# CORRELATION BETWEEN DYNAMIC CONE PENETROMETER (n-VALUE) AND ALLOWABLE BEARING PRESSURE OF SHALLOW FOUNDATION USING MODEL FOOTING

By

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A Thesis submitted to the Department of Civil Engineering, Kwame Nkrumah University of Science and Technology, Kumasi in partial fulfilment of the requirements for the award of

Master of Philosophy

Faculty of Civil and Geomatic Engineering College of Engineering

## DECLARATION

I hereby declare that this submission is my own work towards the MPhil in Civil Engineering and that, to the best of my knowledge, it contains no material previously published by another person nor material which has been accepted for the award of any other degree of the University, except where due acknowledgment has been made in the text.

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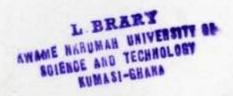
# DEDICATION

KNUST

I dedicate this work to my late mother

# **ESTHER AFI ADEKPO**

of blessed memory who can not see the successful end of this work



#### ACKNOWLEDGMENT

I am grateful to the LORD ALMIGHTY GOD for making it possible for me to complete this landmark project in my life.

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#### ABSTRACT

An allowable bearing pressure is one of the most important basic parameters to be determined before the design and construction of foundations for civil engineering structures. The conventional methods of estimating this parameter is becoming relatively expensive and time consuming for small scale projects such as residential buildings.

The DCP is a versatile equipment that may be applied to obtain the bearing capacity. However, currently, there are no reliable correlations between the DCP test results and the bearing capacity. This project was undertaken to develop a reliable correlation between the Dynamic Cone Penetrometer (DCP) n-value (blows/100mm) and the allowable bearing pressure qall (kN/m2) for shallow foundations using a model footing. In this work, compacted soil sample of different dry densities in a mould was loaded with a model footing until the sample yielded. On the same sample, DCP testing was performed at two locations to determine the average Dvalue (mm/blow). Triaxial samples were also taken for triaxial test. Results from the triaxial test were used to calculate ultimate bearing capacity using Terzaghi bearing equation. The measured DCP D-value (mm/blow) was processed into n-value (blows/100mm) which is the standard form of recording the DCP test results in the field. The results were analysed and a correlation  $q_{all}=48n + 57$ , with a coefficient of correlation, R2=0.98 was obtained for the model footing. correlation was similar to the correlation between the n-value and the allowable bearing pressure computed using the Terzaghi approach, except that the Model underestimated the allowable bearing pressure by a constant value of 165 kN/m2 for

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# LIST OF SYMBOLS AND ABBREVIATIONS

A = Cross-Sectional Area of Cone Base

ASTM = America Society of Testing and Materials

b = Mould Width

B = Footing Width

B = Skin Resistance Reduction Factor

BC-15 = Name of the Test for 15 rammer blow on Model Sample

BS = British Standard

CBR = California Bearing Ratio

CPT = Cone Penetrometer Test

c<sub>u</sub> = Undrained Coefficient of Cohesion

D = Penetration per Blow

DCP = Dynamic Cone Penetrometer

D<sub>f</sub> = Foundation Depth

E<sub>50</sub> = Stiffness of Triaxial Sample at Half the Maximum Deviator Stress

EMC = Existing Moisture Content

€50 = Strain at Half the Maximum Deviator Stress

FS = Factor of Safety

g = Acceleration due to Gravity

h = Mould Height

H = Height of Hammer Fall

k<sub>o</sub> = Plane Strain

λ = Tip Resistance Reduction Factor

1 = Mould Length

L = Footing Length

LL = Liquid Limit

M = Mass of the Hammer

MDD = Maximum Dry Density

M<sub>R</sub> = Resilient Modulus,

M<sub>s</sub> = Mass of Sample

n-value= Number of DCP test Blows per 100mm Penetration

N-value= Number of SPT Blows per 300mm Penetration

N<sub>e</sub> = Bearing Capacity Factor due to Contribution of Cohesion

N<sub>1e</sub> = Bearing Capacity Factor due to Contribution of Cohesion for Local Shear Failure

Nγ = Bearing Capacity Factor due to Contribution of Unit weight

N<sub>1y</sub> = Bearing Capacity Factor due to Contribution of Unit weight for Local Shear Failure

N<sub>q</sub> = Bearing Capacity Factor due to Contribution of Overburden Pressure

N<sub>1q</sub> = Bearing Capacity Factor due to Contribution of Overburden Pressure for Local Shear Failure

OMC = Optimum Moisture Content

\*p = Average Confining Pressure at Failure during the Triaxial Test

P = Mass of the Penetrometer without the Sliding Hammer

PCPT = Piezocone Cone Penetrometer Test

PL = Plastic Limit

PI = Plasticity Index

φ<sub>u</sub> = Undrained Internal Angle of Friction

φ<sub>1</sub> =Transformed internal friction angle for local shear failure

Q = Load

q<sub>all</sub> = Allowable Bearing Pressure,

qc = Static Cone Penetration Resistance of Soil

qf = Ultimate Failure Pressure using Model Footing

q<sub>f(1)</sub> = First-Failure Pressure

quit = Ultimate Bearing Capacity by Terzaghi Approach

Quh(1) = First-Failure Load

qy = Yield Stress

R<sub>d</sub> = Dynamic Resistance of Soil

s = Skin (rods) Resistance

s<sub>c</sub> = Bearing Capacity Shape Factor for Cohesion Component

S<sub>f</sub> = Maximum Deviator Stress

s<sub>γ</sub> = Bearing Capacity Shape Factor for Unit Weight Component

s<sub>q</sub> = Bearing Capacity Shape Factor for Surcharge Component

S<sub>Y</sub> = Yield settlement

t = Tip (cone) Resistance

U<sub>100</sub> = Undisturbed 100mm Diameter Sample

UCS = Unconfined Compression Strength

V = Yield Stiffness

 $\rho$  = Density

 $\sigma_1$  = Axial Pressure

 $\sigma_3$  = Confining Pressure

#### CHAPTER ONE

#### INTRODUCTION

## 1.1 Background

One important objective in site investigation for the design and construction of foundations for civil engineering structures is the estimation of the allowable bearing stress for the foundation system proposed. Improper estimation of the allowable bearing stress leads to either overestimation or underestimation of the soil's load carrying ability, with consequent safety or economy implications on the project respectively.

The conventional approach to estimation of allowable bearing stress is through site investigation techniques. By this approach in cohesive soils, U<sub>100</sub> samples are obtained using a drilling rig or for shallow depths, through trial pits. The samples obtained are sent to the laboratory, and subjected to triaxial test to determine the strength parameters, c<sub>u</sub> and  $\phi_u$  Depending on the proposed footing configuration, the allowable bearing stress is then estimated using Terzaghi's bearing capacity equation. On the other hand, for cohesionless soils, the method of determining the allowable bearing capacity consists of the use of the Standard Penetration Test (SPT) which is an in-situ test. The SPT, however, is performed as part of site investigations using the percussion drilling rig whose cost do not make it any more economical than the use of samples. For example, a typical site investigation within Acera, for simple structure such as 4 bedroom single-storey house will cost about GH¢ 3,000.00 for 3 boreholes not exceeding 10.0m deep and will require at least 8 days for the fieldwork alone. This cost is considered too high for average developers of these types of buildings with estimated total project cost of about GH¢30,000.00 (The site

investigation cost therefore represent about 10% of the total project cost). In the light of these challenges, there is the need to find cost effective ways of undertaking site investigation for these categories of simple structures. The potentials of the DCP are therefore being explored for evolving allowable bearing capacity for shallow foundations. The dynamic cone penetrometer (DCP) is simple and versatile equipment which in combination with other methods has the potential of simplifying site investigations.

# 1.2 Objective

The objective of the study was to establish a correlation between the Dynamic Cone Penetrometer Test n-value (number of blows per 100mm) and allowable bearing stress for shallow foundations using a laboratory model footing and a sandy *CLAY* material.

#### 1.3 Justification

The DCP is simple and cost effective equipment and has been recommended to be an appropriate technology for use in developing countries (Sanglerat, 1972). However, up to date, not much research has been conducted into the use of DCP in general and its use in estimating of allowable bearing pressure for shallow foundation design in particular. Even of the few works done regarding the DCP, the focus has been on its use in pavement design and in agricultural engineering for the determination of the consistency of potential cultivatable land and grazing fields (Amini, 2003; Burham and Johnson 1993; Edil and Benson, 2004; Gabr et al., 2000; Karunaprema and Edirisinghe, 2002; Singh et al., 1973; Uddin, 2002; Vanagas et al. 2004;). In spite of the very limited research on DCP test for estimating allowable

bearing stress for shallow foundations design and construction, Sowers and Hedges (1966), Sanglerat (1972), Cearns and McKenzie (1988) and Ampadu (2005) have made significant strides in that respect.

Notwithstanding the few attempts to find suitable ways of using the DCP testing effectively for the estimation of allowable bearing stress, the absence of a credible correlation is leading to a situation where some practitioners in the country are attempting to use any kind of DCP specification to estimate the allowable bearing stress of foundation system basing the analyses mainly on the works of Sanglerat (1972). These weaknesses are leading to inconsistent outputs for the users of the DCP testing for estimating allowable bearing stress. This study, however, sought to correlate the DCP n-values with directly measured allowable bearing stress using a laboratory model footing on a sandy *CLAY* material.

# 1.4 Scope of Work

The scope of work was limited to:

- (i) field work which consisted of manual excavation of test pit to recover sample for the tests and logging the test pit;
- (ii) laboratory work consisting of the characterisation of the test material and performance of unconsolidated-undrained triaxial test to derive the undrained strength parameters,  $c_u$  and  $\phi_u$
- (iii) The laboratory work also covered a model footing test on compacted sample and DCP testing of the model ground.

#### CHAPTER TWO

#### LITERATURE REVIEW

#### 2.1 Introduction

The use of penetrometers in geotechnical site investigation is widespread (Harison, 1987). Different types of penetrometers are used for site investigation in response to varying needs. Usually, during the initial exploration stage, penetration tests are employed to determine the stratigraphy, thickness, the soil type in terms of consistency, and the lateral extent of the lithologies. At the detailed site investigation stage, penetration tests are still employed to determine some geotechnical design parameters (Mayne et al, 1995). In general, there are two categories of penetrometers; static and dynamic penetrometers. Available literature indicates that, the use of static penetrometers have been extensively researched into far more than the dynamic penetrometers.

#### 2.2 The Pocket Penetrometer

This is a static cone penetrometer. It is commonly used on split spoon and thin walled tube samples to evaluate consistency and approximate unconfined compressive strength of saturated cohesive soils. It is also used for the same purpose in freshly excavated test pits and trenches. The pocket penetrometer, Figure 2.1 is extremely small and portable; typically weighing less than 300g and approximately 15cm long. It is a spring-loaded penetrometer. The spring is calibrated against unconfined compressive strength (in kg/cm²). Prior to testing, the indicator ring is positioned at the top of the barrel. Next, the probe is slowly inserted into the soil until the calibration mark is levelled with the soil surface. The movement of the

indicator ring corresponds to the spring compression. The chamber is marked with unconfined compressive strengths that have been calibrated with the internal compression of the spring. The mark at which the indicator ring is located is taken as the unconfined compressive strength of the soil. It must however, be noted that, the readings obtained from the Penetrometer should not replace laboratory test results due to the fact that a small area of penetration test could give misleading results. The instrument should not be used for obtaining final foundation design data. It should be noted that pocket penetrometer readings are only approximations of actual strengths and accuracy of about 1/2 division is possible (Humboldt Manufacturing Company, 2002). One 8- mm interval on the scale is equivalent to 1kg/cm²=98.1kN/m² unconfined compressive strength



Figure 2.1 Pocket Penetrometer (after Humboldt Manufacturing Company, 2002)

#### 2.3 The Cone Penetrometer Test

The Cone Penetration Test (CPT) is traditionally used by most European and American geotechnical engineers. Original specification for its use has been designated ASTM D3441 and was adopted in 1974 by the American Society of Testing and Materials. In this test, a small cone is attached to a rod and pushed steadily into the ground at a rate of 20.32mm of displacement per second. The penetrometer is attached to a vehicle that provides the large jacking forces required

during testing. The 60° apex angle cone has a base area of 1032mm<sup>2</sup>, with a diameter of 35.56mm and a height of 30.48mm. The cone is attached to a friction sleeve having a diameter of approximately 35.6mm as in Figure 2.2. The CPT equipment is steadily pushed into the ground and has been used to characterize soils to depths of 30m below the ground surface. The tip of the cone is typically instrumented to measure the tip resistance using strain gauges, while the sleeve above the cone is instrumented to measure the sleeve friction. These parameters are used to obtain a profile of the variation of soil strength and soil type with depth (Brouwer, 2002). An improvement over the original CPT in 1995, led to the new electronic CPT called Piezocone Cone Penetrometer Test (PCPT) which has been produced and its operation outlined in ASTM D-5778. The PCPT combines both electronic friction cone and piezocone penetrometers. This device produces computerized log of tip and sleeve resistance, induced pore pressure just behind the cone tip, pore pressure ratio (change in pore pressure divided by measured pressure) and lithologic interpretation of each 2 cm intervals of subsoil. The data acquisition systems typically include a portable computer, analog-digital converter, storage media (hard drive, floppy drives), and strip chart recorder or printer and printed out (Mayen et al, 1995).

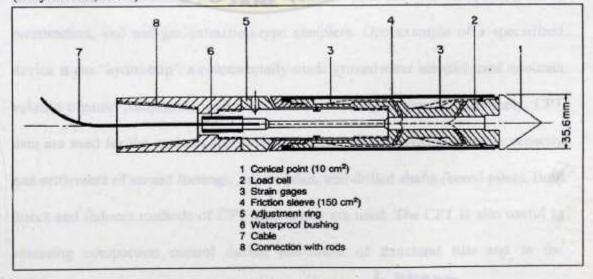


Figure 2.2 Cone Penetration Tip (after D5778)

## 2.3.1 Applications

The results of the PCPT are used for delineating soil strata and for evaluating the geotechnical engineering parameters of the subsurface layers. Its exceptional resolution makes the detection of thin seams and lenses possible, particularly via the pore pressure channel. This facet is very important and is used in slope stability evaluations. Considerable effort has been made to derive soil engineering properties from the results of cone and piezocone data. Methodologies have been developed using empirical and statistical methods, back-calculations, analytical studies, and numerical simulation (Mayen et al, 1995). Mayen et al, (1995) report that many recent developments in CPT have centered on its use for geo-environmental concerns. The incorporation of additional sensors within the penetrometer to instantaneously and continuously monitor phenomena offers significant potential for evaluating subsurface chemical and biological conditions. For example, cone data are used to infer or detect the presence of anomalies such as contaminants in the pore fluid. In addition, modifications and add-on modules to the standard cone penetrometer, such as, direct-push technology has led to other specialized probes for sampling and testing groundwater and soil during environmental site characterization, including: push-in soil samplers, push-in water samplers, push-in piezometers, and soil-gas extraction-type samplers. One example of a specialized device is the "hydro-trap", a commercially-made groundwater sampler used to obtain volatile organic compounds under controlled confining pressures. Routinely, CPT data are used for the analysis and design of foundations, including bearing capacity and settlement of spread footings, driven piles, and drilled shafts (bored piles). Both direct and indirect methods of CPT assessment are used. The CPT is also useful in assessing compaction control during placement of structural fills and in the

evaluation of effectiveness of ground modification techniques (e.g., vibroflotation, dynamic compaction) and site improvement works (Mitchell, 1986).

#### 2.4 The Standard Penetration Test

This is a dynamic cone penetrometer designed to provide information on the engineering properties of soils. The test procedure is described in the British Standard BS 1377: Part 9:1990. A typical test operation is as shown in Figure 2.3. The test uses a split thick-walled sample tube, with an outside diameter of 51mm and an inside diameter of 35mm and a minimum length of 457mm, Figure 2.4. This is driven into the ground at the bottom of a borehole by blows from 63.5kg hammer falling through a distance of 760mm. The sample tube is initially driven 150mm into the ground (seat-in-drive) and then the number of blows needed to further penetrate each of 75mm of the tube markings to a depth of 300mm is recorded. The total number of blows for the 300mm penetration is the standard penetration resistance N-value. In cases where 50 blows are insufficient to advance the tube through the 300mm interval, the penetration after 50 blows is recorded and indicated as refusal. The main purpose of the test is to provide an indication of the relative density of granular deposits, such as sand and gravels from which it is virtually impossible to obtain undisturbed samples for strength testing. The great merit of the test and the main reason for its widespread use is that, it is simple and relatively inexpensive. It has been observed that interpretation of the SPT results depends on the soil type, with fine-grained sands giving the most useful results, with coarser sands and silty sands giving reasonably useful results, and clays and gravelly soils yielding results which may be very poorly representative of the true soil conditions.

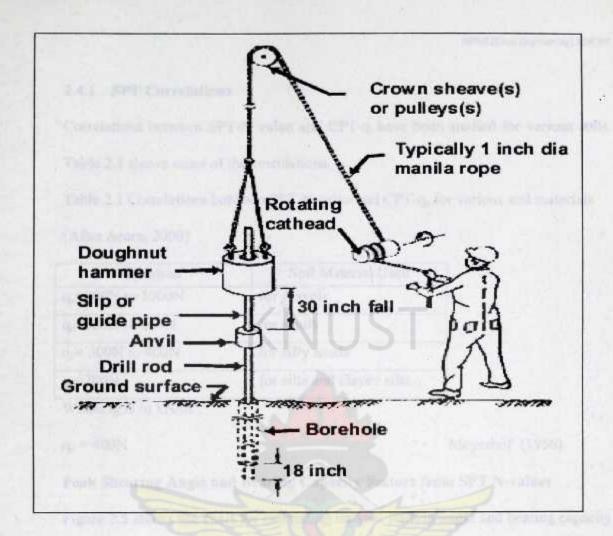


Figure 2.3 Typical Operation of the SPT (after www.en.wikipedia.org)

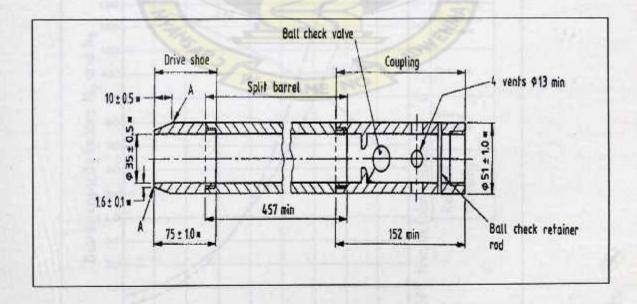


Figure 2.4 Split Spoon Sampler, dimensions in mm (after BS 1377: Part 9: 1990)

#### 2.4.1 SPT Correlations

Correlations between SPT-N value and CPT-q have been studied for various soils.

Table 2.1 shows some of the correlations.

Table 2.1 Correlations between SPT N-value and CPT-q<sub>c</sub> for various soil materials (After Arora, 2000)

Soil Material Used	
for gravels	
for sands	
for silty sands	
for silts and clayey silts ,	

Where qc is in kN/m2,

 $q_c = 400N$ 

Meyerhof (1956)

#### Peak Shearing Angle and Bearing Capacity Factors from SPT N-values

Figure 2.5 shows the chart for estimating internal friction angle and bearing capacity factors of a cohesionless soil from SPT N-values.

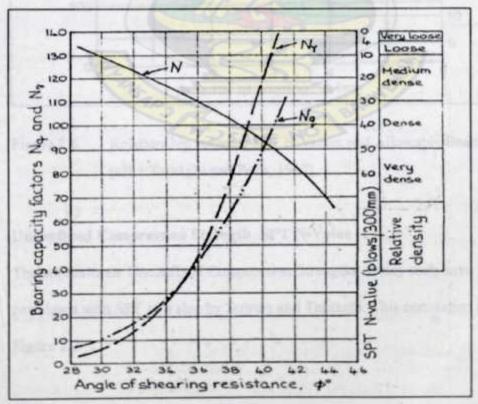


Figure 2.5 Relationship between φ, N<sub>q</sub> and Nγ with SPT N-values (After Peck, Hanson & Thornburn, 1973)

## Allowable Bearing Pressure- SPT N-Values

Figure 2.6 is a chart for estimating allowable bearing capacity of sand based on SPT N-values

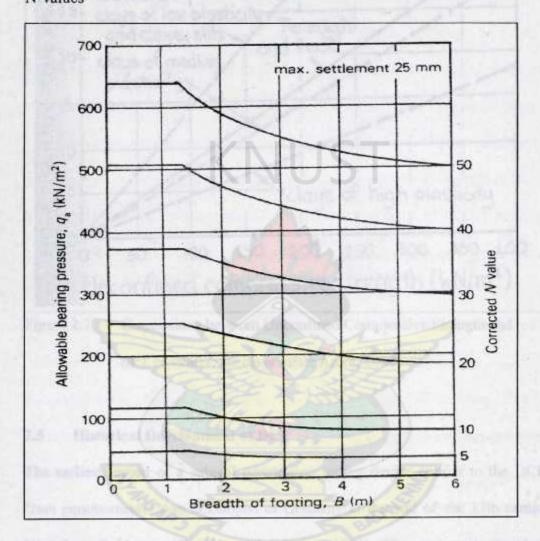


Figure 2.6 Relationship between SPT N-values and Allowable Bearing Pressure

(after Terzaghi and Peck, 1967)

# Unconfined Compression Strength -SPT N-Value

The approximate Unconfined Compression Strengths of clay soils have been correlated with SPT N-Value by Sowers and Terzaghi. This correlation is shown in Figure 2.7

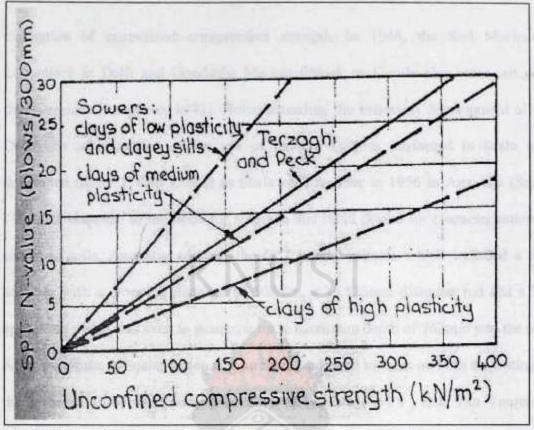


Figure 2.7 Correlations between Unconfined Compressive Strength and SPT N-values, for clays (after Arora, 2000)

# 2.5 Historical Development of DCP

The earliest record of a subsoil penetration testing device similar to the DCP, the "ram penetrometer" was developed in Germany at the end of the 17th century by Nicholaus Goldmann (Burham and Johnson, 1993). The next major development again came from Germany, when, Kunzel in 1936 developed what was known as ":Prufstab". This device was later used by Paproth in 1943 and eventually became standardized in 1964 as the "Light penetrometer", the German Standard DIN 4094. Concurrent with the German standardization of the "Light Penetrometer", several other countries and individuals developed their own penetration devices, such as : Collin of France in 1846; the Swedish Railroads standardized theirs in 1917, Danish Railroads in 1931 came out with the pocket penetrometer used for approximate

evaluation of unconfined compression strength. In 1946, the Soil Mechanics Laboratory in Delft and Gondsche Machinefabriek oa Gouda also came out with their versions (Sanglerat, 1972). Notwithstanding the historical development of the DCP, the advancement of the use of modern DCP is attributed to Scala who developed the DCP also known as Scala penetrometer in 1956 in Australia (Scala, 1956). In response to the need for a simple and rapid device for characterization of subgrade soils. Australia adopted the DCP used by Scala which included a 9kg hammer with a dropping distance of 508mm, a 15.875mm diameter rod and a 300 apex cone which was used to penetrate up to maximum depth of 762mm into the soil. After the works of Scala, various researchers continued to work on both the testing of the instrument and the testing procedure. In the late 1960's , D.J. Van Vuuren in Zimbabwe continued with the development of the DCP(Van Vuuree, 1969). He used a similar device as Scala, except for some differences in dimensions: a 10kg hammer was dropped from a height of 460mm, forcing a 300 apex cone connected to a 16mm diameter rod into the soil for a maximum depth of 1000mm (Burnham and Johnson, 1993). Kleyn (1982) contribution to the development of DCP was in the area of its application to determine in-situ properties of road pavement layers and subgrade in South Africa. Other applications included identification of potential collapsible soils, compaction construction especially when high lift rates render normal compaction control techniques to time be consuming to be effective. He further considered the use of DCP in pavement evaluation and monitoring in terms of structural evaluation of pavement, comparing in-situ CBR and structural monitoring. Some of the many versions of the DCP used by various people are tabulated in Table 2.2.

Table 2.2 Some Versions of the DCP (adapted from Ampadu, 2005)

Туре	Cone Diameter (mm)	Mass of Hammer (Kg)	Height of Fall (mm)	Energy per Blow per Cone Area (KN-m/m²)
Sowers &	solid cont	Salid cone		
Hedges (1966)	38	6.8	508	30
Scala (1956)	20	9.08	508	144
Kleyn (1975)	20	8	575	144
Borros Penetrometer	50	63	750	231
Singh (1973)	35	10	500	51
Ampadu (2005)	20	10	460	144
This Study	20	8	578	144

## 2.6 The Dynamic Cone Penetrometer

The dynamic cone penetrometer in operation consists of a drop weight, an anvil, rods and a cone. It is traditionally used in testing of subgrade soils for road projects. It is also used during reconnaissance exploration of soils at shallow depths for other civil engineering projects. The consistency of cultivatable lands is determined using DCP testing prior to ploughing. During testing, the drop weight is allowed to fall through a specified height. As the drop weight hits the anvil, kinetic energy is imparted into the DCP system and the cone is driven into the ground. The depth of penetration with each blow is measured or derived as the D-value. The soil layering profile and strength are interpreted from the D-values. The test procedure is described in the ASTM D6951-03, The Transport Research Laboratory(TRL) Research Report 361(2004) and in the recommendations of the study group created in 1957 by committee of the International Society for soil Mechanics and foundation engineering to analyse test methods involving the dynamic and static penetrometers (Sanglerat 1972).

A comparism of the main features of SPT and DCPT is shown in table 2.3.

Table 2.3 comparison between SPT and DCPT Used

DCPT
Dynamic Impact
Solid cone
Weight of hammer is 8kg
Height of hammer fall is 578mm
Diameter of cone 20mm
Energy per blow 144kN-m/m <sup>2</sup>

## 2.6.1 Development of Semi Empirical Equations

The principle of the dynamic cone penetrometer test is based on dynamic resistance offered by soil to deformation caused by dynamic penetrometer. The degree of the resistance is a measure of the soil's shear strength and hence its bearing capacity. Thus a penetrometer of constant energy (blow) would develop higher D-value in a soil layer of low shear strength as compared to a soil layer of higher shear strength (Sanglerat, 1972; Singh et al., 1973; Peck et al., 1973 and Braja, 1985). The theory of dynamic cone penetrometer however can be attributed to Aimé Nadal, a French civil engineer of "Ponts et Chaussées" in Paris and the Dutch. The work of Aimé Nadal was based on empirical evidence gathered over many years of working on penetrometer tests and driving of piles. It was observed that the dynamic resistance of soil, R<sub>d</sub> to advancement of penetrometer was a function of the shear strength of the soil by the dynamic expression;

$$R_d = \lambda t(c, \phi) + Bs(c, \phi) \qquad (2.1)$$

Where t and s are the tip(cone) and skin(rods) resistances respectively;  $\phi$  and c are angle of internal friction and cohesion factors respectively, whiles  $\lambda$  and  $\beta$  are

reduction factors whose values depend on the nature of the soil into which the penetrometer is being driven; and the design of the penetrometer.

Because the design of the DCP used in this study is such that, the diameter of the cone base (20mm) is larger than the diameter of the rod behind it (16mm), the total resistance is theoretically provided by the cone resistance only.

Hence, 
$$\lambda = 1$$
 and  $\beta = 0$ 

$$\Rightarrow$$
 R<sub>d</sub>= t(c,  $\phi$ )......(2.2)

The Dutch Formula is a dynamic formula which is based on the assumption that the kinetic energy delivered by the hammer during penetration by a dynamic penetrometer is equal to work done by the entire penetrometer system. It is founded on the Newton's principle of energy and impact motion. Upon releasing the hammer, the mobilized maximum potential energy is progressively transferred into kinetic energy which attains maximum value just at the point of striking the anvil. At point of strike, most of the kinetic energy is transferred from the hammer to the anvil which in turn is transmitted to the lower rods and finally to the cone. Theoretically, when the cone eventually comes to rest, the total kinetic energy imparted has been dissipated in the soil. Therefore by the law of conservation of energy, the total work done by the soil to stop the advancing cone and wasted (lost) energy to environment such as sound and wave energy equals the maximum potential energy attained by the hammer at the highest point, neglecting sliding friction and air resistance of the hammer. The main components of energy losses in DCP operation are; sound energy, heat energy generated, wave energy propagated through the soil and skin friction energy of the rod.

The Dutch formula is hence expressed as;

$$R_{\rm d} = \frac{M^2 g H}{\left[ A D (M+P) \right]} \tag{2.3}$$

Where R<sub>d</sub>= dynamic resistance in kN/m<sup>2</sup>

M=mass of the hammer in kg;

H= height of hammer fall in mm;

A= cross-sectional area of the cone base in m<sup>2</sup>

D-penetration per blow in mm;

g- acceleration due to gravity, in ms-2

P=mass of the penetrometer (without the sliding hammer) in kg; (Sanglerat, 1972).

It should be noted that for a particular dynamic penetrometer, all the parameters in equation (2.3) are constant except D=penetration per blow which is a variable dependent on the shear strength of the soil being tested.

Substituting the following specifications of the dynamic cone penetrometer used,

M = 8kg, H=578mm, A =3.142 x  $10^4$  m<sup>2</sup>, P= 3.864kg, D=penetration in mm per blow;

equation 2.3 reduces to

$$R_d = \frac{97351}{D} kN/m^2$$

Sanglerat (1972) recommends that for shallow foundations, the allowable bearing pressure,  $q_{all,}$  may be estimated by dividing  $R_d$  by a factor of 20,

Thus, 
$$q_{all} = \frac{R_d}{20}$$

$$q_{nH} = \frac{4867.55}{D} kN/m^2$$
, For n-value=100mm/D

$$\Longrightarrow_{\text{q all}=48.7\text{n}}$$
 ....(2.4)

#### 2.6.2 Previous Works

Previous attempts to establish relation between n-value and allowable bearing pressure for the design and construction of shallow foundations includes, Sowers and Hedges (1966), Sanglerat (1972), Cearns and McKenzie (1988) and Ampadu (2005). It must however be noted that, the DCP penetration increment of the Sowers and Hedges(1966) study was only 44 mm instead of the standard 100mm and also the energy per blow per unit area of cone was only 21% of that used in used in this work. Virgin Piedmont soil was used for the research. The Sanglerat (1972) work involved the use of the Dutch formula for the dynamic resistance, R<sub>d</sub>. The obtained R<sub>d</sub> was divided a factor of 20 to obtain the allowable bearing pressure in a cohesionless soils. Based on the specifications of the DCP used in this work, as shown in Table 2.2, Sanglerat (1972) correlates q<sub>all</sub> with n-value as in equation (2.4).

The Cearns and McKenzie (1988) on the other hand, used the Borros penetrometer that had energy per blow per unit area of cone of about double of that used in this study. The Ampadu (2005), work involved establishing a correlation between n-value and allowable bearing capacity. He estimated the allowable bearing capacity from the strength parameters c<sub>u</sub> and φ<sub>u</sub> obtained from triaxial test and then used Terzaghi bearing capacity formula. The n-values of DCP testing were correlated with an allowable bearing stresses obtained. He used sandy to silty *CLAY* material in his work to obtain the correlation equation (2.5);

$$q_{all} = 164n - 504$$
, for  $n > 6$  .....(2.5)

# 2.6.3 Empirical DCP Test and Strength Correlations

Relationship between D-value and other soil parameters have been established by many researchers on empirical basis. Below are some of the correlations.

Relationship between D-value and unconfined compression strength (UCS) in kN/m<sup>3</sup> have been studied in the laboratory for lime stabilised soil and relation below has been established;

$$Log (UCS) = 3.21 - 0.809log D$$

(after McElaveney and Djatnika, 1991)

Many studies have been conducted to formulate correlation between D-value and resilient modulus, M<sub>R</sub>. The results of George and Uddin (2000) shows that;

$$M_R (MN/m^2) = 235.3D^{-0.475}$$
  
 $M_R (MN/m^2) = 532.1 D^{-0.492}$ 

for coarse-grained soil for fine-grained soils

#### Allowable Bearing Pressure - n-value

The relationship between n-value and allowable bearing pressure shown below has been derived by Ampadu (2005) from ultimate bearing capacity calculated using Terzaghi's bearing capacity equation for sandy to silty CLAY materials.

$$q_{all}(kN/m^2) = 164n - 504$$

for n-value(blows/100mm.) > 6

#### DCP and CBR

Correlations between D-value and California bearing ratio (CBR) have been investigated by many researchers using various materials. The Table 2.4 and Table 2.5 show some of the correlations between D-value and California Bearing Ratio and the materials used.

Table 2.4 Some Relationships between D-value and CBR

	Correlation	Material	Investigator	
1	$\log (CBR) = 1.145 - 0.336 \log D$	disturbed soaked CBR	Karunaprema	
		Undisturbed- unsoaked	and Edirisinghe	
2	$\log (CBR) = 1.671 - 0.577 \log D$	clayey or silty sand	(2002)	
3	$\log (CBR) = 2.182 - 0.872 \log D$	disturbed unsoaked CBR	(7 <d<75)< td=""></d<75)<>	
4	$\log (CBR) = 2.56 - 1.16 \log D$	Claylike soils	To be sent to	
5	Log(CBR)=2.81-1.32 log D	Claylike-Well graded Gravels	Harison(1987)	

Table 2.5 Some Relationships between D-value and CBR (after Amini, 2003)

	Correlation	Material	Investigator
1	$\log(CBR) = 2.45 - 1.12 \log D$	Granular and cohesive	Livneh et al. (1992)
2	$\log(CBR) = 2.46 - 1.12 \log D$	Various soil types	Webster et al. (1992)
3	$\log(CBR) = 2.62 - 1.27 \log D$	Unknown	Kleyn (1975)
4	log(CBR) = 2.44 - 1.07 log D	Aggregate base course	Ese et al. (1995)
5	log(CBR) = 2.60 - 1.07 log D	Aggregate base course and cohesive	NCDOT (Pavement, 1998)
6	$\log(CBR) = 2.53 - 1.14 \log D$	Piedmont residual soil	Coonse (1999)

#### 2.6.4 Capabilities of the DCP

The DCP has many capabilities, the following are some of them:

- (i) It is not expensive to acquire and can be manufactured in-house; it is costing about \$1,400.00 from Salem Tool Company, USA as compared to Forager-55 Cable Percussion Rigs cost £14,192.00 (Consallen Group Sales Ltd, 2008). Locally, the DCP is costing about GH¢1,000.00.
- (ii) It is portable and suitable when access and space become a constraint especially in confined areas such as inside buildings to be rehabilitated or

at congested sites that would prevent the use of traditional boring equipment;

- (iii) It is easy to learn how to operate it in matter of minutes;
- (iv) It is a simple device to operate, requiring 3 people for its efficient operation;
- (v) It is fast to conduct, leading to large amounts of data over an area;
- (vi) Its data are easy to process, especially when used with appropriate software;
- (vii) The test results have been correlated to other soil parameters (CBR, unconfined compressive strength, shear strength and SPT N-values) by various researchers;
- (viii) It is cost-effective to operate, especially when compared with other traditional site characterization methods (borings and laboratory/field tests).

#### 2.6.5 Limitations of the DCP

In spite of the many advantages of the DCPT, there are some associated limitations that have been identified. For example:

- (i) It does not permit groundwater conditions to be readily evaluated;
- (ii) No samples are obtained for either visual inspection or further analysis;
   and
- (iii) It is not suitable for gravel soils and hard formations such as highly weathered and fresh rock formations. This is because, the DCP rod may bend.

# 2.7 Bearing Capacity of Shallow Foundations

The ultimate bearing capacity of a soil, quit is classically defined as the value of bearing pressure that will cause a large increase in settlement of the foundation without further increase in bearing pressure, resulting in shear failure. The ultimate bearing capacity of shallow foundations is usually calculated from the Terzaghi bearing capacity equation by incorporating appropriate soil parameters (cohesion, internal friction and unit weight), footing details and foundation depth. The allowable bearing pressure (qall) is the maximum bearing pressures that can safely be applied to a foundation such that, it is safe against instability due to shear failure and maximum tolerable settlement. The allowable bearing pressure is normally calculated from the ultimate bearing capacity by using appropriate factor of safety, F<sub>s</sub>. Typical value of factor of safety for shallow foundation—is 3.0.

# 2.7.1 Factors Affecting Bearing Capacity of Shallow Foundations

A number of factors have been studied and found to affect the bearing capacity of shallow foundations. Some of these factors are;

- i. The unit weight, shear strength and deformation characteristics of the soil.
- ii. The size, shape, depth and roughness of the footing.
- Groundwater level and initial stresses in the foundation soil (Dunn et al., 1980).

## 2.7.2 Terzaghi Ultimate Bearing Capacity Equation

The present Terzaghi formula which is a modified form of Prandtl (1921) plastic failure theory (Radhakrishnan and Ramanathan, 1965) is used for evaluation of the ultimate bearing capacity of shallow foundations. Terzaghi,s theory is based

on the observation that the failure surface in soil at ultimate load takes the form shown in Figure 2.8, thus forming a triangular elastic wedge zone directly under the footing and two bulges delineated by two log spiral curves extending to the ground surface. In the theory, the angles  $\alpha$  between the footing base and the sides of the triangular zone were taken to be equal to the soil internal friction angle,  $\phi$ .

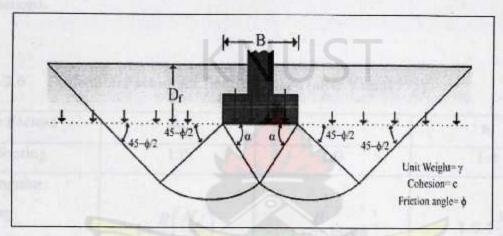


Figure 2.8 General Shear Failure Pattern (after Terzaghi Analysis)

Using the equilibrium analysis, Terzaghi expressed the ultimate bearing capacity for shallow strip foundation in its simplest form as;

$$q_{ult} = cN_c + \gamma D_f N_q + \frac{1}{2} \gamma B N_y$$
 (2.6)  
Where  $N_c$ ,  $N_q$  and  $N_\gamma$  are the Terzaghi bearing capacity factors.

Equation (2.6) is subject to the following conditions:

- general shear failure;
- strip (continuous) footing for which L>5B;
- · rough foundation base; and
- shallow foundation (D<sub>f</sub> < 4B).</li>

On the basis of comparative loading tests with footings of different shapes, equation (2.6) has been modified into equation (2.7) by incorporating correction factors for footing shape configuration (Vesic, 1973).

$$q_{ult} = cN_c s_c + \gamma D_f N_q s_q + \frac{1}{2} \gamma B N_\gamma s_\gamma$$
 ....(2.7)

Where sc, sq and sy are shape factors

Table 2.6 gives the correction factors for footing shape configuration for shallow foundations.

Table 2.6 Correction Factors for Footing Shape (after Vesic1973)

Shape Factors	s <sub>c</sub>	Sq	Sy
Strip footing	1.0	1.0	1.0
Rectangular Footing Limitation $B \le L$	$1 + \frac{B}{L} \left( \frac{N_q}{N_c} \right)$	$1+\frac{B}{L}\tan\phi$	$1-0.4\frac{B}{L}$
Circle and square	$1+\frac{N_q}{N_c}$	1+tan¢	0.60

L and B are length and breadth of the footing.

For foundations that exhibit local shear failure, equation (2.6) is modified into;

$$q_{ult} = 2/3c N_{1c}s_c + q N_1qs_q + \frac{1}{2}\gamma B N_{1\gamma} s_{\gamma}$$
 (2.8)

where the values of  $N_{1c}$ ,  $N_{1q}$  and  $N_{1\gamma}$  are bearing capacity factors obtained after modifying  $\phi$  using equation (2.8).

$$\phi_1 = \tan^{-1} (2/3 \tan \phi)$$
....(2.9)

### 2.7.3 Modes of Foundation Failures

According to experimental results of foundations resting on sands, the mode of failure likely to occur in any situation depends on the size of the footing relative to the foundation depth and the relative density of the soil (Vesic, 1973). It must be stated that while the above stated factors have been observed to be responsible for a particular mode of failure, there is at present no general numerical criteria that can be used to predict a particular failure mode. An attempt by Vesic (1973) to link relative density and depth of foundation relative to footing width to failure mode is shown in Figure 2.9.

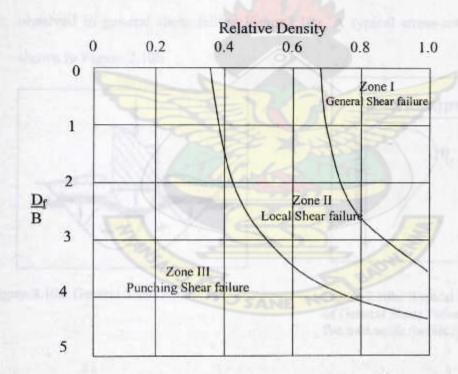


Figure 2.9 Failure Mode Chart (after Vesic, 1973)

Vesic (1973) classified bearing capacity failures into three main categories for shallow foundations depending on how the yield surface developed as explained below.

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#### 2.7.4 General Shear Failure

When a load (Q) is gradually applied to a foundation, settlement occurs which is almost elastic to begin with. At the ultimate load, general shear failure occurs when a plastic yield surface develops under the footing, extending outward and upward to the ground surface and catastrophic settlement and/or rotation of the foundation occurs. The load per unit area at the point of failure is the ultimate bearing capacity, qf for foundation using model footing. This type of failure has been noted to be common in highly incompressible soils of definite shear strength such as very dense sand and hard overconsolidated clays. A heave on the side is always observed in general shear failure Figure 2.10a. A typical stress-settlement graph is shown in Figure 2.10b

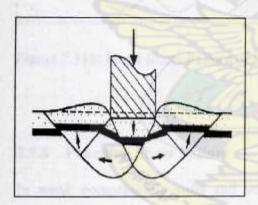


Figure 2.10a: General Shear Failure of Soil

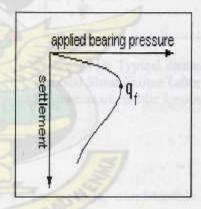


Figure 2.10b: Typical Stress-Settlement Graph of General Shear Failure (after www. fbe.uwe.ac.uk /public /geocal/foundations)

#### 2.7.5 Local Shear Failure

In medium dense sand and clays of medium consistency, significant vertical settlement may take place due to local shear failure. The local shear results from partial development of the state of plastic equilibrium close to the lower edges of the footing. The yield surfaces often do not reach the ground surface and may only slightly heave Figure 2.11a. Several yield developments may occur accompanied by

settlement in a series of jerks. The bearing pressure at which the first yield takes place is referred to as the first-failure pressure,  $q_{f(1)}$  Figure 2.11b-the. The term first-failure load  $Q_{f(1)}$  is also used (Malandraki and Toll, 1996). Local shear is characterised by relatively large settlements and the ultimate bearing capacity is not clearly defined (www. fbe.uwe.ac.uk /public /geocal/foundations), in these cases settlement is the major factor in the foundation design.

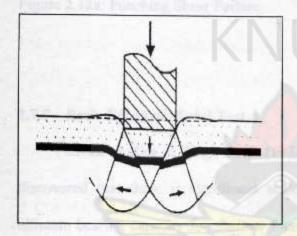


Figure 2.11a: Local Shear Failure of Soil

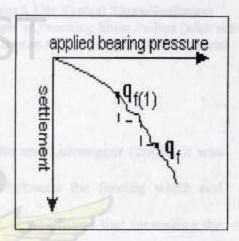


Figure 2.11b: Typical Stress-Settlement Graph of Local Shear Failure (after www. fbe.uwe.ac.uk /public /geocal/foundations)

#### 2.7.6 Punching Shear Failure

In weak compressible soils and very loose to loose soils, considerable vertical settlement may take place with the yield surfaces restricted to vertical planes immediately adjacent to the sides of the foundation; The ground surface may be dragged down, Figure 2.12a. There is no heaving of the ground surface away from the edges and no tilting of the footing occurs. After the first yield has occurred, the stress - settlement curve will become slightly steep, but remain fairly flat Figure 2.12b. This is referred to as a punching shear failure.

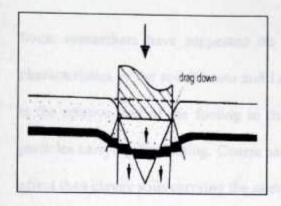


Figure 2.12a: Punching Shear Failure

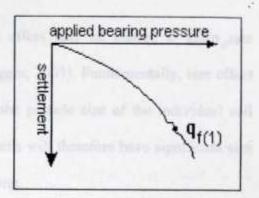


Figure 2.12b: Typical Stress-Settlement Graph of Punching Shear Failure (after www. fbe.uwe.ac.uk /public /geocal/foundations)

#### 2.7.7 Scale Effect in Model Test

In the studies of Singh et al. (1973), Cerato and Lutenegger (2003), it was discovered that there is significant scale effect between the footing width and ultimate bearing capacity for shallow foundations. It was found that increasing the footing width results in higher ultimate bearing stress for the same soil material. Similarly, in the works of De Beer (1965) and Vesic (1973), it was realised that model footings used for bearing capacity studies of width, B less than 150mm gave results that were higher. This was due to the fact that, the magnitude of N<sub>γ</sub> decreased significantly with B until B ≥ 50mm when N<sub>γ</sub> becomes fairly constant as illustrated in Figure 2.13. By this, a condition of lower limit on footing size to use in model footing experiments is suggested.

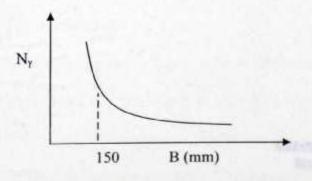


Figure 2.13: Variation of Ny with Footing Width (after Das and Omar 1993)

Some researchers have suggested the scale effect may be related to grain size characteristics of the soil (Cerato and Lutenegger, 2003). Fundamentally, size effect is the relative size of the footing to that of the particle size of the individual soil particles carrying the footing. Coarse sandy soils will therefore have significant size effect than clayey soils carrying the same footing.

Model footing sizes used in previous model footing studies are shown in Table 2.7.

Table 2.7 Some Model Footings Used in Various Studies

Authors	Mould Dimension 1 x b x h (mm³)	Dimension L x B (mm²)	Footing Clearance from Mould Edge	Soil Material Used
Patra et al (2005)	800 x 365 x 700	360 x 80	1.8B	Geogrid Re- inforced SAND
Shin et al (1999)	1000 x175 x 800	175 x 70	0.8B	Crude oil con- terminated SAND
Das & Omar (1994)	1960 x 305 x 914	304.8 x 127	0.7B	SAND
Das & Maji (1994)	610x 610 x 610	76.2 x 76.2	3.5B	Geogrid Re- inforced SAND
Ko & Davidson(1973)	1540x102x 480	100 x 76	0.2B	SAND
This Study	609 x 300 x 376	300 x 150	0.5B	Sandy CLAY

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#### CHAPTER THREE

#### METHODOLOGY

#### 3.1 The Fieldwork

The fieldwork was carried out near the College of Engineering within the site earmarked for KNUST/DFR collaborative pavement engineering studies on KNUST Campus, Kumasi. The work involved manual excavation of a test pit using pick axe and shovel and then recovery of sample from the test pit. Samples were recovered from a depth of 0.30m to 0.80m. Two samples were taken, a bulk sample and then a small sample in an airtight plastic sample bag for the determination of natural moisture content. The test pit was then logged, after which the samples were taken to the laboratory for further works.

## 3.2 The Laboratory Work

## 3.2.1 Sample Preparation

The bulk sample was air dried for about 72 hours to almost constant moisture content. It was then sieved through 19mm aperture sieve to remove cobble materials. Index property and compaction tests were then performed on a representative sample. The remaining sample was bagged into five sacks for storage awaiting the model tests.

## 3.2.2 Sample Characterisation

- The sample for the natural moisture content was taken directly from the test pit in an air tight plastic bag in accordance with BS 1377: Part 2:1990.
- The particle size analysis was performed in accordance with BS 1377: Part
   2:1990 but first, the sample was sieved through a 19mm aperture sieve in order

to remove cobble sized particles. Sedimentation by the hydrometer method was used for the fines.

- Atterberg limits test was performed on the bulk sample according to BS 1377:
   Part 2:1990. The Casagrande method was used for the determination of the liquid limit.
- Compaction characteristics were determined according to BS 1377; Part 4:1990 using 2.5kg rammer (Standard Proctor).

## 3.3 Model Test Set -Up

## 3.3.1 The Perspex Mould and Wooden Footing

The mould consisted of 16mm thick perspex of internal dimensions, 609mm (length) x 300mm (width) x 376mm (height). To minimise lateral yielding during model soil compaction and to ensure plane strain (k<sub>o</sub>) condition during the loading test, the mould was braced on the outside by three (3no.) 5mm thick by 25mm wide steel as shown in Figure 3.1.

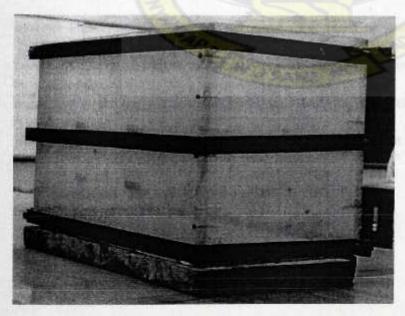


Figure 3.1 The Perspex Mould used for the Footing Model Test

The model footing was made of wooden block of dimensions, 300mm (length) x 150mm (width) and 50mm thick. The base of the model footing was made rough by gluing 60-grade sandpaper onto it as shown in Figure 3.2, to satisfy the rough footing base condition of Terzaghi's bearing capacity formula derivation.



Figure 3.2 Wooden Footing with 60-Grade Sandpaper glued to the Base

## 3.3.2 Loading and Measuring System

The loading system used was strain controlled set to a rate of 0.375mm penetration per minute (resulting in approximately 2hours per loading test). The loading system consisted of a system of gears that push up the loading system. This system raised the whole loading platform against the fixed frame Figure 3.3. The proving ring readings at 0.5mm penetration intervals were taken. Prior to the start of the test, the proving ring was calibrated. The details of the calibration are presented in Appendix III.

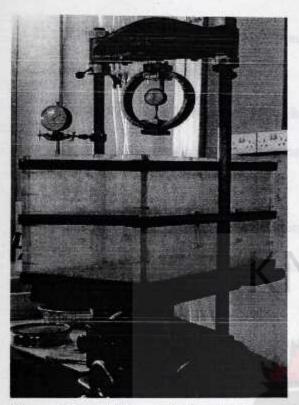


Figure 3.3 Picture of Triaxial Frame Used for loading test with the Mould in it

## 3.4 Model Sample Preparation

 The existing moisture content (EMC) of the sample for each run of the model compaction was determined. The amount of water to be added to achieve optimum moisture content was computed from;

Amount of water added = 
$$\frac{OMC - EMC}{100 + EMC} \times M_s$$

where; OMC-Optimum moisture content,

EMC- Existing moisture content,

Ms- Mass of sample used.

The following procedure was followed for the model compaction;

 About 120 kg of the stored sample was weighed and divided into four parts of about 30kg per part.

- The amount of water required to achieve OMC was added to each part and mixed thoroughly in a large pan.
  - 4. After mixing all the four parts, the mixture was heaped together in a large pan and covered with polythene sheet for about 16 hours for curing in order to equilibrate the moisture content.
- The sample was then compacted in 6-layers in the perspex mould with each layer receiving 20 blows of the rammer, dropping through approximately 0.50m height.
- The rammer consists of 190mm x 190mm square base wooden hammer with 1.36m long handle, weighing 6.06kg.
- 7. Steps 1-6 were repeated for 15, 30, 35, 100 and 150 blows respectively.

Figure 3.4 shows details of the wooden rammer used for the model's compaction.

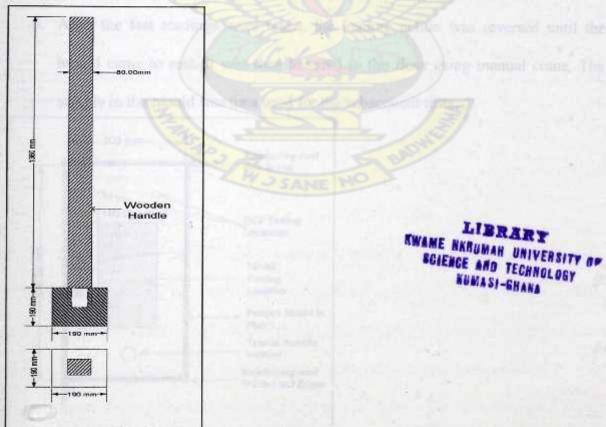


Figure 3.4 Wooden Rammer Used for the Model Compaction in Section and Plan

# 3.5 The Model Testing Procedure

- The compacted sample in the mould was placed in the triaxial frame. The wooden footing was centrally positioned on the compacted sample as shown in Figure 3.5. and Figure 3.6.
- The proving ring was lowered to make contact with the footing and the dial gauge was set fixed to the triaxial frame and making contact with the mould.
- The loading system was then engaged and maintained at a rate of 0.375mm penetration per minute for the loading test until a settlement of at least 25% of the footing width (37.5mm) was attained.
- 4. The proving ring readings were taken at 0.0mm, 0.5mm, 1.0mm, 1.5mm, 2.0mm, until minimum penetration of 37.5mm was achieved. The dial gauge readings for the corresponding settlement were taken and recorded.
- 5. After the last readings were taken, the loading action was reversed until the mould came to rest. It was then lowered to the floor using manual crane. The sample in the mould was then used for the subsequent tests.

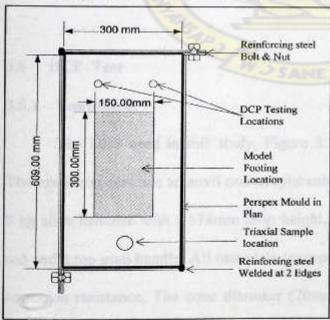


Figure 3.5 Plan of location of tests positions

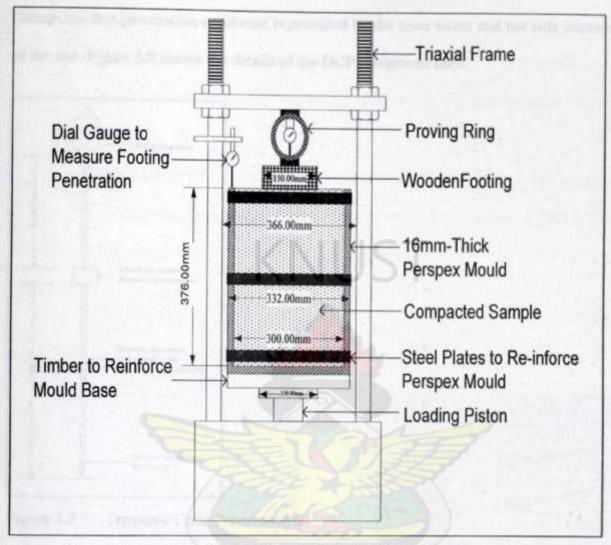


Figure 3.6 The Loading Test Set-Up with Mould filled in Section

### 3.6 DCP Test

## 3.6.1 Equipment

The DCP used in this study, Figure 3.7, consists of two 16mm diameter rods. The lower rod contains an anvil and a replaceable 60° apex cone. The upper rod contains 8 kg slide hammer with a 578mm drop height, an end plug for connection to the lower rod and a top grab handle. All materials (except the drop hammer) are stainless steel for corrosion resistance. The cone diameter (20mm) is made wider than the rod diameter

(16mm) so that penetration resistance is provided by the cone alone and not side friction of the rod. Figure 3.7 shows the details of the DCP equipment used.

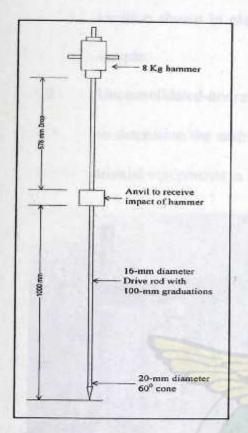


Figure 3.7 Dynamic Cone Penetrometer

#### 3.6.2 Test Procedure

- 1. The DCP set up was held vertically and position at the desired test location.
- 2. The hammer was lifted and dropped from the specified height to initiate the test.
- Following each blow of the hammer, the depth of penetration was measured and recorded as the D-value (mm/blow).
- Steps 2 and 3 were continued until the desired depth of testing was reached.
   After the testing was completed, a special adapted jack was used to extract the device.

#### 3.7 Triaxial Test

- A 38mm x 76mm sampling thin tube was pushed into the sample at a location shown in plan in Figure 3.5 in the mould, to obtain a triaxial test sample.
- 2 Unconsolidated-undrained triaxial tests were conducted on the cored samples to determine the undrained shear strength parameters (c<sub>u</sub> and φ<sub>u</sub>) using the triaxial equipments in Figure 3.8 and Figure 3.9.

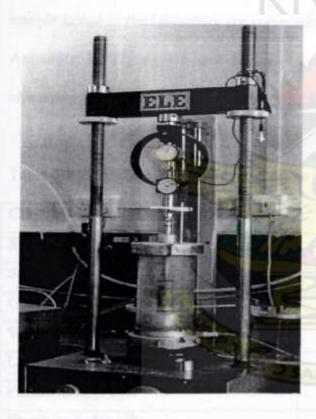


Figure 3.8 Triaxial Test Set Up Used

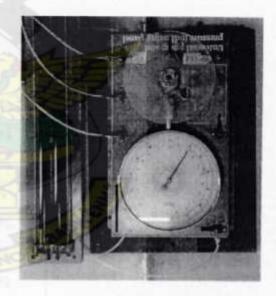


Figure 3.9 Confining Pressure Unit

#### CHAPTER FOUR

# RESULTS ANALYSIS AND DISCUSSIONS

## 4.1 Material Properties

The formation from which the sample was obtained for this investigation was reddish brown, sandy *CLAY* with some silt and muscovite mica. The log of the test pit is presented in Appendix I. The index properties and compaction characteristics of the sample tested in the laboratory are summarised in the Table 4.1 below and the details in Appendix II. Figure 4.1 and Figure 4.2 show the grading curve and compaction characteristics of the test sample respectively.

Table 4.1: Summary of Materials Properties

Clay	46%
Silt	12%
Sand	39%
Gravel	3%
% passing 75μm sieve	58%
Liquid limit (LL)	56%
Plastic limit (PL)	29%
Plasticity Index (PI)	27%
Specific gravity	2.65
Maximum Dry Density (MDD)	
(Standard Proctor)	1.68Mg/m <sup>3</sup>
Optimum Moisture Content(OMC)	19%

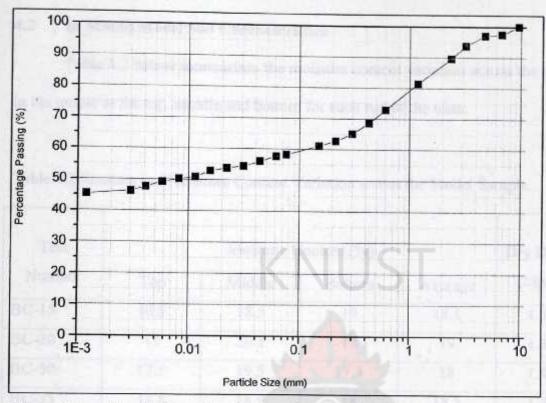


Figure 4.1 Particle Size Distribution Curve of the Test Sample

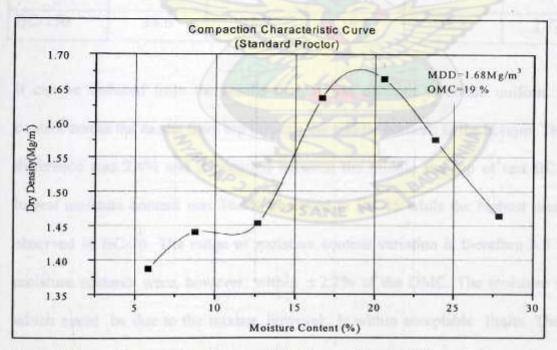


Figure 4.2 Compaction Characteristic Curve of the Test Sample

Based on the unified soil classification system (ASTM D2487-00), the soil is classified as inorganic sandy *CLAY* with high plasticity (CH).

# 4.2 In Mould Model Soil Characteristics

Table 4.2 below summarises the moisture content variation across the model soil in the mould at the top, middle and bottom for each run of the tests.

Table 4.2 Summary of Moisture Content Variation across the Model Sample

Test Number		Moisture Content (%)							
	Тор	Middle	Bottom	Average	(Mg/m³)				
BC-15	16.8	18.5	19	18.1	1.326				
BC-20	19	20.1	18	19	1.345				
BC-30	17.1	19.5	17.3	18	1.451				
BC-35	18.8	18.9	18	18.5	1.500				
BC-100	19.7	19.5	19.1	19.4	1.648				
BC-150	18.6	18.7	19.1	18.8	1.775				

It can be deduced from the results that, it was difficult to obtain uniform moisture content across the model from top through the middle portions to the bottom. The largest difference was 2.4% and it occurred between the middle and top of test BC-30. The lowest moisture content was 16.8% observed in BC-15 while the highest was 20.1 % observed in BC-20. The range of moisture content variation is therefore 3.3 %. The moisture contents were, however, within ±2.2% of the OMC. The moisture variation which could be due to the mixing, however is within acceptable limits. The sample may therefore be said to have been prepared at the OMC. Figure 4.3 shows the mositure content variation across the in mould sample.

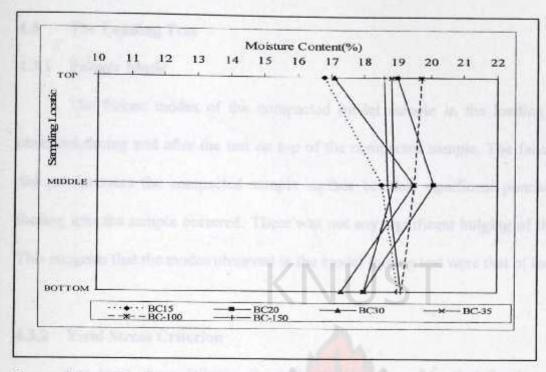


Figure 4.3 Mositure Content variation of the Model Sample

Figure 4.4 shows a plot of the dry density and average moisture content across the sample on the compaction curve. Figure 4.4 shows that, the objective of the model sample preparation for this study, which was to produce samples of different dry densities, was achieved within the range of 1.326 Mg/m<sup>3</sup> to 1.775 Mg/m<sup>3</sup>. These correspond to relative compaction of between 79% and 106%.

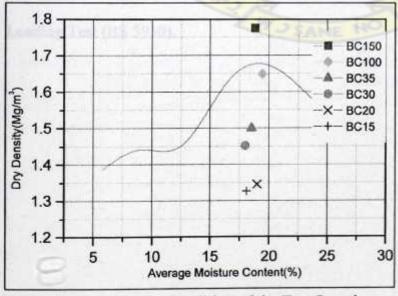


Figure 4.4 Starting Condition of the Test Sample
David Dzitse-Awuku 42

## 4.3 The Loading Test

#### 4.3.1 Failure Mode

The failure modes of the compacted model sample in the loading test were observed during and after the test on top of the compacted sample. The failure surface did not intersect the compacted sample surface besides, significant punching of the footing into the sample occurred. There was not any significant bulging of the surface. This suggests that the modes observed in the model footing test were that of local shear.

## 4.3.2 Yield Stress Criterion

For local shear failures, the ultimate load is not well defined, it therefore becomes difficult to establish the ultimate load. In this study, the concept of yield stress,  $Q_Y$  criterion is used to define the limit of safe contact pressure at which the creeping settlement curve increases considerably with the corresponding yield settlement,  $S_Y$ . This pressure is the intersection point of the two slopes illustrated in the Figure 4.5 below. This pressure is also considered the allowable bearing pressure in this study. This definition is consistent with the definition of ultimate load in the evaluation of Plate Loading Test (BS 5930).

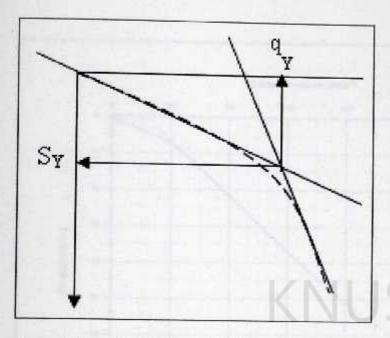


Figure 4.5 Definition of Yield Pressure

## 4.3.3 Model Footing Test Results

Six successful set of tests were conducted. For each set of tests, different compaction energy was used resulting in different dry density. A summary of the load test results is given in Table 4.3. Figure 4.6 to Figure 4.11 show the load-settlement curves for the various model tests with the details given included in Appendix IV.

Table 4.3: Summary of Loading Test Results

Test Number	Dry Density  pdry(Mg/m³)	Average Moisture Content (%)	Yield Bearing stress q <sub>Y</sub> (kN/m <sup>2</sup> )	Yield settlement s <sub>Y</sub> (mm)	Yield Stiffness $V = \frac{q_{\gamma}}{S_{\gamma}}$ $(kN/m^{2}/mm)$
BC-15/1	1.326	18.1	41	7.3	5.6
BC-20/1	1.345	19.0	83	12.6	6.6
BC-30/1	1.451	18.0	178	8.6	20.7
BC-35/1	1.500	18.5	183	8.6	21.3
BC-100/1	1.648	19.4	266	7.1	37.5
BC-150/1	1.775	18.8	331	8.1	40.9

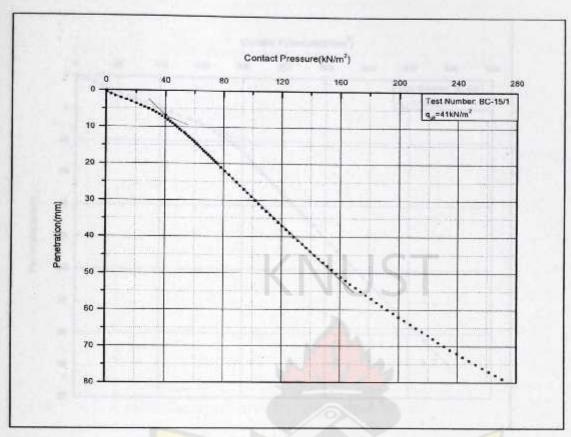


Figure 4.6 Load-Settlement Curve for Experiment BC-15

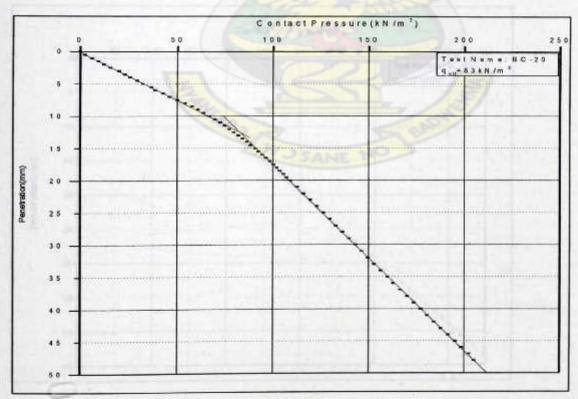


Figure 4.7 Load-Settlement Curve for Experiment BC-20

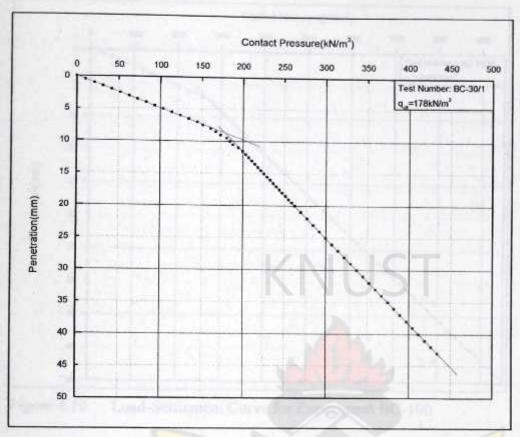


Figure 4.8 Load-Settlement Curve for Experiment BC-30

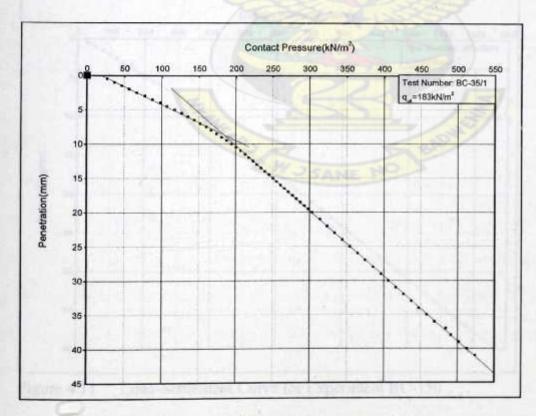


Figure 4.9 Load-Settlement Curve for Experiment BC-35

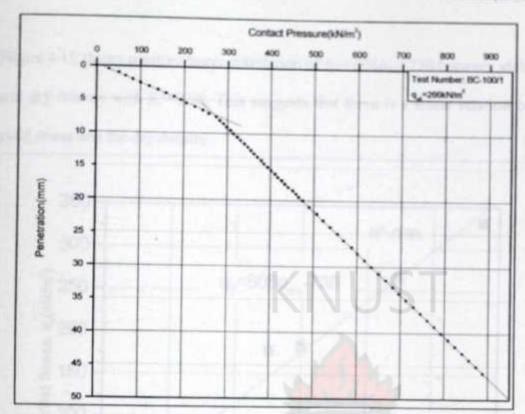


Figure 4.10 Load-Settlement Curve for Experiment BC-100

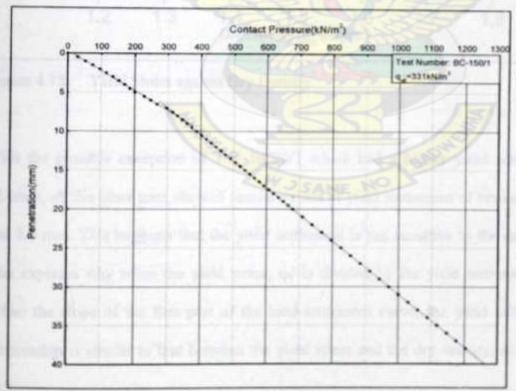


Figure 4.11 Load-Settlement Curve for Experiment BC-150

Figure 4.12 shows positive linear correlation of  $q_Y = 608\rho_{dry}$ -736 between yield stress,  $q_Y$  and dry density with  $R^2$ =0.98. This suggests that there is a linear relation between the yield stress and the dry density.

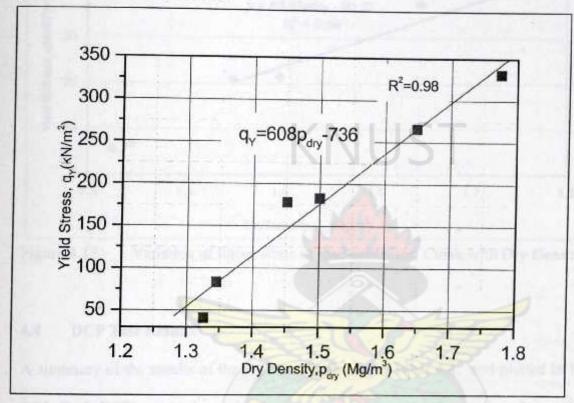


Figure 4.12 Yield Stress against Dry Density

With the possible exception of Test BC20/1 which had a higher yield settlement of 12.6mm, all the other tests showed similar values of yield settlement of between 7.1mm and 8.6 mm. This suggests that the yield settlement is not sensitive to the dry density. This explains why when the yield stress, q<sub>Y</sub> is divided by the yield settlement, S<sub>Y</sub> to define the slope of the first part of the load-settlement curve, the yield stiffness, the relationship is similar to that between the yield stress and the dry density, as in Figure 4.13

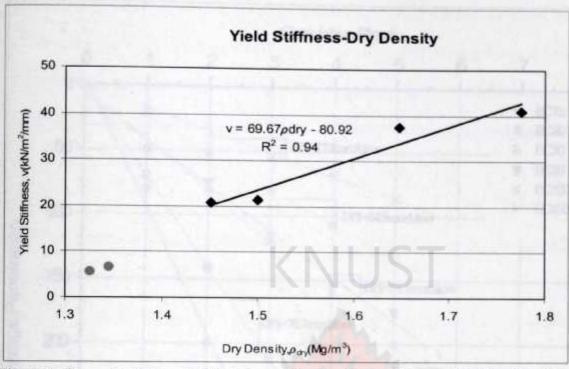


Figure 4.13 Variation of Initial slope of load-settlement Curve with Dry Density

#### 4.4 DCP Test Results

A summary of the results of the DCP test is shown in Table 4.4. and plotted in Figure 4.14. Each DCP penetration value is the average of two readings taken from adjacent test locations. The average D-values also known as the Dynamic Penetration Index (DPI), is the penetration produced by one drop of the sliding hammer and it is obtained as the gradient of the line of best fit of the graph of cumulative blows against penetration in mm as shown Figure 4.14. In field, however, the number of blows required to advance the cone by 100mm into the soil is what is measured and recorded as the n-value. The n-values (blows/100mm) in this study were derived from the D-values by dividing 100mm by the D-values. The details of the DCP test results are presented in Appendix V.

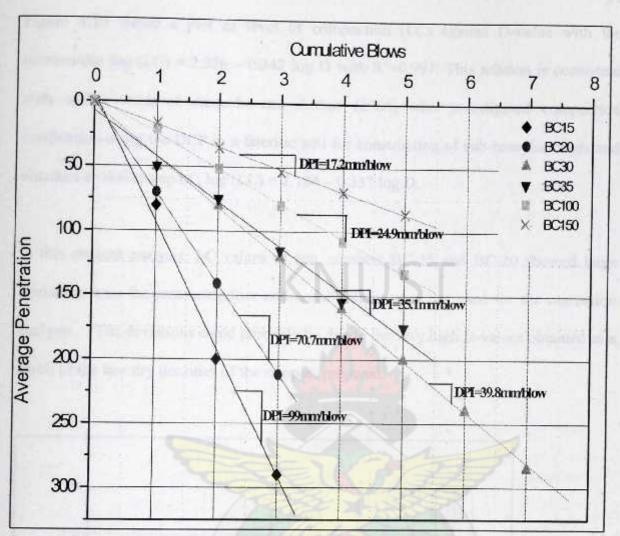


Figure 4.14 Plot of Average DCP Penetration against Cumulative No. of Blows

Table 4.4: Summary of DCP Test Results

Test Number	D-value (mm / blow)	Water Content (%)	Dry Density, $\rho_{dry}(Mg/m^3)$	Level of Compaction, LC (%)	
BC-15	99	18.1	1.326	79	
BC-20 70.7		19	1.345	80	
BC-30	39.8	18	1.451	86	
BC-35	35.1	18.5	1.500	89	
BC-100	24.9	19.4	1.648	98	
BC-150	17.2	18.8	1.775	106	

Figure 4.15 shows a plot of level of compaction (LC) against D-value with the relationship  $\log (LC) = 2.326 - 0.242 \log D$  with  $R^2$ =0.997. This relation is consistent with earlier work of Ampadu and Arthur (2006) who investigated compaction verification using the DCP in a lateritic soil for construction of sub-base for roads and obtained a relationship of ,  $\log (LC) = 2.184 - 0.337 \log D$ .

In this present analysis, LC values of test numbers BC-15 and BC-20 showed large deviation from the remaining four and consequently were not used for the regression analysis. The deviations could probably be due to the very high D-values obtained as a result of the low dry densities of the samples prepared.

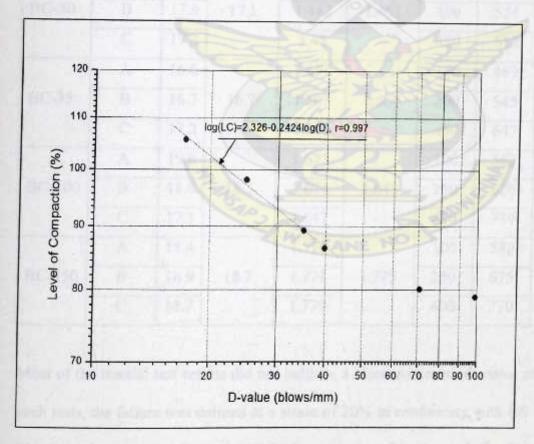


Figure 4.15 Plot of level of compaction (LC) against D-value

# 4.5 Allowable Bearing Pressure from the Triaxial Test

A summary of the triaxial tests results is shown in Table 4.5 below. The detailed triaxial tests data and graphs are shown in Appendix VI.

Table 4.5 Summary of the Triaxial Results

TEST NO.	Sample	Cor	Moisture Content (%)		Dry Density (Mg/m³)		σ <sub>1</sub> -σ <sub>3</sub> (kPa)	C <sub>u</sub> (kN/m <sup>2</sup> )	6
	Number	_	Average		Average				1
	A	16.1	division.	1.325	III. OFFICIAL	150	372	Eron Ze	
BC-15	В	16.5	16.3	1.327	1.326	300 469		126	11
	C	16.2		1.326	ILIC	450	517		
	A	16.2		1.348	00	150	420		
BC-20	В	16.8	16.5	1.342	1.345	300	504	135	13
	С	15.1	15.1	1.367	1.367	450	533		
	A	16	COLY 11	1.461		150	448		
BC-30	В	17.9	17.1	1.442	1.451	300	524	144	13
	С	17.3		1.45		450	618		
	A	16.6		1.497	(F)	150	469	162	
BC-35	В	16.3	16.7	1.503	1.500	300	545	149	13
Nami	С	17.2	1	1.501		450	647	No.	
	A	19.6	1	1.623		100	567	700 21	
BC-100	В	18.8	18.5	1.674	1.648	250	624	199	13
Mary Mary	C	17.1	500	1.647	.00	400	739	100 40	
	A	18.4		1.776	E NO	100	589		
BC-150	В	18.9	18.7	1.771	1.775	250	675	208	13
The same	C	18.7	Land	1.779	NSB .	400	770	001 35	

Most of the triaxial test results did not indicate a clear maximum deviator stress, S<sub>f</sub>. For such tests, the failure was defined at a strain of 20% in conformity with BS 1377 part 7. The triaxial test result of sample BC-20 C was not used in the analysis. This was due to drastic loss of moisture prior to the triaxial test. From the stress – strain graphs of the

triaxial test, the deviator stresses at failure,  $S_f$  and confining pressures were used to plot Mohr circles. From the Mohr circles, the undrained cohesion intercept,  $c_u$  and angle of internal friction  $\phi_u$  were deduced, and used to calculate the allowable bearing pressure.

The stiffness of the samples tested is defined in terms of the E<sub>50</sub> which is the stiffness at half the deviator stress. In order to take into account the effect of confining pressure, the stiffness values have been divided by the mean pressure, p. These values have been plotted against the dry density in Figure 4.16. From the graph, there seems to be no correlation between the two parameters in this work.

Table 4.6 Stress-Strain Analysis of the Triaxial Results

Test	Moisture Content	Dry Density,	σ3	*p	S <sub>f</sub>	€50	E <sub>50</sub>	E <sub>so</sub>
Number	(%)	(Mg/m³)	(kN/m²)	(kN/m²)	(kN/m²)	(%)	(kN/m²)	p
BC-15-150	16.1	1.325	150	274	372	7.5	2,480	9.1
BC-15-300	16.5	1.327	300	456	469	7.4	3,169	6.9
BC-15-450	16.2	1.326	450	622	517	6.8	3,801	6.1
BC-20-150	16.2	1.348	150	290	420	2.9	7,241	25.0
BC-20-300	16.8	1.342	300	468	504	2.3	10,957	23.4
BC-30-150	16.0	1.461	150	299	448	2.5	8,960	29.9
BC-30-300	17.9	1.442	300	475	524	1.9	13,789	29.1
BC-30-450	17.3	1.45	450	656	618	1.1	28,091	42.8
BC-35-150	16.6	1.497	150	306	469	1.8	13,028	42.5
BC-35-300	16.3	1.503	300	482	545	1.7	16,029	33.3
BC-35-450	17.2	1.501	450	666	647	1.4	23,107	34.7
BC-100-100	19.6	1.623	100	289	567	4.5	6,300	21.8
BC-100-250	18.8	1.674	250	458	624	3.1	10,065	22.0
BC-100-400	17.1	1.647	400	646	739	3.1	11,919	18.4
BC-150-100	18.4	1.776	100	296	589	4.3	6,849	23.1
BC-150-250	18.9	1.771	250	475	675	2.3	14,674	30.9
BC-150-400	18.7	1.779	400	657	770	1.7	22,647	34.5

$$*p = \frac{\sigma_1 + 2\sigma_3}{3}$$

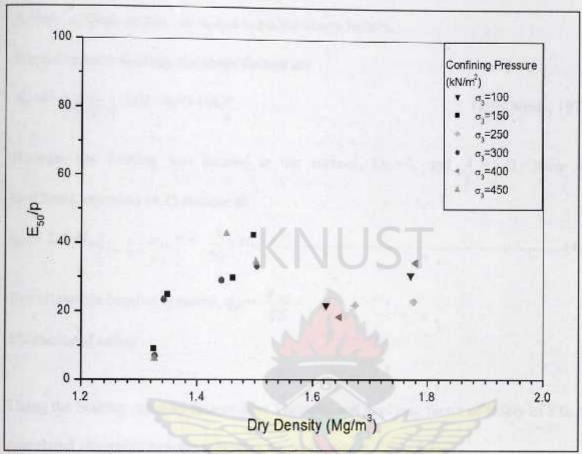


Figure 4.16 Variation of Stiffness with Dry Density

## 4.5.1 Allowable Bearing Pressure

The allowable bearing pressure from the triaxial results was calculated using the Terzaghi bearing capacity equation for local shear failure, taking the footing shape factors into consideration and applying a factor of safety to the ultimate bearing capacity, quit to obtain q<sub>all</sub>.

$$q_{ult} = 2/3c N_{1c}s_c + \gamma D_f N_1 q s_q + \frac{1}{2} \gamma B N_{1\gamma} s_{\gamma}$$
 (4.1)

where the values of  $N_{1c}$ ,  $N_{1q}$  and  $N_{1\gamma}$  are obtained by replacing  $\phi$  by  $\phi_1$  and reading the bearing capacity factors from Terzaghi bearing capacity chart.

Being a local shear failure mode,

 $\varphi_1{=}tan^{\text{-}1}$  (2/3tan  $\varphi)$  and  $-s_c,s_q$  and  $s_\gamma$  are the shape factors.

For rectangular footing, the shape factors are

$$S_c=1+\frac{B}{L}\left(\frac{N_{\gamma}}{N_{c}}\right)$$
 and  $S_{\gamma}=1-0.4\frac{B}{L}$  (after Vesic, 1973)

Because the footing was located at the surface,  $D_f = 0$ , and  $\frac{B}{L} = \frac{1}{2}$  (L=30cm and B=15cm), equation (4.1) reduces to

$$q_{ult} = 2/3cN_{1c} \left( 1 + \frac{1}{2} x \frac{N_{1q}}{N_{1c}} \right) + \frac{3}{50} \gamma N_{1\gamma}$$
 (4.2)

But allowable bearing pressure,  $q_{all} = \frac{q_{ult}}{FS}$ ,

FS-Factor of safety

Using the bearing capacity factors after Terzaghi and applying factor of safety of 3.0, the calculated allowable bearing pressures are as shown in Table 4.7.

Table 4.7 Computation of Allowable Bearing Pressure.

Test Number	C <sub>u</sub>	φ (°)	φ <sub>1</sub> (°)	Niu	N <sub>1q</sub>	Niz	Ybulk kN/m³	q <sub>ult</sub> (kN/m²)	q <sub>all</sub> (kN/m <sup>2</sup> )
BC-15	126	11	7	8	1.8	0.9	15.1	748	249
BC-20	135	13	9	9	2.4	1	15.4	918	306
BC-30	144	13	9	9	2.4	1	16.7	979	326
BC-35	149	13	9	9	2.4	1	17.2	1013	338
BC-100	199	13	9	9	2.4	1	19.2	1353	451
BC-150	208	13	9	9	2.4	1	20.7	1415	472

# 4.6 Correlation between D-value, n-value and Allowable Bearing Pressure, qall

A summary of model yield stress, allowable bearing pressure by Terzaghi approach, the D-value and n-values have been presented in Table 4.8. The model yield stress is take to be the allowable bearing pressure for the model

Table 4.8 Summary of the Allowable Bearing pressures, D-value and n-value.

Test Number		earing Pressure N/m²)	D-value	n-Value (blows/100mm)	
	Model	Terzaghi Approach	(Mm/blow)		
BC-15	41	249	99	1.0	
BC-20	83	306	70.7	1.4	
BC-30	178	326	39.8	2.5	
BC-35	183	338	35.1	2.8	
BC-100	266	451	24.9	4.0	
BC-150	331	472	17.2	5.8	

In order to establish the relevant correlations, the allowable bearing pressures have been plotted against the D-values and n-values in Figure 4.18 and Figure 4.19 respectively.

The regressions for the correlation equations did not include the two data points for the very soft samples. The regression analysis of the data points gave the following correlations;

$$q_{all \text{ (model)}} = 48.3 \text{ n} + 57.3,$$
  $R^2 = 0.988$ 

$$q_{all(model)} = 881-443.9 Log_{10}$$
 (D),  $R^2 = 0.993$ , by model footing, and

$$q_{all(Terzaghi)} = 46.4n + 222$$
 R<sup>2</sup>=0.921,

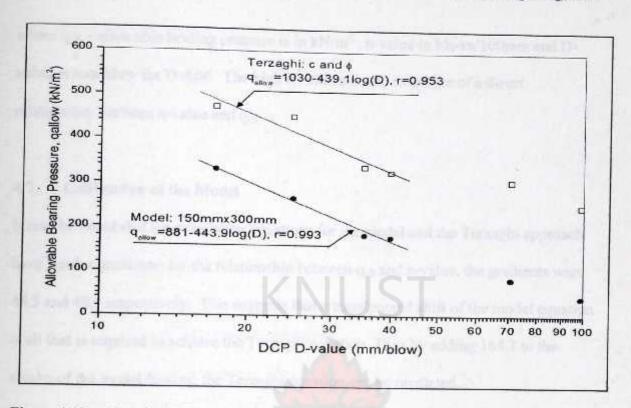


Figure 4.17 Correlation between Allowable Bearing Pressure and D-value for the Model and Terzaghi Approach

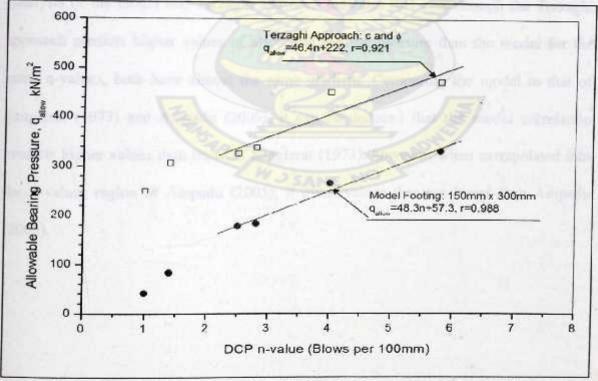


Figure 4.18 Correlation between Allowable Bearing Pressure and n-value for Model and Terzaghi Approach

where  $q_{all}$  – allowable bearing pressure is in kN/m<sup>2</sup>, n-value in blows/100mm and D-value in mm/blow for D  $\leq$  00. The high R<sup>2</sup> indicates the existence of a direct relationship between n-value and  $q_{all}$ .

## 4.7 Calibration of the Model

It may be noted that the correlation equations for the model and the Terzaghi approach have similar gradients; for the relationship between q<sub>all</sub> and n-value, the gradients were 48.3 and 46.4 respectively. This suggests that a translational shift of the model equation is all that is required to achieve the Terzaghi equation. Thus by adding 164.7 to the results of the model footing, the Terzaghi equation can be predicted.

## 4.8 Comparison of the Model to Similar Works

Analysis of the model and Terzaghi approach indicates that even though the Terzaghi approach predicts higher values of allowable bearing pressure than the model for the same n-values, both have almost the same gradient. Comparing the model to that of Sanglerat (1973) and Ampadu (2005), it can be deduced that the model correlation predicts higher values than those of Sanglerat (1973), however, when extrapolated into the n-values region of Ampadu (2005), it gives values that are lower than Ampadu (2005).

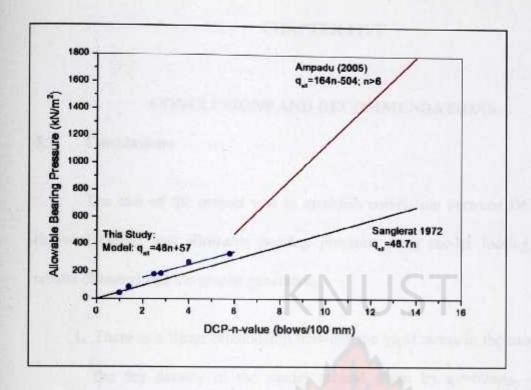


Figure 4.19 Comparisons between Ampadu (2005), Sanglerat (1972), and the Model

#### CHAPTER FIVE

#### CONCLUSIONS AND RECOMMENDATIONS

#### 5.1 Conclusions

The aim of the project was to establish correlation between DCPT n-value (blows/100mm) and allowable bearing pressure using model footing. From the results obtained and the graphs generated,

- 1. There is a linear relationship between the yield stress in the model test and the dry density of the model ground given by  $q_Y=608\rho_{dry}$  736 with  $R^2=0.98$ .
- There is a log log relationship between the level of compaction of the model ground and the DCP D-value given by log (LC) =2.326 0.242 log(D) with R<sup>2</sup> = 0.997.
- The stiffness of the model ground normalized by the mean pressure as measured in the triaxial test appears insensitive to the dry density.
- 4. There is a linear relationship between the allowable bearing pressure and the DCP n-value and this equation has similar gradient for allowable bearing pressure obtained by both the model footing and by the Terzaghi approach.
- 5. The relationship between n-value and allowable bearing pressure for the calibrated model is given by  $\mathbf{q_{all}} = \mathbf{48n} + \mathbf{57}$ , for 2 < n < 6This relationship predicts marginal higher allowable bearing pressure values than that of Sanglerat (1972).

#### 5.2 Recommendations

It is recommended that;

- A simple loading system should be procured for this type of work in the future. The existing loading system makes the work too cumbersome requiring a lot of physical strength.
- Additional works should be done with higher n-values for similar sandy Clay
  material so as to validate the established correlation over wider range.



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## APPENDICES

II	NATURAL MOISTURE CONTENT
	SPECIFIC GRAVITY
	GRADING
	ATTERBERG LIMITS
	COMPACTION TEST
Ш	CALIBRATION CURVES
IV	LOADING TEST
v	DYNAMIC CONE PENETROMETER TEST
VI	UNCONSOLIDATED-UNDRAINED TRIAXIAL TEST
	TRIAXIAL STRESS-STRAIN CURVES

TEST PIT LOG

MOHR CIRCLE

# APPENDIX I

TEST PIT LOG

#### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT -TEST PIT LOG

Site Location: Near Engineering Snack Bar, KNUST

Date: 10/10/2006

Method of Excavation: Manual

Depth	Strata	Strsta Description	Natural Moisture
m	Symbol	VINITATION	Content (%)
	וווו	NAUS I	
	וווו	Loose to medium dense, moist, dark	
	וווו		
	וווו	brown silty SAND. Decayed organic	
	וווו		
0.30	וווו	matter in the matrix (Topsoil).	
1			
7			the barrier
1		Firm, moist, reddish brown, lateritic sandy	20.50
-		Firm, moist, regulsir brown, faterrice sandy	20.50
			19 AL 19
1		CLAY with little silt, rich in	
7			
1		muscovite mica	
			I well
	CHECK STATE OF STREET,	(Residual Soil)	

END OF TEST PIT

## APPENDIX II

NATURAL MOISTURE CONTENT
SPECIFIC GRAVITY
GRADING
ATTERBERG LIMITS
PLASTICITY CHART
COMPACTION TEST

## CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT

NATURAL MOISTURE CONTENT

Date: 10/10/2006

Test Number	1	2
Container Number	B23	A33
Container Mass(g)	3.70	3.62
Container + Wet Sample Mass(g)	28,98	34,00
Container + Oven Dried Sample Mass(g)	24.71	28.78
Moisture Content Mass(g)	4.27	5,22
Dry Sample Mass (g)	21.01	25,16
Moisture Content (%)	20.3	20.7
Average Moisture Content %	2	0.5

Date:17/10/2006

			-	·	~		200	600	Sec. 1	
СD	E	as a	чи	100.0	684	100	CO N		100	v
31			an r	6270		100				

SPECIFIC GRAVIII	A STATE OF	A	В
Sample Number		1	2
Pycnometer Number	10	846.20	843.10
Empty Pycnometer Mass (g)	M1		1322.10
Prenometer + Oven Dried Sample Mass (g)	M2	1329.70	The second second
Pycnometer + Oven Dried Sample + Water Mass (g)	M3	2401.10	2405.60
Pschometer Full of Water Mass (g)	M4	2100,80	2106,60
	M2-M1	483.50	479,00
Oven Dried Sample Mass (g)	M4-M1	1254.60	1263,50
Mass of Water Filling the Empty Pycnometer (g)	M3-M2	1071.40	1083.50
Mass of Water in Pyenometer over and above Dry Sample (g)	(M4-M1)-(M3-M2)	183.20	180,00
Mass of Water having the same Volume as Dry Sample (g)			2.66
Specific Gravity	(M2-M1)	2.64	2,00
specific Gravity	(M4-M1)-(M3-M2)		
		2.	65
Average Specific Gravity			

#### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT PARTICLE SIZE DISTRIBUTION

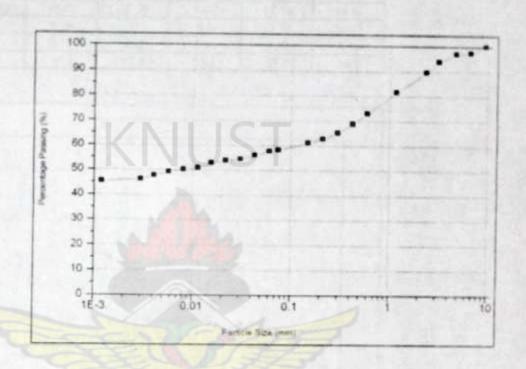
ple Depth: 0.30-0.80m

Date 19/10/2006

#### WET SIEVING

| Dry Mass (g)= 200

-			
1	Mass	26	1/4
	Retain	Retain	Passing
Ł	(g)		
£		0.00	100.00
	3.16	2.58	97.42
1	0.84	0.42	97.00
į.	6.52	3.26	93.74
	8.36	4.18	89.56
,	16.20	8.10	81.46
b	16.88	8.44	73.02
5	8.72	4.36	68,66
Đ.	6.96	3.48	65.18
2	4.76	2.38	62.80
WICKS CHOW	3.28	1.64	61.16
3	5.80	2.90	58.26



#### HYDROMETER TEST

Total Dry	Mass	(g)=	50.0	
-----------	------	------	------	--

Gs= 2.65

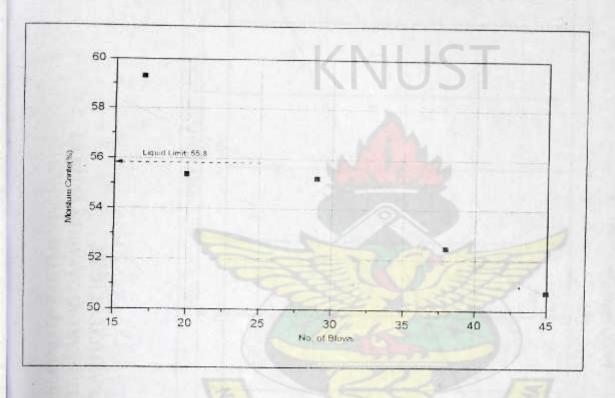
Time	Elap- sed Time, t (min)	Temp. T (°e)	Direct Hydrometer Reading, R <sub>R</sub>	Transformed Hydrometer Reading R <sub>b</sub>	True Hydrometer Rending  R <sub>b</sub> =R <sub>b</sub> +C <sub>m</sub>	Effective Depth, H <sub>i</sub> (mm)	Visco- sity D (mPa s)	Equivalent Drameter, D (mm)	Temp Correc- tion to Health- matter Realing. 3.8	Modified Hydrometer Reading R <sub>2</sub> -R <sub>2</sub> -M <sub>2</sub> -R <sub>3</sub>	Particles D. Kt*a)
9.39am	0.5	29.0	1.0190	19.0	195	123.58	0.809	0.0609	2 0591	17.96	57.69
9:40am	1.0	29.0	1.0185	18.5	19.0	125.55	0.809	0.0434	2.0591	17.46	56.08
9.41am	2.0	29.0	1.0180	180	18.5	127.53	0.809	0.0309	2.0591	16.96	54.47
9.43am	4.0	29.0	1.0178	17.8	18.3	128.32	0.81	0.0219	2.0591	16.76	53.83
9:47am	8.0	29.0	1.0175	17.5	18.0	129.50	0.81	0.0156	2.0591	16.46	52.87
9.34am	15.0	28.5	1.0170	17.0	17.5	131.48	0.82	0.0115	1.9218	15.82	50.82
10:09am	30.0	28.5	1.0168	16.8	17.3	132.27	0.82	0.0082	1.9218	15.62	50.18
10:39am	60.0	28.5	1.0165	16.5	17.0	133.45	0.82	0.0058	1.9218	15.32	49.22
11:39am	120	28.5	1.0160	16.0	16.5	135.43	0.82	0.0041	1.9218	14.82	47.61
1.39pm	240	28.0	1.0157	15.7	16.2	136.61	0.83	0.0030	1.7872	14.39	46.21
9.39pm	1440	29.5	1.0150	15.0	155	139 38	0.80	0.0012	2.1991	14.10	45.29

### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT ATTERBERG LIMITS DETERMINATION

Date:25/10/2006

Liquid Limit

CONTRACTOR OF THE PROPERTY OF					
Container Number	B29	A43	A7	A24	C4
Container Mass (g)	3:75	3.79	3.9	3.71	3.66
Number of Blows	45	38	29	20	17
Wet Sample+ Container Mass(g)	18.76	17.51	19.11	21.31	20.75
Dry Sample + Container Mass(g)	13.71	12.79	13.7	15.04	14.39
Dry Sample Mass(g)	9.96	9	9.8	11.33	10.73
Moisture Content Mass (g)	5.05	4.72	5.41	6.27	6.36
Moisture Content (%)	50,7	52.4	55.2	55.3	59.3



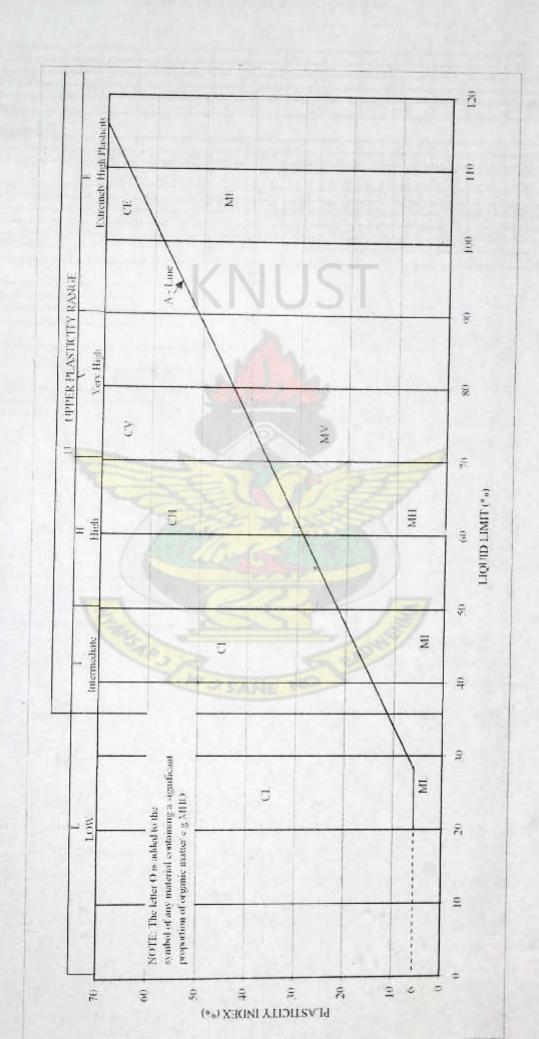
#### Plastic Limit

E3 3.57	101
3.57	
	3-62
9.68	8 94
8.29	7.76
4.72	4.14
1.39	1.18
29.4	28.5
29	.0
	9.68 8.29 4.72 1.39 29.4

Plastic Limit =29%

Liquid Limit (%)	56
Plastic Limit (%)	29
Plasticity Index (%)	27

CIVIL ENGINEERING DEPARTMENT-KNUST
DCPT-ULTIMATE BEARING CAPACITY PROJECT
PLASTICITY CHART



#### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT COMPACTION TEST (STANDARD PROCTOR)

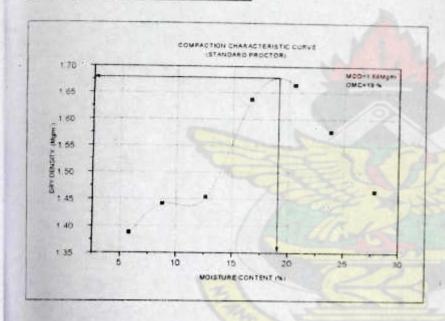
NAME OF THE PREPARATION	1.00	7	1				
Mould Manage	4293	4293	25/1	4201	4203	5	1
Lord - Wet Sample Manage	5073	5771	5817	(4904	0186	6134	4291
Wet Sample Meorgi	1384	1478	194	1801	1893	1841	(4)50
Wet Hulk Density (Mg m)	1.4670	1.665	Links	1908	2004	19812	1.9717

#### Moisture Content Determination

Test Number		1		1		1		1	_					
Tontamer Number	B21	ART	F3	1020	122	1 10	124	1	112	1	-	19		7
Continue Stangs	1.7	2.65	3.43	1.71	2.2	No.	114	W	A42	0.4	.47	.543	234	401
container + Wet Sample Likeway	25.08	34	Sec 184	34.27	1264	31.00	120	30.74	41 20	51.08	52.41	132	3.75	3.61
outnines - Dry Sample Mossey	24.71	32.18	34.29	31.7%	20.24	75 04	33.8	11.4	34 37	42.98	-	17.00	27.01	04.32
Pr. Sample Mannig)	23 01	28.76	30.72	28.60	13.30	74.74	29117	37.66	71.79	50 50	42.65	41.00	45.74	48.57
feasture Content Mass (g)	1.27	167	33.5	* 40	5.1	1.01	-	27.00A	11 -2	30.29	38.93	4117	41 (6)	45.08
Location Content (%)	0.04	4.63	861	2.75	7.4	2774	4.8	4.00	6.42	2.1	9.76	9.68	11.87	1249
Versee Monture Content (%)	10.00	2.09	3.63	N NO.	13.31	11.95	16.51	16.84	20.42	20.62	25107	22.60	28.27	27.45
C. C		3	. 8	3	1.	6	16	1.7	21	0	3 31	1	* *	43
Ore Denome (Mg m')	1.3	57	14	141	1.	65	1.6	(Jei		664	1.5	26	1	144

#### Mould Characteristics

Mould Height (cm)	11.633
Mould Enumeter cm)	1016
Should Volume (cm²)	943.5



## APPENDIX III

PROVING RING CALIBRATION CURVES



## CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT CALIBRATION OF PROVING RING

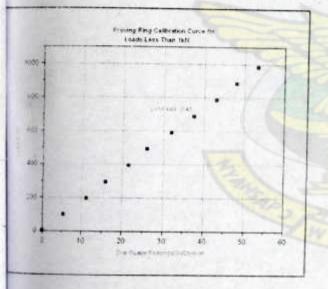
For Loads Less Than IkN

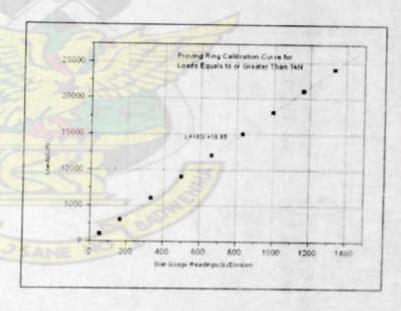
Dial Gua	ge Readings	(G)/Division	
Reading I	Reading 2	Average Reading	Load (L)/N
0.0	0.0	0.0	0.0
5.5	5.5	5.5	98.1
11.0	11.5	11.3	196.2
16.0	16.0	16.0	294.3
21.5	21.5	21.5	392.4
26.0	26.0	26,0	490,5
32,0	32.0	32.0	588,6
37.5	37.5	37.5	686.7
43.0	43.0	43.0	784.8
48.0	48.0	48.0	882.9
53.0	53.5	53.3	981.0

Date: 18/04/2007

For Loads Equals to or	Greater	Than IkN	
------------------------	---------	----------	--

	ings(G)	Guage Readi					
Load (L)/N	Average Reading	Reading 2	Reading 1				
1000	55.5	55.0	56.0				
3000.0	166.0	166.0	166.0				
6000.0	333.5	334.0	333.0				
9000.0	499.0	499.0	499.0				
12000.0	664.5	665.0	664.0				
15000.0	832.5	832.0	833.0				
18000.0	999.0	999.0	999.0				
21000.0	1166,0	1165.0	1167.0				
24000.0	1335.5	1335.0	1336.0				





## APPENDIX IV

LOADING TEST



#### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT LOADING TEST

Te	st No.: BC	-15/1					E
Sr	1 Dial	Provin	2 F	orce	Penetrat	ion	Stre
No	Guage di	v) Ring(di	8) (	N)	(mm)		(kN/a
1	0.0	0.0		0	0.0		0.0
12	50.0	2.0	3	4.5	0.5		0.8
3	100,0	8.0		5.2	1.0		3.2
1	150.0	15,0	manufacture and the	4.2	1.5		6.1
5	200,0	24.0		0.2	2.0		9.8
6	250.0	33.0		6.2	2.5		13
7	300.0	41.0		3.7	3.0		16.
8	350.0	49.0	90	1.2	3.5	-	20.0
9	400.0	57.0		12.3	4.0	-	23
10	450.0	63,0		0.2	4.5	-	25.0
11	500,0	70.0		6.2	5.0	-	28
12	550.0	77.0		2.1	5.5	+	31.2
13	600.0	82.0	149	11400011	6.0		33.2
14	650,0	88.0		0.0	6.5	+	35.0
15	700.0	93.0	168		7.0	+	37.6
16	750.0	98.0	177		7.5	+	39,6
17	800.0	102.5	186	2000	8.0	+	41.4
18	850.0	106,5	193	2.8	8.5	+	43.0
19	900.0	111.0	201	3.7	9.0	+	44.7
20	950.0	115.0	208		9.5	+	46,3
21	1000.0	119.0	215		10.0	+	47.9
22	1050.0	123.0	222		10.5	+	49.5
23	1100.0	126.0	228		11.0	+	50.7
24	1150,0	131.0	237		11.5	+	52.7
25	1200,0	134.0	242	44.5	12.0	+	53,9
26	1250.0	138.0	2490		12.5	+	55.5
27	1300.0	141.0	2553		13.0	+	56.7
28	1350.0	145.0	2625		13.5	+	58.3
29	1400.0	148.5	2688		14.0	+	59.7
30	1450,0	152.0	2751		14.5		61.1
31	1500.0	155.0	2805		15.0	+	62.3
32	1550.0	158.5	2868		15.5		63.7
33	1600.0	162.0	2931	3	16.0		65.1
34	1650.0	165.0	2985	2	16.5		66.3
35	1700.0	168.5	3048		17.0		67.7
36	1750.0	172.0	3111		17.5	-	69.1
37	1800.0	175.5	3174		18.0		70.5
38	1850.0	179.0	3237		18.5		71.9
39	1900,0	182.0	3291.		19.0		73.1
40	1950.0	185.0	3345		19.5		74.3
41	2000.0	188.5	3408.	11.1	20.0		75.7
42	2100.0	194.5	3515.		21.0		78.1
43	2200.0	201.5	3641.	12.1	22.0		30.9
44	2300.0	208.0	3758.		23.0		3.5
45	2400.0	214.5	3875.		24.0		6.1
46	2500,0	221.0	3992.		25.0	0	8.7
47	2600,0	227,0	4100,	_	26.0		
18	2700.0		-	-		-	1.1
49		234.0	4226.	_	27.0	-	3.9
_	2800.0	240.0	4334.:	-	28,0	_	6.3
50	2900.0	246.5	4451	-	29,0	9	8.9
51	3000,0	253.0	4568.3		30.0	10	1.5
52	3100.0		4676	-	31.0	-	13.9
53	3200.0		4784.2	_	32.0	-	6.3
54.	3300.0			_		_	
55	The second second	THE RESERVE OF THE PARTY OF THE	4919.1	-	33.0	_	19,3
	3400.0	279.0	5036.1	1	34.0	1.1	1.9

	VI.		The state	Date: 23/0	4/2007
	rl Dial	Proving	Force	Penetration	Stress
N	1000 000 000		INI	(mm)	
5			5153.0	35.0	114.5
	7 3600,0	292.5	5278.9	36.0	117.3
5		299.0	5395.9		119 9
59	The second secon	306,0	5521.8	38.0	122.7
-00		312,5	5638.7	39.0	125.3
6.		319.0	5755.7	40.0	127.9
6.		326.5	5890.6	41.0	130.9
63		334.0	6025.5	42.0	133.9
6-	The second secon	341.0	6151.4	43.0	136.7
6.5		348.0	6277.4	44.0	139.5
66		355.5	6412.3	45.0	142.5
67		362.5	6538.2	46.0	145.3
68		370.0	6673.2	47.0	148.3
69	7.57 (5.57)	377.5	6808.1	48.0	151.3
70	The second second second	385.0	6943.0	49.0	154.3
71	5000.0	393.0	7086.9	50.0	157.5
72	5100.0	400.0	7212.9	51.0	160.3
73		408.5	7365.8	52.0	163.7
74	The state of the s	416.0	7500,7	53.0	166.7
75	5400.0	425,5	7671.6	54.0	170.5
76	5500.0	434.0	7824.5	55,0	173.9
77	5600.0	443.0	7986.4	56,0	177.5
78	5,700.0	452.0	8148.3	57.0	181.1
79	5800.0	461.0	8310.2	58.0	184.7
80	5900.0	470,0	8472,2	59.0	188.3
81	6009.0	479.0	8634.1	60,0	191.9
82	6100.0	489.0	8814.0	61.0	195.9
83	6200.0	498.0	8975.9	62.0	199.5
84	6300.0	508.0	9155.8	63.0	203.5
85	6400.0	518.0	9335.7	64.0	207.5
86	6500.0	528,0	9515.6	65.0	211.5
87	6600,0	538.0	9695.5	66.0	215.5
88	6700,0	548.0	9875.4	67.0	219.5
89	6800.0	556.5	10028.3	68.0	222.9
90	6900.0	566.0	10199.2	69,0	226,6
91	7000.0	576.0	10379.1	70.0	230.6
92	7100.0	586.0	10559.0	71.0	234.6
93	7200.0	597.5	10765,9		239.2
94	7300.0	608.0	10954.8	73,0	243,4
95	7400.0	619,0	11152.7		247.8
96	7500,0	630.0	11350.6	75.0	52.2
9.7	7600.0	641.5	11557,4		56.8
98	7700.0	652.5	11755.3		61.2
99	7800.0	664.0	11962.2		65.8
100	7900,0	675.0	12160.1	79.0 2	70.2

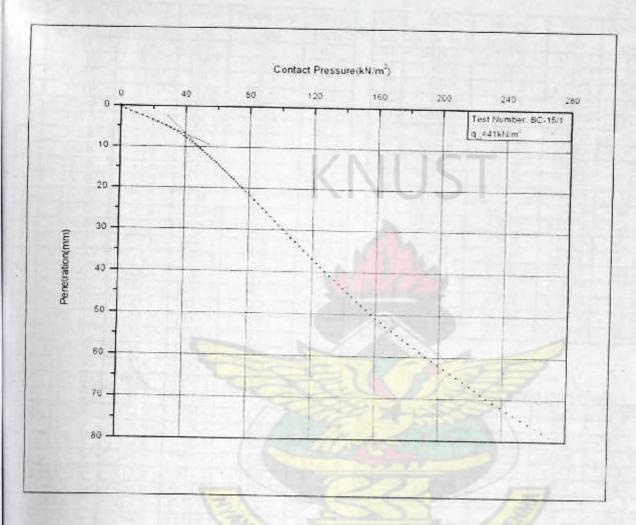
#### Moisture Content Determination

Sample Location	Тор	Middle	Bottom
Constiner Number	X4	D9	C6
Container Mass (g)	18.17	18.1	17.8
Container + Wet Sample Mass(g)	122.11	102.5	112.2
Container + Dry Sample Mass (g)	107.14	89.3	97.2
Moisture Content (g)	14.97	13.14	15.09
Dry Sample Mass (g)	88.97	71.18	79,32
Moisture Content (%)	16.8	18.5	19.0
Average Moisture Contenting	) =	18.1	9100000

### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT LOADING TEST GRAPH

Test No.: BC-15/1

Date: 23/04/2007



RWAME RKRUMAH URIVERSITY OF SCIENCE AND TECHNOLOGY KUMASI-GPC.

## CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT LOADING TEST

Test No.: BC-20/1

Date: 19/04/2007

100	il Dial		Prov	ıng	Fo	rce	Penetra	non	Stres
-	lo: Guage(e	liv)	Ring	div)	(1)	()	timo		(kN/m
	0,0		0,0		0.	()	0.0		0.0
	2 50.0		6.0		105		0.5		2.4
1	100,0		15.		27-	1.2	1.0		6.1
	The second secon		23.		421		1.5		9.4
			30,		550		2.0		12.2
7	W. C		38.		698		2.5		15.5
8			49.0		901	.2	3.0		20.0
0			56.0		102-		3.5		22.8
10	700,0	-	63.0		1150		4.0		25.6
11		-	72.0	)	1312		4.5		29.2
12		-	79.0		1438		5,0		32.0
13			90.0		1636		5.5		36.4
14	The state of the s	-	95.0	10 mm	1725		6,0		38.4
15		-	105,0	1.7	1905		6.5		42.4
16		+	114.(		2067		7.0		45.9
17	800.0	-	124.0		2247		7.5		49.9
18	850.0	-	134.0		2427		8.0	. 0	53.9
19			143,0	0.00	2589		8.5		57.5
20	950.0		151.0	_	2733,		9.0		60.7
21	1000.0		158.0	_	2859.		9.5		63.5
22	1050.0		166.0		3003.		10.0		66.7
23	1100.0		73.0	-	3129.		10.5		69.5
24	1150.0		80.0		3255.		11.0		72.3
25	1200.0		86.0		363,0		11.5		74.7
26	1250.0		92.0		470.		12.0	1	77.1
27	1300.0		98.0	-13	578.	1	12.5		79.5
28	1350.0		04.0		686.8		13.0		81.9
29	1400.0		15.0	_	785.8		13.5		84.1
30	1450.0		20.0		884.7		14.0		86,3
31	1500.0	1 3	25.5		974.7		14.5		88.3
32	1550.0	1 3	30.0		073.6		15.0		90.5
33	1600.0		35.0		154.6		15,5		92.3
34	1650.0		0.0		244.5		16.0	the second second	14.3
35	1700.0		5.0		334.5 124.4	-	16.5	_	06.3
36	1750.0		9.5		05.4	-	17.0		28.3
37	1800.0		4.0	_	86.3	-	17.5		00.1
38	1850.0		8.0		58.3	-	18.0		01.9
39	1900.0	1000	3.0	_	48.2		18,5	10	03.5
40	1950.0		7.0		20.2	_	19.0		15.5
41	2000.0		2.0	-	ACCUPATION AND ADDRESS OF		19.5		)7.1
42	2100.0		).5		10.1		20.0	A comment of the	19,1
43	2200.0		0.0	-	63.0	_	21.0		2.5
44	2300.0	298		-	16.0		22.0	the state of the s	5.9
45	2400.0	300			77.9		23.0		9.5
46	2500.0			Address of the last	30.8		24.0		2.9
47	2600.0	315		-	83.7		25.0		6.3
18	2700.0	323		-	6.6		26.0	_	9.7
49	-	331	-	-	30,5		27.0	13,	2.9
50	2800.0	340		613	3.5		28.0	130	6.3
-11	2900.0	348	5	628	6.4		29.0	134	CONTRACT AND ADDRESS

Srl	Dral	Proving	Force	Penetration	Stress
No.	Guage(div)	Ring(div)	(N)	(mm)	(kN/m2)
51	3000.0	357.0	6439.3	30.0	143.1
52	3100,0	365.0	6583.2	31.0	146.3
53	3200,0	373.0	6727.1	32.0	149.5
54	3300.0	381.5	6880.0	33.0	152.9
55	3400.0	390.0	7033.0	34.0	156.3
56	3500.0	398.5	7185.9	35.0	159.7
57	3600,0	406.0	7320.8	36.0	162.7
58	3700.0	415,0	7482.7	37.0	166.3
50	3800.0	424.0	7644.6	38.0	169.9
60	3900,0	433.0	7806.5	39,0	173.5
61	4000.0	442.0	7968.4	40.0	177.1
62	4100.0	450.0	8112.4	41.0	180 3
63	4200.0	459.0	8274.3	42.0	183.9
64	4300.0	468.0	8436.2	43.0	187.5
65	4400.0	478.0	8616.1	44.0	191.5
00	4500.0	487.0	8778.0	45.0	A. 7. 10 . 10.
57	4600,0	495.0	8921.9	46.0	195.1
8	4700.0	505.0	9101.8		198,3
9	4800.0	512.0	9227.7	710	202.3

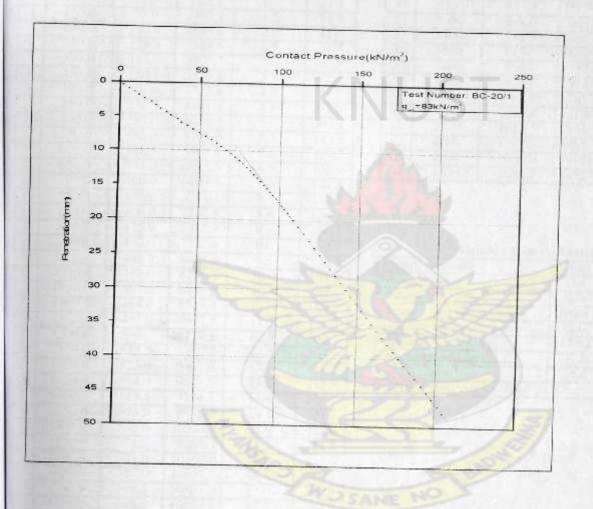
#### Moisture Content Determination

Тор	Middle	Bottom
X4		C6
18.2	100000	17.8
126.4	103.5	105.2
109.1	893	919
		13.3
90.9	1.0	74 1
19:0	-	18.0
	X4 18.2 126.4 109.1 17.3	X4   D9   18.2   18.1   126.4   103.5   109.1   89.3   17.3   14.3   90.9   71.1

### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT LOADING TEST GRAPH

Test No.: BC-20/1

Date: 19/04/2007



## CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT LOADING TEST

Test No.: BC-30/1

Date: 26/04/2007

1 2 3 4 5 6 7 8 9 10 11 12 13 14 15 16 17 18 19 20 21 22 23 24 25 26	Guage(div) 0.0 50.0 100.0 150.0 200.0 250.0 300.0 350.0 400.0 450.0 500.0 650.0 700.0 750.0 800.0	0.0 25.5 51.5 77.0 103.0 128.5 155.0 206.0 231.5 257.0 283.0 308.5 334.0	0 467.9 947.3 1402.1 1869.8 2328.6 2805.3 3255.1 3722.8 4181.5 4640.3 5108.0	(mm) 0,0 0,5 1,0 1,5 2,0 2,5 3,0 3,5 4,0 4,5	0.0 10.4 21.1 31.2 41.6 51.7 62.3 72.3 82.7 92.9
2 3 4 5 6 7 8 9 10 11 12 13 14 15 16 17 18 19 20 21 22 23 24 25 26	50.0 100.0 150.0 200.0 250.0 300.0 350.0 400.0 450.0 500.0 600.0 650.0 700.0	25.5 51.5 77.0 103.0 128.5 155.0 180.0 206.0 231.5 257.0 283.0 308.5 334.0	467.9 947.3 1402.1 1869.8 2328.6 2805.3 3255.1 3722.8 4181.5 4640.3 5108.0	0.0 0.5 1.0 1.5 2.0 2.5 3.0 3.5 4.0 4.5	0.0 10.4 21.1 31.2 41.6 51.7 62.3 72.3 82.7
3 4 5 6 7 8 9 10 11 12 13 14 15 16 17 18 19 20 21 22 23 24 25 26	100.0 150.0 200.0 250.0 300.0 350.0 400.0 450.0 500.0 650.0 700.0 750.0	51.5 77.0 103.0 128.5 155.0 180.0 206.0 231.5 257.0 283.0 308.5 334.0	947.3 1402.1 1869.8 2328.6 2805.3 3255.1 3722.8 4181.5 4640.3 5108.0	1,0 1,5 2,0 2,5 3,0 3,5 4,0 4,5	21.1 31.2 41.6 51.7 62.3 72.3 82.7
4 5 6 7 8 9 10 11 12 13 14 15 16 17 18 19 20 21 22 23 24 25 26	150.0 200.0 250.0 300.0 350.0 400.0 450.0 500.0 650.0 700.0 750.0	77.0 103.0 128.5 155.0 180.0 206.0 231.5 257.0 283.0 308.5 334.0	1402.1 1869.8 2328.6 2805.3 3255.1 3722.8 4181.5 4640.3 5108.0	1.5 2.0 2.5 3.0 3.5 4.0 4.5	21.1 31.2 41.6 51.7 62.3 72.3 82.7
5 6 7 8 9 10 11 12 13 14 15 16 17 18 19 20 21 22 23 24 25 26	200.0 250.0 300.0 350.0 400.0 450.0 500.0 650.0 700.0 750.0	103.0 128.5 155.0 180.0 206.0 231.5 257.0 283.0 308.5 334.0	1869.8 2328.6 2805.3 3255.1 3722.8 4181.5 4640.3 5108.0	1.5 2.0 2.5 3.0 3.5 4.0 4.5	31.2 41.6 51.7 62.3 72.3 82.7
6 7 8 9 10 11 12 13 14 15 16 17 18 19 20 21 22 23 24 25 26	250.0 300.0 350.0 400.0 450.0 500.0 550.0 650.0 700.0 750.0	128.5 155.0 180.0 206.0 231.5 257.0 283.0 308.5 334.0	2328.6 2805.3 3255.1 3722.8 4181.5 4640.3 5108.0	2.0 2.5 3.0 3.5 4.0 4.5	41.6 51.7 62.3 72.3 82.7
7 8 9 10 11 12 13 14 15 16 17 18 19 20 21 22 23 24 25 26	300.0 350.0 400.0 450.0 500.0 550.0 600.0 650.0 700.0	155.0 180.0 206.0 231.5 257.0 283.0 308.5 334.0	2805.3 3255.1 3722.8 4181.5 4640.3 5108.0	2.5 3.0 3.5 4.0 4.5	51.7 62.3 72.3 82.7
8 9 10 11 12 13 14 15 16 17 18 19 20 21 22 23 24 25 26	350.0 400.0 450.0 500.0 550.0 600.0 650.0 700.0 750.0	180,0 206,0 231,5 257,0 283,0 308,5 334,0	3255.1 3722.8 4181.5 4640.3 5108.0	3.0 3.5 4.0 4.5	62.3 72.3 82.7
9 10 11 12 13 14 15 16 17 18 19 20 21 22 23 24 25 26	400.0 450.0 500.0 550.0 600.0 650.0 700.0 750.0	206,0 231,5 257,0 283,0 308,5 334,0	3722.8 4181.5 4640.3 5108.0	3.5 4.0 4.5	72.3 82.7
10 11 12 13 14 15 16 17 18 19 20 21 22 23 24 25 26	450.0 500.0 550.0 600.0 650.0 700.0 750.0	231,5 257,0 283,0 308,5 334,0	4181.5 4640.3 5108.0	4.5	82.7
11 12 13 14 15 16 17 18 19 20 21 22 23 24 25 26	500.0 550.0 600.0 650.0 700.0 750.0	257,0 283,0 308,5 334,0	4181.5 4640.3 5108.0	4.5	Annual Control of the
12 13 14 15 16 17 18 19 20 21 22 23 24 25 26	550.0 600.0 650.0 700.0 750.0	283.0 308.5 334.0	4640.3 5108.0		A CONTRACTOR OF THE PARTY OF TH
13 14 15 16 17 18 19 20 21 22 23 24 25 26	600.0 650.0 700.0 750.0	308.5 334.0	5108.0	5.0	103.1
14 15 16 17 48 19 20 21 22 23 24 25 26	700.0 750.0	308.5 334.0		5.5	113.5
15 16 17 18 19 20 21 22 23 24 25 26	700.0 750.0	334,0	5566.8	6.0	123,7
16 17 18 19 20 21 22 23 24 25 26	750.0		6025.5	6,5	133.9
17 18 19 20 21 22 23 24 25 26		360.0	6493.3	7.0	144.3
18 19 20 21 22 23 24 25 26		376.0	6781.1	7.5	150,7
19 20 21 22 23 24 25 26		400.0	7212.9	8,0	160,3
19 20 21 22 23 24 25 26	850.0	415.0	7482.7	8.5	166,3
20 21 22 23 24 25 26	900.0	430.0	7752.6	9.0	
21 22 23 24 25 26	950,0	448.0	8076.4	9.5	172.3
22 23 24 25 26	1000.0	458,0	8256.3	10.0	
23 24 25 26	1050.0	467.0	8418.2	10.0	183.5
24 25 26	1100.0	484.0	8724.0	11.0	187.1
25 26	1150.0	497.0	8957.9	11.5	193,9
26	1200.0	506.0	9119.8	12.0	199.1
	1250.0	515.5	9290.7		202 7
	1300.0	525.0	9461.6	12.5	206.5
	1350.0	534.0	9623.5	13.0	210.3
10000	1400.0	543.5	9794.4	13.5	213,9
100-100-0	1450.0	553.0	9965.3	14.0	217.7
SEPTIME SHOW	1500.0	562.0		14.5	221.5
ACCOUNTS OF THE PARTY OF	1550.0	571.5	10127.2 10298.1	15.0	225.0
	1600.0		Charles and the Control of the Contr	15,5	228.8
the same of the same of	1650.0	581.0 590.0	10469.0	16.0	232.6
	1700.0	599.5	10631.0	16.5	236.2
	1750.0	609.0	10801.9	17.0	240.0
	1800.0	A CONTRACTOR OF THE PARTY OF TH	10972.8	17.5	243,8
		618.0	11134.7	18.0	247.4
	1850.0	627.5	11305.6		251.2
All the second	1900.0	637.0	11476.5		255.0
1000	1950.0	646.0	11638.4		258,6
manufacture of the second	2000.0	655.5	11809.3	20.0	262.4
	2100.0	674.0	12142.1		269.8
43 2	2200.0	693.0	12483.9		277.4
	2300.0	711.5	12816.7	23.0	284.8
	2400.0	730.0	13149.6		292.2
	2500.0	749,0	13491.4		299.8
	2600.0	767.5	13824.2		307.2
	2700.0	786.0	14157.0		314.6
49 2	Annual Control of the last	804.5	14489.8		22.0
50 2		The second second	14831.6	# O. W	44.1/

Sil	Diaf	Proving	Force	Penetration	Stress
No	Guagerdiyi	Ring(div)	(N)	(mm)	(kNom
51	3000	842.0	15164.4	30.00	337 ()
52	3100	860.5	15497.2	31.00	344.4
53	3200	879.5	15839.1	32,00	352.0
54	3300	898.0	16171.9	33.00	359.4
55	3400	916.5	16504.7	34,00	366.8
56	3500	935.5	16846.5	35.00	374.4
57	3600	954.0	17179.3	36.00	381.8
58	3.700	972.5	17512.1	37.00	389.3
59	3800	991.5	17853.9	38.00	396.8
60	3900	1010.0	18186.8	39.00	404.2
61	4000	1028.5	18519.6	40.00	411.5
62	4100	1047.0	18852.4	41.00	418.9
63	4200	1066.0	19194.2	42.00	426.5
64	4300	1084.5	19527.0	43,00	433.9

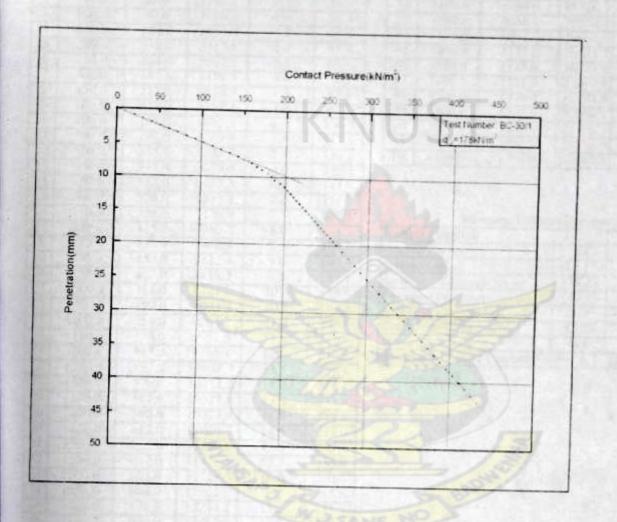
Moisture Content Determination

Sample Location	Top	Middle	Botton
Constiner Number	X4	D9	C6
Container Mass (g)	18.17	18:13	17.8
Container + Wet Sample Mass(g)	107,67	108.52	119.1
Container - Dry Sample Mass (g)	94.63	93.77	104.2
Moisture Content (g)	13.04	14.75	14.95
Dry Sample Mass (g)	76.46	75,64	86.32
Moisture Content (%)	17.1	19.5	173
Average Moisture Contenti"	<sub>i</sub> ) =	18.0	

### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT LOADING TEST GRAPH

Test No.: BC-30/1

Date: 26/04/20



### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT LOADING TEST

58

59

(91)

61

62

3800.0

3900.0

+000,0

4100.0

Test No.: BC-35/1

Stress Force Penetration Proving Dial Sil (mm) (kN/m) (N) Ring(div) Guage div No. 406.1 30.0 1015.0 18276. 3000.0 51 416.1 31.0 18726.5 1040.0 3100.0 52 426.5 32.0 19194.2 1066.0 3200.0 53 436.9 33.0 19661.9 1092.0 3300.0 54 446.9 34.0 20111.7 1117.0 3400.0 55 458.1 35.0 20615.4 1145.0 3500.0 56 468.5 36.0 21083.1 1171.0 3600.0 57 484.1 21784.8 1210.0 3700.0

22090.6

22576.3

23080.0

23565.8

Date: 1/05/2007

38.0

39.0

40.0

41.0

490.9

501.7

512.9

523.7

Srl	Dial	Proving	Force	Penetration	Stress
(888)	Guage(div)	Ring(div)	(N)	(mm)	(kN/m <sup>*</sup> )
Ve.	0.0	0.0	0	0.0	(),()
1	50.0	65.0	1186.2	0.5	26,4
2	100.0	90.0	1636.0	1.0	36.4
3	The second section is a second	114.0	2067.7	1.5	45.9
4	150.0	139.0	2517.5	2.0	55.9
5	200.0	164.0	2967.2	2.5	65.9
6	250.0	192.0	3470.9	3.0	77.1
7	300.0	217.0	3920.7	3.5	87.1
8	350.0	243.0	4388.4	4.0	97.5
9	400.0		4802.2	4.5	106.7
10	450.0	266.0	5269.9	5.0	117.1
11	500.0	292.0	5683.7	5.5	126.3
12	550.0	315.0	6079.5	6.0	135.1
13	600.0	337.0	6457.3	6.5	143.5
14	650.0	358.0	0437.3	7.0	151.5
15	700.0	378.0	6817.1	7.5	159.5
16	750,0	398.0	7176.9	8.0	166.3
17	800.0	415.0	7482.7	8.5	173.5
18	850.0	433.0	7806.5	9.0	180.3
19	The second secon	450.0	8112.4	9.5	187.1
20		467.0	8418.2		193.9
21	1000.0	484.0	8724.0	10.0	199.9
22	1,1000000000000000000000000000000000000	499.0	8993.9	10.5	205.9
23		514.0	9263.7	11.0	212.3
24		530.0	955L6	11.5	217:5
25		543.0	9785.4	12.0	
26		557.0	10037.3	12.5	223.1
27		570.0	10271.2	13.0	228.2
28		584.0	10523.0	13,5	233.8
21		598.0	10774.9	14.0	239.4
100		612.0	11026,7	14.5	245.0
30		625.0	11260.6	15.0	250.2
3		637.0	11476.5	15.5	255.0
3.	The state of the s	651.0	11728.3	16.0	260.6
3		-	11962	16.5	265.8
3		678.0	12214.	17.0	271.4
		100000	12465,9	17.5	277.0
3		1	12699.8	18.0	282.2
1000		The second secon	12933.	7 18.5	287.4
	8 1850.0		13185	5 19.0	293.0
1 Acres	9 1900.0 0 1950.0	The second second	13419.	19.5	298.2
-			13653.	3 20.0	303.4
	1 2000.0		14139.	0 21.0	314.2
	2 2100.0	72.000 75	14570.	8 22.0	323,8
	3 2200:0				333,8
	2300.0	100 - 100 - 45	-		344.2
-	15 2400.0	-	200		354.2
	2500.0	0110			364.0
_	17 2600.0	100000	10000		374.0
L	48 2700.0		-		385
	19 2800.		10000		
-	50 2900.	0,000	17827	79,0	1 3 444

Medicture C	ontent Determi	nation	
	Top	Middle	Botton
sample Location	- X4	B3	Co
onatiner Number	18.17	14.2	17.8

1227.0

1254.0

1282.0

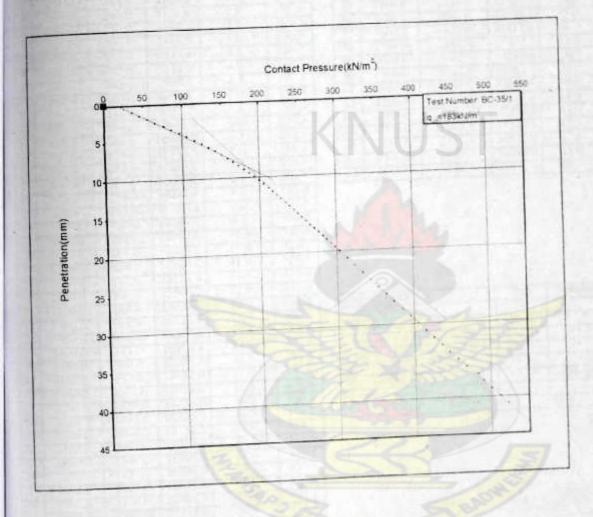
1309,0

Sample Location Constiner Number Container Mass (g) Co 17.8 1498 133,44 88.1 Container + Wet Sample Mass(g) Container + Dry Sample Mass (g) Moisture Content (g) 129.7 76.4 115.19 11.73 20.09 18.25 111.86 62.16 97.02 Dry Sample Mass (g) 180 18.9 18.8 Moisture Content (%) Average Moisture Content(%) = 18.5

## CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT LOADING TEST GRAPH

Test No.: BC-35/1

Date: 1/05/2007



#### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT LOADING TEST

Test No.: BC-100/1

Date 25/05/2007

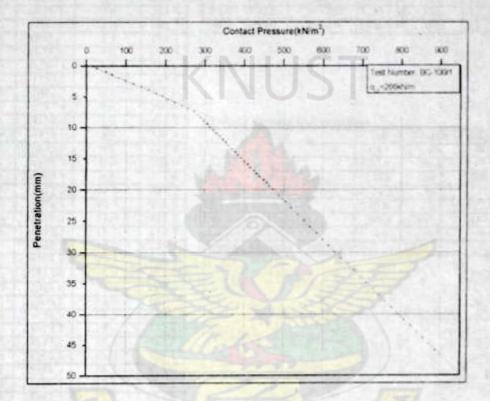
Sil	Dial	Proving	Force	Penetration	Suess
No.	Gunge(div)	Ring(div)	(N)	(mm)	(kN/m
1	0.0	0,0	0	0,0	0.0
2	50,0	68.0	1240.2	0.5	27.6
3	100.0	125.0	2265.6	1.0	50.3
+	150.0	164.0	2967.2	1.5	65,9
5	200.0	205.0	3704.8	2.0	82.3
6	250.0	250,0	4514.4	2.5	100.3
7	300.0	300.0	5413.9	3.0	120.3
8	350.0	350.0	6313.4	3,5	140.3
0	400.0	395.0	7122.9	4.0	158.3
10	450.0	442.0	7968.4	4.5	177.1
11	500.0	480.0	8652.1	5.0	192.3
12	550.0	512.0	9227.7	5.5	205.1
13	6,000	560.0	10091.3	6.0	224.3
14	650.0	606.0	10918.8	6.5	242.6
15	700.0	640.0	11530.5	7.0	256.2
16	750.0	682.0	12286.0	7.5	273.0
17	800,0	703.0	12663,8	8.0	281.4
18	850.0	724.0	13041.6	8,5	289.8
19	900.0	743.0	13383,4	9.0	297.4
201	950.0	766.0	13797.2	9.5	106.6
21	1000.0	781.0	14067.0	10.0	312.6
22	1050.0	805.0	14498.8	10.5	322.2
23	1100.0	824.0	14840.6	11.0	329.8
24	1150.0	847.0	15254.4	11.5	339.0
25	1200.0	866.0	15596.2	12.0	346,6
26	1250.0	884.0	15920.0	12.5	353,8
27	1300.0	901.0	16225.8	13.0	360,6
28	1350.0	922.0	16603.6	13.5	369.0
19	1400,0	944.0	16999.4	14.0	377.8
30)	1450.0	960.0	17287.3	14.5	384.2
317	1500,0	982.0	17683.0	15.0	393.0
32	1550,0	1001.0	18024.8	15.5	400.6
33	1600.0	1025.0	18456.6	16.0	410.1
34	1650.0	1040,0	18726.5	16.5	416.1
33	1700,0	1064.0	19158.2	17.0	425.7
36	1750.0	1081.0	19464.0	17.5	432.5
37	1800.0	1100.0	19805.9	18.0	440.1
38	1850.0	1127.0	20291.6	18.5	450.9
19	1900.0	1145.0	20615.4	19.0	458,1
40	1950.0	1160,0	20885.3	19,5	464.1
41	2000.0	1185.0	21335.0	20.0	474.1
12	2100.0	1222.0	22000.6	21.0	488.9
43	2200.0	1266.0	22792.2	22.0	506.5
14	2300,0	1301.0	23421.8	23.0	520.5
45	2400.0	1342.0	24159.4	24.0	536.9
46	2500.0	1384.0	24915.0	25.0	553.7
47	2600.0	1420.0	25562.7	26.0	568.1
48	2700.0	1465.0	26372.2	27,0	586.0
49	2800,0	1500.0	27001.9	28.0	600.0
50	2900.0	1540.0	27721.5	29.0	616.0

Srl	Dial	Proving	hotce	Penetration	Stress
No	Chagerdix	Ringidiya	IN.	(Him)	(kN m)
51	3000.0	1579.0	28423.1	30.0	631.6
52	3100.0	1620,0	29160,7	31.0	648.0
53	3200.0	1664.0	29952.2	32.0	665.6
54	3300.0	1704.0	30671.8	33.0	681.6
55	3400.0	1740.0	31319.5	34.0	696,0
56	3500.0	1780.0	32039.1	35.0	712.0
57	3600.0	1823.0	32812,6	36,0	729.2
58	3700.0	1863.0	33532.2	37.0	745.2
59	3800,0	1902.5	34242.8	38.0	761.0
60	3900.0	1942.5	34962.4	39.0	776.9
61	4000,0	1981.5	35664.0	40.0	702.4
62	+[00]+	2022.0	36392.6	41.0	808.7
63	4200.0	2061.5	37103.2	42.0	824.5
64	4300.0	2102.5	37840.8	43.0	840.9
65	4400.0	2141.5	38542.4	44.0	856.5
66	4500.0	2182.5	39280,0	45.0	872.9
67	4600.0	2221.0	39972.6	46.0	888.3

Sample Location	Тор	Middle	Bottom
Constiner Number	DI	7.3	18.0
Container Mass (g)	18.32	17.9	17.7
Container + Wet Sample Massign	132.89	141.9	150.4
Container + Dry Sample Mass (g)	114.05	124.2	129.2
Moisture Content (g)	18.84	20,69	21.25
Dry Sample Mass (g)	95.73	106.25	111,46
Moisture Content (%)	19.7	10.5	19.1
Average Moisture Content(%	6)=/	19.4	

#### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT LOADING TEST GRAPH

Test No.: BC-100/1 Date: 25/03/2007



#### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT LOADING TEST

Test No.: BC-150/1

Date 30/05/2007

Sel	Dial	Proving	Force	Penetration	Stress
No.	Guage(div)	Ring(div)	(N)	(mm)	(kN/m²
1	0.0	0.0	.0	0.0	0.0
2	50.0	73.0	1330.1	0.5	29.6
3	100,0	130.0	2355.6	1.0	52.3
4	150.0	185.0	3345.0	1.5	74.3
5	200.0	238.0	4298.5	2.0	95.5
6	250.0	286.0	5162.0	2.5	114.7
7	300.0	334.0	6025.5	3.0	133.9
8	350.0	378.0	6817.1	3.5	151.5
9	400.0	421.0	7590.6	4.0	168.7
0	450.0	460,0	8292.3	4.5	184.3
11	500.0	498.0	8975.9	5.0	199.5
2	550.0	571.0	10289.1	5.5	228.6
3	600.0	619.0	11152.7	6.0	247.8
4	650,0	667.0	12016.2	6.5	267,0
5	700.0	741.0	13347.4	7.0	296.6
6	750.0	785.0	14139.0	7.5	314.2
7	800.0	821.5	14795,6	8.0	328.8
8	850.0	858.0	15452.3	8.5	343,4
9	900.0	894.5	16108.9	9.0	358.0
10	950.0	931.0	16765.5	9.5	372.6
	1000.0	967.5	17422.2	10.0	387.2
2	1050.0	1004.0	18078.8	10.5	401.8
13	1100.0	1040.5	18735.4	11.0	416.3
4	1150.0	1077.0	19392.1	11.5	430.9
25	1200.0	1113.5	20048.7	12.0	445.5
	1250.0	1150.0	20705.4	12.5	460.1
26	1300.0	1186.0	21353.0	13.0	474.5
28	1350.0	1223.0	22018.6	13.5	489.3
-	1400.0	1259.0	22666.3	14.0	503.7
29 30	1450.0	1296.0	23331.9	14.5	518.5
11	1500.0	1332.0	23979.5	15.0	532.0
12	1550.0	1369.0	24645.2	15.5	547.7
13	The second secon	The second secon			
	1600,0	1405.0	25292.8	16.0	562.1
14	1650.0	1442.0	25958.4	16.5	576.9 591.2
15	1700.0	1478.0	26606.1	17.0	and the second second second
36 17	1750.0	1514.5	27262.7	17.5	605.8
	1800.0	1551.0	27919.3	18.0	620.4
19	1850,0	1587.5	28576.0	18,5	635.0
5095	1900,0	1624.0	29232.6	19.0	649.6
101	1950,0	1660,5	29889.2	19.5	664.2
11	2000.0	1697.0	30545.9	20.0	678.8
12	2100,0	1770.0	31859.2	21.0	708,0
13	2200,0	1843.0	33172,4	22.0	737.2
H	2300.0	1916.0	34485.7	23,0	766.3
15	2400.0	1988.5	35790.0	24,0	795.3
46:	2500.0	2061.5	37103.2	25,0	824.5
47	2600,0	2134.5	38416.5	26.0	851.7
48	2700.0	2207.5	39729.8	27:0	882.9
19	2800.0	2280.5	41043.0	28.0	912.1
50	2900.0	2353.0	42347.3	29.0	941.1

Sil	Dial	Proving	Force	Penetration	Stress
No.	Conspecdivi	Ring day)	(N)	110000	(kNm)
51	3000,0	2426.0	43060.6	30.0	970,2
52	3100,0	2499.0	44973.9	34.0	999.4
53	3200.0	2572.0	46287.1	32.0	1028.6
54	3300.0	2645.0	47600.4	33.0	1057.8
55	3400,0	2718.0	48913.7	34.0	1087.0
50	3500.0	2791.0	50226.9	35.0	1116.2
57	3600.0	2864.0	51540.2	36.0	1145.3
58	3700.0	2937.0	52851.5	17.0	11743
50	3800.0	3010.0	54166.8	38.0	12017

# **(NUST**

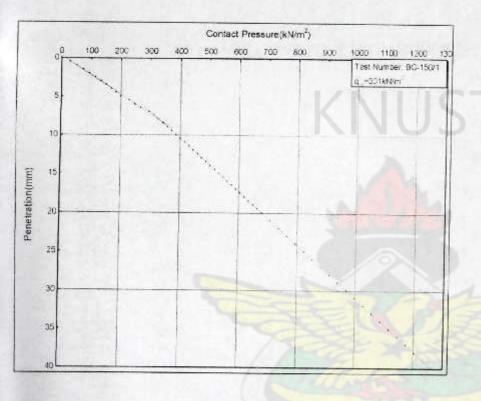
Moisture Content Determination

Sample Location	Тор	Middle	Hottom
Conatinar Number	C17	40	11
Container Massergy	17.8	18.4	17.8
Container - Wet Sumple Massiers	144.0	1(4) 5	160.1
Container - Dry Sample Mass (g)	124.2	138-1	137.2
Moisture Content (g)	19.8	22.4	22.8
Dry Sample Mass (g)	106,4	119.8	119.5
Mousture Content (***)	18.6	18.7	19.1
Average Moisture Contentt"	o) =	18.8	

#### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT LOADING TEST GRAPH

Test No.: BC-150/1

Date: 30/05/2007



KWAME NKRUMAN UNIVERSITY OF SCIENCE AND TECHNOLOGY KUMASI-GHANA

## APPENDIX IV

DYNAMIC CONE PENETROMETER TEST



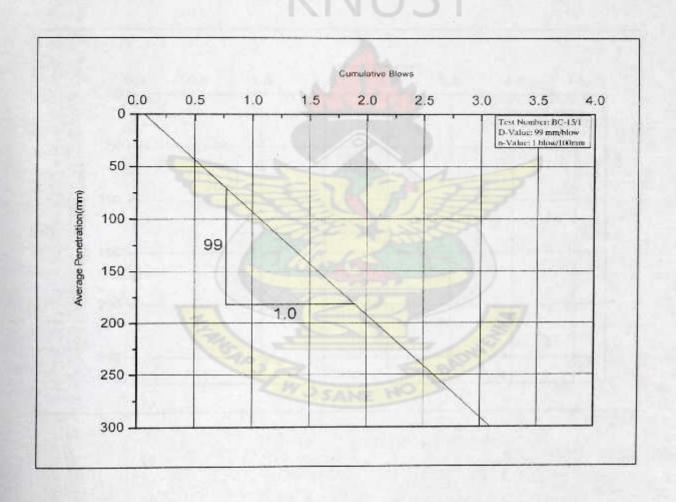


#### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT DYNAMIC CONE PENETROMETER TEST

Test No.: BC-15/1

Date: 23/04/2007

Cumulative	Penetration(mm)			
Blows	d <sub>1</sub>	d <sub>2</sub>	Aveg.d= $(d_1+d_2)/2$	
0	0	0	0	
1	70	90	80	
2	220	180	200	
3	320	260	290	

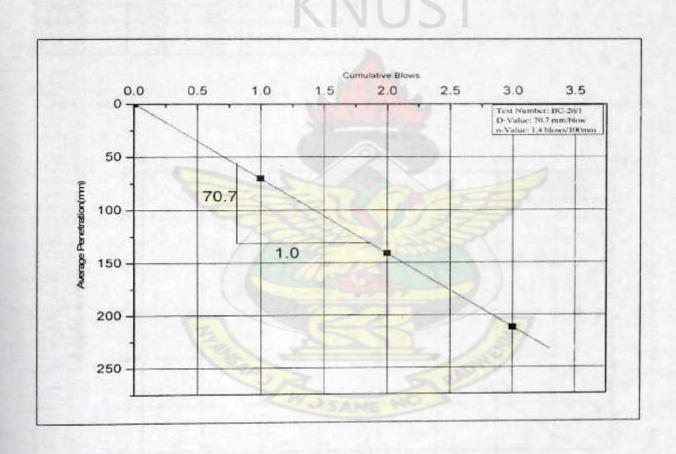


#### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT DYNAMIC CONE PENETROMETER TEST

Test No.:BC-20/1

Date: 19/04/2007

Cumulative	Penetration(mm)			
Blows	d <sub>1</sub>	d <sub>2</sub>	Aveg.d= $(d_1+d_2)/2$	
0	0	0.0	0	
1	50	90.0	70	
2	112	170.0	141	
3	180	244.0	212	

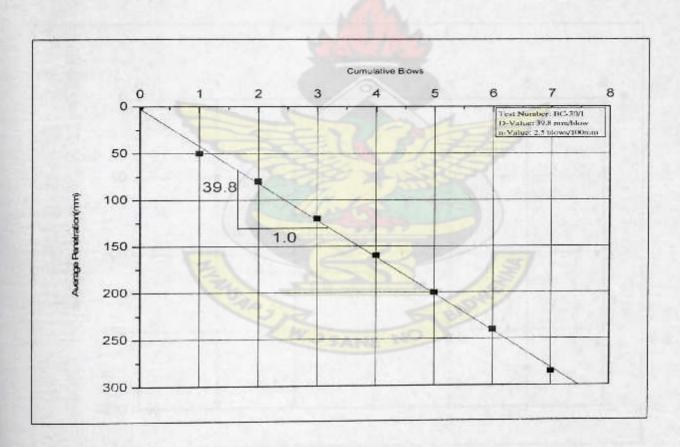


#### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT DYNAMIC CONE PENETROMETER TEST

Test No.: BC-30/1

Date: 26/04/2007

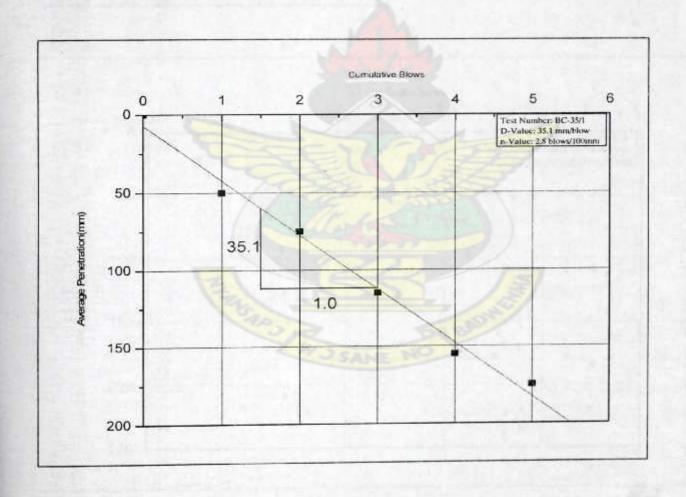
Cumulative Blows	Penetration(mm)		
	d <sub>1</sub>	d <sub>2</sub>	Aveg.d= $(d_1+d_2)/2$
0	0	0	0
1	60	40	50
2	90	70	80
3	130	110	120
4	160	160	160
-5	200	200	200
6	230	250	240
7	270	300	285



### CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT DYNAMIC CONE PENETROMETER TEST

Test No.: BC-35/1 Date: 1/05/2007

Cumulative	Penetration(mm)					
Blows	d <sub>1</sub>	d <sub>2</sub>	Aveg.d= $(d_1+d_2)/2$			
0	0	0	0			
1	50	50	50			
2	80	70	75			
3	120	110	115			
4	160	150	155			
5	180	170	175			

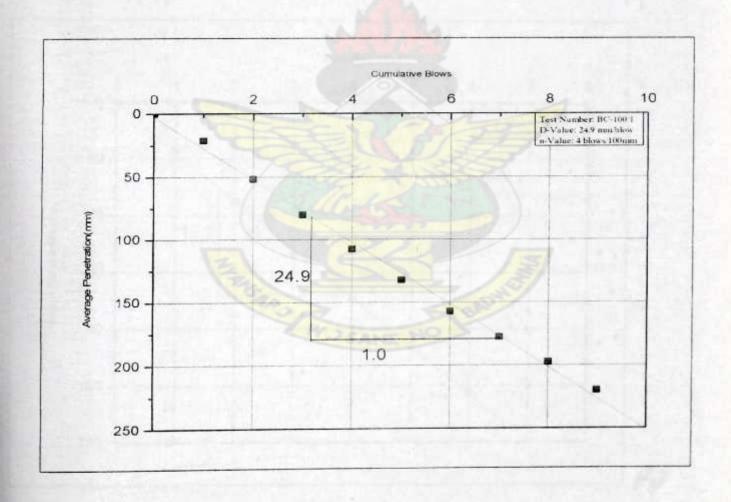


# CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT DYNAMIC CONE PENETROMETER TEST

Test No.: BC-100/1 Date: 25/05/2007

Cumulative		Penetra	ition(mm)
Blows	d <sub>1</sub>	d <sub>2</sub>	Aveg.d= $(d_1+d_2)/2$
0	0	- 0	0.0
1	20	22	21.0
2	55	48	51.5
3	80	80	80.0
4	110	105	107,5
5	135	130	132,5
6	160	155	157.5
7	185	170	177.5
8	200	195	197,5
9	220	220	220,0



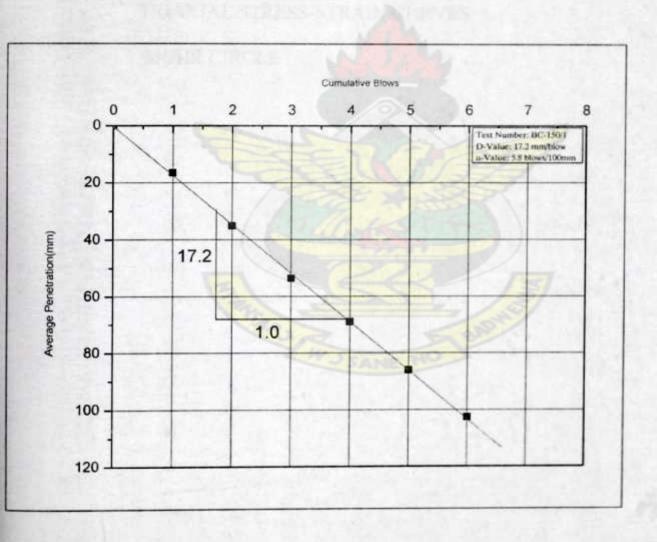


# CIVIL ENGINEERING DEPARTMENT-KNUST DCPT-ULTIMATE BEARING CAPACITY PROJECT DYNAMIC CONE PENETROMETER TEST

Test No.: BC-150/1

Date: 30/05/2007

Cumlative	Penetration(mm)						
Blows	d <sub>i</sub>	d <sub>2</sub>	Aveg. $d=(d_1+d_2)/2$				
0	0	0	0				
1	18	15	16.5				
2	35	35	35				
3	52	55	53.5				
4	68	70	69				
5	85	87	86				
6	100	105	102.5				



# APPENDIX VI

UNCONSOLIDATED-UNDRAINED TRIAXIAL TEST
TRIAXIAL STRESS-STRAIN CURVES
MOHR CIRCLE

KWAME MERUMAH UNIVERSITY OF SCIENCE AND TECHNOLOGY KUMASI-GHANA

Test No.: BC-15/1

SAMPLE CHARACTERISTICS

Date :24/4/2007

Sample Number	A	В	С	Averages
Initial Sample Height (cm)	7.60	7.60	7.60	Viverages
Initial Sample Diameter (cm)	3.80	3.80	3.80	
Initial Sample Volume(cm <sup>3</sup> )	86.19	86.19	86.19	
Initial Wet Sample Mass (g)	132.60	133.21	132.85	
Wet Bulk Density (g/cm³)	1.538	1.546	1.541	1.542
Container Number	5	D9	D19	110.12
Container Mass (g)	16.45	16.01	18.24	
Cont. + Final Wet Sample Mass (g)	146.70	148.81	150.62	
Container + Dry Sample Mass (g)	128.67	129.99	132.13	
Moisture Content Mass (g)	18.03	18.82	18.49	10
Dry Sample Mass (g)	112.22	113.98	113.89	
Moisture Content (%)	16.1	16.5	16.2	16.3
Dry Density (g/cm³)	1.325	1.327	1.326	1.326

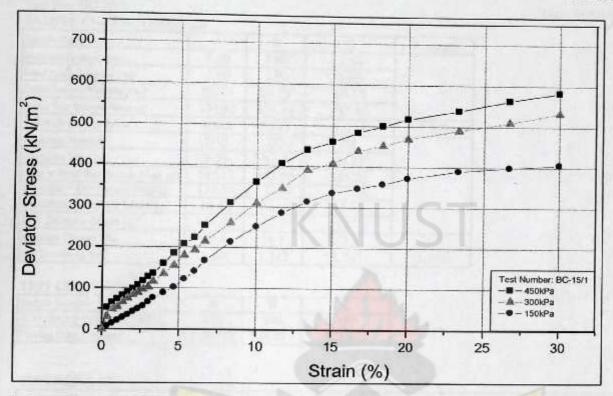
#### TEST CONDITION

Sample Number	A	В	C	
Confining Pressure (kpa)	150	300	450	
Proving Ring Constant	0.00816	0.00207	0.00199	

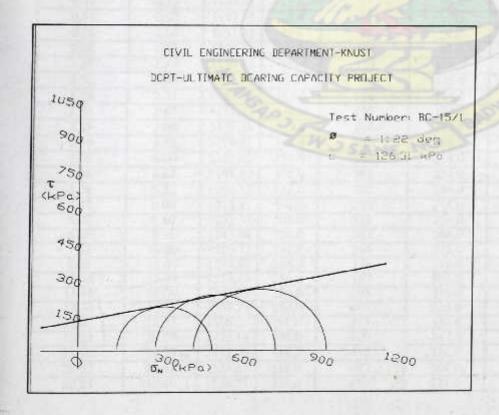
			Sample Num	ber A (σ <sub>3</sub> =150kPa	Sample Number	B (σ <sub>3</sub> =300kPa)	Sample Number	C (03=450kPa)
Change in Length (mm)	Strain (%)	New area (mm²)	Ring Reading (div)	Deviator Stress q=σ <sub>1</sub> -σ <sub>3</sub> (kPa)	Proving Ring Reading (div)	Deviator Stress q = σ <sub>1</sub> -σ <sub>3</sub> (kPa)	Proving Ring Reading (div)	Deviator Stres q =σ <sub>1</sub> -σ <sub>3</sub> (kPa)
0.00	0.00	1140.09	0.0	0	0.0	0	0.0	0
0.25	0.33	1143.91	1.0	7	17.0	31	30.0	52
0.51	0.67	1147.75	2.0	14	27.0	49	37.0	64
0.76	1.00	1151.61	3.0	21	30.0	54	42.0	73
1.02	1.33	1155.50	4.0	28	37.0	66	47.0	81
1.27	1.67	1159.42	5.0	35	42.0	75	52.0	89
1.52	2.00	1163.36	6.0	42	47.0	84	57.0	98
1.78	2.33	1167.33	7.0	49	50.0	89	63.0	107
2.03	2.67	1171.33	8.0	56	55.0	97	69.0	117
2.29	3.00	1175.36	9.5	66	58.0	102	75.0	127
2.54	3.33	1179.41	11.0	76	66.0	116	80.0	135
3.05	4.00	1187.60	13.0	89	77.5	135	95.0	159
3.56	4.67	1195.90	15.0	102	89.5	155	111.0	185
4.06	5.33	1204.33	18.0	122	105.0	180	126.0	208
4.57	6.00	1212.87	21.0	141	113.0	193	136.0	223
5.08	6.67	1221.53	25.0	167	127.0	215	155.0	253
6.35	8.33	1243.74	32.5	213	156.0	260	193.0	309
7.62	10.00	1266.77	39.0	251	189.0	309	229.0	360
8.89	11.67	1290.67	45.0	285	216.0	346	263.0	406
10.16	13.33	1315.49	50.5	313	248.5	391	291.0	440
11.43	15.00	1341.29	55.0	335	264.0	407	310.0	460
12.70		1368.11	58.0	346	289.5	438	332.0	483
13.97	18.33	1396.03	61.0	357	305.0	452	351.0	500
15.24		1425.12	65.0	372	323.0	469	370.0	517
17.78		1487.08	71.0	390	352.0	490	402.0	538
20.32	26.67	1554.67	76.0	399	383.0	510	440.0	563
22.85	30.00	1528.71	76.0	406	393.5	533	448.0	583

Test Number.: BC-15/1

Date :24/04/2007



σ <sub>3</sub> (kPa)	$(\sigma_1 - \sigma_3)$ kPa	(o <sub>1</sub> )kPa
150	372	522
300	469	769
450	517	967



Test No.: BC-20/1

SAMPLE CHARACTERISTICS

Date :20/04/2007

Sample Number	Λ	В	C	Averages
Initial Sample Height (cm)	7.60	7.60	7.60	- The same of
Initial Sample Diameter (cm)	3.80	3.80	3.80	
Initial Sample Volume(cm <sup>3</sup> )	86.19	86.19	86.19	
Initial Wet Sample Mass (g)	134.92	135.14	135.66	
Wet Bulk Density (g/cm <sup>3</sup> )	1.565	1.568	1.574	1.569
Container Number	D12	D2	D1	
Container Mass (g)	9.39	9.47	20.21	
Cont. + Final Wet Sample Mass (g)	142.11	152.90	155.74	
Container + Dry Sample Mass (g)	123.65	132.26	137.91	
Moisture Content Mass (g)	18.46	20.64	17.83	10-
Dry Sample Mass (g)	114.26	122.79	117.70	
Moisture Content (%)	16.2	16.8	15.1	16.5
Dry Density (g/cm³)	1.348	1.342	1.367	1.345

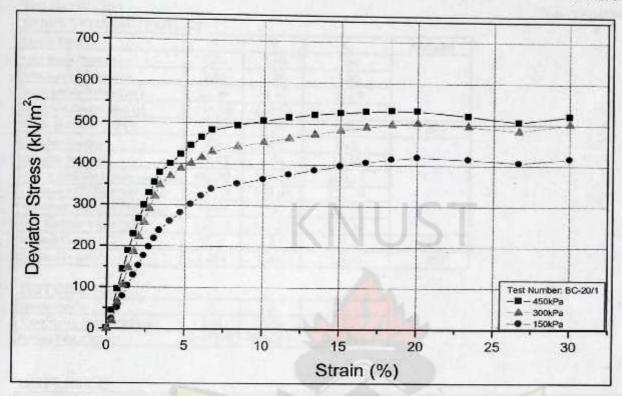
#### TEST CONDITION

Sample Number	A	В	C
Confining Pressure (kpa)	150	300	450
Proving Ring Constant	0.00816	0.00207	0.00199

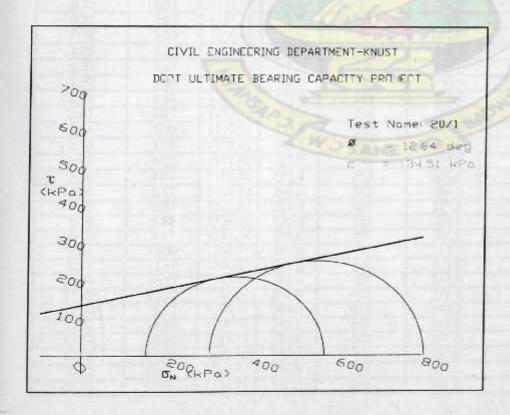
		-	Sample Num	ber A (σ <sub>3</sub> =150kPa )	Sample Number	B (03=300kPa)	Sample Number	C (03=450kPa)
Change in Length (mm)	Strain (%)	New area	Ring Reading (div)	Deviator Stress $q = \sigma_1 - \sigma_3$ (kPa)	Proving Ring Reading (div)	Deviator Stress $q = \sigma_1 - \sigma_3$ (kPa)	Proving Ring Reading (div)	Deviator Stres q =σ <sub>1</sub> -σ <sub>1</sub> (kPa)
0.00	0.00	1140.09	0.0	0	0.0	0	0.0	0
0.25	0.33	1143.91	13.0	23	10.0	18	25.0	43
0.51	0.67	1147.75	29.0	50	35.0	63	55.0	95
0.76	1.00	1151.61	45.0	78	59.0	106	83.0	143
1.02	1.33	1155.50	60.0	103	82.0	147	109.0	188
1.27	1.67	1159.42	75.0	129	104.0	186	133.0	228
1.52	2.00	1163.36	89.0	152	125.0	222	155.0	265
1.78	2.33	1167.33	103.0	176	145.0	257	175.0	298
2.03	2.67	1171.33	116.0	197	164.0	290	193.0	328
2.29	3.00	1175.36	129.0	218	182.0	321	209.0	354
2.54	3.33	1179.41	141.0	238	198.0	348	224.0	378
3.05	4.00	1187.60	155.0	260	212.0	370	239.0	400
3.56	4.67	1195.90	169.0	281	224.0	388	254.0	423
4.06	5.33	1204.33	183.0	302	234.0	402	269.0	444
4.57	6.00	1212.87	196.0	322	244.0	416	283.0	464
5.08	6.67	1221.53	208.0	339	254.0	430	296.0	482
6.35	8.33	1243.74	220.0	352	266.0	443	309.0	494
7.62	10.00	1266.77	232.0	364	278.0	454	322.0	506
8.89	11.67	1290.67	244.0	376	290.0	465	334.0	515
10.16	13.33	1315.49	256.0	387	302.0	475	345.0	522
11.43	15.00	1341.29	268.0	398	314.0	485	355.0	527
12.70	16.67	1368.11	280.0	407	326.0	493	365.0	531
13.97	18.33	1396.03	291.0	415	337.0	500	374.0	533
15.24	20.00	1425.12	301.0	420	347.0	504	382.0	533
17.78	23.33	1487.08	310.0	415	356.0	496	389.0	521
20.32	26.67	1554.67	318.0	407	364.0	485	395.0	506
22.85	30.00	1528.71	320.0	417	371.0	502	400.0	521

Test Number.: BC-20/1

Date: 20/04/2007



σ <sub>3</sub> (kPa)	$(\sigma_1 - \sigma_3)$ kPa	(o <sub>1</sub> )kPa
150	420	570
300	504	804
450	533	983



L BRAKE KWAME NKRUMAH UNIVERSITY OF SCIENCE AND TECHNOLOGY KUMASI-GHANA

Test No.: BC-30/1

SAMPLE CHARACTERISTICS

Date :27/04/2007

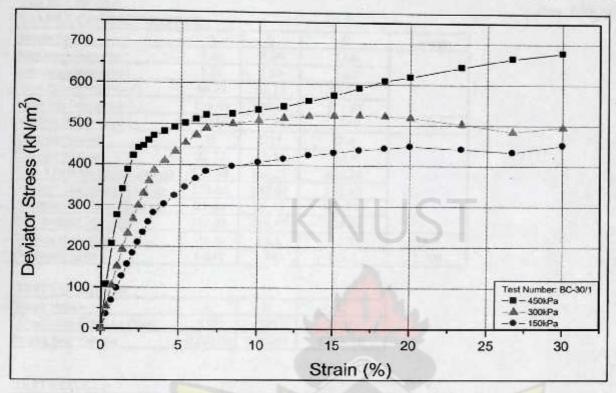
Sample Number	Α	В	C	Averages
Initial Sample Height (cm)	7.60	7.60	7.60	
Initial Sample Diameter (cm)	3.80	3.80	3.80	
Initial Sample Volume(cm³)	86.19	86.19	86.19	
Initial Wet Sample Mass (g)	146.08	146.52	146.65	
Wet Bulk Density (g/cm³)	1.695	1.700	1.701	1.699
Container Number	K5	D15	D12	
Container Mass (g)	17.45	6.44	14.40	
Cont. + Final Wet Sample Mass (g)	162.33	175.27	160.24	1
Container + Dry Sample Mass (g)	142.34	149.66	138.70	
Moisture Content Mass (g)	19.99	25.61	21.54	16
Dry Sample Mass (g)	124.89	143.22	124.30	
Moisture Content (%)	16.0	17.9	17.3	17.1
Dry Density (g/cm³)	1.461	1.442	1.450	1.451

TEST CONDITION

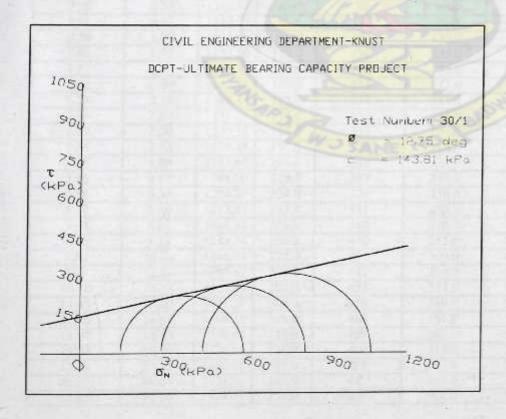
Sample Number	A	В	С	
Confining Pressure (kpa)	150	300	450	
Proving Ring Constant	0.00199	0.00207	0.00816	

			Sample Nun	ber A (σ <sub>3</sub> =150kPa )	Sample Number	$B(\sigma_y=300kPa)$	Sample Number	C (#3=450kPa)
Change in Length (mm)	Strain (%)	New area (mm²)	Ring Reading (div)	Deviator Stress q =σ₁-σ₁ (kPa)	Proving Ring Reading (div)	Deviator Stress q = σ <sub>1</sub> -σ <sub>3</sub> (kPa)	Proving Ring Reading (div)	Deviator Stres
0.00	0.00	1140.09	0.0	0	0.0	0	0.0	0
0.25	0.33	1143.91	20.0	35	30.0	54	15.0	107
0.51	0.67	1147.75	39.0	68	58.0	105	29.0	206
0.76	1.00	1151.61	57.0	98	84.0	151	39.0	276
1.02	1.33	1155.50	74.0	127	108.0	193	48.0	339
1.27	1.67	1159.42	91.0	156	130.0	232	55.0	387
1.52	2.00	1163.36	107.0	183	150.0	767	60.0	421
1.78	2.33	1167.33	123.0	210	169.0	300	63.0	440
2.03	2.67	1171.33	138.0	234	187.0	330	64.0	446
2.29	3.00	1175.36	153.0	259	204.0	359	66.0	458
2.54	3.33	1179.41	167.0	282	220.0	386	68.0	470
3.05	4.00	1187.60	181.0	303	235.0	410	70.0	481
3.56	4.67	1195.90	195.0	324	250.0	433	72.0	491
4.06	5.33	1204.33	209.0	345	264.0	454	74.0	501
4.57	6.00	1212.87	223.0	366	277.0	473	76.0	511
5.08	6.67	1221.53	236.0	384	289.0	490	78.0	521
6.35	8.33	1243.74	248.0	397	301.0	501	80.0	525
7.62	10.00	1266.77	259.0	407	312.0	510	83.0	535
8.89	11.67	1290.67	270.0	416	322.0	516	86.0	544
10.16	13.33	1315.49	281.0	425	331.0	521	90.0	558
11.43	15.00	1341.29	291.0	432	339.0	523	94.0	572
12.70	16.67	1368.11	301.0	438	346.0	524	99.0	590
13.97	18.33	1396.03	311.0	443	352.0	522	104.0	608
15.24	20.00	1425.12	321.0	448	357.0	519	108.0	618
17.78	23.33	1487.08	330.0	442	361.0	503	117.0	642
20.32	26.67	1554.67	339.0	434	364.0	485	126.5	664
22.85	30.00	1528.71	347.0	452	366.0	496	127.0	678

Test Number: BC-30/1 Date:27/04/2007



σ3 kPa	σ <sub>1</sub> -σ <sub>3</sub> kPa	$\sigma_1$ kPa	
150	448	598	
300	524	824	
450	618	1068	



Test No.: BC-35/1

SAMPLE CHARACTERISTICS

Date :2/05/2007

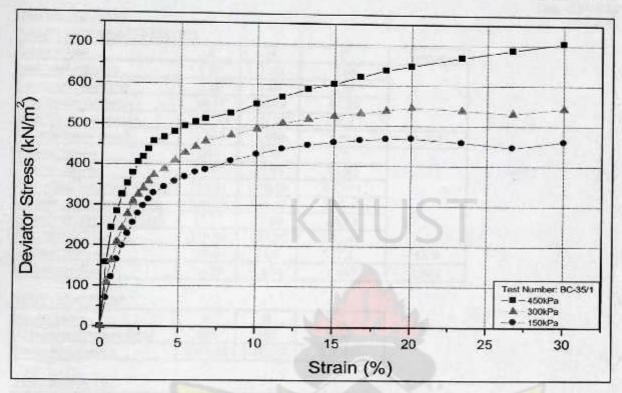
Sample Number	A	В	C	Averages
Initial Sample Height (cm)	7.60	7.60	7.60	777.50
Initial Sample Diameter (cm)	3.80	3.80	3.80	
Initial Sample Volume(cm3)	86.19	86.19	86.19	
Initial Wet Sample Mass (g)	150.49	150.66	151.61	
Wet Bulk Density (g/cm³)	1.746	1.748	1.759	1.751
Container Number	В	D21	D14	
Container Mass (g)	22.72	15.59	9.67	
Cont. + Final Wet Sample Mass (g)	173.04	166.21	160.24	
Container + Dry Sample Mass (g)	151.60	145.06	138.13	1
Moisture Content Mass (g)	21.44	21.15	22.11	110
Dry Sample Mass (g)	128.88	129.47	128.46	
Moisture Content (%)	16.6	16.3	17.2	16.7
Dry Density (g/cm³)	1.497	1.503	1.501	1.500

#### TEST CONDITION

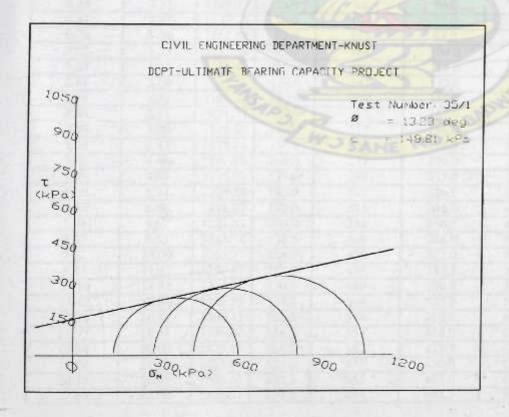
Sample Number	A	В	C
Confining Pressure (kpa)	150	300	450
Proving Ring Constant	0.00199	0.00207	0.00816

			Sample Nun	iber A (03=150kPa)	Sample Number	$B(\sigma_1=300kPa)$	Sample Number C ( $\sigma_x$ =450kPa	
Change in Length (mm)	Strain (%)	New area (mm²)	Proving Ring Reading	Deviator Stress q=σ <sub>1</sub> -σ <sub>1</sub> (kPa)	Proving Ring Reading (div)	Deviator Stress $q = \sigma_1 - \sigma_2$ (kPa)	Proving Ring Reading (div)	Deviator Stress $q = \sigma_1 - \sigma_3$ (kPa)
0.00	0.00	1140.09	0.0	0	0.0	0	0.0	0
0.25	0.33	1143.91	40.0	70	60.0	109	22.0	157
0.51	0.67	1147.75	70.0	121	90.0	162	34.0	242
0.76	1.00	1151.61	95.0	164	115.0	207	40.0	283
1.02	1.33	1155.50	115.0	198	135.0	242	46.0	325
1.27	1.67	1159.42	133.0	228	155.0	277	50.0	352
1.52	2.00	1163.36	149.0	255	173.0	308	54.0	379
1.78	2.33	1167.33	163.0	278	183.0	325	58.0	405
2.03	2.67	1171.33	175.0	297	193 0	341	60.0	418
2.29	3.00	1175.36	185.0	313	203.0	358	63.0	437
2.54	3.33	1179.41	195.0	329	213.0	374	66.0	457
3.05	4.00	1187.60	205.0	344	223.0	389	68.0	467
3.56	4.67	1195.90	215.0	358	237.0	410	70.5	481
4.06	5.33	1204.33	224.0	370	250.0	430	73.0	495
4.57	6.00	1212.87	232.0	381	260.0	444	75.0	505
5.08	6.67	1221.53	238.0	388	270.0	458	77.0	514
6.35	8.33	1243.74	256.0	410	285.0	474	80.5	528
7.62	10.00	1266.77	272.0	427	300.0	490	85.5	551
8.89	11.67	1290.67	286.0	441	314.0	504	90.0	569
10.16	13.33	1315.49	298.0	451	327.0	515	95.0	589
11.43	15.00	1341.29	309.0	458	339.0	523	99.0	602
12.70	16.67	1368.11	319.0	464	351.0	531	104.0	620
13.97	18.33	1396.03	328.0	468	363.0	538	109.0	637
15.24	20.00	1425.12	336.0	469	375.0	545	113.0	647
17.78	23.33	1487.08	343.0	459	387.0	539	122.0	669
20.32	26.67	1554.67	349.0	447	399.0	531	131.0	688
22.85	30.00	1528.71	354.0	461	402.0	544	132.0	705

Test Number: BC-35/1 Date :2/05/2007



σ3 kPa	σ <sub>1</sub> -σ <sub>3</sub> kPa	σιkPa	
150	469	619	
300	545	845	
450	647	1097	



Date:28/05/2007

Test No.: BC-100/1

SAMPLE CHARACTERISTICS

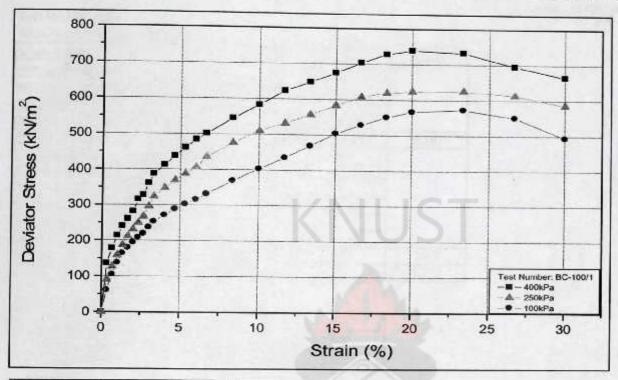
Sample Number	A	В	C	Averages
Initial Sample Height (cm)	7.60	7.60	7.60	7114,4840
Initial Sample Diameter (cm)	3.80	3.80	3.80	i
Initial Sample Volume(cm³)	86.19	86.19	86.19	
Initial Wet Sample Mass (g)	167.33	171.48	166.17	1
Wet Bulk Density (g/cm³)	1.941	1.990	1.928	1.953
Container Number	D11	X10	C9	
Container Mass (g)	17.90	17.58	17.92	
Cont. + Final Wet Sample Mass (g)	184.25	188.21	183.54	-
Container + Dry Sample Mass (g)	156.98	161.15	159.40	10
Moisture Content Mass (g)	27.27	27.06	24.14	
Dry Sample Mass (g)	139.08	143.57	141.48	
Moisture Content (%)	19.6	18.8	17.1	18.5
Dry Density (g/cm³)	1.623	1.674	1,647	1.648

#### TEST CONDITION

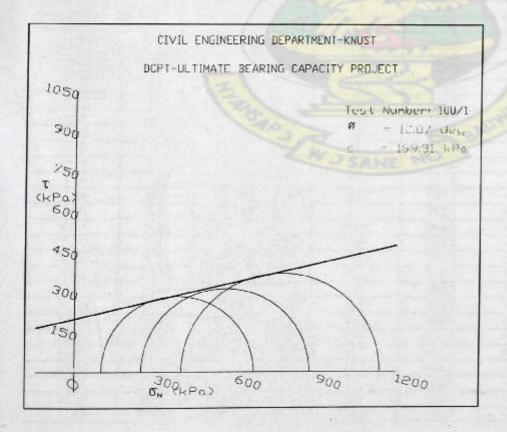
Sample Number	A	В	C	
Confining Pressure (kpa)	100	250	400	
Proving Ring Constant	0.00199	0.00207	0.00816	

F-H-LATE	4 - 7 - 54			Sample Number A ( $\sigma_3$ -100kPa)		B (σ <sub>3</sub> =250kPa)	Sample Number C ( $\sigma_1$ =400kPa	
Change in Length (mm)	Strain (%)	New area	Proving Ring Reading	Deviator Stress q =σ <sub>1</sub> -σ <sub>3</sub> (kPa)	Proving Ring Reading (div)	Deviator Stress q = \sigma_1 - \sigma_3 (kPa)	Proving Ring Reading (div)	Deviator Stress q =σ <sub>1</sub> -σ <sub>3</sub> (kPa
0.00	0.00	1140.09	0.0	0	0.0	0	0.0	0
0.25	0.33	1143.91	35.0	61	50.0	90	19.0	136
0.51	0.67	1147.75	60.0	104	70.0	126	25.0	178
0.76	1.00	1151.61	80.0	138	88.0	158	30.0	213
1.02	1.33	1155.50	95.0	164	104.0	186	34.0	240
1.27	1.67	1159.42	105.0	180	118.0	211	37.0	260
1.52	2.00	1163.36	114.0	195	130.0	231	40.0	281
1.78	2.33	1167.33	122.0	208	141.0	250	45.0	315
2.03	2.67	1171.33	129.0	219	151.0	267	47.0	327
2.29	3.00	1175.36	140.0	237	168.0	296	52.0	361
2.54	3.33	1179.41	150.0	253	184.0	323	56.0	387
3.05	4.00	1187.60	162.0	271	199.0	347	60.0	412
3.56	4.67	1195.90	173.0	288	213.0	369	64.0	437
4.06	5.33	1204.33	183.0	302	226.0	388	68.0	461
4.57	6.00	1212.87	192.0	315	238.0	406	72.0	484
5.08	6.67	1221.53	203.0	331	256.5	435	75.0	501
6.35	8.33	1243.74	230.5	369	286.0	476	83.0	545
7.62	10.00	1266.77	256.5	403	312.0	510	90.5	583
8.89	11.67	1290.67	282.0	435	331.5	532	98.5	623
10.16	13.33	1315.49	310.0	469	354.0	557	104.5	648
11.43	15.00	1341.29	339.5	504	377.5	583	111.0	675
12.70	16.67	1368.11	363.5	529	402.0	608	118.0	704
13.97	18.33	1396.03	386.5	551	418.0	620	124.5	728
15.24	20.00	1425.12	406.0	567	429.5	624	129.0	739
17.78	23.33	1487.08	428.0	573	450.0	626	133.5	733
20.32	26.67	1554.67	430.5	551	461.0	614	132.5	695
22.85	30.00	1528.71	379.5	494	432.5	586	124.5	665

Test Number: BC-100/1 Date :28/05/2007



o3 kPa	σ <sub>1</sub> -σ <sub>3</sub> kPa	σ <sub>1</sub> kPa	
100	567	667	
250	624	874	
400	739	1139	



Test No.: BC-150/1

SAMPLE CHARACTERISTICS

Date:31/05/2007

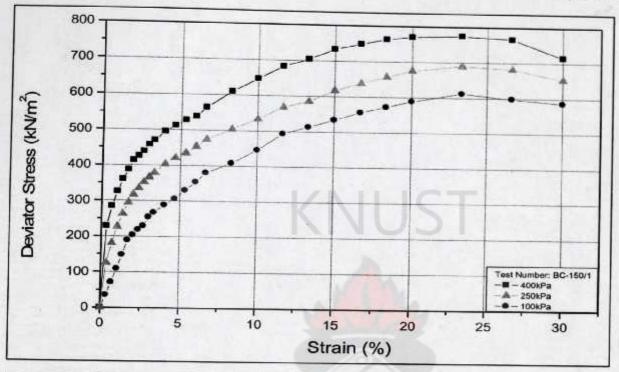
Sample Number	A	В	С	Averages
Initial Sample Height (cm)	7.60	7.60	7.60	- Tringe
Initial Sample Diameter (cm)	3.80	3.80	3.80	
Initial Sample Volume(cm3)	86.19	86.19	86.19	
Initial Wet Sample Mass (g)	181.28	181.46	182.00	
Wet Bulk Density (g/cm3)	2.103	2.105	2.112	2.107
Container Number	D8	D6	X7	
Container Mass (g)	18.02	18.60	18.14	
Cont. + Final Wet Sample Mass (g)	197.62	196.32	198.11	
Container + Dry Sample Mass (g)	169.67	168.07	169.80	IC
Moisture Content Mass (g)	27.95	28.25	28.31	
Dry Sample Mass (g)	151.65	149.47	151.66	10
Moisture Content (%)	18.4	18.9	18.7	18.7
Dry Density (g/cm³)	1.776	1.771	1.779	1.775

#### TEST CONDITION

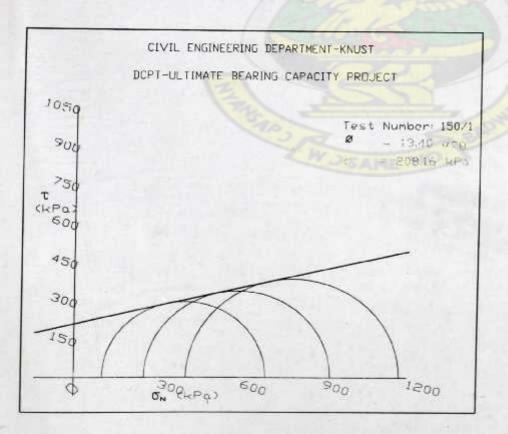
Sample Number	A	В	C
Confining Pressure (kpa)	100	250	400
Proving Ring Constant	0.00199	0.00207	0.00816

Change in Length (mm)			Sample Number A (03=100kPa)		Sample Number B (03=250kPa)		Sample Number C (\sigma_3=400kPa)	
	Strain (%)	New area (mm²)	Proving Ring Reading	Deviator Stress $q = \sigma_1 - \sigma_3$ (kPa)	Proving Ring Reading (div)	Deviator Stress q = \sigma_1 - \sigma_3 (kPa)	Proving Ring Reading (div)	Deviator Stres
0.00	0.00	1140.09	0.0	0	0.0	0	0.0	0
0.25	0.33	1143.91	20.0	35	69.0	125	32.0	228
0.51	0.67	1147.75	41.0	71	100.0	180	40.0	284
0.76	1.00	1151.61	63.0	109	126.0	226	46.0	326
1.02	1.33	1155.50	86.0	148	147.0	263	51.0	360
1.27	1.67	1159.42	110.0	189	165.0	295	55.0	387
1.52	2.00	1163.36	118.5	203	178.0	317	59.0	414
1.78	2.33	1167.33	128.0	218	189.0	335	61.0	426
2.03	2.67	1171.33	135.0	229	199.0	352	63.0	439
2.29	3.00	1175.36	150.0	254	208.0	366	66.0	458
2.54	3.33	1179 41	158.0	267	216.0	379	68.0	470
3.05	4.00	1187.60	172.0	288	231.0	403	72.0	495
3.56	4.67	1195.90	184.0	306	243.0	421	75.0	512
4.06	5.33	1204.33	200.0	330	253.0	435	78.0	528
4.57	6.00	1212.87	216.0	354	266.0	454	80.0	538
5.08	6.67	1221.53	233.0	380	280.0	474	84.5	564
6.35	8.33	1243.74	255.0	408	303.0	504	93.0	610
7.62	10.00	1266.77	284.0	446	327.0	534	100.5	647
8.89	11.67	1290.67	320.0	493	354.0	568	108.0	683
10.16	13.33	1315.49	339.0	513	372.0	585	113.5	704
11.43	15.00	1341.29	360.0	534	400.0	617	120.5	733
12.70	16.67	1368.11	382.0	556	421.5	638	125.5	749
13.97	18.33	1396.03	401.0	572	443.0	657	130.5	763
15.24	20.00	1425.12	422.0	589	464.5	675	134.5	770
17.78	23.33	1487.08	457.0	612	496.0	690	141.0	774
20.32	26.67	1554.67	467.0	598	512.0	682	146.0	766
22.85	30.00	1528.71	450.0	586	482.0	653	134.0	715

Test Number: BC-150/1 Date :31/05/2007



σ3 kPa	σ <sub>1</sub> -σ <sub>3</sub> kPa	σ <sub>1</sub> kPa 689	
100	589		
250	675	925	
400	770	1170	



# KNUST

