specialized drilling slurry for the full depth of the caissons through clay, as well as through granular layers overlying the hardpan and bedrock. In addition, the use of full-length steel casing may be preferred for isolated foundation elements in areas of known soil and groundwater contamination (particularly in the upper fill layers), as discussed earlier in this report. For larger groups of drilled pier elements, a perimeter cutoff wall will likely be more cost effective.

Based on local experience with subsurface conditions in the Detroit Area, as well as observations from test boring programs on nearby parcels, the possibility of random occurrence of toxic, noxious, and explosive gases in caisson excavations cannot be precluded, although proper gas monitoring will minimize the risk associated with such events.

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10.3 DEEP (SLENDER) FOUNDATION ELEMENTS FOR SECONDARY FOUNDATIONS

Based on the concepts developed in the Bridge Conceptual Engineering Report, the secondary and approach pier foundation elements may be supported on long slender deep foundation elements. For the purposes of this report, such foundation elements will consist of drilled piers, or pre-drilled or driven piles. In all cases, multiple pier or pile elements would be required to support a main pile cap, in turn supporting the pier or other structural element.

10.3.1 Drilled Piers

Drilled piers would extended through the upper fill, silty clay, granular soil layers, and be founded at least 2 feet (0.6 meters) into the underlying hardpan soils, resulting in depths of approximately 90 to 100 feet (27 to 30 meters). The drilled pier should be designed for end bearing in the hardpan. For a drilled shaft constructed in this fashion, the nominal end resistance should be approximately 40 tsf (3.8 MPa) for conceptual design purposes, which corresponds to a settlement of approximately 5 percent of the shaft end diameter. A resistance factor of 0.55 should be used. Invariably, some load will be distributed into the overburden soils along the drilled pier. This load is difficult to quantify with certainty, and will be small relative to the end resistance mobilized in the hardpan. As a result, overpredicted settlements are likely due to some load transfer into the overburden soils.

10.3.2 Driven Piles

Based on the current bridge concepts, some bridge foundation elements (most likely approach piers) could potentially be supported on piles, which could involve various sizes of mandrel or top-driven concrete filled pipe piles, or top driven H-piles.

10.3.2.1 Pipe Piles

Pipe piles for this project could consist of concrete filled steel pipes, which are typically equipped with a driving boot. The pipe piles are then pre-drilled to within approximately 5 feet (1.5 m) of hardpan, mandrel driven to bear within the hardpan, and then filled with concrete. The concept engineering should include provisions for the use of drilling slurry for the full depth of the pre-drilled hole through clay, as well as through granular layers overlying the hardpan and bedrock.

Table 10-6 summarizes the nominal and factored pile driving resistance values for pipe piles recommended in MDOT BDM. The dynamic resistance factor (ϕ_{DYN}) presented by the MDOT BDM is equal to 0.4, and assumes that pile driving criteria will be developed by using the FHWA-modified Gates Dynamic formula.

Pile	Axial (tons)		
The	R _{NDR} (t)	$\phi_{\rm DYN}R_{\rm NDR}$	
12" O.D. 0.25" wall	175	70	
14" O.D. 0.312" wall	200	80	
14" O.D. 0.438" wall	250	100	
$\phi_{DYN} = 0.4$ based on using the FHWA-modified Gates dynamic formula to establish driving criteria.			

Table No. 10-6: Conceptual Driving Resistance Values for C.I.P. Pipe Piles.

We recommend considering the use of dynamic testing in developing driving criteria. If AASHTO guidelines for dynamic testing are followed, a dynamic resistance factor of 0.65 may be used instead of 0.4.

The pipe piles would need to be spaced at least 3 diameters apart to prevent significant reduction in capacity and increases in settlement due to grouping effects.

10.3.2.2 H-Piles

H-piles for this project would be top driven into the hardpan. Table 10-7 summarizes the nominal and factored pile driving resistance values for driven H-piles recommended in F:\646294_DRIC_Study\Final_Eng_Report_Sept_08\Submittals\Engineering Report\Engineering Report_FINAL_Nov 08\Main Bridge Structure Study\Appendix D-Geotechnical\NTH_Report_FINAL_11-21-08.docx the MDOT BDM. As with pipe piles, the dynamic resistance factor (ϕ_{DYN}) presented by the MDOT BDM is equal to 0.4, and assumes that pile driving criteria will be developed by using the FHWA-modified Gates Dynamic formula.

Table No. 10-7: Conceptual Driving Resistance Values for C.I.P. Pipe Piles.

Pile	Axial (tons)		
T He	R _{NDR} (t)	$\phi_{DYN}R_{NDR}$	
HP 12x53	200	80	
HP14x102	400	160	
$\phi_{\text{DYN}} = 0.4$ based on using the FHWA-modified Gates dynamic formula to establish driving criteria.			

We recommend considering the use of dynamic testing in developing driving criteria. If AASHTO guidelines for dynamic testing are followed, a dynamic resistance factor of 0.65 may be used instead of 0.4.

Due to the relatively deep depth to the hardpan and because pre-drilling is not typically practical for H-piles, sweep of the piles during driving operations could be problematic. The piles would need to be spaced at least 5 dimensions apart in consideration of the potential for sweep, and to prevent significant reduction in capacity and increases in settlement due to grouping effects.

10.3.2.3 Environmental Considerations for Piles

Research indicates that there is a potential for pile foundations to enhance vertical migration of contamination through aquitards. The research indicates that a variety of factors may influence the potential for vertical migration, including:

- Pile shape (round, square, or H)
- Pile diameter
- Shape of pile bottom (flat or pointed)
- Installation method (driven or cast in place)
- Pile material (steel, concrete, or wood)

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- Soil type of aquitard (stiff clay or soft clay)
- Thickness of aquitard
- Amount of aquitard penetration (partial or full)

Research indicates that installation of certain types of piles under certain conditions may lead to a ten times increase in vertical contaminant flow through the aquitard. In general, bored piles may lead to more vertical flow than driven piles. Also, H piles that are driven may lead to more vertical flow than round piles that are driven.

During the brine well and geotechnical investigations, oversized steel environmental casings and specialized cementing methods were used to prevent vertical migration. Since these investigations required open boreholes, this environmental casing was deemed an appropriate precaution. However, for permanent piles, such preventative measures may or may not be required. Due to the numerous factors that may affect the potential for increased vertical migration, it would not be appropriate to make a recommendation at this time. However, based on the concerns of the MDEQ and former liable owner, and the high visibility of this project; the potential for piles to increase vertical migration should be considered during the final foundation design.

10.4 DEEP EXCAVATIONS FOR FOUNDATION INSTALLATION

For several of the bridge foundation concepts that have been developed (particularly for installation of a suspension bridge anchorage), a relatively large excavation will need to be extended into the bedrock. Based on the concepts developed as part of the Bridge Conceptual Engineering Report (as discussed above), such excavations could potentially consist of a circular shaft 165 feet (50 m) in diameter or two rectangular shafts 35 by 200 feet (10 by 60 m) in plan dimension. Two hybrid designs are also proposed that consist of smaller diameter circular and rectangular caissons, in combination with drilled piers. In any case, the excavation support systems would extend to the top of (or into) rock, which can be expected to be approximately 100 feet (30 m) below ground surface. Such

excavations would require a significant earth retention system, which could potentially consist of an internally braced tangent pile shaft, structural slurry wall, or sinking caisson. In addition, internally braced steel sheeting has been considered, although this option is not considered to be practical. These options are discussed below.

10.4.1 Tangent Pile Shaft

A tangent pile shaft involves augering a series of holes, tangent to one-another (usually staggered in plan view), through the overburden soils to bedrock in pattern around the planned shaft location. Heavy steel members or reinforcement rods are then placed vertically into the holes, with the annulus tremie-grouted with concrete. The steel effectively provides resistance to bending in the vertical direction. As excavation proceeds, steel reinforcing beams (rectangular shaft) or reinforced concrete "ring beams" (circular shaft) are cast-in-place around the inside perimeter of the tangent pile walls. The steel bracing or concrete ring beams act as horizontal compression members, resisting the resulting soil pressures acting on the outside of the shaft.

Due to the large soil loads, the steel bracing or concrete beams would need to be placed at close vertical spacing (probably in the range of 5 to 10 feet or 1.5 to 3 meters), resulting in a large number of bracing levels over the entire depth of the shaft. Also, the shaft must be designed such that the tangent pile walls can resist lateral earth loads below the base of the excavation during construction. Such forces for the expected necessary size shaft would be large, with associated steel members also large. A major risk of this method involves inward creep of the tangent piles and squeezing of soil through gaps between tangent piles; both issues are due to high soil pressure and associated high overload factors as the shaft depth approaches the hardpan. A generalized plan view of a typical tangent pile shaft is shown in Figure 10-8.

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Figure No. 10-8: Schematic Plan View of a Typical Tangent Pile Wall.

A hybrid of this alternative, termed a secant pile shaft, consists of a similar installation of augered bore holes that are spaced having a clear distance less than one bore diameter away from each other. These initial spaced holes are backfilled with lean concrete and have no steel reinforcing. After these initial lean concrete piles have cured, the space between the lean concrete piles is augered, along with a portion of the adjacent lean concrete piles. Reinforcing steel and structural concrete is placed in these intersecting (i.e., secant) piles. The main support derives from the reinforced piles, while the lean concrete piles act as lagging. Similar risks are associated with a secant pile shaft as were noted for the tangent pile shaft. The benefit of using a secant pile shaft is that the amount of soil squeeze between structural support elements should be reduced. A generalized plan view of a typical secant pile shaft is shown in Figure 10-9.



Figure No. 10-9: Schematic Plan View of a Typical Secant Pile Wall.

10.4.2 Steel Sheeted Shaft

A steel sheeted shaft would be similar in concept to a tangent pile shaft, with the steel sheeting most likely being driven within a starter pit to reduce sheeting lengths. Similar to the tangent pile shaft concept, as the excavation proceeds, steel bracing or reinforced "ring beams" are then placed around the inside perimeter of the sheeting. Also similarly, the supports would need to be placed at close vertical spacing and the sheeting would need to be designed to resist lateral earth loads below the base of the excavation during construction. Major risks involved with this method involve jumping of interlocks in the sheets during driving, resulting in squeezing of soil into the shaft through gaps, as well as inward creep due to high ground pressure. In addition, pre-drilling may be required to remove or dislodge cobbles and thus allow adequate sheeting penetration. The noted risks probably make the use of this method impractical for the DRIC project.

10.4.3 Slurry Wall Shaft

A slurry wall shaft, also known as a diaphragm wall, is constructed by excavating to or into bedrock a series of deep trenches, each of a short finite length, to the designed bottom depth of the shaft. Each trench is excavated and kept open using slurry material, usually a mixture of bentonite, water, and additional additives that modify the slurry's dynamic properties. Following excavation, steel reinforcement is placed in the trench, and the trench is then tremie grouted with structural concrete. Following the concrete placement, the next trench is excavated, and the process is repeated. Such walls can be constructed in a circular plan configuration, which resists soil and water loads primarily through compression, with some secondary bending affects. However due to large compression loads and buckling effects, larger shafts constructed in this manner often utilize internal support, such as internal ring beams. For a rectangular shaft, a concrete slurry wall would probably require installation of internal support as the shaft excavation proceeds.

When considering the size of the proposed shaft and the prevailing soil conditions, slurry wall construction methods offer a few distinct disadvantages. Although these types of walls can carry significant load in compression, they are sensitive to unsymmetrical

loading. This is critical when considering the trench excavations would take place in soft soils. Slurry trenches in soft soils have historically resulted in poor control of squeezing within deep excavations, although this can usually be controlled through careful construction practices. Another potential disadvantage is that slurry wall construction is a technique somewhat unfamiliar to the local construction community.

10.4.4 Sinking Caisson Construction

A sinking caisson could either be constructed in as a circular shaft or rectangular shaft. A circular shape is much more common and is very efficient structurally, because the walls act as a circular compression beam. A rectangular or square caisson typically requires very thick and highly reinforced walls, although this can sometimes be reduced through the use of internal diaphragm walls cast between the outside long-span walls to provide support.

For this project, the caisson would be constructed through variable fill soils, soft to medium cohesive soils, loose to very compact silty/sandy soils, very stiff to very hard clayey hardpan, then very stiff to stiff silty clay, and finally bedrock. During excavation, bentonite injection is typically be used to advance the shaft downward. The general sequence of the installation would then involve completing the shaft by underpinning the shaft to bedrock. Excavating and/or blasting the underlying rock will be necessary to install the lower portion of the shaft into competent bedrock. The general concept for the sinking caisson construction method by stage is shown on Figure 10-10 and Figure 43 in Attachment A, and is discussed in greater detail in the following paragraphs.

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Figure No. 10-10: Sinking Caisson Construction Schematic.

The sinking caisson construction method generally begins with the removal of any existing structures, sewer pipes, piles, and other obstructions within the proposed shaft area, followed by a bedrock and soil/rock interface grouting program (discussed previously). Following the grouting program, a cutoff wall may be installed for groundwater control purposes, and a steel sheet pile wall retaining system may be used to execute the launching pit excavation.

Given the existence of variable fill material, a concept design involving a sinking caisson should include the caisson being launched from a shaft starter pit excavated larger than the shaft footprint. The launching pit for this project would almost certainly be constructed within a braced excavation, and due to groundwater, environmental and face instability concerns of the fill soils, the excavation bracing would probably consist of steel sheet piling. Given the large diameter of the caissons, it likely that internal bracing for a launching pit will consist of cast-in-place concrete ring beams. All fill soils within the shaft starter pit would be removed prior to constructing the launching platform, and a pad of engineered fill would be constructed on native or otherwise suitable fill soil. The launching pit excavation would then be backfilled to a specified elevation such that flooding during construction can be avoided. Steel sheeting installed to cut-off potential connection to the river and near surface groundwater (perched) may also be installed.

The concept design should consider that any sheets that are driven through the fill soils will encounter multiple obstructions and result in very difficult driving, misalignment, and jumping sheet pile interlocks. As such, the concept design should include provisions that, prior to pile driving, the line of sheeting be pre-excavated and filled in limited sections to remove major obstructions.

Once the launching pit is completed, construction is started on the first couple of lifts of the caisson, on the launching platform within the starter pit. The first lift contains a "cutting shoe" that consists of a tapered edge to allow the tip to cut into the soil as soil is removed from the interior of the shaft. After the first and second lifts have gained

appropriate strength, the launching platform is segmentally removed and the caisson begins to sink into the soil profile generally via a controlled soil failure at the cutting shoe tip. The shaft is subsequently sunk to a pre-determined depth by excavating within the shaft to reduce bearing of the cutting shoe and frictional resistance on the shaft wall, adding concrete lifts, and injecting bentonite to the shaft exterior to decrease skin resistance. The actual sinking of the caisson occurs whenever the total weight of the constructed portion exceeds the soil bearing support of the cutting shoe, hydrostatic uplift pressure, and the overall skin resistance of the soil on the shaft walls. During construction, a proper balance between these forces must be maintained to ensure shaft sinking is controlled.

Sinking the caisson in the wet has been considered to reduce the groundwater cutoff requirements, although this method is often problematic when sinking through harder soil layers at great depth, and is not recommended for this project.

Caissons sink incrementally and the large shaft diameters required for the primary foundation elements will require substantial effort to maintain plumbness. Maintaining roundness is also an issue for circular sinking caissons, since the wall elements are typically very thin in comparison to the diameter, and may be subject to buckling if unbalanced loading occurs. Rectangular caissons typically have thicker walls with respect to the wall length, and are less susceptible to bucking. These issues are generally controlled by excavating in particular areas of the caisson interior as required. At times the excavation may be asymmetrical to achieve the desired result in the caisson structure.

Sinking a caisson through the anticipated loose to very compact granular layers encountered above the hardpan soils in some locations will require the use of a cut-off wall to form a barrier between the caisson sinking operation and the existing groundwater and contaminated soils (also expected to be variably impacted). This could consist of a slurry wall as discussed above, or a jet grout cut-off wall extending down into hardpan expected near Elevation 500 feet (Elevation 152.4 meters). If used, such a wall should be designed to also intercept the intermittent granular layers above/below the hardpan where encountered (specifically on Crossing X-10). The jet grout cut-off wall would have such dimensions as to allow sufficient room for caisson sinking activities and subsequent ground settlement impacts. In any case, provisions for the concept design should include monitoring the effectiveness of the cut-off wall, such as monitoring wells on each side of the cut-off wall screen within the granular soils zone.

Potential risks associated with the sinking caisson method include the potential that untimely injection of the bentonite lubricating fluid may result in undesirable movement of the caisson; there is some potential for ground settlement; and that the success of the system is strongly dependent on development of, and adherence to a well thought-out sinking plan.

Although there are a number of local contractors that are very familiar with the sinking caisson method, there are others less familiar. Since the method requires diligent control of the ground and careful excavation to maintain plumbness (and roundness in the case of circular caissons), it would be prudent for the final design, that very stringent qualification requirements be included in the project specifications.

10.4.5 Shaft Wall Construction at Soil/Rock Interface

Specialized construction methods will be required to construct the shaft walls at the bedrock interface and to the desired elevation within the bedrock. The shaft will probably be advanced by interior soil excavation until rock is encountered beneath the shaft wall at approximately Elevation 485 to 494 feet (Elevation 147.8 to 150.6 meters)

For the Sinking Caisson option discussed in the previous section, the shaft would then be stabilized by replacing the lubricating bentonite on the exterior of the caisson with cement grout injected through the bentonite injection pipes. As the bedrock surface is expected to vary beneath the shaft, ground stabilization (jet grouting or ground freezing) would then be employed to stabilize the zone between the caisson shoe and the bedrock interface prior to underpinning the caisson. The same method would also hold true for a structural slurry wall designed shaft. Depending on the continuity and success of the bedrock grouting program, the ground stabilization must be designed for groundwater cutoff, partial support of the shaft weight during underpinning (sinking caisson option only), and soil/hydrostatic pressures.

Depending on the final anchorage design, the sinking caisson shaft may require advancement into the bedrock, either for bearing, or to provide added resistance to lateral load. This is typically accomplished by excavating into the rock after the caisson itself is underpinned and locked into place. The entire shaft bottom, or only the perimeter, can then be excavated into the rock, depending on the bearing requirements and lateral passive resistance requirements. Passive resistance may also be developed by constructing key-ways in the rock.

Where passive resistance is needed to provide a reaction to lateral anchorage loads, the following lateral passive equivalent fluid pressure values are suggested for conceptual design purposes.

Soil Stratum	Equivalent Fluid Pressure
Son Suatum	(pounds per cubic foot)
Overlying soil material	135
Hardpan	257
Bedrock	330

This may be combined with a coefficient of frictional resistance component between rock and the concrete anchorage foundation of 0.65. It is expected that these values may be refined based on further testing and evaluation that would be conducted during the final design.

10.5 FOUNDATION COST CONSIDERATIONS

For development of concept level cost estimates for piles and drilled piers, the use of RSMeans (annual construction cost estimating publication) or local bid tabulation data is

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recommended. Costs for tangent pile walls and braced sheeting are also probably best developed from local bid tabulations for similar work.

Cost estimates for slurry wall foundations can be developed from RSMeans. However, this type of construction is not well known to the local construction community, and as a result, costs in the Detroit area may be greater than those in areas where this method is commonly used. For a recent project in the Detroit area, the bid tabulation indicated a unit cost for a slurry wall of similar depth in similar soils at \$100 per square foot.

Based on our experience, caisson construction is generally more cost-effective than other methods for projects of this type in similar soil and groundwater conditions. For the purposes of developing conceptual cost estimates, examples of cost comparisons for caisson construction are available in "Sinking Caissons as an Effective Means of Construction Shafts", 1997 Rapid Excavation and Tunneling Conference in Las Vegas, Nevada, June 22-25, 1997, presented as Figure No. 23 in Attachment A. The paper presents the results of a study of case histories on projects where sinking caissons were selected to deal with the impact of soil and groundwater conditions (commonly called bad ground conditions) on the performance and constructability of different shaft systems. The paper also looks at factors that have led to the success or failure of the systems. Cost data for five projects have been reviewed and compared to cost of shaft installation by other means.

For cost estimating drilled piers, Means may also be used, although local bid tabulation information may be more useful to provide up-to-date regional unit cost data. Based on recent projects we are familiar with in the Detroit area involving large diameter drilled piers extending to rock and drilled under slurry, the following unit costs for developing a concept-level cost estimate for the drilled piers are suggested:

• Assuming full length steel cages and slurry installation (necessary for the known site conditions), assume about \$1,400/cy of concrete, in-place. If the steel cage

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can be reduced to only extend about 50 feet deep, the unit rate can be reduced by 10 percent.

• For piers to be socketed into rock (necessary for the main piers), assume an additional \$2,700/cy for the rock excavation and removal (under slurry). For the main piers, it is suggested herein that the piers be socketed at least 5 feet. For estimating purposes, a slightly deeper socket may be considered to provide for variable conditions.

For all foundation installation, the presence of contamination in the upper water bearing granular layers should be considered in developing cost estimates. As indicated elsewhere in this report, the most practical method for preventing cross contamination of upper aquifers is, most likely, the use of a perimeter cutoff wall such as steel sheeting or slurry wall, surrounding each group of piles or drilled piers. The cost of such a wall can be developed from RSMeans or similar cost data, or from local bid tabulations for similar construction.

In any case, concept level cost estimates for foundations should consider what may be fluctuating prices for steel, concrete and fuel, which have resulted in several recent projects being bid significantly higher than the design estimate.

10.6 FOUNDATION CONCLUSIONS AND RECOMMENDATIONS

Various methods for constructing the several different foundation elements are discussed above. Although a number of different construction methods are considered feasible for the various foundation elements, certain methods of construction appear to offer advantages in terms of least risk and probable lowest cost. The following sections provide recommended foundation types and construction methods for the individual foundation elements. It should be noted that these recommended methods are for the purpose of this feasibility report and for feasibility-level cost estimation. It is expected that upon final design, these suggested methods will be refined and modified to reflect the final design requirements.

10.6.1 Cable Stay and/or Suspension Towers

Drilled piers with pier cap and tie-beams are considered the most likely foundation type for these elements. The piers should be drilled under slurry, with the upper fill (20 to 40 feet, depending on location) cased to reduce the possibility collapse of unstable material. The pier area should be isolated using a cutoff wall, to prevent cross contamination of upper aquifers. The piers should be socketed at least 5 feet into rock. Final pier diameter and rock socket length should be determined based on the applied loading and settlement tolerance.

10.6.2 Suspension Anchorages

All of the options presented in the DRIC Engineering Report (and in Section 10.1, above) are considered feasible for installation of the suspension anchorages. However, the specific anchorage loads and sizes are not known at this time and thus detailed cost estimating is not possible at this time.

10.6.3 Bridge Approaches

Drilled piers with a pier cap are considered the most likely foundation type for these elements. Similar to the main pier foundations, the piers should be drilled under slurry, with the upper fill (20 to 30 feet, depending on location), cased to reduce the possibility for collapse of unstable material. Where these foundation elements are isolated or not closely spaced, it may be practical to design the upper casing method to satisfy the requirement for prevention of cross-contamination of aquifers (i.e., a perimeter cutoff wall may not be the most cost effective solution for such cases). The piers may be supported in hardpan or on the top of rock, depending on the required settlement limitations. Final pier diameter should be determined by applied loading and settlement tolerance.

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11.0 SEISMIC DESIGN CRITERIA

The 2003 Michigan Building Code (MBC) states that the site shall be classified as one of the site classes defined in Table 1615.1.1. Where the soil shear velocity is not known, site class shall be determined, as permitted in Table 1615.1.1, from standard penetration resistance or from soil undrained shear strength, calculated in accordance with Section 1615.1.5. The site soil profile does not contain any soils having one or more of the characteristics that would require the site to be classified as Site Class F. Therefore, according to Section 1615.1.5.1 of the MBC, and for the purposes of the concept level design, the site should be classified as Site Class E. A site-specific investigation including determination of shear wave velocities for the various soil layers should be performed prior to actual design of foundation elements to confirm this assessment.

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12.0 LIMITATIONS

The evaluations and preliminary recommendations presented in this report have been formulated on the basis of generalized data in the vicinity of the proposed bridge crossings, together with current preliminary concepts for the bridge and foundations. As such, all of the preliminary conclusions presented herein are considered appropriate for concept-level evaluations of the design and for concept-level cost estimating. Experience indicates that the actual sub-soil conditions at the actual final locations of all the primary and secondary foundation elements may vary from those explored and presented in this report. Therefore, a comprehensive final design-specific geotechnical investigation should be performed to provide geotechnical exploration and analysis at the locations of each primary and secondary foundation element.

The scope of the present investigation was limited to the preliminary site specific evaluation of subsurface conditions for the support of the proposed bridge foundations. Considerations relating to environmental concerns beyond those specifically mentioned in this text, or other possible regulatory restrictions on development, were not included in the scope of this investigation.

Respectfully Submitted,

NTH Consultants, Ltd.

Craig R. Johnson Project Engineer Geotechnical Investigation Donald C. Wotring, Ph.D., P.E. Project Engineer Geotechnical Analysis

Fritz J. Klingler, P.E. Project Manager

JNS/CRJ/FJK

Attachments



				NIN CONSULTANTS, LTG.		
- 11		7	CAD FILE NAME: D05001457	INCEP DATE: 10 MAR 2008	DRAWING SCALE: 1" = 300'	PLOT DATE: 15 SEP 2008
	1		NTH PROJECT No.: 15-050014-12	DESIGNED BY: CRJ	DRAWN BY: SHB	CHECKED BY: FJK
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			FIG	URE	No.	





GENERAL NOTES

TERMINOLOGY

Unless otherwise noted, all terms utilized herein refer to the Standard Definitions presented in ASTM D 653.

PARTICLE SIZES

CLASSIFICATION

Boulders Cobble	-	Greater than 12 inches (305mm) 3 inches (76.2mm) to 12 inches (305mm) 3/4 inches (10.05mm)	The major soil constituent is t clay, silt, sand, gravel. The se constituent and other minor co reported as follows:	he principal noun, i.e., econd major soil onstituents are
Sand - Coarse Medium	-	No. 4 - $3/16$ inches (19.05mm) to 5 inches (70.2mm) No. 4 - $3/16$ inches (4.75) to $3/4$ inches (19.05mm) No. 10 (2.00mm) to No. 4 (4.75mm) No. 40 (0.425mm) to No. 10 (2.00mm)	Second Major Constituent (percent by weight)	Minor Constituents (percent by weight)
Fine	-	No. 200 (0.074mm) to No. 40 (0.425mm) 0.005mm to 0.074mm	Trace - 1 to 12%	Trace - 1 to 12%
Clay	-	Less than 0.005mm	Adjective - 12 to 35% (clayey, silty, etc.)	Little - 12 to 23%
			And - Over 35%	Some - 23 to 33%

COHESIVE SOILS

If clay content is sufficient so that clay dominates soil properties, clay becomes the principal noun with the other major soil constituent as modified; i.e., silty clay. Other minor soil constituents may be included in accordance with the classification breakdown for cohesionless soils; i.e., silty clay, trace of sand, little gravel.

Consistency	Unconfined Compressive Strength (psf)	Approximate Range of (N)
Very Soft Soft Medium Stiff Very Stiff Hard Very Hard	Below 500 500 - 1000 1000 - 2000 2000 - 4000 4000 - 8000 8000 - 16000 Over 16000	$\begin{array}{rrrrrrrrrrrrrrrrrrrrrrrrrrrrrrrrrrrr$

Consistency of cohesive soils is based upon an evaluation of the observed resistance to deformation under load and not upon the Standard Penetration Resistance (N).

COHESIONLESS SOILS

Density Classification	Relative Density %	Approximate Range of (N)	
Very Loose	0 - 15	0 - 4	
Loose	16 - 35	5 - 10	
Medium Compact	36 - 65	11 - 30	
Compact	66 - 85	31 - 50	
Very Compact	86 - 100	Over 50	

Relative density of cohesionless soils is based upon the evaluation of the Standard Penetration Resistance (N), modified as required for depth effects, sampling effects, etc.

SAMPLE DESIGNATIONS

AS	-	Auger Sample - directly from auger flight
BS	-	Miscellaneous Sample - bottle or bag
S	-	Split Spoon Sample - ASTM D 1586
LS	-	Split Spoon Sample S with Liner Insert 3 inches in length
ST	-	Shelby Tube Sample - 3 inch diameter unless otherwise noted
PS	-	Piston Sample - 3 inch diameter unless otherwise noted
RC	-	Rock Core - NX core unless otherwise noted
CS	-	Continuous Sample - from rock core barrel or continuous sampling device

STANDARD PENETRATION TEST (ASTM D 1586) - A 2.0" outside-diameter, 1-3/8" inside-diameter, split barrel sampler is driven into undisturbed soil by means of a 140-pound weight falling freely through a vertical distance of 30 inches. The sampler is normally driven three successive 6-inch increments. The total number of blows required for the final 12 inches of penetration is the Standard Penetration Resistance (N).



SUMMARY OF ROCK LOG NOMENCLATURE

RUN NUMBER

The number of the individual coring interval starting at the rock interface.

ROCK TYPE/DESCRIPTION

Description of the color, grain size or texture, bedding, foliation, lithology and mineralogy.

Color - When describing the color, use only common colors such as gray, brown, green, etc., or simple combinations of these (e.g., yellow-brown). The degree of color (light vs. dark) should also be employed.

Grain Size/Texture - Terminology used to identify size, shape, and arrangement of the constituent elements: e.g., porphyritic, glassy, amygdaloidal, etc.

Where applicable, the following size classification is utilized:

- Amorphous Particles too small to be seen with the naked eye.
- Fine grained Particles barely seen with naked eye.
- Medium grained Particles barely seen with naked eye to 1/8 in.
- Coarse grained Particles between 1/8 in. and 1/4 in.
- Very coarse Particles greater than 1/4 in.

Bedding or Foliation - A bed (or foliation) is the smallest diversion of a stratified series, and marked by a well defined divisional plane from strata or layers above and below. Bedding is the collective term signifying the existence of beds or laminae.

The relative thickness of the bedding planes shall be described as follows:

Bedding Planes Spacing

- Laminated Less than 0.4 in. (1 cm)
- Very thin 0.4 inch (1 cm)
- Thin 2 to 12 inches
- Medium 1 to 3 feet
- Thick 3 to 10 feet

Lithology – Rock name or classification and modifiers such as Limestone, Shaly Limestone, Shale, Calcareous Shale, etc.



WEATHERING/ALTERATION

Weathering (alteration) of the rock (mineral fabric) is caused by mechanical and chemical action (temperature variations, water, bacteria, physical and chemical attack) and produces deterioration of the rock fabric leading eventually to a disaggregated mass resembling soil. The terms used to describe the relative degree of weathering are as follows:

- F Fresh Rock fresh, crystals bright, few joints may show slight staining. Rock rings under hammer if crystalline. SW - Slight Discoloration indicates weathering of rock material and discontinuity surfaces may be somewhat weaker externally than in its fresh condition. MW - Moderate Less than half the rock material is decomposed and or disintegrated to a "soil". Fresh or slight weathered rock present either as a continuous framework or as corestones. Large pieces cannot be broken by hand. More than half the rock material is decomposed and/or disintegrated to HW - High a soil. Rock so weakened by weathering that fairly large pieces can be crumbled by hand. Fresh or discolored rock (slight) may be present as a discontinuous framework or as corestones. CW - Complete Rock reduced to "soil". Rock "fabric" not discernible or discernible only in small scattered locations. The original minerals of the rock have been entirely altered to **RS** - Residual Soil
 - secondary minerals and the original rock fabric is not apparent.

FIELD HARDNESS

A measure of resistance to scratching or abrasion. The descriptions of the relative degrees of hardness are as follows:

•	S - Soft	Reserved for plastic material only.
•	F - Friable	Easily crumbled by hand, pulverized or reduced to powder and is too soft to be cut with a pocket knife.
•	LH - Low Hardness	Can be gouged deeply or carved with a pocketknife.
•	MH - Moderately Hard	Can be readily scratched by a knife blade. Scratch leaves heavy trace of dust and scratch is readily visible after the powder is blown away.
•	H - Hard	Can be scratched with difficulty; scratch produces little powder and is often faintly visible; traces of the knife steel may be visible.
•	VH - Very Hard	Cannot be scratched with pocketknife; leaves knife steel marks on surface.

GRAPHIC LOG OF FRACTURES

A scaled representation of fractures and discontinuities observed along the length of the core run. Fracture angles with respect to the longitudinal axis of the core run shall be noted were applicable.



DESCRIPTION OF ROCK DEFECTS

Description of rock defects shall include information regarding **discontinuities** as well as **solution cavities** or **voids**.

Discontinuities - Surface representing breaks or fractures separating the rock mass into discrete units.

The types of discontinuities are as follows:

•	Crack	A partial or incomplete fracture.
•	Joint	A simple fracture along which no visible shear displacement has
		occurred. May occur with parallel joints to form a joint set.
•	Shear	A fracture along which differential movement has taken place
		parallel to the surface sufficient to produce slickensides, striations
		or polishing. May be accompanied by a zone of fractured rock
		(shear zone).
•	Fault	A major fracture along which there has been
		measurable/observable displacement; often accompanied by clayey
		gouge and/or a severely fractured adjacent zone of rock.
•	Shear or Fault Zone	A band or zone of parallel or sub-parallel shears and/or faults.

Discontinuity Spacing – The spacing should be measured in feet to the nearest tenth perpendicular to the plane in the set.

- IF Intensely Fractured <0.3ft
- CF Closely Fractured 0.3 to 1.0ft
- MF Moderately Fractured 1.0 to 3.0ft
- WF Widely Fractured 3.0 to 6.0ft
- VWF Very Widely Fractured >6ft

Surface Roughness - The terms used to describe the relative degree of surface roughness of the discontinuity are as follows:

•	VR - Very Rough	Near "vertical" steps and ridges occur on the discontinuity
	surface.	
•	R - Rough	Some ridges and side-angle steps are evident; asperities are clearly visible; discontinuity surface feels very abrasive.
•	SR - Slightly Rough	Asperities on the discontinuity surface are distinguishable and can be felt.
•	S - Smooth	Surface appears smooth
•	SLK - Slickensides	Visual evidence of striations or a smooth glassy-appearing finish.

Other terms used for surface roughness can include stepped, planar, and undulating.



Dip/Attitude - The terms used to describe the angle of inclination of the discontinuities with respect to the plane normal to the longitudinal axis of the core run are as follows:

- Horizontal 0 to 5 degrees
- Low Angle 5 to 35 degrees
- Moderately dipping 35 to 55 degrees
- Steep or high angle 55 to 85 degrees
- Vertical 85 to 90 degrees

Discontinuity Infilling - A description of the mineralogy, thickness and hardness of observed discontinuity infilling should be noted.

The terms used to define the relative degree of infilling are as follows:

- ST Surface stain
- Sp Spotty
- P Partially filled; half of surface or opening is filled
- F Filled (partially)
- H Healed

Solution Cavities and Voids - Open spaces in the subsurface are generally due to removal of rock material by chemical dissolution or the action of running water. Since most of these voids result from the action of groundwater, the openings are not usually equi-dimensional, but rather are elongated in the horizontal plane.

The relative size of voids and cavities are as follows:

- Pit or pitted Voids barely seen with the naked eye to 1/4 in.
- Vug Voids 1/4 in. to 2 in. in diameter
- Cavity Holes 2 in. to 2 ft. in diameter
- Cave Holes 2 ft. and larger in diameter



PERCENT CORE RECOVERY

The amount of core actually recovered divided by the length of the run (expressed as a percentage). Both intact and weak rock including gravel sized pieces are included in the percent recovery.

RQD (ROCK QUALITY DESIGNATION)

Total length of all "<u>intact</u>" pieces of core greater than 4-inches in length measured along the <u>centerline</u> of the core, divided by the total length of the run. Mechanical discontinuities such as those resulting from the core operation or handling of the core sample should not be included in the length measurements for RQD.

FRACTURES/FOOT

The number of naturally occurring fractures observed over the length of the recovered core divided by the length of the total core run.

CORE BOX NUMBER

The box number in which the core is stored.

COMMENTS

Comments include information on drilling water losses, reasons for core loss or fracture, gas readings, average pull-down pressure used to advance the run, total time required to complete the run and any other data pertinent to the core operation and/or condition of the core.

Miscellaneous Features - Any additional characteristics to further identify and evaluate the rock from the standpoint of engineering properties: secondary mineralization, fossils, swelling and slaking properties, etc.



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LOG OF TEST BORING NO: TB-106		Ň	11	NT	ΉCα	onșu	itanté	i, Lto	1.
Project Name: LiktC Preliminary Foundation Investigation		(K	し え	1	NTH Pr	oj No.:	15 0500	114.00	
Project Location: Detroit. Att	_	, <u> </u>	9	(Checke	d Ry		-	
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Figure No. 9

LOG OF TEST BORING NO: TB-107		N	TH	NT	ΉCα	onsul	tants	s, Lto	l.
Project Name: DRIC Preliminary Foundation Investigation		(n	الأ	1	NTH Pr	oj. No.	15.050	973-00	
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Project Name – 1960, Proliminary Foundation Investigation		-			NUMPO	oj No	15-050	n4-00	
Project Location: Detroit, MI			Ş	(Chocke	d By:			
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Figure No. 10



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20										460	Light Orey Amorphous LIMESTONE, Lamineted Bocking, Occasional Hosells and Pite. Rivoliths along core length, especially between 136-1 feet and 120-3 feet. Chert Banding and Nodules along core length, ospecially between 716 feet and 120-3 teel.	1100 1100 1104 1104			SR Sr Sr	•
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190	,	132	100 0	45	04	4/151			450	Light Gray Amorphous I (MESTON), Laminaled to Thinly Beaded, Chr. assonal Pits and Cossils - A Victus noted at 128.7 feet A Stychols nuted at 103.13 feet Chert Notule a noted at 128 83 feet. Potrolnum Starning, throughout	197 e 177 e 198 h	< L <	1 2 -	N VR VN	
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40,	¢	147	w/ U	∎D.	1 11	Мл	,		445	DETROIT RIVER GR (LUCAS FORMATION) Light Gray Amorphous LIMES (ONE, Laminated Berling, Occasional Possils and Vuga, numerous Pits Styolite noted at 134 h (set Chart noted between 138.35 feet and 138.8 feet. Numerous hasing Gracks noted between 140 / 5 feet and 142 (set 1/4-inch Clay or Ash Interbed (possible Kawkawin Bentonite) noted at 137.77 feet. Potroioum Stammy (moughout	130 0 130 0 137 3 137 8 137 8 138 4 138 4 138 4 138 4 140 0 141 2 141 4 141 9	••••••••••••••••••••••••••••••••••••••		A SERRER S SER	
45									405		143 4 143 4 143 0	101	LI.	iste Ijan Fe	
50.	2	157	. 100 0	63	.08	, Mi1 .			430.	Gray Amorphous LIMESTONE with Laminated Dadding bacoming Dolomitic briew 146.57 feet Occasional Linna H. Styollies noted at 134.6 feet, 140 foot and 150.7 feet. Vertical Crack noted between 146.57 and 149.18 (aat with a midpoint at 147.93 leat A Vertical Fracture is noted between 149.6 and 160.18 foot with a midpoint at 740.8 feet. Unconformity noted at 149.95 feet. Turbidite Bedding noted between 149.96 and 100.6 feet.	147 L 145 R 165 7 150 7	0	> >=> L	SH SH SH VR S	
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109	1	100		, <i>4</i>	12.	MH .	۴			485	Light Gray Amorphous LIMESTONE, Laminated Rending, Occasional Pite, Intersoly Fractured Detween 60.05 and 100 feet. Petrolaum Staining Throughout	_				
106,	7	100	. 70 U	an	1 m	мн 		10 8		480	Light Gray Amorphous LIME STONE, Lawinated Rending, Chori banding noted between 103.58 and 104.3 (set Styolites noted between 103.7 and 104.1 (set	103 1 104 9 104 2	г - г - г	7 7 T I	11 × 11	5P 20
110	3	110	100 0	_97.	00	MI	. "	יפי.		475	Light Gray Amorphous LIME STONE, Leminaled (incluing, Occasional Pils, especially between 305 and 105.25 fnot Styollies at 107.25 feet, 107.75 feet, and 1 <u>09.15 feet. Petrol</u> eum staining throughout	107.6 108.9 110.4 110.4 110.5 110.6	сіс г с	I 7 217	38 58 58 58 58 58 58 58 58 58 58 58 58 58	ън -
115	4	115	1000	- 33	27.	"мн.	. r		1 1 1 1 1 1 1		Light Gray Amorphoue I IMICSTONE, Lemmand Bricking, Occasional Pits, copromity between 111-3 and 112-75 feet - Calorie Intilling located at 114-08 feet - Percleum <u>Ste</u> ining (traughout Fod of Doring at 115 feet	1110		ISTITUS:	**************************************	58 58 58 58
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	1.	107 5	29.0	22.	2.07	<u>. M-H</u>	cw	22.4		485	Light Gray Amorphous to Fine Grained CIMLSTONE, Leminated Radding, Occasional Fite and Losaits Stypictes noted heween 101-1 and 101-25 feat Intensely Fractured 00 to 99-5 feet and between ±01.7 to 102.3 test	100 B 100 Q 101 U 101 0 101 0 101 0		M - 2 2 2 3	5.M 5.M 5.M 5.M	
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1Q	,	11/ 2	WU U	1	u 4 .	Alter				475	Light Gray Amorphones to Fine Grained LIMESTONE, Laminated (withing, Occasional Foseils, Pile and Vuge noted from 107.1 to 108 fent and 117.1 to 112.3 feet Numerus alyoites noted along core length, especially between 103.3 and 103.5 feet. Chert Sands noted at 104 feet. 104.5 feet, and 109.2 feet. Petroleum Silamed	1100	· · ·	II II	н 907 902 811	
10										470	Light Gray Amarchans to First Grained LIMESTON:	1 85 5 1 16 0 1 18 9 1 17 0			SH SN SR SR SR	
20 ;									L L		Laminated Northing, Occasional Fossila, Occasional Pile and Vuga, requiring between 112.6 and 117.3	114 0	-	!!	5	
	3	122 1	ne 0 .	90.	ov.	λητι	•	. 15 4 ;		465	Feet, Numerous Stychtvis Holed stong core length Occasional Chart Humin and Nodules noted stong core length. Petroleum Stained, capacially between 112 5 feet and 117 3 feet	121 1 121 4 122 6 122.5 122.5 122.6		I I I I I	5 0 0 0 0 0 0 0 0 0 0 0 0 0	
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LOG OF CORE BORING NO: TB-104E



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30	•	1.32 11	י כטו	24	10	. MII	,			400	Light Gray Annophous to Fine Grained LIMESTONE, Laminated Bodding, Occasional Foreits and Pite. Styolites noted at 122.9 feet, 125.25 fmil, and from 227.4 to 130.85 feet. Occasional Grant rodules Polityleum Stathod	129 0 129 7 130 3 130 6 130 7 131 4				20. 40
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									1.1		DETROIT NIVER OR (LUGAS FORMATION)	126.2	, J	н	50	5
										450.	Light Gray Could to Gray Amorphous to Fine Grained LIVE STUNE', Lammated Sedding, Occasional Feasile and Pile noted along core - Interactly Fractured	137 e 130 e		- 15	Six Six	5
40											between 140 4 and 140 9 feet. Vertical Crack onter between 140 04 and 141.75 feet, with a midpoint at 141 35 feet. Horizontal Cracks noted between 130.55 and 139 9 feet. Occasional Chart Bands and Nodvies	140 P 140 P 140 P	202	122	NH Sir H	2002
	,	142.5	180 0	78	12	мн	. * .			446	noted glong core length Petroleum Steined	142 (L 142 (L	1	ΞI	5 314	;
45	ļ								ļ,		Light Gray Amorphous to Fine Grained LIMESTONE, Leonoured (souther Lightenione grading to Colomite	140 B 144 2 144 7	3	TIT	9 99 9	5
				İ					L T		Occasional Fossile and Pits, noted optwoon 142.5 and 147.5 faal Stypilio noted at 145.67 feet A marty	145.0	11	V I I	5M 5M 5M	1
									11	440_	between 145 foot and 145.4 foot, with a midpoint at 145.2 foot. Another many Vertical Crack is located between 150.5 feet and 152.5 feet with a midpoint at	147 6	ŝ	4	5	
50									11		151 5 feet Numerous filled and healed Cracks are noted stong core length, especially between 143.5 to	140 D 149 B	;	н	á sn	5
		1020.	rių 47	ъv	12	мн					146 feet, and 152 25 to 152 5 feet. Turomito zone la noted between 146 88 and 146 feet. An Unconformity exists at 146 feet. The Lintestone is grading to Determite below 150 feet.	150 B	L I	Ξī	SPR 21	
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115	. 1	. 117	17KI <u>E</u>	νņ	() m3	MPI				470	Light Gray Amorphous LIMESTONE, Laminated Bodong, Occasional Pitting, noted between 111 and 112 lost. Check Hunding 0.25 and 2 inches thick present between 113.96 and 117 feet. Styliftee noted at 112.75 feet, 113.4 feet, 113.73 feet. 114.3 feet, and 114.6 to 116.7 feet. Petroteum staining predominately between 111 feet_ass 112.0 feet. End of Bolling at 117 feet.	113.0 114.6 115.6 116.6		I LIL I	OM SR OM SP SP	
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103	_2	1.00	100.0		1.	, MPI ,	sw			BQ.	Light Grey Amorphous, LIMESTONE, Leminated Indding, Occasional I'lts and Fossis noted. Styckes noted at 104-1 and 104-58 feet. Finticioum Steroing Detwoon 100 <u>5</u> and 105 feet.	100 m 101 8 102 0 102 0 102 4 104 4 104 4 104 5 107 4 107 4 108 4 100 1			SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR RR SR SR SR SR SR SR SR SR SR SR SR SR SR S
110		110	0 # 0.	14	12	MH	г			/5	Light to Dark Gray Amorphous, LIMESTONE, Lemmated Bodding. Pite and Vugs noted throughout, but most prevalent between 105 and 110 fact. Chort bands between 114 04 to 114 17 feet and from 114.4 to 114.5 feet. Alymbias noted between 105 3 and 106 21 fact at 107.1 fact. 107.4 feet, between 105 04 and 108 6 feet, at 110 2 feet. 112.35 feet, and between 112 66 and 114 07 feet. Patroleum Steining present logiween 105 and 109 5 feet.	110 2 111 4 111 6 111 6 112 0 112 0 112 0 113 2			SR SR SR SR SR SR SR SR SR
120,										70	Light Gray Amorphous, LIMES (ONE, Lammated Redding, Pita and Cossila noted throughout, Chert Banding, noted between 115 and 117 8 feet. Styollian noted between 115 and 118 feet, at 110 feet. 4t 122.17 feet, and at 122 7 feet. A strong (~80 coproc) Augle Joint is located between 118 1 and 110 7 feet with a midpoint at 118.4 feet. Wavy Carbonaceus (sending	115 0 116 0 116 0 117 0 117 0 117 0 117 0 117 0 117 0 118 0 118 0 118 0 118 0			********
125	4.	126	a / n	וה		мн	. ' ;		1 48 1 1	95	noted between 119 27 and 123 35 from with a recipient at 121 31 (set. Two nearly Vertical Joints noted at 118.4 and 121.31 (set. Intense Fractiving noted between 115 54 and 117 5 linit, and between 120 4 and 121 13 feet. Petroleum Stalning noted between 116 2 and 125 feet.	1 19 5 1 19 8 1 20 6 1 20 4 1 20 4 1 20 6		-1221-	VRAR SON

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30	ő	, •30	. 199 0.	e Q	. 1 .	41-1	, r			-60	Light Gray Amorphous, LIMI'STONE, Laminaled Bedding, Pits and Losells noted throughout. Chen noted at 125.3 feet. (296)(cell at 125.27 feet. 127.87 feet, ftom 126.58 to 124.7 feet, 129.29 feet. and 129.71 feet. Petroleum Staining throughout, but most provetont between 127.7 and 129.8 feet.	1224 3 1224 3 1226 1 1226 1 1226 4 1226 4 1206 4 10		12-222222	*********
35	U	137	100 0.	v 0 ,	. 0.4	MH	,		1 1-1 -1 1 1 1 1	A55	Gray Amorphous, LIMESTONE, Laminaled Dadding Pits are present throughout with Vuge at 132 feet and 134.25 feet, 131.5 feet, and 131.5 feet, 130.4 feet and 130.5 feet, 131.5 feet, and 131.5 feet, Potroleum Signed from 131.25 to 132.5 feet and 133.4 to 134.5 feet	134 5		N N	54
40	7	1+0	420	1 1	175	мн	r			450	Light Grey Amorphous LIMESTONT. Lominated Dedding Pilling thorughout, but most provisiont Detween 136-2 and 137-5 (set Chort nond between 135 and 135-4 feet A stdep (~75 degree) to combu Vertice/ Joint is noted between 138 and 139 feet with a midpoint at 135-5 feet Interview noted between 136-2 and 136-50 feet Interview noted between 136-2 and 136-50 feet Interview noted between 137-2 to 137-63 feet and from 138-66 to 139 feet Thebolaum Standing throughout, but most	•354 1965 1576 1576 1581 1384 1380 1401			54 54 20 20 20 20 20 20 20 20 20 20 20 20 20
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46.	•	мя.	.≈2 0 .	m o .	14	м н ,	C_		1		141 and 142 38 foot with midpoint at 141 99 foot. A Varical Crack is noted between 144 35 and 144 9 feet with a midpoint at 144 64 feet. Interse Cracking along laminations between 143.85 and 144.25 feet. Patroinum Stationg, aspecially between 140 and 140.92 feet.	143 41 144 3 144 0 144 0	0.444	<121	995 -
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Johnson Inspector Subsertice (Lacation) Milling Inspector Subsertice (Milling) Milling Inspector Milling Inspector Dunce (Linestrone) Milling Milling Inspector Milling Inspector Dunce (Linestrone) Milling Milling Inspector Milling Inspector Milling Milling Milling Inspector Milling Milling Milling Milling Milling Inspector Milling Milling Milling Milling Milling Inspector Milling <	Clining: Detroit, MI Inspector N. Emory/D. Adler Proj. No: 15 050014-00 Contractor DLZ ng Date: 4/24/2008 - 4/30/2008 Driller. D. Hoskins i Size: NO Checked By: C. Johnson i Size: Size: Size: Size: i Size: Size: Size: Size: i Size: Size: Size:	ECHARDE: Detroit, MI Inspector N. Emory/D. Adler Proj. No: 15 050014-00 Contractor DLZ Ing pate: 4/24/2008 - 4/30/2008 Driller. D. Hoskins Size: NO Checked By: C. Johnson Size: NO Size: Size: No: Size: Size: Size: Size: Size: Size: Size: Size: Size: Size: Size: Size: Size: <td>Periol. No: Delirolit, MI Inspector N. Emory/D. Adler Proj. No: 15 050014-00 Contractor DLZ Inspector N. Emory/D. Adler Proj. No: 15 050014-00 Contractor DLZ Inspector N. Emory/D. Adler Discontinuities Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Size: Discontinuities Discontinuities I Size: NO Size: Size: Discontinuities Discontinuities I Size: Size: Size: Size: Discontinuities</td>	Periol. No: Delirolit, MI Inspector N. Emory/D. Adler Proj. No: 15 050014-00 Contractor DLZ Inspector N. Emory/D. Adler Proj. No: 15 050014-00 Contractor DLZ Inspector N. Emory/D. Adler Discontinuities Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Checked By: C. Johnson Discontinuities I Size: NO Size: Discontinuities Discontinuities I Size: NO Size: Size: Discontinuities Discontinuities I Size: Size: Size: Size: Discontinuities

LC	G	OF	C	DRI	EE	OF	RIN	GI	NO:		TB-108 NTH NTH CONSU	LΤ/	AN'	TS,	LT	D.
Pro Pro NT	ject ject H Pr	Nar Loc oil N	ne: atior lo:	DF ז: [15-	NC - Detro	Pre bit, N	limir Al -00	läry	Fou	ndaf	lon Investigation Checked By: C. Johnson					
Depth (2)	<u>م</u>	Esen il	Receiver	i i i i i i i i i i i i i i i i i i i	Fractures "por	Hathess	Westervy (Uncompile		jën K	Connect Continue Filmation - 677-7	194 194	Dinc R R	ontinu X	Sugness of	e.
135		120						17.4		445						
140	•	-56	. */ 0	. 17	. 0.	м	aw.			440	Gray Line Grained FOSSILIFEROUS LIMESTONE, Laminated Rodding, Petroleum Odor and Occusional Pits note <u>d th</u> youghout run. Eng of Boring at 139 feet	; 				
+45																
. 150.																
155																
100																
165																
NO	TES															



D050014PZ101 12 AUG 2008



Project No. 15-050014-00

NTH Consultants, Ltd.

DRIC Preliminary Foundation Investigation

TABULATION OF LABORATORY TEST DATA

			p (ft)	ive			(H)		F	PARTIC	CLE SIZ	e dist	RIBUT	ION (%	b)	AT LI	TERBE MITS ('	.RG %)	vity		ion
Boring / Designation	Sample Number	Depth of Sample (ft)	Elevation of Sample Ti	Unconfined Compress Strength (PSF)	Failure Strain (%)	Natural Water Content (% of dry weight)	In-Place Dry Density (I	Permeability (cm/sec)	Colloids	Clay (passing 200)	Silt	Fine Sand	Medium Sand	Coarse Sand	Gravel	Liquid Limit	Plastic Limit	Plasticity Index	Apparent Specific Grav	Loss on Ignition (%)	Unified Soil Classificat
TB-101	LS-11	44	537			19.3	109.1			3	23	74	0	0	0						
	LS-14	59	522	1350	14.7	21.6	107.3			47	33	12	6	1	1						
	LS-18	75	506	820	14.5	35.9	86.5			72	26	1	1	0	0						
	LS-21	90	491	14050	8.8	14.0	120.3			39	36	14	8	3	0						
TB-102	LS-4	20	561			40.9	78.0													5.4	
	LS-7	34.5	546.5			19.4	110.7			2	3	90	5	0	0						
	LS-10	49.5	531.5			19.9				2	23	75	0	0	0						
	LS-14	68	513	870	14.5	25.9	100.4														
	ST-1	70	511			25.0	101.3														
	LS-18	90	491	20370	11.7	9.1	136.1														
TB-103	LS-6	29.5	552			14.2	106.9			1	1	16	46	17	19						
	LS-11	54.5	527	550	14.4	22.1	106.7														
	ST-1	63	518.5			21.5	106.8														
	LS-13	64.5	517	1160	14.5	22.7	106.3														
TB-104	LS-5	25	568			25.4	100.4			3	8	61	27	1	0						

FIGURE NO. 21

Project No. 15-050014-00

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DRIC Preliminary Foundation Investigation

TABULATION OF LABORATORY TEST DATA

			ip (ft)	ive			PCF)		F	PARTIC	CLE SIZ	e dist	RIBUT	ION (%)	AT LI	TERBE MITS ('	.RG %)	vity		ion
Boring / Designation	Sample Number	Depth of Sample (ft)	Elevation of Sample Ti	Unconfined Compress Strength (PSF)	Failure Strain (%)	Natural Water Content (% of dry weight)	In-Place Dry Density (I	Permeability (cm/sec)	Colloids	Clay (passing 200)	Silt	Fine Sand	Medium Sand	Coarse Sand	Gravel	Liquid Limit	Plastic Limit	Plasticity Index	Apparent Specific Grav	Loss on Ignition (%)	Unified Soil Classificat
TB-104	LS-6	30	551	770	14.7	29.8	95.6														
	LS-7	35	546	410	14.6	44.1	77.1			61	38	1	0	0	0						
	LS-10	50	531			22.4	110.4														
	LS-14	70	511			24.6															
	LS-19	95	486			11.3	127.1														
TB-105	LS-5	15	578.5			137.2	32.0													32.2	
	LS-9	30	563.5	240	14.6	52.5	69.2														
	LS-17	70	523.5	880	14.7	25.7	102.7														
	LS-21	90	503.5			9.3				33	34	16	9	3	5						
	LS-22	93.5	500			13.2				6	13	38	21	10	12						
TB-106	LS-5	20		1970	14.7	25.3	101.0														
	LS-9	40				29.9															
	LS-16	75				21.5															
	LS-18	85		920	14.8	27.0	98.1														
TB-107	LS-2	20	566	600	14.9	26.9	99.2			35	64	1	0	0	0						

Project No. 15-050014-00

NTH Consultants, Ltd.

DRIC Preliminary Foundation Investigation

TABULATION OF LABORATORY TEST DATA

			p (ft)	ive			PCF)		F	PARTIC	CLE SIZ	E DIST	RIBUT	ION (%))	AT LI	TERBE MITS (9	RG %)	vity		ion
Boring / Designation	Sample Number	Depth of Sample (ft)	Elevation of Sample Ti	Unconfined Compress Strength (PSF)	Failure Strain (%)	Natural Water Content (% of dry weight)	In-Place Dry Density (I	Permeability (cm/sec)	Colloids	Clay (passing 200)	Silt	Fine Sand	Medium Sand	Coarse Sand	Gravel	Liquid Limit	Plastic Limit	Plasticity Index	Apparent Specific Grav	Loss on Ignition (%)	Unified Soil Classificat
TB-107	LS-4	30	556	350	14.6	46.1	71.6														
	LS-10	60	526			22.6	105.4														
	LS-15	85	501			42.3	80.2														
	LS-16	90	496			19.9	107.3			9	69	22	0	0	0						
TB-108	LS-3	15	562	280	14.7	40.8	81.8														
	ST-1	38	539			21.2	107.4									26	16	10			
	LS-8	40	537	550	14.6	21.0	108.0														
	LS-9	45	532			20.9	110.5														
	LS-19	95.0	482			7.9	142.1			8	19	22	12	10	29						



NTH Consultants, Ltd. Southeast Michigan Laboratory

Telephone: 248. 553.6300 Fax: 248.324.5179

aare	aate	Soil Test Report		Report No: I	MAT:15-050014	-00-S03 Issue No:
lient:	The C	orradino Group		This labora of State Hig (AASHTO)	nory is accredited by Ame ghway and Transportation The test(s) reported have	rican Associati Officials
Project:	Detro Geote	it River Inter. Crossing echnical Engineering		IT BCCORdar	noe with the lerms of the e	ccreditation.
ob No:	15-05	60014-00		Date of Is Approved	sue: 6/19/2008 Signatory: Zeerak Pa	ydawy
Sample [Detail	S	Other Test Resul	lts	an that significantly a significant	
oring No:	le No:	TB-101	Description Sand Gravel Description	ASTM D 422	Result	Limits
ample Dep ate Sampl	oth: ed:	44	Shape Hardness			
ampled By	/:	Nathan Emery	Dispersion Device		4	
imple Loca	ation:	Detroit River International Crossing	Moisture Content (%) Wet Density (lb/ft ³)	ASTM D 221	16 19.3 130.2	
Particle	Size C	Distribution				
article :	Size C	istribution		Method: A	STM D 422	
article	Size C	Distribution		Method: A Drying by: O	STM D 422 oven	
Particle \$	Size C	Distribution		Method: A Drying by: O	STM D 422 Ven	
article : %Pess	Size C	Distribution		Method: A Drying by: 0	STM D 422 oven	
article : %Pas 100 T · · · ·	sing	Distribution		Method: A Drying by: 0 Sleve Size 1in (25.0mm)	STM D 422 oven % Passing 100	Limits
article : %Pas 100 - · · · ·	Size C	Distribution		Method: A Drying by: 0 Sleve Size 1in (25.0mm) 3⁄in (19.0mm)	STM D 422 oven	Limits
article : %Pas 100 - · · · 80 - · · · 80 - · · ·	Size C	Distribution		Method: A Drying by: 0 Sleve Size 1in (25.0mm) ¾in (19.0mm) 3/8in (9.5mm) No 4 (4.75mm)	STM D 422 oven % Passing 100 100 100	Limits
article : %Pas 100 - · · · 80 - · · · 80 - · · ·	Size C	Distribution		Method: A Drying by: O Sleve Size 1in (25.0mm) 3/8in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm)	STM D 422 oven % Passing 100 100 100 100 100	Limits
article : %Pes 100 - · · · 80 - · · · 80 - · · · 70 - · · ·	Size C	Distribution		Method: A: Drying by: O Sleve Size 1in (25.0mm) ½in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm)	STM D 422 Ven % Passing 100 100 100 100 100 100	Limits
article : %Pes 100 - · · · 80 - · · · 80 - · · · 80 - · · · 80 - · · · 80 - · · · 80 - · · ·	Size C	Distribution		Method: A Drying by: O Sleve Size 1in (25.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.4 (4.75mm) No.10 (2.0mm) No.20 (850µm) No.20 (850µm) No.40 (425µm)	STM D 422 over % Passing 100 100 100 100 100 100 100	Limits
article : %Pes 100 - · · · 50 - · · · 60 - · · · 50 - · · ·	Size C	Distribution		Method: A Drying by: ○ Sleve Size 1in (25.0mm) ¾in (19.0mm) ¾in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.20 (850µm) No.40 (425µm) No.60 (250µm) No.100 (150µm)	STM D 422 oven % Passing 100 100 100 100 100 100 100 100 100 10	Limits
article : %Pes 100 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · · 90 - · · ·	Size C	Distribution		Method: A: Drying by: O Sleve Size 1in (25.0mm) ¾in (19.0mm) ¾in (19.0mm) ¾in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.40 (425µm) No.60 (250µm) No.100 (150µm) No.200 (75µm)	STM D 422 oven % Passing 100 100 100 100 100 100 100 100 100 10	Limits
Particle : %Pas 100 90	Size C	Distribution		Method: A Drying by: O Sleve Size 1in (25.0mm) ½in (19.0mm) ¾in (19.0mm) ¾in (19.0mm) ¾in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.40 (425µm) No.60 (250µm) No.100 (150µm) No.100 (150µm) No.200 (75µm) 0.049 mm	STM D 422 oven % Passing 100 100 100 100 100 100 100 100 100 10	Limits
Particle : %Pas 100 90	Size C	Distribution		Method: A: Drying by: O Sleve Size 1in (25.0mm) ½in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.40 (425µm) No.60 (250µm) No.100 (150µm) No.200 (75µm) No.200 (75µm) 0.035 mm O.035 mm 0.023 mm	STM D 422 oven % Passing 100 100 100 100 100 100 100 100 100 10	Limits





Infrastructure Engineering and Environmental Services

NTH Consultants, Ltd. Southeast Michigan Laboratory

Telephone: 248. 553.6300 Fax: 248.324.5179

Aggregate/Soil Test Report				Report No: MAT:15-050014-00-S03			
Client:	The C	Corradino Group		This laboral of State Hig (AASHTO)	ory is accredited by Ame hway and Transportation The least(s) reported has	Hican Associa Officials	
Project:	Detroit River Inter. Crossing Geotechnical Engineering			Becordance with the terms of the accreditation.			
Job No:	15-05	50014-00		Date of Issue: 6/19/2008 Approved Signatory: Zeerak Paydawy			
Sample	Detail	S	Other Test Resu	lts			
Boring No:		TB-101	Description	Method	Result	Limits	
ield Samp ample Dep ate Sampl ampled By WO No: ample Loc	le No: pth: led: y: cation:	LS-14 59 Nathan Emery 000307 Detroit River International Crossing	Unconfined Compressive Streng Shear Strength (lb/ft ²) Ave. Rate Strain to Fail Strain at Failure(%) Average Height (in.) Average Diameter (in.)	ure(%)	6 1353 677 0.9 14.7 2.728 1.339		
			Init. Dry Dens. Init. Water Content (%) Liquid Limit		2.0		
			Plastic Limit				
Particle	Size C	Distribution	Plastic Limit	Method: AS Drying by: Ov	STM D 422 /en		
Particle :	Size C	Distribution	Plastic Limit	Method: AS Drying by: Ov	STM D 422 /en		
Pes 100	Size C	Distribution	Plastic Limit	Method: AS Drying by: Ov	STM D 422 /en		
Pes %Pes 100 · · · 80 · · ·	Size C	Distribution		Method: AS Drying by: Ox Sleve Size 1in (25.0mm) %in (19.0mm)	STM D 422 /en % Passing 100 100	Limits	
Pess 100 • • • • 80 • • •	Size C	Distribution		Method: AS Drying by: Ov Sleve Size 1in (25.0mm) ¾in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm)	STM D 422 /en % Passing 100 100 100 99	Limits	
Pess 100 · · · 80 · · · 80 · · · 80 · · ·	Size C	Distribution		Method: As Drying by: Ox Sleve Size 1in (25.0mm) ½in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm)	STM D 422 /en % Passing 100 100 100 99 98	Limits	
Particle : %Pes 100	Size C	Distribution		Method: As Drying by: Ox Sleve Size 1in (25.0mm) ¾in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.20 (850µm)	STM D 422 /en % Passing 100 100 100 99 98 98 94 02	Limits	
Pes %Pes 100	Size C	Distribution		Method: As Drying by: Ox Sleve Size 1in (25.0mm) ¼in (19.0mm) ¾in (19.0mm) ¾in (19.0mm) ¾in (9.5mm) No.4 (4.75mm) No.40 (2.0mm) No.40 (425µm) No.40 (425µm) No.60 (250µm) No.60 (250µm)	STM D 422 Ven % Passing 100 100 100 99 98 98 94 92 89	Limits	
Pes %Pes 100 • • • 80 • • • 80 • • • 80 • • • 80 • • • 80 • • • 80 • • • 80 • • • 80 • • • •	Size C	Distribution		Method: As Drying by: Ox Sleve Size Ox 1in (25.0mm) Yin (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.4 (4.75mm) No.10 (2.0mm) No.20 (850µm) No.20 (850µm) No.40 (425µm) No.60 (250µm) No.100 (150µm) No.100 (150µm)	STM D 422 Ven % Passing 100 100 100 99 98 98 94 92 89 85	Limits	
Particle : %Pas 100 1 · · · · 80 · · · ·	Size C	Distribution		Method: As Drying by: Ox Sleve Size 1in (25.0mm) ½in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.40 (425µm) No.60 (250µm) No.100 (150µm) No.100 (150µm) No.200 (75µm) Ox Ox Ox	STM D 422 /en % Passing 100 100 100 99 98 94 92 89 85 80 76	Limits	
Particle : %Pes 100		Distribution		Method: As Drying by: Ox Sleve Size 1in (25.0mm) ½in (19.0mm) ¾in (9.5mm) 3/8in (9.5mm) No.4 (4.75mm) No.4 (4.75mm) No.10 (2.0mm) No.20 (850µm) No.40 (425µm) No.40 (425µm) No.60 (250µm) No.100 (150µm) No.200 (75µm) 0.039 mm 0.028 mm	STM D 422 Ven % Passing 100 100 100 99 98 94 92 89 85 80 76 74	Limits	

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NTH Consultants, Ltd. Southeast Michigan Laboratory

Telephone: 248. 553.6300 Fax: 248.324.5179

Aggre	gate/Soll lest Report		Report No: MAT:15-050014-00-S0: Issue No:				
Client:	The Corradino Group		This laboral of State Hig (AASHTO)	lory is accredited by Ame hway and Transportation The test(s) moonted has	rican Associa Officials		
Project:	Detroit River Inter. Crossing		ut accordan	ce with the terms of the a	coreditation.		
	Geotechnical Engineering			Bound Parte	dere.		
Job No:	15-050014-00		Date of Ist Approved	sue: 6/19/2008 Signatory: Zeerak Pa	ydawy		
Sample	Details	Other Test Resul	ts				
Boring No:	TB-101	Description	Method	Result	Limits		
leid Samp	le No: LS-18	Unconfined Compressive Strengt	h (16/112) ASTM D 216	8 823			
ample Dep	oth: 75	Shear Strength (lb/ft ²)		412			
ate Sampl	ed:	Ave. Rate Strain to Fallu	ıre(%)	0.9			
ampled By	A Nathan Emery	Strain at Failure(%)		14.5			
WU NO:	000307 Detroit Diver International Creasing	Average Height (in.)		2.751			
ample Loc	ation: Detroit River International Crossing	Average Diameter (In.) Height-Diameter Potio		1.347			
		Init Dry Dens		2.0			
		Init. Water Content (%)					
		THE FIGLOT OUTLOTE [70]					
		Liguld Limit					
Particle &	Size Distribution	Liquid Limit Plastic Limit					
article	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or	STM D 422 ven			
Particle &	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or	STM D 422 ven			
Particle &	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or	STM D 422 ven			
article &	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or Sieve Size	STM D 422 ven % Passing	Limíts		
article &	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or Sieve Size 1in (25.0mm)	STM D 422 ven % Passing 100	Limíts		
article 3 %Pas 100 ····	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or Sieve Size 1in (25.0mm) ¾in (19.0mm)	STM D 422 ven % Passing 100 100	Limits		
article : %Pas 100 ···· 50 ····	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or Sieve Size 1in (25.0mm) 3/8in (19.0mm) 3/8in (9.5mm)	STM D 422 ven % Passing 100 100 100	Limits		
article 3 %Pas 100 ···· 50 ···· 50 ····	Size Distribution	Liquid Limit Plastic Limit	Method: A: Drying by: O: Sieve Size 1in (25.0mm) ¼in (19.0mm) ¾in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm)	STM D 422 ven % Passing 100 100 100 100	Limits		
article : %Pas 100 ···· 50 ···· 50 ····	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or Sieve Size 1in (25.0mm) 3/8in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm)	STM D 422 ven % Passing 100 100 100 100 100	Limits		
article : %Pas 100 50 50 60	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or Sieve Size 1in (25.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm)	STM D 422 ven % Passing 100 100 100 100 100 100	Limits		
Pass 100 100 100 100 100 100 100 100	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or Sieve Size 1in (25.0mm) 3/8in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.20 (850µm) No.40 (425µm)	STM D 422 ven % Passing 100 100 100 100 100 100 100 100 100	Limits		
Pass 100 90	Size Distribution	Liquid Limit Plastic Limit	Method: A: Drying by: O: Sieve Size 1in (25.0mm) ¾in (19.0mm) ¾in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.20 (850µm) No.40 (425µm) No.60 (250µm)	STM D 422 ven % Passing 100 100 100 100 100 100 100 99 99 99	Limits		
Pass 100 100	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or Sieve Size 1in (25.0mm) ¾in (19.0mm) ¾in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850μm) No.20 (850μm) No.20 (850μm) No.60 (250μm) No.60 (250μm)	STM D 422 ven % Passing 100 100 100 100 100 100 100 99 99 99	Limits		
Pass 100 100	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or Sieve Size 1in (25.0mm) ¾in (19.0mm) ¾in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.20 (850µm) No.20 (850µm) No.60 (250µm) No.60 (250µm) No.100 (150µm) No.200 (75µm) 0.036 mm	STM D 422 ven % Passing 100 100 100 100 100 100 99 99 99 99 99	Limits		
Particle 3 %Pas 100 80	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or Sieve Size 1in (25.0mm) ¾in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850μm) No.20 (850μm) No.20 (850μm) No.40 (425μm) No.60 (250μm) No.60 (250μm) No.100 (150μm) No.200 (75μm) 0.036 mm 0.025 mm	STM D 422 ven % Passing 100 100 100 100 100 100 99 99 99 99 99 99	Limits		
Particle 3 %Pas 100 80	Size Distribution	Liquid Limit Plastic Limit	Method: As Drying by: Or Sieve Size 1in (25.0mm) 3/in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.40 (425µm) No.40 (425µm) No.40 (425µm) No.40 (425µm) No.60 (250µm) No.100 (150µm) No.200 (75µm) 0.036 mm 0.025 mm 0.016 mm No.16 mm	STM D 422 ven % Passing 100 100 100 100 100 100 99 99 99 99 99 99 99 99 99 99 99 99 9	Limits		



Form No: 18909.V1.00

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Infrastructure Engineering and Environmental Services

NTH Consultants, Ltd. Southeast Michigan Laboratory

Telephone: 248. 553.6300 Fax: 246.324.5179

laare	aate	e/Soil Test Report		Report No: I	MAT:15-05001	4-00-S(
Client:	The C	Corradino Group		This labora	lory is accredited by Am	arican Assoc
				(AASHTO)	The lest(s) reported ha	ve been per
Project:	Detro	olt River Inter. Crossing		In accordan	The with the terms of the	accreditation
	Geot	echnical Engineering		ZZA STREET	Sound for je	and the
lob No:	15-05	50014-00		Date of Is Approved	sue: 6/19/2008 Signalory: Zeerak P	aydawy
Sample	Detail	S	Other Test Result	ts		
Soring No:		TB-101	Description	Method	Result	Limit
ield Samp	le No:	LS-21	Unconfined Compressive Strengt	h (1b/fl2) ASTM D 216	14053	10497194
ample De	pth:	90	Shear Strength (lb/ft ²)		7027	
ate Samp	led:		Ave. Rate Strain to Failu	re(%)	0.9	
ampled B	y:	Nathan Emery	Strain at Failure(%)		8.8	
WO No:		000307	Average Height (in.)		2.827	
ample Lo	cation:	Detroit River International Crossing	Average Diameter (in.)		1.362	
			Height-Diameter Ratio		2.1	
			Init. Dry Dens.			
			Init. Water Content (%)			
			Liquid Limit			
				Method: A	STM D 422	
				Drying by.	VEII	
%Pas	ssing					
100 I · ·	• • • • • •					
so				Sieve Size	% Passing	Limits
+				1in (25.0mm)	100	
80			************	¾in (19.0mm)	100	
-				3/8in (9.5mm)	100	
w1				No.4 (4.75mm)	100	
60				No.10 (2.0mm)	97	
ł				No.20 (850µm)	92	
50 + • •		*******************************	*****	No.40 (425µm)	89	
	2 (g. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1.			No.60 (250µm)	86	
40 1 ***		************************************	Num Vinner	No.100 (150µm)	81	
30				No.200 (75µm)	75	
~				0.040 mm	70	
20				0.029 mm	67	
1				0.018 mm	63	
10 +		*************************************	*********	0.013 mm	58	



Form No: 18909.V1.00

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NTH Consultants, Ltd. Infrastructure Engineering and Environmental Services	NTH Consultants, Ltd. Soulheast Michigan Laboratory Telephone: 248. 553.8300 Fax: 248.324.5179	
Andregate/Soil Test Report	Report No: MAT:15-050014-00-S0	017
Client: The Corrading Group	This laboratory is accredited by Amarican Assoc	dation
chent. The contactino croup	(AASHTO) The lest(s) reported have been per	formed
Project: Detroit River Inter. Crossing	Bay Bay Bay	
Geotechnical Engineering	LASHIN + 10	
Job No: 15-050014-00	Date of Issue: 5/21/2008 Approved Signatory: Zeerak Paydawy	
Sample Details	Other Test Results	
Boring No: TB-102	Description Method Result Limit	ts
Field Sample No: LS-7 Sample Depth: 34.5 Date Sampled: Sampled By:	Sand Gravel Description ASTM D 422 Shape Hardness Dispersion Device Dispersion Period	
Sample Location: Detroit River International Crossing	Moisture Content (%) ASTM D 2216 19.4	
	Wet Density (lb/ft³)132.2Dry Density (lb/ft³)110.7	-
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7	
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7 Method: ASTM D 422 Drying by: Oven	
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7 Method: ASTM D 422 Drying by: Oven	-
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7 Method: ASTM D 422 Drying by: Oven	
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7 Method: ASTM D 422 Drying by: Oven Sieve Size % Passing Limits 1in (25.0mm) 100	5
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7 Method: ASTM D 422 Drying by: Oven Sieve Size % Passing Limits 1in (25.0mm) 100 % in (19.0mm) 100 3/8in (9.5mm) 100	
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7 Method: ASTM D 422 Drying by: Oven Sieve Size % Passing Limits 1in (25.0mm) 100 ¾in (19.0mm) 100 3/8in (9.5mm) 100 No.4 (4.75mm) 100	
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7 Method: ASTM D 422 Drying by: Oven Sieve Size % Passing Limits 1in (25.0mm) 100 % In (19.0mm) 100 % No.4 (4.75mm) 100 No.4 (4.75mm) 100 No.20 (850µm) 100	5
Particle Size Distribution	Wet Density (lb/ft³) 132.2 110.7 Method: ASTM D 422 Drying by: Oven Sieve Size % Passing Limits 1in (25.0mm) 100 ¾in (19.0mm) 100 ¾in (9.5mm) 100 No.4 (4.75mm) 100 No.10 (2.0mm) 100 No.20 (850µm) 100 No.40 (425µm) 95	
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7 Method: ASTM D 422 Drying by: Oven Sieve Size % Passing 1in (25.0mm) 100 3/8in (19.0mm) 100 3/8in (9.5mm) 100 No.4 (4.75mm) 100 No.10 (2.0mm) 100 No.20 (850µm) 100 No.40 (425µm) 95 No.60 (250µm) 35	5
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7 Method: ASTM D 422 Drying by: Oven Sieve Size % Passing 1in (25.0mm) 100 ¾in (19.0mm) 100 ¾in (19.0mm) 100 3/8in (9.5mm) 100 No.4 (4.75mm) 100 No.10 (2.0mm) 100 No.40 (425µm) 95 No.60 (250µm) 35 No.100 (150µm) 9 No.200 (75µm) 5	
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7 Method: ASTM D 422 Drying by: Oven Sieve Size % Passing 1in (25.0mm) 100 ¾in (19.0mm) 100 ¾in (19.0mm) 100 3/8in (9.5mm) 100 No.4 (4.75mm) 100 No.10 (2.0mm) 100 No.20 (850µm) 100 No.40 (425µm) 95 No.60 (250µm) 35 No.100 (150µm) 9 No.200 (75µm) 5 0.052 mm 4	5
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7 Method: ASTM D 422 Drying by: Oven Sieve Size % Passing Limits 1in (25.0mm) 100 3/8in (19.0mm) 100 3/8in (9.5mm) 100 No.4 (4.75mm) 100 No.10 (2.0mm) 100 No.20 (850µm) 100 No.40 (425µm) 95 No.60 (250µm) 35 No.100 (150µm) 9 No.200 (75µm) 5 0.052 mm 4 0.037 mm 3	
Particle Size Distribution	Wet Density (lb/ft³) 132.2 Dry Density (lb/ft³) 110.7 Method: ASTM D 422 Drying by: Oven Sieve Size % Passing Limits 1in (25.0mm) 100 ¾in (19.0mm) 100 ¾in (19.0mm) 100 3/8in (9.5mm) 100 No.4 (4.75mm) 100 No.10 (2.0mm) 100 No.20 (850µm) 100 No.40 (425µm) 95 No.60 (250µm) 35 No.100 (150µm) 9 No.200 (75µm) 5 0.052 mm 4 0.037 mm 3 0.023 mm 3	

0		Bin +	0.4	10+	20+	4 3 9		E	0.010 mm 0.007 mm	3
	P.,	ſĊ	Z	No	No	z z ż	0.007 0.007 0.007 0.007 0.007 0.007 0.007 0.007	0.001	0.005 mm 0.003 mm 0.001 mm	2 1 1
Comments N/A				-						
Form No: 18909.V1.00				15 <u>8 de -</u>	11		(c) 2005-2007 QESTLab by Specin	aQEST.com		Figure No. 2

3

NH	NTH Consultants, Ltd.
	Infrastructure Engineering and Environmental Services

Telephone: 248. 553.6300 Fax: 248.324.5179

\sim	gale/son rest kepon				ssue No: 1
Client:	The Corradino Group		This laborate of State High (AASHTO),	way and Transportation The lesi(s) reported hav	ican Association Officials e been performed
Project:	Detroit River Inter. Crossing Geotechnical Engineering		TASHIC FIL	Jorma Duyd	coreditation. 1 NI.4, · /
Job No:	15-050014-00		Date of iss Approved	ue: 5/21/2008 Signatory: Zeerak Pa	ydawy
Sample	Details	Other Test Resul	ts		
Boring No:	TB-102	Description	Method	Result	Limits
Field Samp Sample De Date Samp Sampled B	ole No: LS-10 pth: 49.5 led: y: 000289	Sand Gravel Description Shape Hardness Dispersion Device Dispersion Period	ASTM D 422	1	
ample Lo	cation: Detroit River International Crossing	Moisture Content (%) Wet Density (lb/ft ³) Dry Density (lb/ft ³)	ASTM D 2216	5 19.9	
Particle	Size Distribution		Mathad- AC	TMD 422	
Particle	Size Distribution		Method: AS Drying by: Ov	TM D 422 /en	
Particle	Size Distribution		Method: AS Drying by: Ov	TM D 422 ven	
Particle %Particle	Size Distribution		Method: AS Drying by: Ov	TM D 422 ven	
Particle %Particle %Particle	Size Distribution		Method: AS Drying by: Ov Sieve Size 1in (25.0mm)	TM D 422 ven % Passing 100	Limits
Particle %Particle 100 T 50 50	Size Distribution		Method: AS Drying by: Ov Sleve Size 1in (25.0mm) 3/8in (9.5mm)	TM D 422 ven % Passing 100 100 100	Limits
Particle %Pa 100 50 80 70	Size Distribution		Method: AS Drying by: Ov Sleve Size 1in (25.0mm) ½in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm)	TM D 422 ven % Passing 100 100 100 100 100	Limits
Particle %Particle 100 50 80 80 80 80 80	Size Distribution		Method: AS Drying by: Ov Sieve Size 1in (25.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.40 (425µm)	TM D 422 ven % Passing 100 100 100 100 100 100 100 100	Limits
Particle %Per 100 - • • 90 - • •	Size Distribution		Method: AS Drying by: Ov Sieve Size 1in (25.0mm) ½in (19.0mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.40 (425µm) No.60 (250µm) No.100 (150µm) No.100 (150µm)	5TM D 422 Yen % Passing 100 100 100 100 100 100 100 100 100 10	Limits
Particle %Pase 100 [• • 50 • • 60 • • 60 • • 60 • • 60 • • 60 • • 60 • • 60 • • 60 • • 60 • • 60 • • 60 • • 60 • • 60 • •	Size Distribution		Method: AS Drying by: Ov Sieve Size 1in (25.0mm) ½in (19.0mm) ¾in (19.0mm) ¾in (19.0mm) ¾in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.40 (425µm) No.60 (250µm) No.100 (150µm) No.200 (75µm) No.200 (75µm) 0.051 mm	TM D 422 ren % Passing 100 100 100 100 100 100 100 100 100 10	Limits
Particle %Pa 100 [• • 50 • • 50 • • 50 • • 50 • • 50 • • 50 • • 50 • • 50 • • 50 • • 50 • • 50 • • 50 • •	Size Distribution		Method: AS Drying by: Ov Sieve Size 1in (25.0mm) ½in (19.0mm) ¾in (9.5mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.10 (2.0mm) No.10 (2.0mm) No.10 (2.0mm) No.10 (2.0mm) No.10 (2.0mm) No.10 (2.0mm) No.10 (2.0mm) No.10 (2.0mm) No.20 (850µm) No.10 (2.0mm) No.20 (850µm) No.20 (850µm) No.60 (250µm) No.100 (150µm) No.100 (150µm) No.200 (75µm) 0.051 mm 0.036 mm 0.023 mm	TM D 422 Yen % Passing 100 100 100 100 100 100 100 100 100 10	Limits



Telephone: 248. 553.6300 Fax: 248.324.5179

Agare	gate	Soil Test Report		Report No: MA	T:15-0 50014	-00-502 ssue No:
Client:	The (Corradino Group		This laboratory of State Highwa	is accredited by Amer ay and Transportation	ican Associa Officials
	D			(AASHTO). The in accordance v	e test(s) reported have with the terms of the a	been perfor creditation.
roject:	Detro	bit River Inter. Crossing		11/2	Borna Dayd	icre,
	Geol	echnical Engineering		Sala alla		
lob No:	15-0	50014-00		Date of Issue Approved Sig	: 5/21/2008 malory: Zeerak Pag	dewy
Sample	Detail	S	Other Test Result	S		
Boring No:		TB-103	Description	Method	Result	Limits
ield Samp	le No:	LS-6	Sand Gravel Description	ASTM D 422		
ample De	pth:	29.5	Shape			
late Samp	led:		Hardness			
ampled B	y:		Dispersion Device			
WO No:		000289	Dispersion Period		1	
ample Lo	cation:	Detroit River International Crossing	Moisture Content (%)	ASTM D 2216	14.2	
Lines			Wet Density (lb/ft ^a)		122.1	
			Dry Density (lb/ft ³)		106.9	
				Method: AST Drying by: Over	M D 422 n	
%Pee	ssing					
100 I · ·	• • • • •		* * * * * * * * * * * * * * * * * * * *			
80	4			Sieve Size %	6 Passing	Limits
t				1in (25.0mm)	100	
80 1				%in (19.0mm)	100	
				3/8in (9.5mm)	93	
				No.4 (4.75mm)	81	
60				No.10 (2.0mm)	64	
				No.20 (850µm)	A	
50 - • •				No.40 (425µm)	48	
		*******************************			48 18	
40			* * * * * * * * * * * * * * * * * * * *	No.60 (250µm)	48 18 3	
ent.			5 * * * * * * * * * * * * * * * * * * *	No.60 (250µm) No.100 (150µm)	48 18 3 2	
30[********************************	*************	No.60 (250µm) No.100 (150µm) No.200 (75µm)	48 18 3 2 2	
		***************************************	* * * * * * * * * * * * * * * * * * *	No.60 (250µm) No.100 (150µm) No.200 (75µm) 0.053 mm	48 18 3 2 2 1	
20	• • • • • • • • • •	************************************	**************	No.60 (250µm) No.100 (150µm) No.200 (75µm) 0.053 mm 0.038 mm	48 18 3 2 2 1 1	
20+	• • • • • • • • •	***********	*************	No.60 (250µm) No.100 (150µm) No.200 (75µm) 0.053 mm 0.038 mm 0.024 mm	48 18 3 2 2 1 1 1	
20 - · · 10 - · ·	• • • • • • • • •	*****	**************	No.60 (250µm) No.100 (150µm) No.200 (75µm) 0.053 mm 0.038 mm 0.024 mm 0.017 mm	48 18 3 2 2 1 1 1 1	



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Aggre	gate	e/Soil Test Report		Report No: M	AT:15-05001	4-00-S039 Issue No: 1
Client:	The (Corradino Group		This laborate of State High	is accredited by American way and Transportation	arican Association n Officials
Project:	Detro	oit River Inter, Crossing		(AASHTO) In accordance	a with the lerms of the	ve been performed accreditation.
	Geot	echnical Engineering			Bernis parte	They
Jah Mai	45.0	50014 00		ine of the	() · · · · · · · · · · · · · · · · · · ·	. 2
JOD NO;	15-0:	50014-00		Approved S	Signatory: Zeerak Pa	aydawy
Sample I	Detail	S	Other Test Result	S		
Boring No:		TB-104	Description	Method	Result	Limits
Field Sampl	le No:	LS-5	Sand Gravel Description	ASTM D 422		
Sample Dep	oth:	25	Shape			
Date Sample	ed:		Hardness			
Sampled By	/:	Nathan Emery	Dispersion Device			
LWO No:		000307	Dispersion Period		1	
Sample Loc	ation:	Detroit River International Crossing	Moisture Content (%)	ASTM D 2215	25.4	
			Wet Density (lb/ft3)		125.9	
			Dry Density (Ib/ft ³)		100.4	
r aiticic t	DIEC L	istribution		Method: AS Drying by: Ove	TM D 422 en	
%Pass	sing					
1007			********			
80		•••••••••••••••••••••••••••••••••••••••	******	Sieve Size	% Passing	Limits
ant				1in (25.0mm)	100	
w]	1 76 27 17 18 18 18 18 18 18 18 18 18 18 18 18 18			%IN (19.0mm)	100	
70+			********	3/8/n (9.5mm)	100	
				NO.4 (4.75mm)	100	
60				No.10 (2.0mm)	99	1
				No.20 (d_{25} μ m)	97	
ະພາ			*************	No 60 (250)	12	
40	******		********	No 100 (250µm)	32	
t				No 200 (75um)	14	
30	* * * 3 * * *			0.051 mm	0	
				0.036 mm	9	
201			************	0.023 mm	9	
10+				0.016 mm	7	
					I	1



Form No: 18909.V1.00

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gate/Soil Test Report	Кер	on No: MAI:15-050014-00-5041 Issue No: 1
The Corradino Group Detroit River Inter. Crossing	1000	This laboratory is accredited by American Association of State Highway and Transportation Officials (AASHTO). The lest(s) reported have been performed in accordance with the terms of the accreditation.
Geotechnical Engineering	TICHIO AU	Sound Prayear
15-050014-00		Date of Issue: 6/19/2008 Approved Signatory: Zeerak Paydewy
	gate/Soil Test Report The Corradino Group Detroit River Inter. Crossing Geotechnical Engineering 15-050014-00	gate/Soil Test Report The Corradino Group Detroit River Inter. Crossing Geotechnical Engineering 15-050014-00

Sample Details

Other Test Results

Boring No:	TB-104	Description	Method	Result	Limits
Field Sample No:	LS-7	Unconfined Compressive Strength (Ib/R	7) ASTM D 21	⁵⁶ 412	
Sample Depth:	35	Shear Strength (lb/ft ²)	M.	206	
Date Sampled:		Ave. Rate Strain to Failure(%)	0.9	
Sampled By:	Nathan Emery	Strain at Failure(%)		14.6	
LWO No:	000307	Average Height (in.)		2.748	
Sample Location:	Detroit River International Crossing	Average Diameter (in.)		1.353	
		Height-Diameter Ratio		2.0	
		Init. Dry Dens.			
		Init. Water Content (%)			
		Liquia Limit Directio Limit			
		Plastic Limit			
Particle Size [Distribution				
allicie Size L	Jacobucion	Mo	thod: A	STM D 422	
		Dr	ving by: 0		
			ying by. O	YGN	
%Passing					
100					
90		Sie	eve Size	% Passing	Limits
.		110	(25.0mm)	100	
au		7411	n (19.0mm)	100	
70		3/8	sin (9.5mm)	100	
		INO	4 (4.7 5mm)	100	
60 - • • • • • • • • •	*******************************	NO NO	20 (850um)	100	
		No	A0 (425um)	100	
50 I · · · · · · · ·	***************************************	No	60 (260um)	100	
A0		NO NO	100 (250µm)	100	
~ +		No	200 (75um)	00	
30			37 mm	03	
t		0,0	27 mm	00 00	
20	**********************************		17 mm	84	
t.		0.0	12 mm	77	
INT					



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ggre	gate	e/Soi	I Te	st Re	port			ē.	Rep	ort No:	: MAT:	15-0 5001	4-00-S029
Client: Project:	The C	Corradino	Group	ssina						This labo of State (AASHT) in accord	oratory is ac Highway an O). 'The les dance with 0	credited by Am d Transportation (s) reported has has terms of the	narican Associatio on Officials ave been parlorm accreditation.
,	Geot	echnical E	Ingineer	ing							300	mk pay	cherry
lob No:	15-05	50014-00								Date of Approve	Issue: 6 ed Signato	/19/2008 ory: Zeerak P	aydawy
Sample	Detail	S				Othe	r Tes	t Resul	ts		de de de con		
Soring No: Tield Samp Sample Dep Sample Sample Sample Loc	ole No; pth: led: y: cation:	TB-105 LS-21 90 Nathan I 000307 Detroit F	Emery River Int	ernationa) C	Crossing	Descri Unconfin Shear Ave. R Strain Averag Averag Height Init. Dr Init. W Liquid Plastic	ed Compre Strengtl ate Stra at Failur de Heigh de Diam Diamet y Dens. ater Cor Limit Limit	issive Strengt n (Ib/ft²) in to Failu re(%) nt (in.) eter (in.) er Ratio	h (lb/ft²)	Method ASTM D 2	166	Result N/O N/O N/O N/O N/O N/O N/O N/O	
100 T · · · · · · · · · · · · · · · · · ·									Sieve 3 1in (25 3/lin (19 3/8in (9 No.4 (4 No.40 (No.20 (No.20 (No.200 0.042 n 0.030 n 0.019 n 0.019 n 0.011 n 0.011 n	Size .0mm) .0mm) .0mm) .0mm) .5mm) .5mm) .75mm) (.75mm) (.250µm) (% P;	assing 100 100 95 92 87 83 79 74 67 61 58 53 49 47 43	Limits
	*** ***	3/Bir No.4	No.10 No.20	Sieve No.100 De.00	0.042 m m	E E E E E E E E E E E E E E E E E E E	0.004 mr 0.003 mr	0.001 mm	0.006 n 0.004 n 0.003 n 0.001 n	nm nm nm		38 33 30 20	

Form No: 18909.V1.00

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Telephone: 248. 553.6300 Fax: 248.324.5179

	Aggre	gate/So	il Test	Report
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The Corradino Group Client:

Detroit River Inter. Crossing Project: Geotechnical Engineering

15-050014-00 Job No:

This laboratory is accredited by American Association of State Highway and Transportation Officials (AASHTO). The test(s) reported have been performed in accordance with the terms of the accreditation.

Issue No: 1



Sound Day child

Date of Issue: 6/19/2008 Approved Signalory: Zeerak Paydawy

Report No: MAT:15-050014-00-S030

Sample Details

Boring No:	TB-105	Description	Method	Result	Limit
Field Sample No:	LS-22	Sand Gravel Description	ASTM D 422		
Sample Depth:	93.5	Shape			
Date Sampled:		Hardness			
Sampled By:	Nathan Emery	Dispersion Device			
LWO No:	000307	Dispersion Period			
Sample Location:	Detroit River International Crossing	Moisture Content (%) Wet Density (lb/ft³) Dry Density (lb/ft³)	ASTM D 221	6 13.2	
Particle Size D	Distribution		Method: AS Drying by: Or	STM D 422 Ven	<u></u>
%Passing					
100 I	······	* * * * * * * * * * * * * * * * * * *	Sieve Size	% Bossine	1 familie
90 T · · · · · · · · ·		* * * * * * * * * * * * * * * * * * *	1in (25 0mm)	100	
80+			%in (19.0mm)	100	
			3/8in (9.5mm)	96	
70			No.4 (4.75mm)	88	
			No.10 (2.0mm)	78	
ω[No.20 (850µm)	67	
50		**********	No.40 (425µm)	57	
			No.60 (250µm)	46	
40		**************	No.100 (150µm)	31	
			No.200 (75µm)	19	
30 + • • • • • • • • • • •			0.047 mm	15	
	X		0.034 mm	14	
		ರಾಜಕಾಲ್ಯಾರ್ ಕಾಂ ಕಾಲಕಾ (ಕ್ಲಾಕ ಜ್ಯಾಕಾಲಕಾಲ್) ಶ್ವಾಕಾಲಕ್ (ಕ್ಲಾಕಾಲಕಾ) ಇಲ್ಲ	0.022 mm	13	
(n.)			0.015 mm	11	

Other Test Results

Field Sample No: LS-22 Sand Gravel Description ASTM D 422 Sample Depth: 93.5 Shape Hardness Date Sampled By: Nathan Emery Dispersion Device Dispersion Period Sample Location: Detroit River International Crossing Moisture Content (%) ASTM D 2216 Wethod: Dispersion Period Dispersion Period Stim D 2216 Particle Size Distribution Wethod:: ASTM D 4 %Passing Oven Sieve Size % Passing 10 Sieve Size % Passing No.4 (4.75mm) 00 No.4 (4.75mm) No.40 (425µm) No.60 (250µm) 00 No.200 (75µm) No.40 (425µm) No.200 (75µm) 00 No.200 (75µm) No.200 (75µm) No.200 (75µm)	Result	Limits
Sample Depth: 93.5 Shape Sampled By: Nathan Emery D00307 Sample Location: Detroit River International Crossing Particle Size Distribution %Passing 10 50 50 50 50 50 50 50 50 50 5		
Date Sampled: Hardness Sampled By: Nathan Emery Dispersion Device WO No: 000307 Dispersion Period Sample Location: Detroit River International Crossing Molsture Content (%) ASTM D 2216 Wet Density (lb/ft ²) Ory Density (lb/ft ²) Ory Density (lb/ft ²) Particle Size Distribution Method: ASTM D 4 %Pessing 1in (25.0mm) 3/8in (9.5mm) %o 3/8in (9.5mm) No.4 (4.75mm) No. 10 (2.0mm) No.40 (425µm) No.40 (425µm) No. 40 (425µm) No.40 (425µm) No.40 (425µm) No. 200 (75µm) No.40 (75µm) No.40 (0.150µm) Mol Dury Density No.40 (150µm) No.40 (150µm)		
Sampled By: Nathan Emery Dispersion Period WO No: 000307 Dispersion Period Sample Location: Detroit River International Crossing Moisture Content (%) ASTM D 2216 Wet Density (lb/ft ³) Dry Density (lb/ft ³) Dry Density (lb/ft ³) Particle Size Distribution Method: ASTM D 4 %Pessing Sieve Size % Pas 10 Sieve Size % Pas 11 (25.0mm) 3/8in (9.5mm) 80 No.40 (42.5µm) No.40 (42.5µm) 90 No.40 (42.5µm) No.610 (250µm) 90 No.100 (150µm) No.100 (150µm) 91 No.100 (150µm) No.200 (75µm) 92 No.200 (75µm) No.47 mm		
WO No: 000307 Dispersion Period Sample Location: Detroit River International Crossing Molsture Content (%) Wet Density (lb/ft*) ASTM D 2218 Particle Size Distribution Method: ASTM D 4 %Passing Sieve Size % Passing 100 Sieve Size % Passing 101 Sieve Size % Passing 102 Sieve Size % Passing 103 No.4 (4.75mm) No.40 (425µm) 104 No.40 (425µm) No.40 (425µm) 105 No.40 (150µm) No.40 (150µm) 101 Sieve Size No.40 (475µm) 102 Sieve Size No.40 (425µm) 103 Sieve Size No.40 (425µm) 104 No.40 (425µm) No.40 (425µm) 105 Sieve Size No.40 (425µm) 106 Sieve Size No.40 (425µm) 107 Sieve Size Sieve Size 108 Sieve Size Sieve Size 109 Sieve Size Sieve Size 100 Sieve Size Sieve Size 100 Sieve Size Sieve Siz		
Sample Location: Detroit River International Crossing Moisture Content (%) Wet Density (lb/ft*) ASIM D 2216 Particle Size Distribution Method: ASIM D 4 Dry Density (lb/ft*) Drying by: Oven %Pessing Sieve Size % Passing 100 Sieve Size % Passing 20 Vin (19.0mm) 3/8in (9.5mm) 80 No.4 (4.75mm) No.40 (425µm) 90 No.40 (425µm) No.40 (425µm) 90 No.40 (425µm) No.40 (425µm) 90 No.40 (475µm) No.40 (425µm) 90 No.40 (425µm) No.40 (425µm) 90	1	
Wet Density (lb/ft ²) Dry Density (lb/ft ²) Particle Size Distribution Method: ASTM D 4 Drying by: Oven %Passing Sieve Size % Pass 10 Sieve Size % Pass 11 (25.0mm) %in (19.0mm) 10 Sieve Size % Pass 10 No.4 (4.75mm) No.4 (4.75mm) 10 No.10 (2.0mm) No.40 (425µm) 10 No.10 (2.0mm) No.60 (250µm) 10 No.10 (150µm) No.10 (150µm) 11 No.10 (150µm) No.200 (75µm) 11 O.047 mm O.047 mm	13.2	
Dry Density (ID/R*) Particle Size Distribution Method: ASTM D 4 Drying by: Oven %Passing Sieve Size % Passing 100 Sieve Size % Passing 200 %Passing 201 Sieve Size % Passing 201 %Passing 202 %Passing 203 %Passing 204 %Passing 203 %Passing 204 %Passing		
Sieve Size % Passing 100 Sieve Size % Passing 100 Sieve Size % Passing 100 Sieve Size % Passing 100 Sieve Size % Passing 100 Sieve Size % Passing 101 Sieve Size % Passing 101 Sieve Size % Passing 101 No.4 (4.75mm) No.40 (2.50mm) 101 No.40 (250µm) No.40 (250µm) 101 No.40 (150µm) No.200 (75µm) 101 Output No.200 (75µm) 102 Signal Signal		
%Passing 100 Sieve Size % Pas 80 1in (25.0mm) 3/8in (19.0mm) 80 3/8in (9.5mm) No.4 (4.75mm) 80 No.10 (2.0mm) No.20 (850µm) 80 No.40 (425µm) No.60 (250µm) 80 No.100 (150µm) No.200 (75µm) 80 0.047 mm 0.047 mm	422	
Sieve Size % Pas 1in (25.0mm) 3/8in (9.5mm) % No.4 (4.75mm) % No.4 (4.75mm) % No.20 (850µm) % No.40 (425µm) No.40 (425µm) No.60 (250µm) % No.200 (75µm) % No.47 mm % Output		
Sieve Size % Pas 1in (25.0mm) 3/8in (19.0mm) 3/8in (9.5mm) 3/8in (9.5mm) No.4 (4.75mm) No.10 (2.0mm) No.20 (850µm) No.40 (425µm) No.40 (425µm) No.60 (250µm) No.100 (150µm) No.200 (75µm) 0.047 mm 0.047 mm		
80 1in (25.0mm) 70 3/8in (9.5mm) 70 No.4 (4.75mm) 80 No.10 (2.0mm) 80 No.20 (850µm) 80 No.40 (425µm) 80 No.60 (250µm) 80 No.200 (75µm) 80 0.047 mm 80 0.047 mm	issing L	_imits
80 9/in (19.0mm) 70 3/8in (9.5mm) 70 No.4 (4.75mm) 80 No.10 (2.0mm) 80 No.20 (850µm) 80 No.40 (425µm) No.60 (250µm) No.100 (150µm) 80 No.200 (75µm) 80 0.047 mm	100	
70 3/8in (9.5mm) 60 No.4 (4.75mm) 60 No.10 (2.0mm) 50 No.20 (850µm) 50 No.40 (425µm) 50 No.60 (250µm) 50 No.100 (150µm) 50 No.200 (75µm) 50 0.047 mm	100	
No.4 (4.75mm) No.10 (2.0mm) No.20 (850μm) No.40 (425μm) No.60 (250μm) No.100 (150μm) No.200 (75μm) 0.047 mm 0.021 mm	96	
 No.10 (2.0mm) No.20 (850μm) No.40 (425μm) No.60 (250μm) No.100 (150μm) No.200 (75μm) 0.047 mm 0.024 mm 	88	
50 50 50 50 50 50 50 50 50 50	78	
50 40 30 30 No.40 (425μm) No.60 (250μm) No.100 (150μm) No.200 (75μm) 0.047 mm 0.024 mm	67	
N0.60 (250μm) No.100 (150μm) No.200 (75μm) 0.047 mm	57	
40 No.100 (150μm) 30 30 30 0.047 mm 0.047 mm	46	
30 NO.200 (75µm) 0.047 mm	37	
0.047 mm	19	
	15	
20	14	
0.022 mm	13	





Telephone: 248. 553.6300 Fax: 248.324.5179

Aggreg	jate	Soil Test Rep	ort	Re	port No: MAT:	:15-050014	-00-S045 ssue No: 1
Client:	The C	Corradino Group			This laboratory is a of State Highway a (AASHTO). The le	nd Transportation	ican Association Officials a been parformed
Project:	Detro	oit River Inter. Crossing			in accordance with	the terms of the ac	ccreditation.
	Geot	echnical Engineering		ENDATIC PILY	all all all all all all all all all all	- /·	
Job No:	15-05	50014-00			Date of Issue: Approved Signal	6/19/2008 lory: Zeerak Pay	/dawy
Sample D	etail	S	Other Test Re	sults			
Boring No:		TB-107	Description		Method	Result	Limits
Field Sample	No:	LS-2	Unconfined Compressive	Strength (IbAP)	ASTM D 2165	598	
Sample Dept	h:	20	Shear Strength (lb/	ft²)		299	
Date Sampler	d:		Ave. Rate Strain to	Failure(%)		0.9	
Sampled By:		Nathan Emery	Strain at Failure(%))		14.9	

LWO No: Sample Location:	000308 Detroit River International Crossing	Average Height (in.) Average Diameter (in.) Height-Diameter Ratio Init. Dry Dens. Init. Water Content (%) Liquid Limit Plastic Limit		2.688 1.381 1.9	
Particle Size D	Distribution				
			Method: Drying by:	ASTM D 422 Oven	
%Pessing					
100 T · · · · · · ·					
807			Sieve Size	% Passing	Limits
	\ \		1in (25.0mm)	100	
801	***************************************		¾in (19.0mm)	100	
70+	* * * * * * * * * * * * * * * * * * * *	A	3/8/n (9.5mm)	100	
			No.4 (4.75mm)	100	
60	***************************************		No 20 (850um)	100	
m	*****************************		No.40 (425µm)	100	
~			No.60 (250µm)	100	
40			No.100 (150µm	100	
<u></u>			No.200 (75µm)	99	
aut			0.037 mm	89	
20+			0.027 mm	82	
			0.018 mm	71	
10			0.013 mm	64	



Form No; 18909.V1.00

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Infrastructure Engineering and Environmental Services

NTH Consultants, Ltd. Southeast Michigan Laboratory

Telephone: 248. 553.6300 Fax: 248.324.5179

	aate	/Soil Test Report		Repo	rt No: M	AT:15-050014	-00-504: ssue No: *
Client:	The Co	orradino Group			This laborator of State Highy (AASHTO). T	y is accredited by Amer vay and Transportation he test(s) reported have	ican Associatio Officials a been perform
Project:	Detroit Geotec	River Inter. Crossing chnical Engineering			IN SCOUGUCE	Jaconski paych	
Job No:	15-050	0014-00			Date of Issu Approved S	e: 6/19/2008 Ignatory: Zeerak Pay	ydawy
Sample	Details		Other Test Re	sults	1020 F.M. D.		
Boring No: Fleid Samp Sample Dep Date Sampi Sampled By	le No: pth: led: y:	TB-107 LS-16 90 Nathan Emery	Description Sand Gravel Descrip Shape Hardness Dispersion Device	ption A	lethod STM D 422	Result	Limits
.WO No: Sample Loc	ation:	000308 Detroit River International Crossing	Dispersion Period Molsture Content (% Wet Density (lb/ft ³) Dry Density (lb/ft ³)	6) A	STM D 2216	1 19.9 128.7 107.3	
Particle	Size D	istribution		Mathad		TM D 422	
Particle	Size Di	istribution		Method: Drying I	AS Dy: Ove	TM D 422 en	
Particle %Pas	Size Di	istribution		Method: Drying I	AS by: Ove	TM D 422 en	
Particle %Pes	Size Di	istribution		Method: Drying I	AS Dy: Ove	TM D 422 en	
Particle %Pas 100 [*** 80 [***	Size Di	istribution		Method: Drying I Sieve S 1in (25.0 3⁄4in (19.	ize Omm) Omm)	TM D 422 en % Passing 100 100	Limits
Particle %Pas 100 [** 80 ** 80 ** 80 ** 80 **	Size D	istribution		Method: Drying I Sieve S 1in (25.0 ¾in (19. 3/8in (9. No.4 (4.	AS by: Ove ize ()mm) 0mm) 5mm) 5mm) 75mm)	TM D 422 en % Passing 100 100 100 100	Limits
Particle %Pas 100 ** 80 ** 80 ** 80 ** 80 **	Size D	istribution	* * * * * * * * * * * * * * * * * * * *	Method: Drying I Sieve S 1in (25.0 ¾in (19. 3/8in (9. No.4 (4. No.10 (2 No.20 (8	AS by: Ove ize (mm) 0mm) 0mm) 5mm) 5mm) 5mm) 50mm) 50µm)	TM D 422 en % Passing 100 100 100 100 100 100 100	Limits
Particle %Pass 100 1 ** 80 ** 80 ** 80 ** 80 ** 80 ** 80 ** 80 ** 80 ** 80 **	Size D	istribution		Method: Drying I Sieve S 1in (25.0 ¾in (19. 3/8in (9. No.4 (4. No.10 (2 No.20 (8 No.40 (4 No.60 (2	AS AS Over ize Omm) Omm) Omm) 5mm) 5mm) 5mm) 50µm) 50µm) 50µm) 50µm) 50µm)	TM D 422 en % Passing 100 100 100 100 100 100 100 100 100	Limits
Particle %Pas 100 ** 80 - **	Size D	istribution		Method: Drying I Sieve S 1in (25.0 ¾in (19. 3/8in (9. No.4 (4. No.10 (2 No.20 (8 No.40 (4 No.60 (2 No.100 (2)	AS AS Over ize Omm) Omm) Omm) Smm) Smm) Smm) 250µm) 250µm) 250µm) 250µm) 250µm) 250µm) 250µm) 250µm) 250µm)	TM D 422 en % Passing 100 100 100 100 100 100 100 100 100 10	Limits
Particle %Pes 100 '' 80 ''	Size D	istribution		Method: Drying I Sieve S 1in (25.0 ¾in (19. 3/8in (9. No.4 (4. No.10 (2 No.20 (8 No.40 (4 No.60 (2 No.20 (8 No.40 (4 No.60 (2 No.100 (2 No.200 (8) No.200	AS by: Ove ize)mm) 0mm) 0mm) 5mm) 5mm) 5mm) 250µm) 250µm) 250µm) 250µm) 250µm) 150µm] 150µm] 150µm] 150µm] 150µm] 150µm] 150µm] 150µm] 150µm] 150µm] 150µm] 150µm] 150µm] 150µm]	TM D 422 en % Passing 100 100 100 100 100 100 100 100 100 10	Limits



Telephone: 248. 553.6300 Fax: 248.324.5179

Aaaro	nate	Soil Test Report		Report No: MA	T:15-050014	-00-S005
Client:	The (Corradino Group		This laboratory is of State Highway	s accredited by Ame	8SUE NO: 1 Ican Association
				(AASHTO). The	lest(s) reported hav	e been performe
Project:	Detro	oit River Inter. Crossing	1	1.1.	z / Dayd	null,
	Geot	echnical Engineering		TETRE AD	Journ in 10	
Job No:	15-05	50014-00		Date of Issue: Approved Sign	5/15/2008 nalory: Zeerak Pa	ydawy
Sample	Detail	S	Other Test Result	S		
Boring No:		TB-108	Description	Method	Result	Limits
Field Samp	ple No:	LS-19	Sand Gravel Description	ASTM D 422		
Sample De	pth:	95	Shape			
Date Samp	led:		Hardness			
Sampled B	iy:	Nathan Emery	Dispersion Device			
LWO No:		000286	Dispersion Period		1	
Sample Lo	cation:	Detroit River International Crossing	Moisture Content (%)	ASTM D 2216	7.9	
			Wet Density (Ib/ft ³)		153.3	
			Dry Density (lb/ft ^a)		142.1	
				Method: ASTN Drying by: Oven	1 D 422	
%Pa	ssing					
100 T • •	·····	***************************************				
90 - • •				Sieve Size %	Passing	Limits
t				1in (25.0mm)	100	
s o1				¾in (19.0mm)	93	
70				3/8in (9.5mm)	85	
~L				No.4 (4.75mm)	71	
60+				No.10 (2.0mm)	61	
				No.20 (850µm)	54	
50 + • •				No.40 (425µm)	49	
				No.60 (250µm)	43	
AUT.			* * * * * * * * * * * * * * * * *	No.100 (150µm)	35	
30				No.200 (75µm)	21	
-				0.042 mm	23	
20		***************************************		0.031 mm	21	
+				0.020 mm	18	
10 + • •				0.015 mm	15	



Earth Mechanics Project Name: n/a Location: n/a Client: NTH Consultar	Institute			CITES C	THE REAL PROPERTY OF			Colo Mini	rado So ng Engin	chool of Mines eering Department	
Date: 7/2/2008		Average	Average	Lenath to		Failure	Failure	Unia	xial		
Sample	Rock Type	Length	Diameter	Diameter	Density	Load	Stress	Compressiv Correcte	e Strength ed (2:1)	Notes	
Ð		(in)	(iii)	174 HO	(lbs/ft ³)	(lbs)	$\sigma_{c}(psi)$	(isd)	(MPa)	(Failure type)	
TB-101@138.85-139.50	Sedimentary	4.121	1.975	2.09	155	33,993	11,101	11,255	78	Non-Structural	
TB-102@106.2-106.8	Sedimentary	4.199	1.954	2.15	159	27,893	9,301	9,459	65	Non-Structural	
TB-103@106.25-106.89	Sedimentary	4.161	1.988	2.09	161	32,148	10,362	10,509	72	Non-Structural	
TB-103@110.6-111.34	Sedimentary	4.191	1.985	2.11	157	33,142	10,715	10,877	75	Non-Structural	
TB-104@103.55-104.0	Sedimentary	4.184	1.946	2.15	169	65,594	22,054	22,430	155	Structural	
TB-104@122.45-123.0	Sedimentary	4.167	1.938	2.15	162	44,576	15,111	15,369	106	Non-Structural	
TB-105@102.8-103.4	Sedimentary	4.124	1.960	2.10	161	39,490	13,095	13,288	92	Non-Structural	
TB-105@115.09-115.64	Sedimentary	4.070	1.964	2.07	153	33,466	11,047	11,191	77	Non-Structural	
TB-106@101.5-102.05	Sedimentary	4.145	1.968	2.11	162	40,170	13,212	13,408	92	Non-Structural	
TB-106@127.91-128.45	Sedimentary	4.101	1.973	2.08	154	27,027	8,844	8,963	62	Structural	
TB-107@95.25-95.7	Sedimentary	4.111	1.960	2.10	160	34,209	11,338	11,501	79	Non-Structural	
TB-107@120.15-120.7	Sedimentary	4.143	1.980	2.09	156	30,237	9,825	9,964	69	Structural	
TB-108@101.15-101.87	Sedimentary	4.150	2.003	2.07	162	48,655	15,441	15,642	108	Non-Structural	
TB-108@135.95-136.4	Sedimentary	4.202	1.987	2.12	162	38,459	12,409	12,598	87	Non-Structural	

 $UCS_{2lcorrection} = \frac{\sigma_c}{0.88 + 0.222(\frac{d}{l})}$

Earth Mechan	iics Institu	ite					Color	ado Scl	nool of	Mines
Project Name: n/a	_				ATHE		Mining	Enginee	ring Dep	artment
Location: n/a					70					
Client: NTH Cons	ultants			NOT OR W						
Date: 7/16/2008		Average	Average	Sample	Densiter	P-Wave	S-Wave	Dynami	c Elastic Co	nstants
Sample	Rock Type	Length	Diameter	Weight	DEIISIIY	Velocity	Velocity	Young's	Modulus	Poisson's
ID		(iii)	(in)	(g)	(Ib / ft 3)	(ft/sec)	(ft/sec)	(ksi)	(GPa)	Ratio
TB-101@105.83-106.4	Sedimentary	4.215	1.974	538.69	159	16,726	9,123	7,364	51	0.29
TB-101@138.85-139.5	Sedimentary	4.121	1.975	514.80	155	14,309	8,177	5,641	39	0.26

Earth Mechanics I	nstitute				C	olorado Sch	ool of Mines
Project Name: Chile				155	2	1 ining Engineer	ing Department
Location: n/a))
Client: NTH Consultant	S		CCORAC	λ.			
8/5/2008		Avgerage	Avgerage	Confining	Triaxial Com	pressive Strength	
Sample	Rock Type	Diameter	Length	Pressure	Pea	k Stress	Notes
D		(in)	(in)	(psi)	(isi)	(MPa)	(Failure type)
TB-101@105.83-106.4_T400	Sedimentary	1.975	4.215	400	13,922	96.0	Non-Structural
TB-102@101.6-107.25_T200	Sedimentary	1.949	4.103	200	13,626	93.9	Structural

School of Mines gineering Department		Notes (Failure type)		Non-Structural	Non-Structural	Non-Structural	Non-Structural	Non-Structural	Non-Structural	
Colorado Mining Eng	Trucilo Cturreth	inguanc ausual (r	(MPa)	4.6	3.4	6.0	6.1	5.2	6.7	
	Indianot (Durrilian		(psi)	669	494	877	890	751	975	
A COLOGICAL COLO	A montoon Discussion	Average Diameter	(in)	1.975	1.978	1.947	1.953	1.961	1.968	
CTOR AND COLOR		Average Leugu	(in)	1.39	1.42	1.47	1.39	1.32	1.45	
s Institute ants		Rock Type		Sedimentary	Sedimentary	Sedimentary	Sedimentary	Sedimentary	Sedimentary	
Earth Mechanic Project Name: n/a Location: n/a Client: NTH Consult:	Date: 7/1/2008	Commits ID	CTT AITITIA	TB-101@105.83-106.4	TB-101@138.85-139.5	TB-102@101.6-107.25	TB-102@106.2-106.8	TB-105@102.8-103.4	TB-106@101.5-102.05	



CrineArd Compressive strength rest results (v s.0) - Project: NTH Location: n/a Rock Name: n/a Characteristic: Core ID: Test 10:28 (06.2-106.8) Filemen: Te-1028 (06.2-106.8) Date Teste: 8/12/2008 Filemen: Te-1028 (06.2-106.8) Date Teste: 8/12/2008 Core Longth Diameter UD Filemen: Te-1028 (06.2-106.8) Core Longth Diameter UD Filemen: Te-1028 (06.2-106.8) Core Longth Diameter UD Filemen: Te-1028 (06.2-106.8) PROJECT: MTH- Mode Mark Non-Structural Prove: TE-1028 (06.2-106.8) Prove: Sware Dynamic Static E- Static Decent: Te-1028 (0	E	a the		Minin	Earth M g Engin	Aechanio eering [cs Institu Departme	ite ent, CSN	1		+	
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Detroit River International Crossing -Cable-Stayed Bridge Option



Figure Modified from Detroit River International Crossing, Bridge Conceptual Engineering Report, November 2007, by Parsons



Figure No. 37



Detroit River International Crossing -Suspension Bridge Option





Figure No. 39



Figure No. 40







Figure No. 42







Proceedings

1997 Hapid Excavation and Tunneling Conference Las Vogas, Nevada, Juna 22-25, 1997

> Editors J.E. Carlson T.H. Budd

Sponsored by Society for Mining, Metallurgy, and Exploration, Inc. American Society of Civil Engineers

Society for Mining, Metallurgy, and Exploration. Inc. Littleton, Colorado • 1997

Chapter 29

SINKING CAISSONS AS AN EFFECTIVE MEANS OF CONSTRUCTING SHAFTS

F.J. Klingler,' K.M. Swaffar,' and J.C. Neyer,' and R. Hausmann² 'NTH Consultants, I.td., 'Walbridge Aldinger Co.

ABSTRACT

This paper presents the results of a study of case histories of projects on which sinking calseons were used to construct shalts for turnel construction and other underground works. Design methodologies are reviewed with respect to the impact of soil and groundwater conditions on the performance and constructability of sinking calseons. Factors which have led to success or failure are reviewed and evaluated, and recommendations for future investigation, design and construction activities are presented. Cost data for five recent projects are reviewed and compared to the cost of shalt installation by other means.

INTRODUCTION

A sinking casson consists of a cricular, square, or rectangular concrete wall structure that is initially formed at the ground surface and then surk into place through the underfying sode, using a systematic approach of adding successive lifts of concrete to the top of the structure walls, while removing soil from the intenor of the casson. This process is continued until the required dopth of excavation is achieved, after which a structural base stab is typically installed. Once surk into position, the casson serves the dual purposes of providing the indeposit during construction, as well as providing the walls of the final structure.

The technology of using anking caissons was originally developed in Europe and brought to North America in the rivel 1800's. The technology was lirst used in North America for the construction of bridge loundations for crossings of the Mississippi River in St. Louis and the East River in New York. These early caissons involved the use of compressed air to mitigate ground water infiltration at the excavation bottom. Excavation proceeded with small construction equipment or by hand, and muck was evacuated from the compressed calcion chamber through pressure relief pipes. The use of north modern construction equipment and materials, as well as batter knowledge of soil conditions and their behavior, has enabled the swking caisson to become an important option for making excavations into and through difficult ground conditions.

The authors combined experience in the tast decade includes design and construction of more than two dozen sinking calcsons. These have typically included projects where sinking calcsons were judged to be the most practical means of construction because of concerns related to either boltom stability or potential tatend ground movements. Ground

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CONSTRUCTING SHAFTS

F. J. Klingler, K. M. Swaffar, and J.C. Neyer, NTH Consultants, Ltd.

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ABSTRACT

This paper presents the results of a study of case histories of projects on which sinking calssons were used to construct shafts for tunnel construction and other underground works. Design methodologies are reviewed with respect to the impact of soil and groundwater conditions on the performance and constructability of sanking calssons. Factors which have led to success or failure are reviewed and evaluated, and recommendations for future investigation, design and construction activities are presented. Cost data for five recent projects are reviewed and compared to the cost of shaft installation by other means.

INTRODUCTION

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The technology of using sinking categors was originally developed in Europe and brought to North America in the mid 1600's. The technology was first used in North America for the construction of bridge foundations for crossings of the Mississippi River in St. Louis and the East River in New York. These early categors involved the use of compressed air to mitigate groupd water infiltration at the excavation bottom. Excavation proceeded with small construction equipment or by hand, and muck was evacuated from the compressed categor chamber through pressure relief pipes. The use of more modern construction equipment and materials. as well as befter knowledge of soil conditions and their behavior, has enabled the sinking carsion to become an important option for making excavations into and through difficult ground conditions.

The authors combined experience in the last decade includes design and construction of more than two dozon sinking calasons. These have typically included projects where sinking calasons were judged to be the most practical means of construction because of concerns related to either bottom stability or potential fateral ground movements. Ground conditions through which these calasons have been constructed include both predominantly granular as well as predominantly coheave deposits. This study examines five recent projects where sinking calasons were used in favor of other systems to construct shafts through soft and very soft clay strata.

DESIGN METHODOLOGY

The phrase "difficult ground conditions" can mean many longs to many people. With respect to shaft construction, difficult ground conditions are typically those in which conventional means of earth support such as the use of internally braced soldier ples and lagging or steel sheet pang are not appropriate. Factors which often contribute to conventional systems or approaches not being teasible within prodominantly clay soils include: high lateral earth pressures: strict ground movement limitations; instability of the excavation bottom as a result of weak clayey soils. The potential for hydrostolic uplift of the shaft bottom due to underlying artesian pressures; or the potential for ground water inflows from isolated granular layers. Factors which sometimes contribute to conventional systems or approaches not being feasible in predominantly granular soils include strict ground movement limitations and the potential for uncontrollable groundwater inflows into the shaft.

For many of these situations, a number of available construction techniques are suitable and have proven to be effective. These techniques include the use of diaphragm wall and tangent pile support wall systems, inplace soil mixing, or soil freezing. With such earth retention systems, the purpose of the system is only to support the earth and adjacent facilities until the final structure is built inside. The sinking calsson method of construction, on the other hand, will provide temporary earth support during construction, and can also be used as the permanent structure.

Sinking Calasons to Support High Lateral Earth Pressures

Many shaft construction methods exist which are appropriate for supporting high fateral earth loads However, where lateral loads become very high, internal bracing may become large and difficult to install, and may obstruct the contractor's access to the shaft. In addition, where overload factors exceed eight, clay soils may be subject to squeezing or sudden cocapse (Broms and Bennermark, 1967). In these cases, sinking calisions may be an effective alternative means of ground support. The calision wall is typically sized to insure sufficient weight to sink the structure, which usually results in a wall thickness greater than necessary to support falcral carth loads.

Sinking Calasions to Reduce Ground Movements

Limitations on ground movements may also preclude the use of conventional means of earth support for shafts. Work by D'Appolonia (1971) suggests that for clay with shear strength less than 2000 psf. tateral movement of sheet pile shaft walls can be expected to be about 2 percent of the height of the wall, while tateral movement of ngid wall systems can be expected to be about 0.2 percent of the height of the wall, while tateral movement of ngid wall systems can be expected to be about 0.2 percent of the height of the wall. Data from studies by Pack (1969) suggests that vertical settlement adjacent to shafts may range from one-half to twice the magnitude of taleral wall movement. These correlations compare well with ground movement data observed by the authors for flexible braced shafts and ngid shafts in the Detroit area. As a result of concern regarding ground movements, agid ground support systems (such as sinking calssons) have been specified for several projects in the Detroit area.

Sinking Calasons to Prevent Bottom Instability

Initial shaft design efforts in clay sols typically include analysis to determine the factor of safety for bottom stability using the given shaft depth and diameter, either by methods after Terzagin (1943) or Bjerrum and Exde (1956). Current industry practice requires that where the factor of safety against bottom instability is lass than 1.3, measures must be taken to resist ground movement below the base of the excavation. Conventional means of earth support such as internally braced steel sheet pile shafts can be designed to extend below the base of excivation, but are generally not capable of providing sufficient resistance to below-invert lateral loads where the factor of safety against bottom instability approaches 1.0. In these cases, use of a logid earth support system (such as a sinking caisson) extending below the base of the excavation is often necessary to provide the required factor of safety against bottom instability.

Sinking Calasons to Control Hydrostatic Uplift And/or Groundwater Inflow

Sinking caissons may also be suitable for conditions where high hydrostatic pressures exist in formabons below the base of the shaft, or where the shaft is to be installed through granular soils below the water table. In each case, the caisson can be suitk while maintaining the water level inside the caisson to balance outside or underlying hydrostatic forces. Soil is excavated from inside the caisson using a clamshell, and the base slab is either placed under water, or placed after lowering the groundwater pressures below the base by means of dowatering wells or other methods.

SINKING CAISSON CASE HISTORIES

Erve case histories of recent projects in which sinking caissons were used to construct shafts through difficult ground conditions were reviewed by the authors with respect to; geotechnical and hydrogeologic conditions; unique project requirements; other shaft construction systems considered; why the sinking caisson method of construction was chosen, and the outcome of the construction effort. The following case histories demonstrate the ability to successfully use the sinking calisson method of construction for livaned types of ground conditions where conventional earth support systems were not considered to be practical and/or economical.

Detroit Wastewater Treatment Plant Pumping Station 2A Project

The Detroit Water and Sewerage Department's Pumping Station 2A project in southwest Detroit included construction of two sinking calason shaft structures. The shaft required for the main pumping station was 38.4 motors (126 feet) inside diameter and approximately 21 meters (70 feet) deep, while the inequity connecting shaft was 9.1 meters (30 feet) inside diameter and approximately 15.2 meters (50 feet) deep. The structures were designed by Metcalf and Eddy. Inc. and constructed by Wateridge-Aldinger, Inc. Construction on the project was begun in 1990 and completed in 1994.

The Pump Stalion 2A project was constructed on the grounds of Detroit's Wastewater Treatment Plant, within close proximity to numerous existing structures and underground facilities. Projection of these facilities, many of which were critical to the operation of the existing plant, was dependent on controlling construction-related ground movements.

Ground conditions at the site consisted of surficial fill and a relatively thin desiccated clay layer, undertain by thick deposits of very soft sity clay with shear strength as low as 17.2 ka (360 psf). The clay deposits were undertain by a relatively this layer of very dense glacial till (locally termed hardpan), which was further undertain by tractured dolomitic lonestone containing artesian groundwater.

Several methods of ground support for construction of the pump station and connecting structure were considered during the design process, including carcular freeze walls, diaphragm walls, and sinking caissons. Ground freezing was considered undesirable due to local contractors: unfamilirarity with this method, its expected expense, and questions regarding potential for ground movements and associated utility disruption occurring during thawing. Diaphragm walls were also considered to be undesirable due to local expense. The size and depth of the structures, combined with poor soil conditions, very low ground movement tolerances, and much local expenses with the sinking caisson technique, made sinking caissons the method of choice for installing the pump station and nearby junction chamber structures.

Preparations for instatation of both carssons included pre-excavating the ground surface approximately 3 meters (10 feet). In addition, a three stage rock grouting program was performed for the pumping station caisson. To prevent hydrogen sulfide taken groundwater from enlenng the excavation as the caisson approached the hardpan and bodycock layors. Due to the presence of anostan water, these proceedures would have been required for the pumping station regardless of the system used. At the connecting shaft location, an approximately 3.6 meter (12 foot) clay separation was expected between the caisson bottom and the hardpan equifer, and as such, entesian water infiltration was not expected and rock grouting was not specified.

The pumping station calsion was constructed with seven foot thick concrete walls which were placed in five lifts (including the calsion shoe). Using mass placement methods, the contractor placed each lift as a single pour, without vertical construction joints. Type II cement was used to combat the potential effects of dissolved sulfides and sulfates in the artesian groundwater. During excavation and sinking, a soil plug ranging in height from 3 to 5.4 meters (10 to 21 feet) above the tip of the shoe was required in order to maintain a stable

excavation bottom through the very solf clay solis. Two levels of bentonite sturry injection pipos were cast into the catsson walls to allow for fubrication of the extenor during sinking. However, the contractor was able to control the sinking adequately by discrete excavation of the interior of the catsson, and bentonite injection was not necessary. Prior to placement of the last lift of the catsson walls, steel pipe landing piles were driven around the permeter of the catsson to onsure that it would be prumb and level when it landed. After landing the catsson with the shoe approximately 0.6 to 1.8 meters (2 to 6 foot) above the sloping rock surface, the catsson with the shoe approximately 0.6 to 1.8 meters (2 to 6 foot) above the sloping rock surface. The catsson where the shoe could not be safely exposed. As such, a frozen sol wall was installed along the outside of the shoe to prevent soil squeezing and/or soil flow below the shoe, while the catsson was underpinned by excavaling approximately 3 meter (10 foot) long zones of soft clay and hardpan, and filling with concrete. The rock grouping only about one gallon per minute of seepage. Construction time for the pumping station catsson, was approximately 32 waeks. Underpinning the shoe and placement of the base stab took approximately 8 additional weeks. An independent estimate by the contractor indicated that a braced steel sheeting shaft would have taken twice as long to construct.



Figure No. 1 - Pumping Station 2A Caisson

The connecting shaft carsson was constructed with 1.5 meter (5 foot) thick walls which were placed in five lifts including the shoe. Type II concrete was also used for the junction chamber carsson. As with the pumping station carsson, excavation and anking were completed by maintaining a soil plug within the carsson walls to maintain bottom slability, and isteel pipe landing ples were driven around the permeter of the carsson to ensure that it would be plumb and level when it landed. After sinking the caisson to the required depth, an artesian spring developed at the shaft invert. This flow was apparently the result of the landing ples ponetrating the clay separation barrier and the underlying ungrouted artesian aquifer. In an effort to maintain a slable excavation bottom, the shaft was flooded. The carsson base slab was later installed underwater, using tremie concrete placement mathods. Construction time for sinking this caisson, from shoe construction to

landing, was approximately 22 weeks, with approximately 2 additional weeks required to install the base slab. However, several months elapsed in between the caisson landing and the base slab installation, due to the artesian flow into the shaft and the subsequent shaft flooding required to maintain a stable bottom.

Rouge Steel Scale Pit

Rouge Steel Company, located in Dearborn, Michigan, began work on a new continuous casting operation in 1987. The new fact-lify required construction of a rectangular mail scale pit, 37.2 meters (122 feet) fong, 14 meters (46 feet) wide, and 15.2 meters (50 feet) deep. The structure was designed by Lockwood Greene Engineers and NTH Consultants. Ltd. and constructed by Wallondge-Aldinger Company.

Ground conditions at the site of the new scale pit were very similar to those at the nearby Detroit. Westewater Treatment Plant. Plants important for the performance of the earth retention system included controlling bottom stability, and minimizing lateral movements and their impects on adjacent structures. Several methods of installing the scale of were considered, including an internally braced sleet sheet pile cofferdam, soldior pile and lagging systems, and frozin ground techniques. The poor ground conditions, together with the use of the calisson walls as part of the permanent structure, made the sinking carsson option the least time construing and the most cost effective in the contractor's view.

Prior to constructing the shoe for the scalo pri caisson, the area was precut approximately 4.6 meters (15 feet) below the prevailing ground surface, and sleet pipe landing piles were driven. The excavation bottom was expected to be within thick clay deposits approximately 6.1 meters (20 feet) above the water bearing hardpan and rock layers. As such, groundwater control measures such as rock grouting were not required.

The 0.9 meter (3 foot) thick caisson walls were placed in 4 lifts including the shoe. Type III concrete was used, since groundwater (and dissolved hydrogen sulfide) inliftration was not expected to be a significant problem. Bottom instability was prevented during sinking the caisson by maintaining a soil plug within the caisson interior. A 1.8 meter (6 foot) thick concrete base stab was installed after the caisson was sunk to design elevation.

Construction time for sarking the causion and installing the base slab was approximately 11 weeks. This compared to an estimate of 20 weeks to install an internaty braced cofferdam and construct the scale pit walls and slab within

Redford CSO Pumping Station and Swirl Concentrator Calasons

The Redford Combined Sewage Overflow (CSO) retention facility includes a 18.3 meter (60 foot) inside diameter reinforced concrete pump station and a 10.7 meter (35 foot) inside diameter reinforced concrete swirl concentrator, constructed within 2.4 meters (8 feet) of each other. Both of these facilities were constructed as sinking calssons with completed base slabs requiring excavation to depths on the order of 16.8 meters (55 feet). These structures were designed by Wade-Trim/ Associates, Inc., for the owner, Charter Township of Redford. The project was constructed by Walbridge. Aldinger Company

The original contract documents as well as subsequent contractor test borings indicated soil conditions consisting of approximately 6 feet of surficial sand underlain by thick deposits of medium consistency silly clay. further underlain by compact predominantly granular layers, then shale bedrock. The soils below the base of the cossion excavations were expected to consist of silt, sand land hardpan layers under ariestan groundwater pressure. Because these layers were expected to be very difficull to dewater, the contract documents required that hydrostatic uplifts forces be balanced during construction by flooding the carsions during excavation. In addition, lite design called for dome shaped carsion base slabs to be placed under water using frome concrete placement methods. The purpose of the domed base slab was to carry permanent hydrostatic loads and eliminate the need for rock anchors within the base slab.

Recognizing the difficultly and expense which would be associated with performing the excavation and base stablic construction under water, the contractor proposed to develop a dewatering plan which would allow construction in the dry. A supplemental geotechnical investigation at the site helped define the presence and extent of the aquifer below the calesons. Based on grain size analysis of the sold samples, it was determined that the squifer was dowaterable. A dowatering plan was then developed which was subsequently accepted by the owner's engineer.



Figure No. 2 - Rectiond CSO Shafts

In order to minimize the number and size of wells, a groundwater pumping test was conducted to further define the hydrogeologic parameters of the aquifer and possibly allow a reduction in the number of dewatering webs estimated to be required based on the grain size analysis. The results of this test allowed modification of the dewatering plan to include four, 0.9 meter (3 toot) diameter gravel packed wells.

The walls of the caissons were constructed in four lifts, including the shoe. Using the modified dowalering plan, the excavation was completed in the dry and the domed base states were placed without incident. The caissons were grouted in place by injecting grout through the bentonde injection papes. Construction time was approximately 20 weeks for the 18-3 meter (50 foot) diameter caisson, and 15 weeks for the 10.7 meter (35

foot) drameter structure, not including the base slab construction. In each case, these slabs required approximately 3 additional weeks to construct.

River Rouge CSO Basin Demonstration Project

The City of River Rouge. Michigan has undertaken a major construction project to control the release of combined sewage into the Rouge River. The major component of the project is a large diameter concrete reinforced sinking casson which will serve as the main combined sewage detention factility. The caisson has seven foot thick walls, with an outside diameter of 45.4 meters (149 feet) and a depth of approximately 22.9 moters (75 feet). This project was begun in the spring of 1996, and is planned to be completed in winter 1999. The project was designed by Sigma Associates. Inc., and is being constructed by Walbridge - Aldinger Company.

Soli conditions at the site consisted of approximately 2.4 motors (8 feet) of surficial sand and debris fith, undertain by an approximately 3 meter (10 foot) thick desiccated day layer. The fill and sand layers typically contained perched groundwater. The desiccated clay is undertain by an extensive deposit of soft silty clay, which is further undertain by the hardpan till layer at a depth of about 20.7 meters (68 feet). The bedrock undertying the hardpan consists of dolomatic timestone, which is typically highly fractured within the upper 1 to 3 meters (3 to 10 feet). Artesian pressure within the hardpan and bedrock was measured at approximately 1.5 meters (5 feet) above the ground surface. The hardpan at this site is considered retailively impermeable, although the fractured bedrock is expected to produce high volumes of groundwater if it is exposed. Where encountered, artesian groundwater is expected to contain high concentrations of dissolved toydrogen sulfide, and moderate concentrations of methane gas

Prior to constructing the caisson, the project site was pre-cut approximately 2.1 to 3 meters (7 to 10 feel) in order to prevent difficulties in sinking the caisson through the upper fill and desiccated clay layers. Perched groundwater was controlled by pumping from open sumps. A four stage rock grouting program was then undertaken to seal the fractured bedrock and to minimize unfiltration of groundwater during excavation. Additionally, a geotechnical instrumentation program was undertaken with the purpose of protecting existing adjacent structures.

The caisson wall construction is planned to be competed in six lifts including the shoe. The excavation from the conter of the caisson has been conducted in the dry, using a damshell and smaller earthmoving equipment working within the caisson. At this writing, the caisson is close to its final elevation, and is planned to be completed within five woeks. Following sinking of the caisson walls to the desired depth, the clay zone below and around the perimeter of the shoe is planned to be subject to ground freezing to prevent soil squeezing below the shoe prior to base stab placement. Rock anchors will then be installed from the rock surface, and post tensioned in place following base stab placement.

The caisson construction is planned to be completed in approximately 28 weeks, not including the base slab construction. The base slab construction (including rock anchor installation) is expected to be completed in an additional 8 weeks.

Northweat Lateral Pipeline

Construction for the Costal Water Authority and City of Houston's Northwest Lateral Pipeline began in 1988 and represented an integral step in the expansion of the city's water supply system. The original design called for approximately 3350 meters (11,000 feet) of tunnel crossing two major bodies of water: the Houston Ship Channel and Greens Bayou. After award of the contract to the Greenfield Momiman Joint Venture, the joint venture retained NTH Consultants to redesign the water conveyance system under a value engineering provision in the contract. As part of the redesign, the tunnel depth was decreased and sinking callsons were used to construct "siphon" type extensions below the Houston Ship Channel and Greene Bayou.

At the Houston Ship Channel crossing, octagonal shaped shafts, which were specified in the contract documents to be either diaphragm walls or freeze walls, were redesigned to be round sinking calasions. This modification allowed a reduction in the wall thickness and reinforcement required, and made sinking calasions the economical choice. At the Greens Bayou crossing, round sinking calasions were designed to replace long sloping funnel sections which extended below the Bayou. As a result of this design, the majority of the tunnel was installed at relatively shallow depth, while only the periods of tunnel directly below the bodies of water were mined at the desper depth. The decreased tunnel depth through most of the tunnel alignment allowed the use of steel in and lagging primary tunnel liner rather than steel liner plate as called for in the orginal design. In addition, sleep tunnel grades which were called for in the orginal design were eliminated.

The four shafts used to construct the sphon extensions were each constructed in similar ground conditions. Subsurface soils of each shaft locations generally consisted of medium to stiff clay with interbedded water bearing zones of floc sand and silt, undertain all depths of 21 to 30 meters (70 to 100 feet) by generally soft clay. Where groundwater was present within the water bearing layers, the potentiometing level was about 3 to 4.6 meters (10 to 15 feet) below ground surface.



Figure No. 3 - Greens Bayou Shaft

The shaft designs at the Greens Bayou and Houston Ship Channel crossings called for two stage shaft construction, consisting of an upper sheet pile supported shaft and a lower concrete sinking casson shaft. The upper steel sheet pile shafts were 15.3 interest (50 foot) doop, and were supported internally by concrete ring wales. The lower sinking casson portions of each shaft were 7.6 meters (25 feel) in internal diameter with 1.2 meter (4 foot) thick walls. The shafts extended to overall depths of approximately 33.5 meters (110 feet) below ground surface. If should be noted that only one of the four sinking casson structures was designed to be incorporated into the permanent structure. The remaining three callssons were used only as temporary structures, and were abandoned in place.

Control of ground movements during construction was critical due to the presence of an active railroad line which was located within 6.1 meters (20 feet) of the edge of one of the shafts. Borehote extensioneters were used to measure ground movement in this area. Within the sinking caissons, bottom instability was typically prevented by maintaining a soil 'plug' within the shaft bottom. Within one of the Greens Bayou shafts however, flooding the causion was also required during sinking to balance hydrostatic pressures and groundwater seepage into the shaft. Typically, bentonite was injected at the shoe during excavation to reduce friction and adhesion on the outside of the causion, and to aid in sinking. Once the causion pipes. The average time to construct each of the shafts was approximately 3 to 4, weeks for the upper sheet pile stage and 10 to 12 weeks for the lower sinking caisson (including the base stab).

COST COMPARISONS

Costs for constructing the serving calssons discussed above were summarized and compared to other construction methods which were considered. All costs were adjusted to 1996 dollars. Based on our review of cost data for those projects, the overall cost for sinking calssons in place (excluding base stab construction) was found to be approximately \$795 per cubic meter (\$610 per cubic yard), which can be equated to approximately \$1450 per square meter (\$135 per square fool) of exposed shaft wall. However, if should be noted that unit area costs varied more widely than unit volume costs due primarily to differences in the wall thicknesses. Other methods which were considered for the above projects ranged from \$1940 to \$2150 per square meter (\$180 to \$200 per square fool) of wall, including the cost of the final concrete shaft walls in place. A summary of cost data for the above discussed projects is presented in Table 1.

Project/Shah	Total Cost (Sinking Caisson Method)	Cost per Unit Volume for Calason Method	Commente
P\$2A · Pumping Station	\$5.080.000 (caisson walls) \$2,213,000 (base slab w/ fréeze & undersia)	\$842/m² (\$643/cy) \$816/m² (\$624/cy)	 Costs for steel sheeting and bracing were quoted at \$2 million higher than sinking carsson.
PS2A - Conn. Shaft	Not Available	Not Available	 Costs for freeze wall were quoted at \$3 million higher than sinking caisson. At least 6 months were saved over atternative methods
Rouge Steel Scale Pit	\$2.065.000 (caisson walls) \$301.000 (base slab)	\$943/m ³ (\$721/cy) \$322/m ⁹ (\$246/cy)	 Savings of \$0.7 million gained over use of braced sheet pilo shaft construction.
Redford CSO	\$337.000 (caisson walls) \$152.000 (base slab)	\$368/m ⁸ (\$297/cy) \$287/m ⁸ (\$296/cy)	
Redford CSO (35'dia)	\$173,000 (caisson walls) \$49,000 (base slab)	\$322/m ¹ (\$246/cy) \$323/m ³ (\$247/cy)	
Rouge CSO Basin	54.717.000 (caisson walls) \$635.000 (base slab)	\$801/m ³ (\$612/cy) \$177/m² (\$135/cy)	 Savings of \$2.5 million gained over uso of braced sheet pile shaft construction. Construction time for caisson estimated to be half that of braced sheeting
NW Lat Pip e (all shefts)	Not Available	Not Available	

TABLE 1 - Cost Comparisons for Sinking Caissons Vs. Other Shaft Const. Techniques

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CONSIDERATIONS FOR DESIGNERS

Several considerations are apparent based on the author's experience in the design and construction of shafts using sinking casesons. These are discussed below:

1) Sinking catsoons are often cost effective when utilized as both the construction shaft and final structure. The volume of concrete used for the casson walls will typically be greater than the volume of concrete required for the walls of the final structure, as a consequence of the weight required to sink the casson walls. As a rough comparison of construction costs between sinking casson construction and temporary earth support with an internally constructed structure, the preliminary design may compare the cost of the additional concrete required in the casson walls (above that amount which would be required for the final structure), to the cost of a temporary earth support system. The designer should be aware, however, that the majority of casson wall construction costs are from the labor for vertical form work, rather than the bulk materials. As such, the costs for thickened casson walls do not increase the overall shaft costs incrementative.

2) Sinking calescons are highly effective in controlling ground movements adjacent to constructed shafts Based on emprical studies of ground movement by Peck and D'Appolonia, as well as project experience from the above case histories, settlement adjacent to sinking calescon shafts is typically on the order of 0.2 percent of the depth of the shaft. This compares to estimates by Peck of 1 percent to 2 percent of the shaft depth for flexible earth support systems such as internally braced steel sheet pulses.

3) The design of sinking caissons in clay solve is highly sensitive to adhesion and friction of the sol surrounding the caisson walls, as well as the shear strength of soils below the base. Methods of evaluation of basal stability (Bjorrum and Erde, and Terzaghi), suggest that failure zones will extend to a depth below the caisson shoe approximately equal to 0.7 times the shaft width. As such, geotechnical investigations should be planned to gather data to depths equal to a minimum depth corresponding to one shaft width below the planned final shaft invert. Where bottom stability concerns require ubizzation of a soil plug within the caisson to control bottom heave during excavation, geotechnical investigations should extend to a depth below the caisson shoe of at least 0.7 times the width of the shaft. Soil testing should include unconfined compression strength evaluations or unconsolidated undrained triaxial tests of relatively undisturbed Shelby tube samples. For soft and very soft clay, in-situ vane shear testing should also be considered. Such samples and/or performed at minimum intervals of approximately 15 feet. Where artesian groundwater is encountered within test borngs, a piezometer should be installed to allow for design provisions to prevent hydrostatic uplift of the shaft base.

4) The use of sinking carsion techniques is often most successful and most cost effective where ground conditions are considered very poor. In the words of a contractor's tradesman. The worse the clay is, the better these sinking carsions work? It is significant that in most of the above cited cases, sinking of the caisson was controlled by discrete excavation, without using the bentonite fubrication systems which were installed.

5) Sinking caissons have proved to be an effective means of construction for both owner-designed and contractor-designed shafts. Advantages to owner-designed shafts include: ability to control ground movements (by specifying construction means such as anking cassons), ability to incorporate the ainking caisson into the permanent structure; speed of construction in appropriate conditions; and the ability to obtain competitive bids for the specific type of shaft construction desired. The primary advantage of allowing the contractor to design

shafts is that creative and innovative solutions can be proposed by the contractor, and the owner can share in the cost savings which result from contractor ingeniety.

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