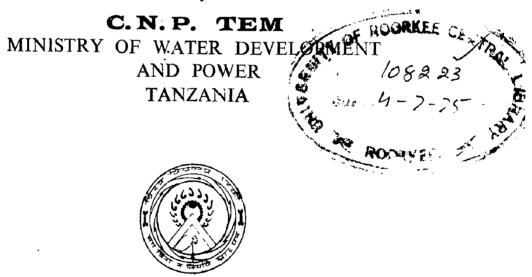
A STUDY OF THE COMPOSITION AND DENSITY OF SLURRY FOR STABILITY OF TRENCHES

A Dissertation submitted in partial fulfilment of the requirements for the Degree of MASTER OF ENGINEERING in WATER RESOURCES DEVELOPMENT

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WATER RESOURCES DEVELOPMENT TRAINING CENTRE UNIVERSITY OF ROORKEE ROORKEE (INDIA) 1975 CONTENTS

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CERTIFICATE

Certified that the dissertation entitled ," A STUDY AND OF THE COMPOSITION OF DENSITY OF SLURRY FOR STABILITY OF TRENCHES" which is being submitted by Mr C. N. P. TEM in partial fulfilment of the requirement for the award of the Degree of Master of Engineering in Water Resources Development of the University of Roorkee, Roorkee, is a record of student's own work carried out by him under my supervision and guidance as internal guide. The matter embodied in this dissertation has not been submitted for the award of any other degree or diploma.

This is further to certify that he has worked for a period of 6 months from 1st October, 1974 to 31st March, 1975 for preparing this dissertation for Master of Engineering (WRD) of this University.

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C.N.P. Tem.

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CHAPTER - I GENERAL

1.1 INTRODUCTION

The techniques of stabilization of excavations by clay suspensions have developed rapidly during the last 15 years. However, the mechanism by which the clay slurry stabilizes such excavations is not completely understood. The stability of circular excavations can be explained adequately by the hydrostatic pressure of the slurry and arching in the soil. With the application to rectangular excavations, in particular continuous trenches, the hydrostatic pressure of the slurry was found insufficient in certain cases, to stabilize the trenches according to classical earth pressure theory. Consequently other stabilizing mechanisms were proposed to account for this discrepancy¹.

Slurries are suspensions which are thix tropic. They have the property of becoming fluid on agitation and of settling to a gel when left undisturbed. Thix tropic suspensions can be made from a variety of materials but the cheapest and the one used very frequently universally is bentonite.

Bentonite slurry is a colloidal suspension formed by mixing powdered natural bentonite, and water, at an appropriate ratio. The bentonite and water must be agitated vigorously to ensure complete hydration of the bentonite particles.

1.2 USES AND ADVANTAGES OF SLURRY TRENCH TECHNIQUE¹⁷

This technique prevents any disturbance to the surrounding soil during work, thus avoiding the sort of damage to nearby structures which often results with the normal techniques of basement excavation. The diaphragm forms part of the final work providing peripheral basement retaining walls.

The system offers the advantage of noise reduction and freedom from vibrations. The construction of diaphragms in populated area has to be such as to give the least annoyance to local inhabitants through unnecessary noise and minimum hazard to adjacent structures by eliminating soil disturbance. Plate 1.1 shows an example of use of diaphragms in a built up area with no likelihood of danger to adjacent structures.

Slurry trench diaphragms solve the problem of constructing underground basement walls for deep excavations adjacent to exsisting shallow foundations. The slurry fills the excavation as work progresses and the costs of dewatering, sheeting or shoring are eliminated. Most of the excavated material may be reused for backfilling the trench after blending it with appropriate proportion of bentonite and suitable soils. This eliminates costs of dumping the excavated material from the work site and the necessity of borrowing special backfilling material.

The most remarkable adv_{antage} and use of slurry trench diaphragms is undoubtedly the impermeable cut off created at relatively cheap cost. The Wanapum Dam cut off¹⁵ (to be referred to later) reduced the coefficient of permeability from 1,000,000 ft/yr to only 0.1 ft/year in addition to offering possibility of working in difficult soils without the need for dewatering and danger of caving. The total cost which amounted roughly to $37 / ft^2$ of the cut off wall was much less than estimated cost for any comparable seepage barrier.

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9 browns for anderpasses of Hyde Park Corner 1.00 PLATE 1.1 USE OF DIAPHRAGM FOR UNDERPASSES

In Australia, at the Grahamstown Dam, due to the great depth to the impervious formation along the central 2,000 metres of the embankment no impervious core was provided during initial construction of the dam, as the cost of providing such a core would have been excessively high¹⁴. Subsurface investigation revealed that a vertical cut off would need to extend 30metres below the crest (maximum height of dam = 15.2 m) of the embankment, with a cutoff area of 43,500 square metres.

Comparison of alternatives for cutoffs were as follows:-TABLE 1.1 COMPARISON OF ALTERNATIVE METHODS TO PROVIDE CUTOFF AT GRAHAMSTOWN DAM, AUSTRALIA

Method	Expected flow reduction %	Cost in A # /m ² and Remarks.
Upstream blanket	50	Nil But dewatering was necessary.
Sheet Piling	. 70 - 90	35 - 60
Mixed in place piles	90 - 75 .	25 - 50
Membrane grouting	90 - 75	20 - 30
Alluvial grouting	90 - 95	150-200.Holes had to be closely spaced.
Slurry Trench (European Method)	90 - 75	35
Slurry Trench (USA Method)	90 - 75	20

Based on economy and achievement of impermeability the bentonite slurry trench method was adopted. At the time of construction, the permeability was reduced to the order of 10^{-5} to 10^{-6} cm/sec.

According to a survey conducted by L.B. Foster Company of Chicago, Illinois, it was revealed that more than half of all construction deaths which occurred in the United States in the first half of 1974, were the result of excavation failure and accidents involving trench cave-ins¹⁶. Though there are no records to indicate what the total fatalities resulting from trench accidents might be world wide, neither is there reason to believe that they occur with any less frequency elsewhere. The slurry trench technique is one of the most economical methods which can be used as a remedy to this hazard. Noise and disturbance reduction, and where applicable, water tight cutoffs for dams, are additional achievements of the method. This explains why this technique has been, in the past 15 years, widely adopted in various engineering projec ts.

1.3 DEVELOPMENT OF THE SLURRY TRENCH TECHNIQUE

The first experimental diaphragms in rectangular trenches were carried out in 1950 in Italy. The method was to enable constructing in-situ continuous reinforced concrete structures to unlimited depths. The structures had to be impermeable and able to carry vertical loads and capable of withstanding bending moments and shearing forces. The construction of such diaphragms h_{ed} to be feasible in difficult soils, such as those with large boulders and it had to be possible to seal off the diaphragm in the underlying bedrock¹⁷.

The slurry trench procedure was used under a permanent dam structure for the first time at Wanapum Dam in Columbia River, Washington State, U.S.A. in 1958¹⁵. The foundation materials at

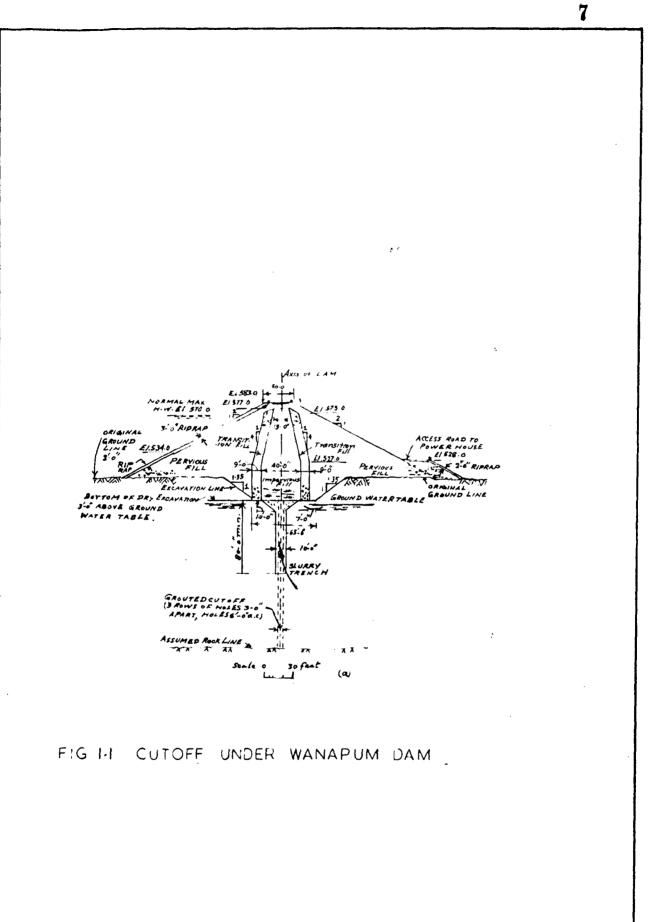
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consisted of an erratic deposits of sands, gravels, site the • and cobbles with some openwork gravel, extending to a depth of more than 80 ft (Fig. 1.1). The coefficient of permeability of the various layers ranged roughly between 100,000 and 3,000,000 ft/year and the average permeability was estimated roughly at 1,000,000 ft/year. Studies indicated that costs of installing a satisfactory sheet pile cutoff or of opening and dewatering a trench for a rolled earth cutoff through the pervious foundation soils would have been very high. A 10 ft wide foundation seepage barrier was constructed to a maximum depth of 80 ft by the slurry trench procedure. The excavation, made with a dragline, was kept continuously full of bentonite rud slurry, which prevented the walls from caving so that no sheeting or shoring had to be used. The excavated material (which consisted of a mixture of sand and gravel with bentonite slurry) was stock piled in windrows adjacent to the trench, and blended by means of dragline. and dozers with a quantity of 15 to 20% of natural silt. This mixture was then dumped from one end of the trench displacing the bentonite slurry until the backfilling was completed.

Construction was carried out with a minimum trouble. Although heavy equipment worked very close to the edge of the trench, no cavings of the trench walls occurred, even though the trench was kept open for lengths of as much as 1000 ft before backfilling commenced. Air pressure tests made in boreholes drilled in the slurry cut off, indicated that the material had a coefficient of permeability of less than 0.1 ft/year.

There has been a continous improvement in the technique since then, at the same time the field of application has

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steadily widened.

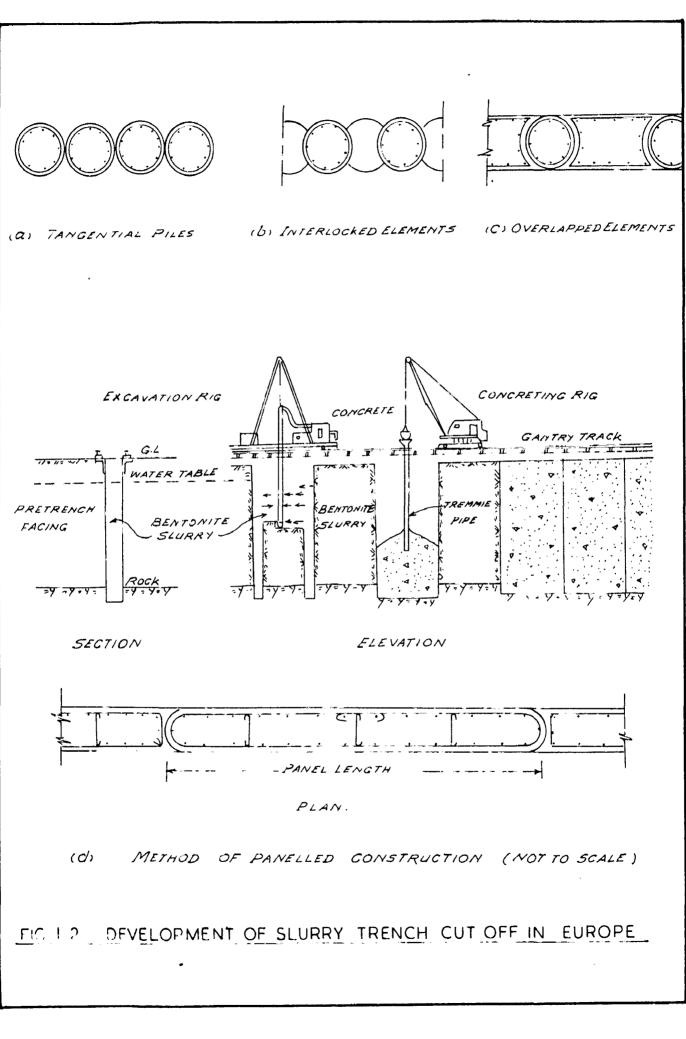
Two methods of construction of slurry trenches are in practice - the European and American methods. In the former method, a rigid concrete diaphragm or wall is installed in section or panels by bentonite slurry trench process as shown in Figure 1.2. This method can be used for cutoff and retaining walls for dams, river works, canals, locks, traffic underpasses, besements to buildings and other load bearing structures.

The American method is mostly used for construction of cutoffs for dams. The trench is excavated continuously in the presence of bentonite slurry without panelling (Fig. 1.3).

Apart from developments and improvements of the slurry trench method from the construction point of view research work on the composition and properties of bentonite slurries have been carried as outlined in Chapter Two.

Besides the Wanapum Dam, the slurry trench method has been used at the following places to provide impermeable cutoffs^{2,6}.

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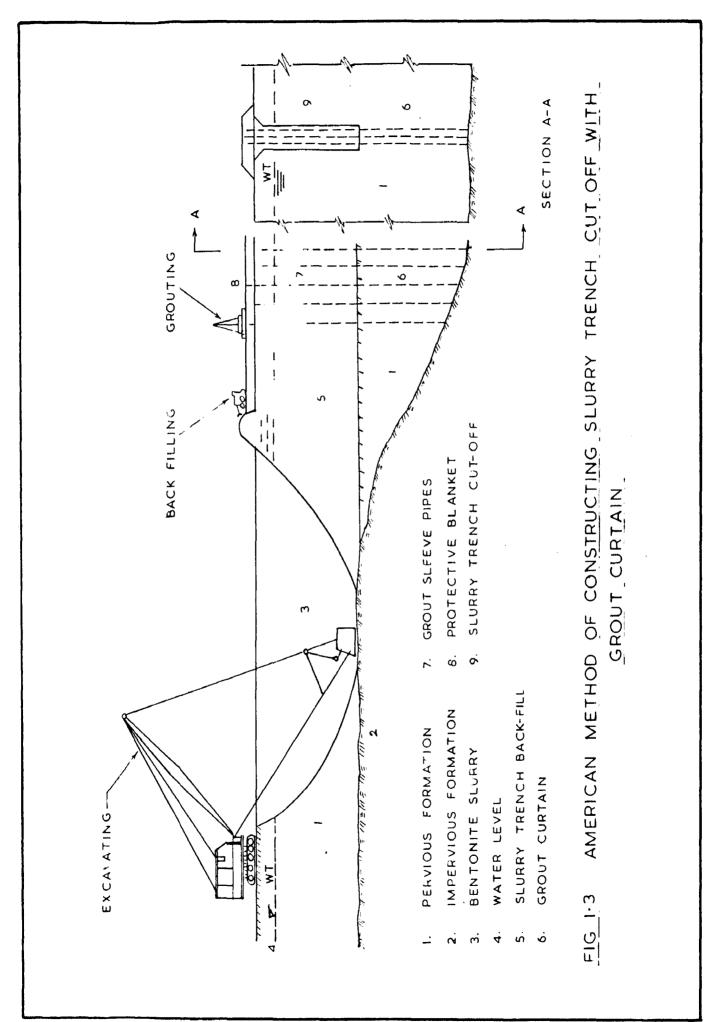


TABLE 1.2 PLACES WHERE SLURRY TRENCH TECHNIQUE HAS BEEN UTILIZED

Sl. No.	Place	Maximum Depth of cut off m
1	Kenewick levees, McNary Dam Project, U.S.	A. 6.7
2	Mangla Closure Dam, Pakistan	6.7
3	Duncan Lake Dam, Canada	18.3
4	West Point Dam, U.S.A.	30.5 (Grout- ing in sound rock below trench)
5	Saylorville Dam, U.S.A.	18.3
6	Brokopond Project, Suriname River, South Africa	4.6
7	Wells Dam, U.S.A.	24.4
8	Yards Creek Lower Reservoir, U.S.A	12.2
₽	Camanche Dam, California, U.S.A.	13.7
10	Bracis, Italy	10.0
11	Pietraporzio, Italy	18.0
12	Quero, Italy	19.0
13	Maria al Lago , Italy	42.0
14.	Isola Serafini, Italy	
	i. Power House	33.0
	ii• Levees	20.0
	iii. Dam	20.0
15.	Pierre Benite, France	28.0
16.	Tanughat, India	16.0
17.	Obra Dam, I. dia	-
18.	Ukai Dam, India	20.0

1.4 SCOPE OF DISSERTATION

It is intended to compile and study the composition and properties of bentonite slurry from research work carried out by different engineers and scientists. The stability phenomenon of slurry trenches, its different factors and the effect of composition and properties of bentonite slurry have also been studied. This is outlined in Chapter Two.

'An experimental work has also been carried out intended to find out the effect of replacing different proportions of bentonite by black cotton soil in four different slurry concentrations. Whereas bentonite is readily available in certain locations only, black cotton soil occurs virtually in all places at large. As found out in the experimental results, replacement of bentonite by black cotton soil in the slurry concentrations studied (1 to 4%), does in fact resultsin improvement of stabilizing capability in addition to the economical gain. Though the results obtained are interesting , yet it is the author's impression that further research work on the subjec t is necessary to help in understanding the problem fully and to enable at least partially, in filling the gap that exists in understanding fully the phenomenon of stabilizing of slurry trenches by bentonite slurries. This gap exists due to the nnmerous variables involved and ample time is needed to study the effect of each and its variations. The work was mostly confined to dry sand through a few tests on saturated sand were also performed. The following is the range of varaiables:

Slurry Concentrations Replacement of Bentonite by black cotton soil 1 % to 4% 0%, 25%, 50%, 75% and 100% for each concentration.

Flow tests were also performed to study slurry characteristics.

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NOTATION

a	=	Half width of trench
C	=	: Cohesive force in soil
Cu	Ę	Unit cohesion in soil (undrained test)
$\mathtt{C}_{\mathtt{a}}$	=	Shearing resistance of slurry saturated zone
Ep	=	: Passive resistance of slurry to resistance. displacement
е	=	void ratio
G	• =	Specific Gravity of soil particles.
g	=	Gravitational acceleration
H		Depth of Slurry in Trench
mH	=	Height of water table above trench bottom
h _w	=	Depth of water table below top of trench
ⁱ e	=	Critical pressure gradient
K _w	=	Permeability of soil to water
R.	<u> </u>	Half thickness of slurry saturated zone
n		Porosity of soil
Pa	-	Active Earth pressure
$^{P}\mathbf{s}$	1	Hydrostatic pressure of slurry
Рр	=	Total passive resistance of slurry
P _G	=	Negative gel pressure
р	=	Unit pressure
q	=	Unit surcharge
R		Resultant earth force
· r	=	Radius of an equivalent pore passage
^s c	=	Slurry concentration
u	=	Pore fluid pressure
v	=	Mean velocity

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W	= Weight of failure wedge
θ	= Inclination of failure wedge to horizontal
Ø	= Developed angle of friction
Ø	= Angle of frictional resistance
Y	= Bulk density of soil
Υ _b	= Submerged density of soil
Υ _s	= Density of slurry
Ϋ́w	= Density of water
η	= Viscosity of water
$\eta_{\rm pL}$	= Plastic viscosity of slurry
T ₀	= True yield stress
r f	= Bingham yield stress
• _m	= Maximum gel strength
T'S	= Structural shear strength of slurry in trench.

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REVIEW OF PROPERTIES AND RESEARCH WORK DONE ON AND SLURRY TRENCH TECHNIQUE

This chapter will deal with the following aspects

(i) Composition and Properties of Slurry Constituents.

(ii) Stability phenomenon of slurry trenches.

2.1 COMPOSITION AND PROPERTIES OF SLURRY CONSTITUENTS

2.1.1. Composition

A slurry fluid may consist of

- (i) Clay
- (ii) Bentonite powder
- (iii) Water
- (iv) Suspended cuttings
- Additives which may include sodium and or barium Sulphate.

The larger solid part of the fluid is bentonite powder, and its concentration is one on which stability and other properties of the slurry depend. Too low concentration causes caving and too dense concentration increases the plastic viscosity and the density and may float the excavating bucket of dragline or clam-shell. For the range of concentration studied, the lower the bentonite concentration (upto a certain limit) the loss the plastic viscosity and the more the slurry saturated zone. This zone, as shown later, is the main contribution to the stabilizing effect of in dry sand trenches. Thus depending on economy of availability of materials, construction convenience and the slurry properties required, different proportions of the bentonite powder can be replaced by chosen soil type. Likewise, the concentration, as well as additive proportion can be varied. The effects of these variations are discussed in para 2.2 which deals with experimental work carried out by different researchers.

Backfill Material⁶

The backfill for the trench should have sufficient fines to fill the voids of the coarser material. In order to limit consolidation, the percentage of fines (minus No. 200 sieve size) should not be excessive. About 10 to 25% of fines along with the entrapped bentonite should be sufficient.

The recommended gradation limits for backfill materials are as below :

Screen size or Number (U.S.A. Standard)	% passing by weight.
3" (76 mm)	80 -100
314" (19 mm)	40 - 100
No.4 (4.75 mm)	30 - 70
No. 30 (500 4)	20 - 50
No.200 (75 4)	10 - 25

2.1.2 Density

A study of available literature (Table 2.1) on field techniques of the slurry trench method reflects that the muds used in practice had densities of between 1.06 and 1.12 T/m³, and the actual values in any particular job depended more on the carrying

capacity of the mud in relation to the particle size of the material being excavated than on the value related to the unit weight necessary to give support. On account of the inherent yield strength, the slurry manifests a higher density in the trench by keeping the cuttings in suspension. This increase in the density of the original suspension is one of the major factors in the stabilizing effect of the slurry and simultaneously economises the use of bentonite clay⁹. The degree of this sanding , i.e., the size of particles of muck remaining in the suspension, can be controlled in a wide range by changing the properties of suspension supplied to the trench. Generally speaking, sanding of the suspension depends on the following factors¹¹: -

i. Structural shear strength (γ_0) of the suspension used.

Method of excavation, and the type of equipment used
Grain size distribution of the soil being excavated.
The methods of handling the suspension(likelihood of separating it from muck etc).

The cuttings are held in suspension by the small yield strength of the slurry. By balancing the submerged weight against the surface drag, CARDWELL has suggested that the diameter, D, of a particle supported by a slurry is given by the equation

$$D = \frac{6 \mathbf{r}_0}{(\gamma - \gamma_s) g} \dots (2.1)$$

where,

D = Equivalent diameter of particle in cm. Y = Density of solid particle in gm/cm³ Y_s = Density of slurry gm/cm³ τ_{o} = Structural shear strength of slurry in dynes/cm²

g = Acceleration due to gravity in cm/sec^2

Figures illustrating increase of slurry density in trench by suspension of cuttings are shown on Table 2.1

TABLE	2.1	SLURRY	DENSITY	INCREASE	WITH	SUSPENSION	OF	CUTTINGS

Project	Original Density of slurry T/m ³	Actual density of slurry observed in trench T/m ³
Wanapum	1.07 to 1.08	1.28
Pierre Benite	1.025	1.2 to 1.25
Ukai	1.04	1.08

Though some slurry rheological properties, like yield stress, do change with time, PIASKOWSKI's observations of the density of mud suspension in an experimental trench, revealed that the density, γ_s , of the slurry did not ' change with passage of time. The trench, 14.7 metres deep, was kept filled with slurry for a period of over 9 months. The observations are reproduced on table 2.2.

TABLE 2.2 VARIATION OF SLURRY DENSITY WITH TIME

Passage of time in days	Slurry density in T/m 3
16	1.310
1249	1.297
247	1.291
261	1.291

Slurry density is very sensitive to fluctuations in ground water level. For higher ground water level denser slurries

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would be required for stability. If mH is the height of water table above the trench (filled with slurry to a height H) bottom, then the required density of the slurry for stability increases with m as shown in figure 2.1. The reason is undoubtedly that the wet soil material below the water table level (together with the water) is heavier than the same dry soil material alone. The water fills up all the pores hitherto occupied by weightless air and the total thrust of the trengh weight side is more. A cutoff under construction at Piere Benite in France had slips when an unexpected flood occurred. In situations where fluctuations of water tableare likely to occur, either a high factor of safety in the selec tion of the slurry density has to be allowed or addition of suitable weighting material to overcome collapse of the trench can be substituted.

2.1.3. Thixotropy⁶

This is the property to undergo isothermal gel-tosol-to-gel transformation upon agitation and subsequent rest. The structural strength of clayey materials which is lost in remoulding is in time slowly removered. The necessary time depends on moisture content, and particularly on the salts in solution in soil water. The gel-sol transformation can be repeated indefinitely without fatigue. Sodium bentonite in suspension has the most pronounced thixotropic properties and consistency for certain ranges of concentration. These properties are mainly due to the large amount of water adsorbed and retained in its structure. The strength of the thixotropic gel formed is dependent on the setting time, the concentration of bentonite

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in the suspension and the chemical composition of the suspending fluid. Figure 2.2 shows the effect of time on the shear strength of bentonite suspensions.

2.1.4 Flow Properties

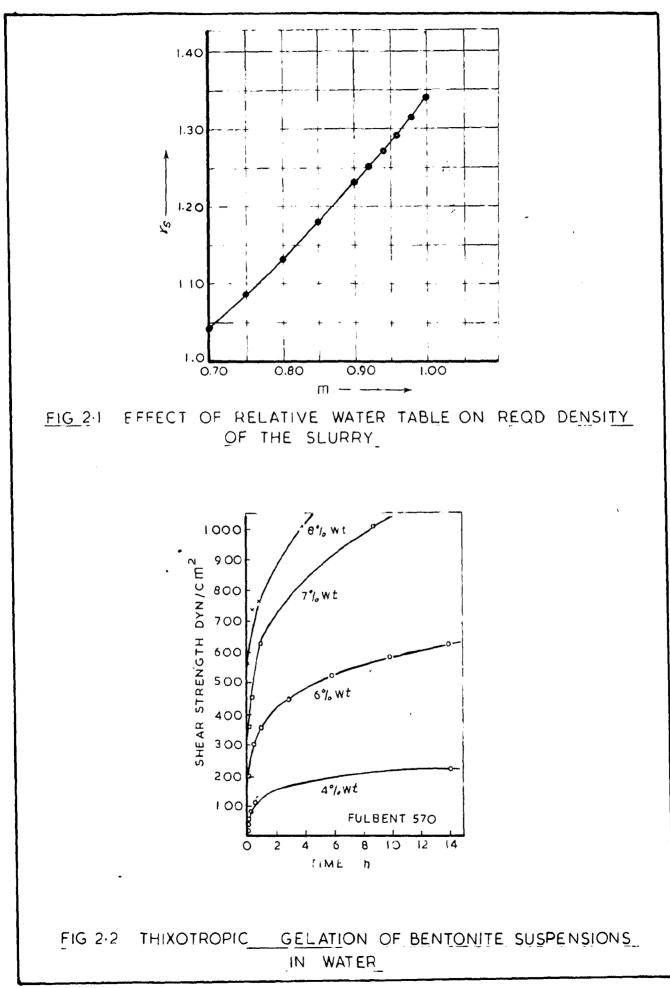
Important flow properties for thixotropic muds are-

- i. Yield stress and plastic viscosity
- ii. Effect of clay concentration
- iii. Gelation properties
 - iv. Yield gradient and limiting penetration
 - v. Effect of additives
 - vi. Activity Number.

The flow properties of bentonite may be attributed to its structure⁵. Its structure is such that addition of water or polar liquids to dry solid bentonite causes interlamellar swelling. The mineral possesses ion-exchange properties, and when the exchangeable base is sodium, bentonite swells in water to give thioxotropic suspensions showing anomalous flow properties at relatively low concentrations.

The characteristics swelling ability of bentonite is associated with the presence of like electrical charges on the critical surface, creating a repulsion between surfaces which are in close proximity. When the magnitude of this repulsion exceeds the attractive forces between atoms in neighbouring crystals, swelling takes place until a fresh equilibrium is established. The swelling takes place in one direction only, normal to the major dimensions of the crystals.

Bentonite suspension in water containing less than 1%



solids are generally free flowing fluids showing no unusual flow properties. When the clay concentration is raised to 1 to 15% flow properties become anomalous and if a solid content of 50% is reached, stiff putty-like masses are formed. A liquid limit of 350-500 is observed.

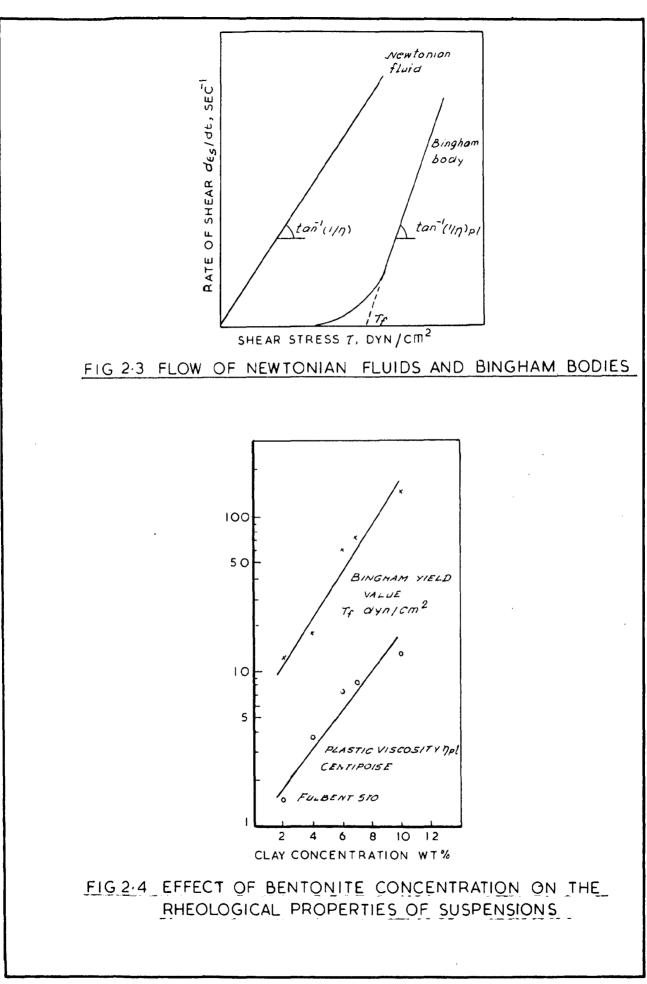
Due to the anomaly in the flow property, bentonite suspensions cannot be characterized by one parameter describing viscosity, as can be done with Newtonian fluids such as water. Bentonite must be described by two parameters, the plastic viscosity η_{PL} , and the yield stress τ_{f} . Both of these factors increase almost exponentially with clay concentration in the suspension in the range 5 to 25 % solids. This anomality of flow exhibited by bentonite is of Bingham bodies. The flow curve no longer passes through the origin as it does pe for a newtonian fluid, but makes an intercept τ_{f} with the shear stress axis as shown in Figure 2.3. Figure 2.4 illustrates the effect of bentonite concentration on the rheological properties of suspensions.

According to BINGHAM,⁷ clay slurries start to flow when a certa in shear stress $\tau_{\rm f}$ (true shear stress) is reached. Thereafter the flow rate is directly proportional to the excess shear stress (T - T_f). The rate of shear strain $\frac{d \Sigma s}{dt}$ (sec⁻¹) at any point in the material is given by

$$\frac{d \mathcal{E}s}{dt} = \frac{1}{\eta_{PL}} (T - T_f) \qquad \dots (2.2)$$

where $\eta_{p_{I_{i}}}$ is the plastic viscosity of the material.

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Clay slurries have the flow curves of the form shown in Figure 2.5. When the shear stress T_R reaches a certain value T_{R1} , shear failure occurs near the wall of the capillary (if flow is through capillaries) and the slurry moves forward as a plug. As the pressure gradient increases, the diameter of the solid plug becomes progressively smaller until the whole of the material in the capillary flows in a manner like a viscous liquid, and the rate of flow then increases linearly with the pressure gradient. If the intercept on the shear stress axis is T_{R1} , then from Figure 2.6 (a),

 $T_{R1} = \frac{R}{2} i_1 \qquad \dots (2.3)$

Where $i_1 = \text{pressure gradient} \frac{\Delta P}{\Delta L}$ at which flow commences and is called the yield gradient.

Flow of bentonite grout through glass capillary and sand were compared by determining the equivalent size of capillary corresponding to a particular sand bed. This was done by comparing Darcy's law for the flow of fluids through a porous body with Poiseuille's equation for capillary flow. This provided means of estimating the yield and flow properties of a grout in any sand from measurements made on the grout in a capillary viscometer, or a single sand bed of known water permeability. Effect of Additives on Flow Properties

(Floculation and Peptization)

Bentonite dispersions are sensitive to the nature of electrolyte in which they are suspended. Addition of sodium chloride to a suspension of pure sodium bentonite in water causes

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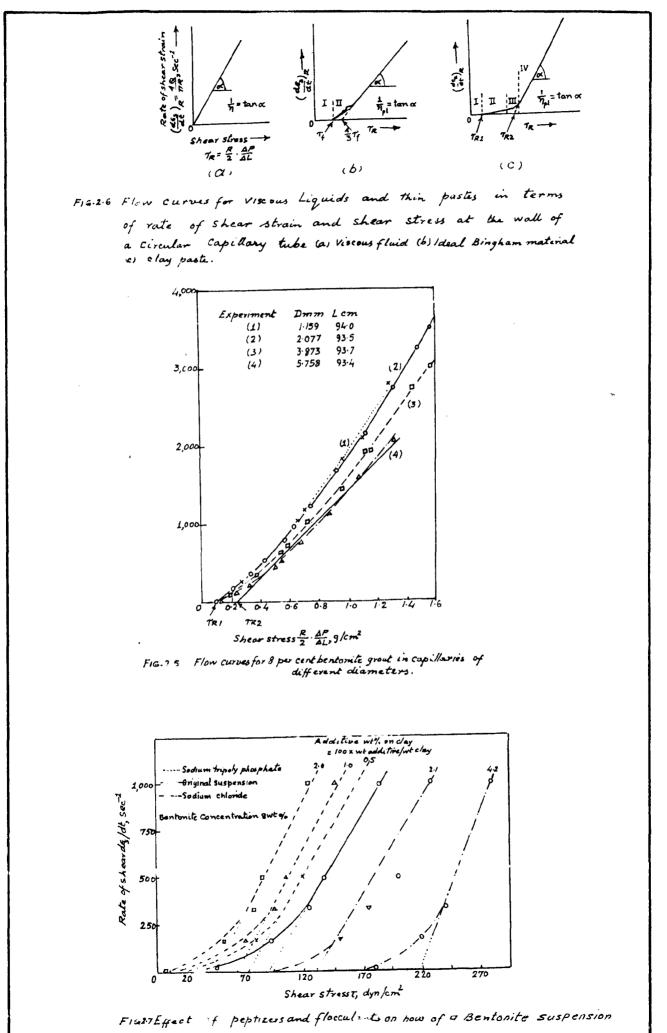
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at first a slight thinning, then a progressive thicknening. If the clay concentration is low it eventually coagulates and water separates at the top of the suspension.Progressively greater sensitivity towards flocullation is noted as the valency of the cation, added to the suspension, is increased.

A similar effect can be noticed when bentonite is added to salt solutions. Swelling does not take place so readily, particularly in calcium chloride or other solutions containing high valency cations, bentonite will completely fail to swell in strong solutions of many salts.

Whereas addition of sodium chloride causes thickening. addition of sodium : salts of polyvalent anions or sodium poly phosphate causes a thinning of bentonite suspensions. The curves of figure 2.7 show the effect of suspensions treated with floculants (e.g. sodium chloride) and peptizers (e.g. sodium polyphosphate). While the viscosity is barely changed, the yield point is markedly influenced. Addition of flocullents to the suspension causes a reduction in the thickness of the electrical double layer existing at the surface of the particles, simultaneously reducing the repulsion between them, and allowing the increased development of links between positive edges and negative surfaces, thereby raising the yield point of the suspension. A further increase in the salt concentration decreases repulsion to the point where particles are able to move closely together causing collapse and coagulation of the dispersion. On the other hand the polyvalent anion of the paptizing agent is sorbed on the edges of the crystals, reducing and even

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perhaps reversing positive charge, eliminating the forces giving mechanical structure to the suspension, and the yield point is reached without coagulation occurring. Thus the flocculating tendency of polyvalent cations may be offset by addition of peptizing agents giving a simple means of controlling the yield value under a variety of environmental conditions. The addition of peptizing agents to bentonite prior to contact with flocculating ions markedly increases their resistance to deterioration.

Relation between rheological characteristics of grouts and their capacity to permeate soils¹³

Mechanics of Permeation

The movement of fluid through the pores space of a soil is resisted by drag at the interface between the grains and the fluid. For true or Newtonian fluids , the drag is proportional to viscosity and shear rate - the shear rate being determined of by means_flow velocity and the geometrical characteristics of the pore space.

Colloidal suspensions such as clays and cement pastes are semi-stable dispersions of particles and are non-newtonian fluids. They are more complex fluids than newtonian, and the resistance to movement is generally related to the rate of shear and the accompanying shear stress. Where shear stress is exhibited, there may remain a significant opposing drag even for a vanishingly small flow rates.

Experiments conducted showed that colloidal fluids with varying bentonite concentrations had their shear rate diminish with time from the instance of mixing. A 5% suspension had an initial shear rate of about 1000 sec⁻¹ and movement ceased after 45 seconds when the radius of permeation was 5 cm. The shear strength then was 10 dynes $/ \text{ cm}^2$. With bentonite concentration of 7%, movement stopped after 15 seconds with radius of permeation of 2.5 cm. The limiting penetration corresponded to a grout with an average shear strength of 25 dynes/ cm².

It is thus evidenced that shear strength of bentonite fluid builds up with time and that the higher the shear strength the lower the shear rate. This is in agreement with G.K. JONES results of figure 2.4. Higher concentrations of bentonite also give rise to higher shear strength in a comparatively short time. This is an indication that the distance of permeation may be approximately taken as inversely proportional to the bentonite concentration (and hence the shear strength of the fluid) and the time elapse after mixing⁷. The type of fluid to be used for a particular soil may be decided on this guidance by considering the permeation required, the stabilizing force of the fluid, and the time of injection after mixing.

2.1.5 Yield Gradient and Limiting Penetration

The last article discusses the factors governing the movement of a fluid through the pore spaces of a soil. Thus if a suspension which has a definite strength is to penetrate a soil formation under constant pressure, the opposing drag due to the corresponding shear stress acting at the growing area of the surface Notted by suspension ultimately becomes equal to the whole of the applied pressure so that none is available to maintain the viscous flow.

¥ 28 −

The numerical value of the opposing drag can be estimated by using a model in which the void passages between granules are regarded as replaced by cylindrical tubes. Raffle and Greenwood deduce that an extra pressure gradient of $\frac{2 r_f}{1}$ must be applied at all parts of the advancing fluid to overcome the shaear strength during penetration, where $\boldsymbol{\tau}_{\mathbf{f}}$ is the Bingham yield stress for the fluid and 'r' is the effec tive radius of an average pore passage. As a first approximation, they assess the value of

Ir! from the corresponding Kozeny equation.

$$\frac{dp}{dx} = \frac{8 \eta_w \bar{v}}{r^2 n} = \frac{\gamma_w g \bar{v}}{K_w} \dots (2.4)$$

where, dp/dx = pressure gradient in dynes/cm³= Viscosity of water in poise η_w $\Upsilon_{\rm W}$ = Density of water in gm/cm³ K_{1,1} = Permeability of soil to water in cm/sec. = porosity of soil formation n v = mean flow velocity per unit area in soil in cm/sec. gravitational acceleration in cm/sec². g =

For an average temperature of $20^{\circ}C$, when $\eta_{\rm w}$ = 0.01 poise,

$$r^2 = \frac{8 \cdot 2 K_w}{n}$$
 x 10⁻⁵ Sq. cm. ...(2.5)

Thus for soils of known permeabilities and porosity, the corresponding values of 'r' can be obtained and hence estimate the limiting value of the pressure gradient $2r_f / r$ for known values of the Bingham's yield stress T_{f} .

In case of permeation from the sides of an excavated

trench, the slurry cannot pemetrate farther than a distance given by

$$X_{L} = \frac{pgr}{2r_{f}}$$

(The gravitational constant g will appear if τ_f is in dynes/om² and p is in gm/cm²).

...(2.6)

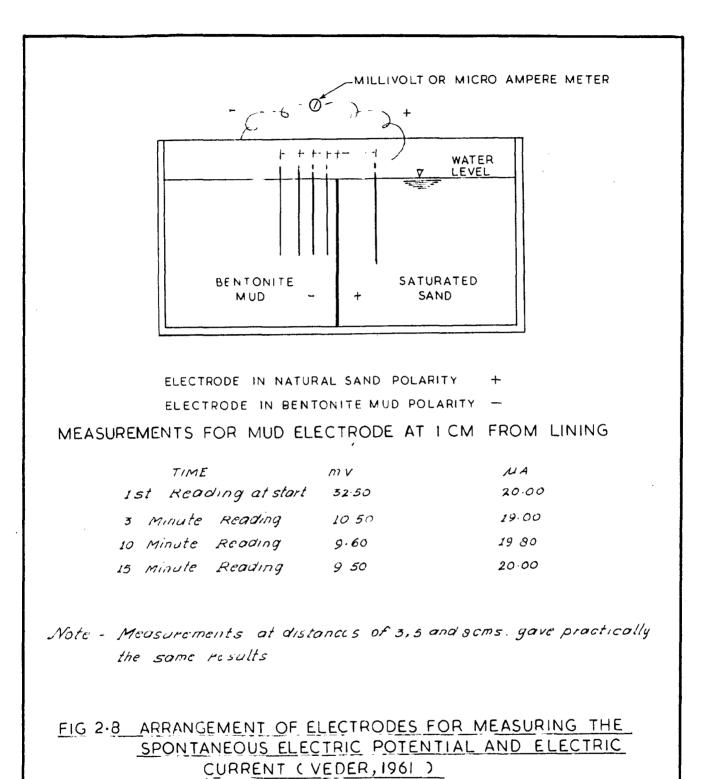
2.1.6 Electrical Properties

M.C. Veder¹⁸ carried out an experiment to determine the current conducting properties of the area of contact between the bentonite mud and cohesionless material and the effect on the interface lining. The experiment was carried out in a tank shown in sketch of figure 2.8. The tank was filled with cohesionless material and a trench was excavated under bentonite mud at constant level. A brass plate electrode was inserted into the bentonite mud and brass rod electrodes into the sand away from the bentonite lining. The electrodes were connected by a 0.8 mm copper wire and the natural existing current was meas ured. Then a direct current was introduced into the circuit and was regularly changed every 4.5 hours.

The results indicated -

i. The natural existing current helped in the formation of the lining whose thickness reached 2.1 cm. The wall collapsed when the level of the mud was lowered by 10cm.
ii. When no current was introduced but the natural current, the thickness of the lining was 1.5 cm as normal and the vertical face collapsed when the level of the mud was lowered by 6 cm.

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iii. Introduction of an opposing current gradually weekened the lining, especially in the arga close to the electrodes. This caused the face of the cohesionless material to collepse although the level of the bentonite mud had not been lowered. However, it led to formation of solid lining of bentonite on the central electrode.

The investigation indicates that the formation of bentonite lining on the vertical face of the cohesionless material can be altered by introducing between the two faces a direct electric current. Formation of the lining and stability will be more favourable if the polarity of the introduced current is the same as that already existing. Opposite polarity weakens the lining and stability of the vertical face of the soil. This offers a method of strengthening stability of slurry trenches if of course, found within economics of scale.

2.2 STABILITY PHENOMENON OF SLURRY TRENCHES

The stability phenomenon of slurry trenches has been studied by different engineers. M.C. Veder, and W.K. Elson have carried out experimental work independently on the subject while Nash and Jones, Mongenstern and Elson have made theoretical approaches. The theoretical approaches are all practically the same, and are based on equating the total disturbing and stabilizing forces on a sliding wedge at the trench face for two dimensional equilibrium.

2.2.1 Experimental Work by Veder and Elson.

2.2.1.1. M.C. Veder

This experiment has briefly been explained on para 2.1.5

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the results of which indicated existence of a distinct polarity between the two phases of bentonite and cohesionless material. This polarity (which gives rise to existence of an electromotive force between the two phases) has a great influence on the formation of a filter cake on the wall of the excavation. Bentonite particles are deposited at the interface by electrophoresis - a process in which colloids suspended in a medium migrate under the influence of an electric field to cathode or anode, according to the sign of their charge. As will be explained later, the filter cake thus formed has some influence on the stability 2.2.1.2. W.K. Elson¹

Tests were carried out on small trenches to find out the factors which consribute to stability of the trenches.

The test trench was constructed in a sectional pressed tank with values for draining the tank . One end of the tank was sealed with a bulkhead fitted with perspex windows to enable complete inspec tion of the trench .

Pressure gauges with sensitivity of 0.01 lb/sq.in(0.00067 kg/cm²) were used to measure the pressure at the bottom of the trench which is the critical one for failure. Dial micrometer gauges calibrated in 0.001 inch (0.025 mm) p.r division measured the changes in levels of the sand and slurry surfaces.

The tank was filled with sand in 7.5 cm layers placed under water. Each shovelful of sand was placed in a regular pattern to reduce the effect of segregation to a minimum. A minimum depth of 15 cms of water was maintained above the sand. Each layer of sand was tamped.

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After filling the tank with slurry the partition between the sand and the slurry was removed slowly to allow sufficient time for the slurry to penetrate into the sand and stabilize the face of the trench. The surface of the slurry was covered with polythene sheet to reduce evaporation. The slurry was then allowed to gel for a period of 4 hours.

Before testing, the slurry level was adjusted so that it was flush with the sand. The water level in the sand was raised at increments of one inch, depending on the stability of the system. At each increment, readings of the slurry, sand and water levels and pressure in the trench were taken at 15 minutes intervals. The water level was raised by a further increment when the readings reached a sensibly constant value. Failure was taken as the water level at which a steady creep occurred which caused a wedge of soil to slip into the trench. If water pressure alone was insufficient to cause failure, a surcharge was applied to the surface of the sand. After failure, the shape of the failure zone, at the surface was measured and sand samples were taken for determining the bulk density. The flow and the thixotropic properties of the slurry were determined with a rotational viscometer.

The angle of shearing resistance φ' was determined by carrying out a series of drained triaxial tests on saturated samples of sand at the various void ratios. The mean angle of shearing resistance and permeability were determined.

The results were analysed using the measured slip surface and compared with the results obtained from Coulombis wedge theory.

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These results are reproduced in para 2.2.4.1 after discussing the factors which affect stability of trenches on para 2.2.2 below.

2.2.2 Factors Which Affect Stabilization of Trenches

2.2.2.1 Trench Dimensions

Width

This is a function of the head on the cutoff and the gradation of the backfill materials and adjacent foundation materials. It will also depend on the type of trenching equipment used, as draglines, clamshells, back hoes or trenching machines. For deep trenches the dragline is the most suitable tool, while a trenching machine is the most economical and suitable equipment for excavating narrow, shallow trenches upto 2m width. The effect of the trench width, 2a, on the trench stability is indicated in equation 2.28 on para 2.2.3.

Depth

This is usually that required to give a positive cutoff and extends to the same depth as the pervious zone. As depth increases, the depth of excavation and cleaning the bottom increase and a grouted cutoff may be combined with the slurry trench to give the most economical cutoff beyond the limiting depth of the slurry trench operations. The limiting depth depends on the equipment used, generally 27m for dragline and 30 m for clem-shell. In the stability point of view the trench depth increases the hydrostatic force of the shurry, the shearing resistance zone, and the activating force composed of the soil and water (if saturated). Equation 2.28 summarizes these effects.

2.2.2.2 Density of Slurry

The ability of slurry to support the adjacent sand mass

which, may be saturated; is attributed partly to the higher density of the slurry than that of water 6 . The slurry exerts sufficient pressure on the walls of the trench to support them in their vertical position. The mud may be considered non-penetrating fluid and support of the excavation is achieved by the action of the hydrostatic pressure of the slurry on the impermeable face of the excavation. As discussed in para 2.1.2, the slurry density is increased by the suspended cuttings and that hardly any change of the achieved slurry density in the trench takes place with time.

The density of the slurry must be properly controlled. If the density is too low, the slurry will not prevent caving of the trench walls. If the slurry is too dense, the dragline bucket or cl_{1} - shell used for excavating will tend to float- high density will also increase the tendency of the backfill material to segreggate. The upper and lower density before placing are, in general 1.44 and 1.04 T/m³ respectively. The final value after setting is higher due to the cuttings from the excavated material.

2.2.2.3 Slurry Properties (other than density)

Quite a number of factors contributing to trench stability come up as a result of the slurry properties discussed in paras 2.1.3, (thixotropy), 2.1.4 (flow properties), 2.1.5 (yield gradient) and 2.1.6 (electrical properties).

(a) Formation of Filter Cake

This has been discussed in Veder's experimental work, and methods of strengthening and accelerating it by electrodes to create an electro-motive force. The impervious filter cake formed

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at the trench side forms a 'membrane' on which the hydrostatic pressure of the slurry can act. Laboratory measurements of the permeability of filter cakes from bentonite muds give values of 2.3×10^{-9} cm/sec, with a porosity of 88%.

The mud slurry will continue to exert hydrostatic pressure against the filter cake if it is stable with respect to time. Mud suspensions have stability, a property which enables them to preserve their original rheological characteristics. Lack of stability is revealed by separation of the constituents, the solid particles falling to the bottom and leaving a variable depth of clear water above them. Bentonite suspensions, being thixotropic in nature, exhibit excellent stability so far as separation of constituents is concerned and there is no sedimentation of suspended particles.

The filter cake formed functions as a vertical memberane exhibiting a resistance to deformation. This bentonite cake(along with the resistance of the slurry saturated zone) is able to withstand the pressure of a wedge of retained soil when the level of the slurry in the trench is lowered. Triaxial tests carried out by Veder on cylindrical samples of sand, on which a cake of bentonite had formed indicated that the resistance to deformation of the filter cake was equivalent to an increase in ambient pressure of 74 gm/cm².

(b) Shearing Resistance

Depending on fluidity of the slurry, it will permeate the trench sides of coarse granular cohesionless soils to form a slurry saturated zone. This zone forms to an appreciable thickness before

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the formation of the filter cake. The slurry saturated zone will have a shearing resistance on account of cohesive strength of the slurry which improves the physical properties of the zone by galling in soil pores under quiescent conditions^{7,9,17}.

On account of the heterogeneity of natural soil formations, it is difficult to estimate the depth of penetration of slurry into trench sides. At the same time, the extent of improvement in the physical properties of slurry saturated zone is another factor which cannot be precisely evaluated. However, if the properties of the trench soil and the slurry are known to a fair degree of accuracy , the following approach can be used to estimate approximately the shearing resistance of the slurry saturated zone. Thickness of Saturated Zone

From equation (2.6) para 2.1.5, it is shown that the limiting depth of penetration of a slurry, having a yield strength of $\gamma_{\rm f}$, is given by

$$X_{L} = \frac{Pgr}{2\tau}$$

where p and r are as defined earlier. Thus for given values of the shear strength $\tau_{\rm f}$, soil permeability $K_{\rm W}$, and pressure differential, the corresponding values of limiting depth of penetration can be calculated. The effective radius of an average pore pressure 'r' is calculated by equation (2.5) from the known viscosity of water of 0.01 poise at 20°C. Table 2.3 shows calculated values of limiting depths of penetration for the values of the variables shown.

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	Limiting depth of penetration in cm					
f dynes∕cm ²	p = 0.2	25 kg/cm ²	199 9 - C. V 1999	p = 0.	5 kg/cm ²	
	K _w =1 cm/sec	K _w =10 ⁻¹ cm/sec	K _w =10 ⁻² cm/sec	K _w =1 cm/sec	K _w =10 ⁻¹ cm/sec	$K_{w}=10^{-2}$ cm/sec
80	26	8.1	2.6	52	16.2	5.2
60	35	10.9	3•5	70	21.8	7.0
40	52	16.2	5.2	104	32.4	10.4
_ 20	104	32•5	10.4	208	65.0	20.8
10	208	65.0	20.8	416	130.0	41.5

TABLE 2.3 n = 0.30, $\eta_{w} = 0.01$ poise

Table 2.3 clearly shows that -

- (i) For the same soil and pressure differential the depth of penetration is inversely proportional to the shear strength of the slurry.
- (ii) For the same slurry and pressure differential the depth of penetration decreases with the reduction in the soil permeability.

(iii) For the same slurry and soil, the depth of penetration is directly proportional to the pressure differential.

Thus in order to increase the depth of penetration-

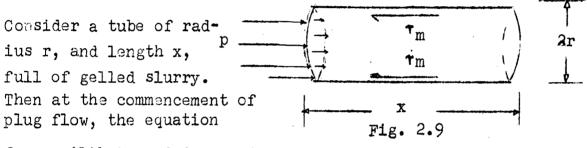
i. Slurry of low shear strength must be used

ii. Porous soil must be used

iii. A high pressure differential must be maintained.

It is not always possible to maintain all these requirements in the field and compromise has to be achieved after taking other stabilizing factors into consideration. The increase in the shearing resistance of the slurry saturated zone can be evaluated by triaxial tests on slurry saturated specimens simulating the field conditions, and then to study the effect of gelation of slurry in soil pores.

According to Elson, for known gel-strength (T_m) of the slurry, thickness of the slurry saturated zone (22), and the effective radius of soil pore (r), the shearing resistance of the slurry saturated zone, C_a , can be determined by the following procedure.



for equilibrium of forces from

Fig. 2.9 is:- $\pi r^2 p = 2 \pi r \tau_m x$

where, p is the pressure necessary to displace the slurry. Thus $p = \frac{2 \tau_m x}{r}$ ($\tau_m \text{ in gm/cm}^2$) ...(2.7)

or in terms of the critical pressure gradient i_c .,

$$\mathbf{i}_{c} = \frac{2 \mathbf{\tau}_{m}}{r} \qquad \dots (2.8)$$

Equation (2.8) represents the maximum pressure gradient that can be resisted by a capillary filled with a gelled slurry, neglecting the effect of creep.

Assuming that the pressure of the slurry in the voids of the soil does not affect the angle of shearing resistance φ' , then the increase in shear strength, C_a , is given by

 $C_a = -u \tan \varphi' \qquad \dots (2.9)$

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where u_i is the negative pore pressure and g' is the angle of internal friction in terms of effective stresses, substituting from equation (2.7)

$$C_a = -\frac{2r_m \ell}{r} \tan \varphi'$$

where (2) is the minimum distance to the permeable boundary of the mass of soil saturated with slurry (Fig. 2.10)

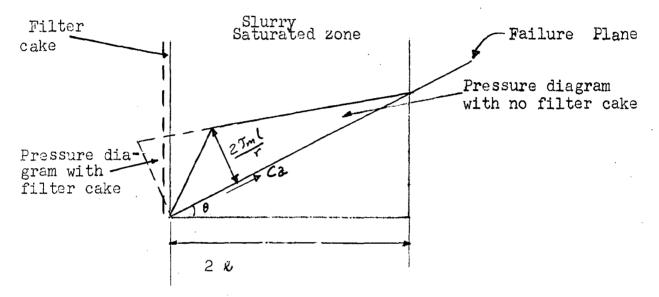


Fig. 2.10 - Pore Pressure Diagram in Slurry Saturated Zone

Consider a vertical zone of soil of thickness $2 \$, saturated with slurry. If the failure plane passes through it at an angle θ to the horizontal, then the pore pressure diagram in the failure plane can be plotted as in Figure 2.10. The shearing resistance of the zone along the rupture surface is given by

$$C_a = (Area of Pore pressure diagram) x tan φ'
 $\begin{bmatrix} 2 \\ - \end{bmatrix} = \begin{bmatrix} 2 \\ - \end{bmatrix} = \begin{bmatrix}$$$

$$= \left[\frac{2 \tau_{m}}{r} \cdot \frac{2 \tau_{m}}{2 \cos \theta} \right] \times \tan \varphi'$$
$$= \frac{2 \tau_{m}}{r} \frac{\varepsilon^{2}}{\cos \theta} \cdot \tan \varphi' \qquad \dots (2.10)$$
$$= P_{a} \tan \varphi'$$

where $P_a = \frac{2\tau_m k^2}{r \cos \theta} = magnitude of negative pore pressure.$ $(<math>\tau_m \text{ in gm/cm}^2$) ...(2.10a)

If a filter cake has formed on one boundary, assuming it has a very large strength compared with the original slurry, then,

$$C_{a} = \frac{4 \tau_{m}}{r} \cdot \frac{k^{2}}{\cos \theta} \cdot \tan \phi' \quad \dots (2.10b)$$

This increased shear resistance of the slurry saturated zone will act along the direction of the rupture surface.

(c) Passive Resistance of the Slurry 8

• Morgenstern suggests that the problem of the stability of slurry trenches should be regarded as one in which the bank material tends to fall in and extrudes into the slurry, but the finite strength of the bendomite mud aids the resistance to this movement.

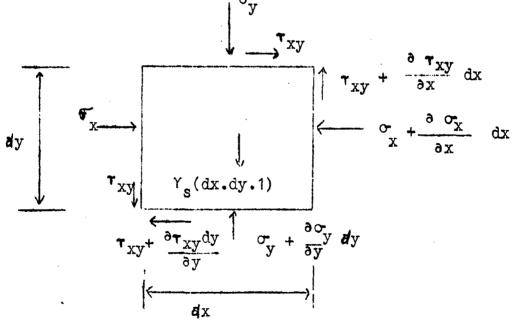


Fig. 2.11 -Equilibrium of a Two Dimensional Element

Considering an infinitesimal element in the clay slurry (Fig.2.11), the equations of equilibrium become :-

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$$\frac{\partial \sigma_{\overline{X}}}{\partial x} + \frac{\partial \tau_{\overline{X}\overline{Y}}}{\partial y} = 0$$

$$\frac{\partial \sigma_{\overline{Y}}}{\partial y} + \frac{\partial \tau_{\overline{X}\overline{Y}}}{\partial x} - \gamma_{\overline{S}} = 0$$

$$(2.11)$$

where Y_s is the density of the slurry. Plastic flow of the slurry will occur when the stresses satisfy the yield criterion.

$$(\sigma_{\bar{x}} - \sigma_{\bar{y}})^2 + 4\tau_{\bar{xy}}^2 = 4\tau_{\bar{s}}^2$$
 ...(2.12)

where τ_s is the shear strength of the bentonite slurry treated as a purely cohesive material.

Subject to the appropriate boundary conditions, the stresses are statically determinate, and using equations (2.11) and (2.12), Morgenstern has shown that for a trench width '2a', the horizontal stress is

$$\sigma_{x} = Y_{s} y + \tau_{s} \frac{y}{a} + \frac{\pi \tau_{s}}{2} \dots (2.13)$$

The total passive resistance of the slurry is

$$P_{p} = \int_{0}^{H} \sigma_{x} dy$$

= $\gamma_{s} \frac{H^{2}}{2} + \frac{\tau_{s}}{2} \frac{H^{2}}{2\pi} + 2\pi \tau_{s} \frac{H}{2}$...(2.14)

where H is the height of the slurry filled trench. The first term on the right hand side of equation (2.14) corresponds to the hydrostatic force exerted by the slurry and the other terms represent the passive resistance of the slurry on account of shear strength τ_s . Thus the passive resistance, E_p , is given by H^2 H^2 H^2

$$E_{p} = \tau_{s} \frac{\pi}{2a} + \pi \frac{\tau_{s}}{2} \frac{H}{2} \dots (2.15)$$

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It is difficult to conceive as to what magnitudes the

shear strength (τ_s) can develop in a practical slurry trench under conditions of considerable disturbance caused by passage of tools up and down . By assuming average values of r_f , r_s , r_m , γ , σ' and G ., T.C. Gupta³ has calculated the magnitudes of the passive resistance E_p , hydrostatic force P_s , and the magnitude of the other forces contributing to stability of trenches. These values are shown in para 2.2.4.2 Table 2.4. Likewise Elson's results on Table 2.5 offer a good comparison of these forces.

(d) Electro - Osmotic Forces

The term electro-osmosis is used to describe the electrokinetic phenomenon of a liquid moving through a porous medium relative to a fixed solid under the influence of an electric field. As shown by Veder, the bentonite suspension is electro= negative relative to the soil and there exists an electric potential between the two 'phases'. Therefore the electro-osmotic forces come into play due to movement of the slurry towards soil.

The equation of the motion of the fluid can be given in the form¹⁴

$$V_{e} = \frac{E \varepsilon_{e}}{\mu \pi \eta L} \qquad \dots (2.16)$$

where,

 V_{ρ} = The fluid velocity = Electrical potential Е = Dielectric constant of the fluid £ = Electro-kinetic potential (or zeta potential) ٤ = Fluid viscosity η = Distance between the electrodes. L The application of this equation to evaluate the effect

of electro-osmotic forces on the trench face is rather difficult because it is not possible to determine ε and ε to any reliable accuracy. At the same time the magnitude of the electric potential E, existing between the two phases (the slurry and the soil) is small and variable being influenced by the ionic concentration in ground water and the water used for making the slurry. Any venture to evaluate electro-osmotic forces is, therefore, not very meaningful. Their presence may be taken as a positive factor in the mechanism of stabilization, where the quantitative effect can be ignored. This assumption is confirmed by a model experiment which was carried out by Nash and Jones¹⁰ in 1963. In this, potential difference between saturated sand and bentonite mud was reduced to zero but no decrease in stability could be detected.

From the foregoing paras, the factors affecting stability of slurry trenches can be summarised to

a) Hydrostatic pressure exerted in the trench sides

$$P_{s} = \frac{1}{2} Y_{s} H^{2}$$

b) The passive resistance of the slurry

$$\mathbf{E}_{\mathrm{p}} = \frac{\mathbf{\tau}_{\mathrm{s}} \mathbf{H}^{2}}{2\mathbf{a}} + \frac{\pi \mathbf{\tau}_{\mathrm{s}} \mathbf{H}}{2}$$

c) The resistance to deformation of the filter cake and
d) The shearing resistance of slurry saturated zone

$$C_a = \frac{2 \tau_m \mathscr{L}^2}{r \cos \theta}$$

The quantitative contribution of these factors to stability is shown in Tables 2.4 - (Elson's experimental results)

and 2.5 (analysis of a particular full scale field slurry trench) in para 2.2.4.

2.2.3 Stability Analysis of Slurry Trenches

The theoretical procedure outlined below is based on . Elson's approach as given by T.C.Gupta^{2a}. The stability of long, vertical trenches in different types of soils has been discussed. The common method of estimating the disturbing and stabilizing forces, and equating them for a translating equilibrium of the comlomb's failure wedge has been adopted. No attempt is made to satisfy the moment equations as the estimation of forces is itself approximate and the points of their application are not precisely known. The main advantages of the method are its simplicity of application to field problems and ease of calculations.

2.2.3.1 Disturbing and Stabilizing Forces

The disturbing forces which cause a slide in the vertical fare of a long continuous trench are

a. The weight W of the sliding wedge, moist or saturated.
b. Surcharge unit load q, if any, on the sliding wedge.
The stabilizing forces which resist the slide are-

- a. Developed frictional forces in the soil mass at the failure plane.
- b. Developed cohesive forces in the soil mass along the failure plane.
- c. Hydrostatic pressure of the mud slurry.

d. Passive resistance of the slurry

e. Shearing resistance of the slurry saturated zone

- f. Resistance to deformation of the filter cake
- g. Electro-osmotic forces

Out of these resisting forces, the effect of the first five can be numerically evaluated, while the effect of the last two cannot be evaluated reliably. However, their contribution towards the overall stability of the trench side is comparatively small, and therefore, their effect in numerical calculations can be ignored. Their effect can be taken to mean existence of some positive factors which improve the calculated value of the factor of safety.

2.2.3.2 Trenches in Clay¹⁰

This case will be met with in cities where concrete diaphragms are required to be constructed for basements of buildings, traffic under passages etc. On account of the fineness of the soil pores, no penetration of slurry into the trench sides is likely to take place. Under undrained conditions, the angle of frictional resistance (ϕ) will be practically zero so that no frictional resistance is developed at the failure plane. The only resisting forces will be the cohesion in clay, the hydrostatic pressure of the slurry and its passive resistance.

The total passive resistance of the slurry is given by

$$P_{p} = P_{s} + E_{p}$$
which from para 2.2.2 is given by
$$P_{p} = \frac{1}{2} Y_{s} H^{2} + \frac{\tau_{s} H^{2}}{2a} + \frac{\pi \tau_{s} H}{2}$$

Consider the equilibrium of a soil wedge under the forces W,C, R and P_p, as shown in Figure 2.12(a). The wedge is at the verge of sliding. Under undrained conditions,

 $\phi = 0$, hence $\phi_{D} = 0$, so that $\theta_{cr} = 45^{\circ}$

The force polygon is shown inFigure 2.12(b). Then, with the defined notation ,

$$W = \frac{1}{2} Y H^{2}$$

$$P_{s} = \frac{1}{2} Y_{s} H^{2}$$

$$F_{p} = \frac{T_{s} H^{2}}{2a} + \frac{\pi T_{s} H}{2}$$

$$C = \int \overline{2} H C_{u}$$

From Figure 2.12(b), resolving horizontally and vertically and eliminating R, we obtain,

$$C = \frac{W - P_p}{\sqrt{2}}$$
 ...(2.17)

If F is the factor of safety , the developed cohosion

$$C = \int 2 H \frac{C_u}{F}$$
 ...(2.18)

From Equations (2.17) and (2.18) we obtain

$$F = \frac{2H - C_{u}}{W - P_{p}}$$

$$= \frac{2 H C_{u}}{\frac{1}{2} YH^{2} - \frac{1}{2} Y_{s}H^{2} - \frac{\tau_{s}H^{2}}{2a} - \frac{\pi_{s}}{2} \frac{\pi_{s}}$$

Equation (2.19) shows that the factor of safety depends on the shear strength and densities of both the soil and the slurry, and the trench dimensions.

Cohesionless Sand¹⁰ 2.2.3.3. Trenches in Dry

In this case, the frictional resistance on account of the normal component of the dry weight of the soll wedge is mobilized at the failure plane, and the resulting factor of safety, after combining the stabilizing effect of hydrostatic pressure of the slurry, is quite large even if the effect of all other factors is ignored. For these conditions, the force polygon gets simplified as shown in figure 2.13(b), where $\phi_{\rm D}$ is the developed angle of friction.

 $W = \frac{1}{2} Y H^{2} \tan (90-6)$ $P_{s} = \frac{1}{2} Y_{s} H^{2}$

Thus,
$$\frac{W}{P_s} = \frac{\Upsilon \tan (90 - \theta)}{\Upsilon_s}$$
 ...(2.20)

Also from the force polygon, under conditions of equilibrium, $\frac{W}{P_s} = \tan \left[90 - (\theta - \phi_D)\right]$...(2.21) From equations (2.20) and (2.21)

 $\frac{Y}{Y_{s}} = \frac{\tan \theta}{\tan (\theta - \phi_{D})}$ Since ϕ_{D} is maximum when $\theta = (45 + \frac{\phi_{D}}{2})$ for the most critical plane $\frac{Y}{Y_{s}} = \frac{\tan (45 + \frac{\phi_{D}}{2})}{\tan (45 - \frac{\phi_{D}}{2})}$ or by trignometry $\tan \phi_{D} = \frac{Y - Y_{s}}{2 \left[Y - Y_{s}\right]} \dots (2.22)$

If we define the factor of safety as F =

 $\frac{\tan \phi}{\tan \phi_{\rm D}}$

where ϕ' is the effective angle of internal friction for the sand, then, $F = \frac{2 (Y.Y_s)^{1/2} \cdot \tan \phi'}{Y - Y_s} \dots (2.23)$

Thus for dry cohesionless sands, the factor of safety is independent of depth.

2.2.3.4 Trenches in Saturated Cohesionless Sands^{1,9}

This is the case of perhaps more general occurrence in river valley projects.

As shown in figure 2.14, a wedge of soil inclined at an angle θ to the horizontal is assumed to be on the verge of sliding. The water level in the soil is mH and the sand is assumed to be saturated up to the top by capillary or otherwise. The effect of capillary forces on the shear strength of soil has been neglected, being small and on safer side. The wedge is separated from the slurry by a thin impermeable membrane.

In figure 2.14, T denotes the tangential force acting along the base of the sliding wedge, N, denotes the reaction normal to the base, and P_a denotes the horizontal force required to stop the wedge from sliding.

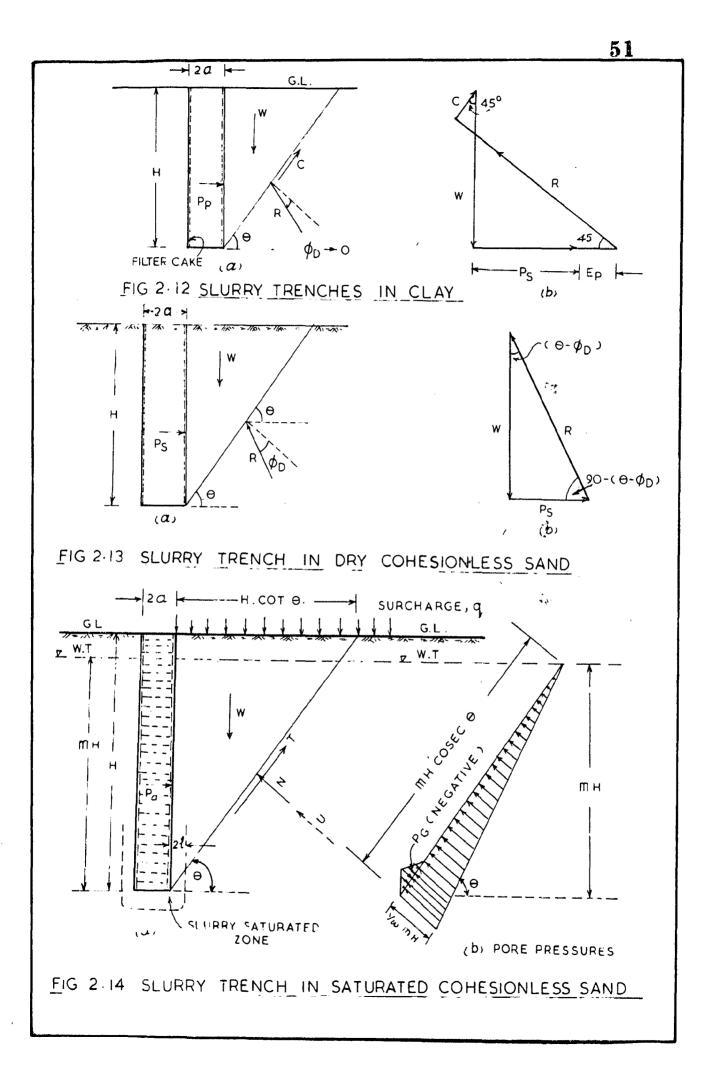
If the wedge is in horizontal equilibrium

 $P_a + T \cos \theta = N \sin \theta$...(2.24) If it is in vertical equilibrium,

 $W = T \sin \theta + N \cos \theta \qquad \dots (2.25)$ The tangential force T is given by $T = N' \tan \phi' = (N - u) \tan \phi' \qquad \dots (2.26)$

where N' is the effective force normal to the slip plane and u denotes the net neutral force on the slip surface, figure 2.14(b). From equations (2.24),(2.25) and (2.26),

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$$P_{a} = \frac{W(\sin \theta - \cos \theta \tan \phi') + U \tan \phi'}{\cos \theta + \sin \theta \tan \phi'} \qquad \dots (2.27)$$
Now $W = \cot \theta \left(\frac{1}{2} + Y + H^{2} + q + H\right)$
 $u = \frac{1}{2} + Y_{W} (mH)^{2} \operatorname{cosec} \theta - P_{G}$
where, $P_{G} = \frac{2 + T_{m} + H^{2}}{r \cos \theta}$ (equation 2.10a)

The total passive resistance offered by the slurry is given by

$$P_{p} = P_{s} + E_{p}$$

$$= \frac{1}{2} Y_{s} H^{2} + (\tau_{s} H^{2} + \frac{\pi \tau_{s} H}{2})$$
If for stability $P_{a} = P_{p}$, then
$$\frac{1}{2} Y_{s} H^{2} + \left[\frac{\tau_{s} H^{2}}{2a} + \frac{\pi \tau_{s} H}{2}\right]$$

$$= \frac{\cot \theta (\frac{1}{2}YH^{2} + qH) (\sin \theta - \cos \theta \tan \phi') + (\frac{1}{2}Y_{w}m^{2}H^{2}\cos ec\theta - P_{c})}{\cos \theta + \sin \theta \tan \phi'}$$

...(2.28)

The two terms on the left hand side , P_s and F_p respectively, have been discussed. The first term on the right hand side represents the active earth pressure of the sliding wedge of the soil (including the effect of saturation) and the

second term represents the stabilizing effect of the slurry saturated zone in horizontal direction.

2.2.4 Quantitative Effect of Factors Affecting Stability of Slurry Trenches

The quantitative contribution of the factors affecting stability of trenches discussed in the foregoing pages will be

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demonstrated by two methods - one involving the results of W.K. Elson's experimental work outlined in para 2.2.1.2 and the second from calculations based on the application of the stability analysis theory discussed in para 2.2.3 on a field trench 3 .

2.2.4.1 Elson's Results

¢'	= 33.1°	Н	= 36.6"
mH	= 34.2"		= 0.0745 lbs/in ³
Y W	$= 0.0362 lb/in^3$, q	= 0
Ύs	= 0.037 lbs/in ³	е	= 0.545,
Tm	= 0.41 lbf/ft^2	۳ _s	= 0.41 lbs $/ft^2$
a	= 2"	r	= 0.0019",
ł	= 1"		•

The calculations are summarised in Table 2.4.

2.2.4.2 Stability of Full Scale Field Trench

T.C. Gupta³ has carried out calculations of the magnitudes of stabilizing forces and their percentage of the disturbing force for the Pierre Benite Trench. These calculations are reproduced on Table 2.5.

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TS OF ELSON'S	and the second se
RESULTS OF	
2.4	
TABLE	

		%L	7% Slurry			F.	4% Slurry	Ŕ.	
Width of Trench	Force	Hansen		Coulomb		Hansen	c	Coulomb	qu
uŗ			Q	4 -	ما	-	ର	~	N N
	Hydrostatic Passive Slurry resistance	77 • 4 18 • 4	77.3. 17.5	80.4 19.1	80.2 18.3	80.2 3.6	77.1 3.7	84.0 3.8	85.3 4.1
4	Sharing resistance of slurry saturated zone		6.8 6.1	- 5 .9	- 2 - 0 - t-	13.6 2.7	18•7 0•5	12.7	16.0 -5.4
	Disturbing	100 . 0	100.0	100.0	100.0	100.0	þ	100.0	100.0
	Hydrostatic	64.1		70.0		76.5		81.4	
·	Passive slurry resistance	27.1		30•0		6.0		6.4	
0	Shearing resistance of slurry saturated zone	4.9		L4 • L4		14.5		13.4	
	Unaccounted	3.9	·	-4.4		3•0		-1.2	
	Disturbing	100.0		100 • 0	ı	100 • 0		100 • 0	
	Hydrostatic Passive slurry resistance		1			67 •5 18 •5		71.5 19.8	
4	Shearing resistance of slurry saturated zone					12.7		11 	
	Unaccounted Di sturbing					100		100	
<u>N .B. (</u> 3	(a) Forces are expressed as % of	f the distal	the disturbing the stabilizing	•	xceeds t	force. force exceeds the disturbing force.	r bing	force.	

(b) Negative sign indicates that the stabilizing force exceeds the disturbing force.

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TABLE 2.5 STABILITY OF PIERRE BENITE TRENCH

Data	Permeability of soil 1.0 t	o 0,1 cm/sec.
		ntonite (Fb $-2-5$)
	Slurry Density Y : i. Initi ii. In tr	al = 1.025 T/m^3 ench=1.2 to 1.25 T/m^3
	Bulk Saturated de asity of fill Y	~
Assumpti	$ion s: \phi' = 30^{\circ}, G = 2.65$	
	$\tau_{\rm f} = 0.05 {\rm gm/cm}^2$, $\tau_{\rm m} =$	0.75 gm/cm ²
	$\mathbf{r}_{s} = \mathbf{r}_{M} = 0.75 \text{ gm/cm}^2$	

<u>Trench Dimensions</u> L = 9 to 25 metres , H = 3.5 metre. 2a = 0.6 metre mH = 3.5 metre

Nature of force	Magnitude T/m	% of Disturbing force.
Total disturbing	7.85	100
Stabilizing forces i. Hydrostatic (P _s)	6.77	86.2
ii. Passive resistance of slurry (E _p)	0.18	2.3
iii. Shearing resistance of slurry saturated zone in horizontal direction	0.67	8•5
Total stabilizing forces	7.62	97.0

Tables2.4 and 2.5 indicate at a glance the quantitative contribution of the major three stabilizing forces of hydrostatic force, the passive resistance of the slurry and the shearing resistance of the slurry saturated zone. The unaccounted forces in Table 2.4 can be considered as the 3% (of the disturbing force) shortage in Table 2.5.

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CHAPTER III

EXPERIMENTAL SETUP, PROCEDURE AND RESULTS

The setup of this experiment was a simplified version of Elson's experiment discussed in para 2.2.1.2..

3.1 APPARATUS

This consisted of a mixer for the slurry, stirrer, a prismatic container fitted with a capillary tube for flow test, and a box in which the loading tests were carried. Loading device, containers like basins , and buckets, measuring cylinders, weighing apparatus, a stop watch and sieving apparatus formed the supplementary apparatus.

3.1.1. Mixer

This is a cylindrical container, about 0.75 litre in capacity in which a stirrer driven by an electric motor could be inserted.

3.1.2 Loading Test Box

This measured 38.6 x 26.0 x 20 cm internally and was divided into two halves by a 20 x 20 cm plate sliding through grooves in the walls and floor of the box. The box walls and floor as well as the partition plate were made from perspex plates to facilitate viewing of the sand and slurry inside. The box walls and floor plates were joined by chloroform and adequately stiffened by steel angles. The joints were made watertight by smearing grease along them. A wooden plank was placed under the bottom of the box to prevent sagging or bending of the floor plate. A perspex plate was also provided to be placed on the sand surface to distribute the load uniformly on the sand surface. A sketch of the box is shown in Figure 3.1(a) and its photograph in plate 3.1.

3.1.3 Flow Test Model

This is also a perspex prismatic box 10 x 10 cm in cross section and 44 cm in height. It was fitted with a capillary tube of 0.195 cm internal radius and 61.3 cm long (Another tube of the same radius and 63.2 cm long was substituted later). A scale was fitted beside the walls of the box to record slurry levels above the capillary tube. A sketch of it is shown in Figure 3.1(b) and a photograph in plate 3.2.

3.2 MATERIALS

Commercial bentonite, black cotton soil, fine sand and water were used. The commercial bentonite had the problem of segregation for pure bentonite slurries of lesser concentration than 4%. Virtually all the slurries between 1 and 4% concentration with replacement of bentonite by black cotton soil displayed the segregation problem.

The black cotton soil was ground and sieved through 75 μ (0.075mm) sieve size.

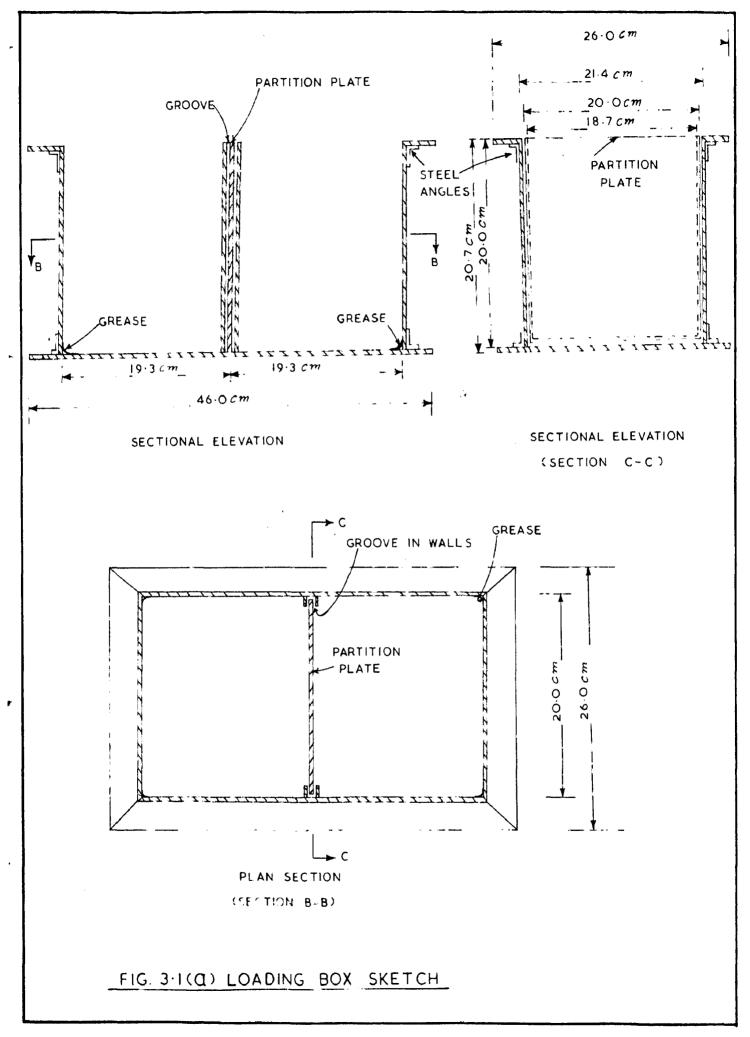
3.3 PROCEDURE

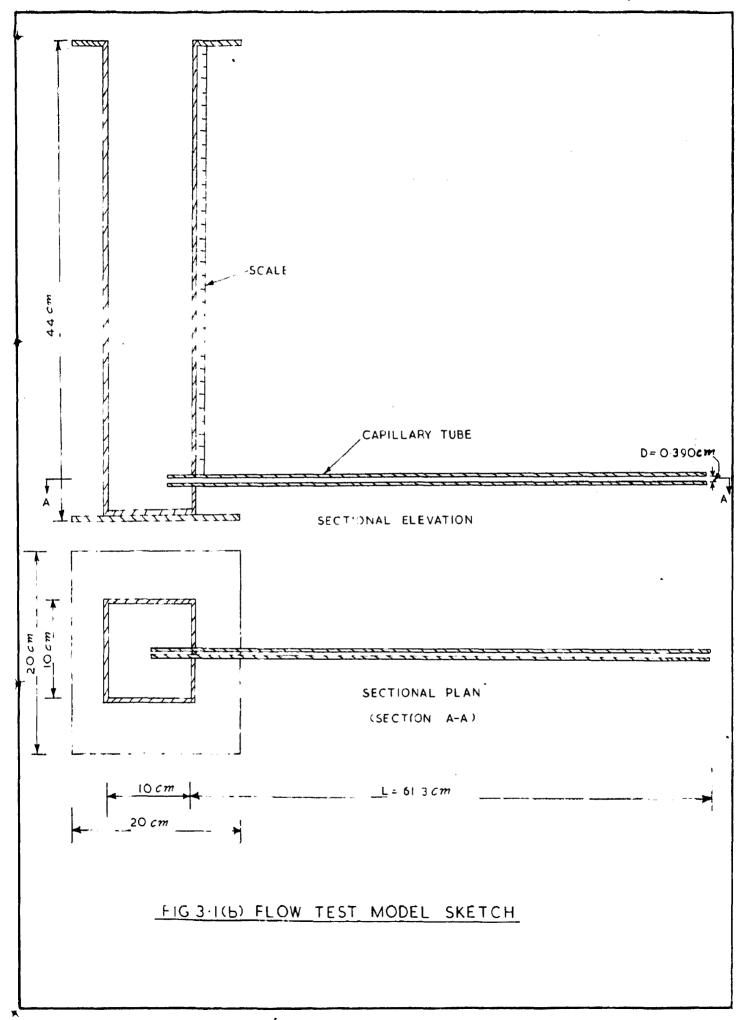
3.3.1. Mixing

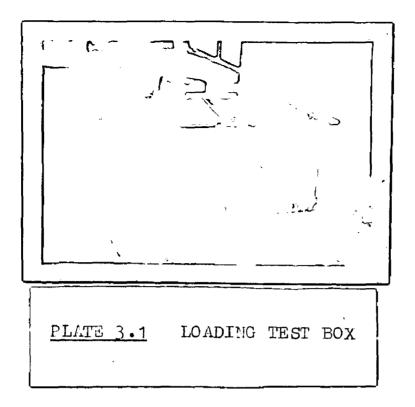
Slurries of 1% to 4% concentration were made. For each concentration, bentonite was replaced by black cotton soil in stages of 25% of the bentonite portion by weight. This resulted in 20 different slurry concentrations.

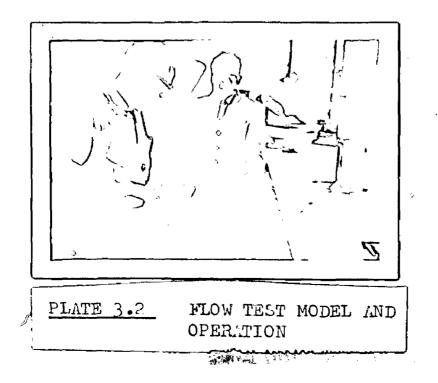
The slurry concentration, SC, was expressed as -



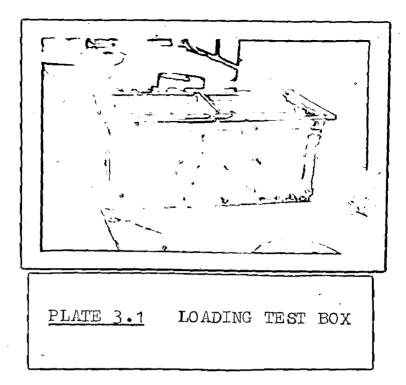


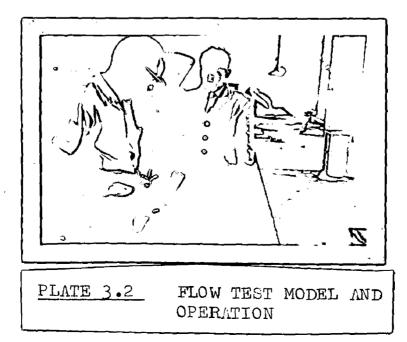






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Weight of solid portion plus weight of water

...(3.1)

In order to have sufficient slurry quantity the amount of water for each slurry was taken as 10 kg (10,000 cc). The requisite amounts of water and bentonite/black cotton soil for each slurry concentration were calculated from equation (3.1) above and the quantities obtained by weighing and measuring. Small portions of the constituents of each slurry concentration were mixed in the mixer, each portion being mixed for one minute. All the portions were discharged into a bucket for further **a**gitation with the stiring rod.

3.3.2 Slurry Density

The slurry density was obtained by dividing the total weight by the total volume taken, taking the specific gravity of solid bentonite and black cotton soil as 2.55. A sample calculation of this is shown on para 3.5.

3.3.3 Flow Test

Portion of the slurry from the bucket was tested for flow properties in the flow test model. The model was filled with slurry and the capillary tube was opened to allow the slurry to run freely. The tube was then closed, and the level of the slurry was noted. Quantities of slurry were collected in a measuring cylinder for a known period of time. (Plate 3.2). The final slurry level was noted The slurry was kept in agitation throughout to prevent segregation. The level for no flow was noted. Flow results and calculations are shown in Table 3.1. Graphs of Rate of Shear strain Versus shear stress are shown in Figure 3.2(a),(b),(c) and (d).

3.3.4 Loading Test

A loading test was carried out for each slurry concentration. With the partition plate in position, one half of the loading test box was filled with dry fine sand (gradation curve shown in Fig. 3.3) . The top of the sand was levelled off and the loading plate was placed on the sand surface. Slurry was then poured into the other half of the box and agitated. The partition plate was then slowly removed. After bulling up a small portion of the plate the slurry was allowed to saturate the sand. The slurry which flowed into the sand was replenished to the level of the sand. Further withdrawal of the plate was continued until all of it was removed. If there was no failure, the loading plate was loaded till failure occurred.

Two methods of loading were used . The first involved placing known loads on the loading plate until failure was achieved. Plate 3.3 shows this operation. For reasons explained later, this method of loading was found unsuitable and unsafe.

The other method was the adoption of the triaxial loading machine. The cell unit was removed leaving the raising floor on which the wooden plank could be placed. The loading test box was placed on the wooden plank and sand poured into the box as explained abov. A small cylindrical wooden piece, and frictionless balls were used to make contact between the loading plate and the calibrated proving ring of the machine. Slurry was poured into the other

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half as before, the machine was levelled for loading and the partition plate was removed **as** explained before. This set up is shown in plate 3.4. The slurry was allowed to penetrate the sand for 15 minutes. The depth of the slurry saturated zone was measured. The floor of the machine was raised by turning the handle manually and slowly as shown in Plate 3.4. This raised the wooden plank and the box and compressed the sand against the loading plate. This compression was thus communicated to the proving ' ring and the load could be recorded. The sand was loaded to failure and the failure load was noted. The weights of the loading plate, and the wooden cylindrical piece were added to the recorded failure load. The loading test results on dry sand and calculations are shown on Table 3.3.

The same loading procedure was repeated for saturated sand. These results are shown on Table 3.4.

3.3.5 Sand Density

As the sand was fine its density could be easily determined by a 25 ml specific gravity bottle. A physical balance was used for weighing.

3.3.6 Sand Sieving

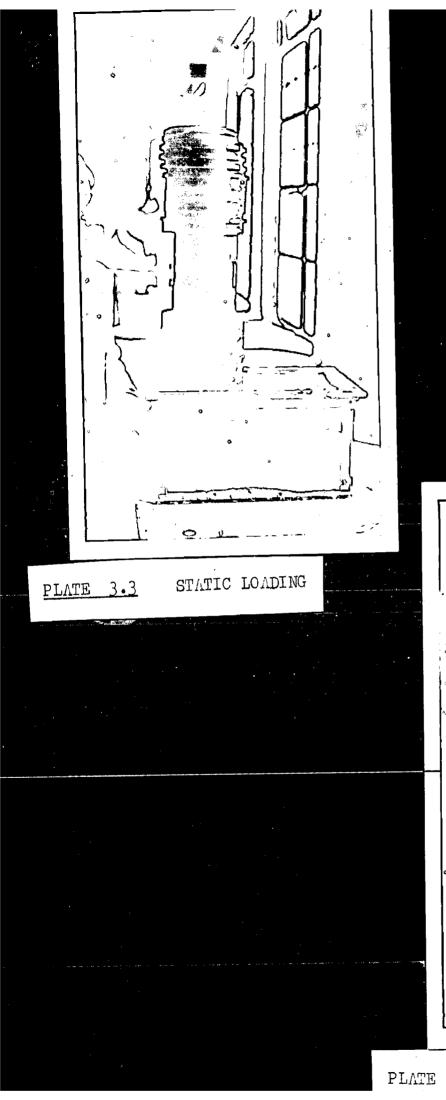
This was done in order to obtain the effective radius of pore passage. The sieving results are tabulated on Table 3.2. and the grain size distribution curve is shown on Figure 3.3.

3.4 RESULTS AND CALCULATIONS

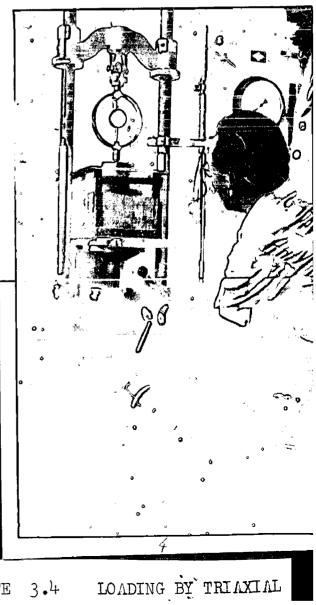
All the observed results and calculations are summarized in para 3.5 Flow test results and calculations are shown on

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14







3•4

Table 3.1, and plotted on Figures 3.2(a), (b),(c) and (d). From these the plastic viscosity \mathbf{T}_{PL} has been determined. Sieve analysis is doalt with on para 3.5.2, and para 3.5.3 deals with the results and calculations of stability. All the results for dry sand loading are summarised on Table 3.3, and on saturated and on table 3.4. Figure 3.4 shows the relation between the failure surcharge and the percentage replacement of bentonite by black cotton soil.

3.5 CALCULATIONS

3.5.1 Flow Test Computations

Length of tube L = 61.30 cm. Radius R = 0.195 cm. Ah= difference in slurry levels in cm. Q = rate of flow in cc/second.

Hydraulic gradient i = $\frac{\Delta h}{\Delta L} = \frac{\Delta h}{61.30} = 0.0164$ h. Rate of shear strain = $\frac{dg}{dt} = \frac{4Q}{\pi R^3} = \frac{4Q}{\pi (0.195)^3} = 171.30$ (Sec⁻¹)

Shear stress
$$r_R = \frac{1}{2} i_1 = \frac{1}{2} \frac{\Delta F}{\Delta L} = \frac{0.195}{2} \frac{\Delta F}{\Delta L} = \frac{1}{2} \frac{\Delta F}{\Delta L} = \frac{1}{2} \frac{\Delta F}{\Delta L} = \frac{1}{2} \frac{1}{2}$$

$$= \frac{0.195}{2} \frac{\Delta h Y_s}{61.30} \times 981 \text{ dynes/cm}^2$$
$$= \frac{0.195 \Delta h Y_s}{2 \times 61.3} \text{ gm/cm}^2$$
$$= 0.0016 Y_s \Delta h \text{ gm/cm}^2$$

The density of the slurry, Y_s , is calculated from the mixture materials, taking the dry specific gravity (G_s) of clay as 2.55.

e.g. For 1% pure bentonite, slurry,

Weight Taken	Volume
Water 10,000 gm	10,000 cc
Rentonite 101 gm	$V = \frac{W}{G_s} = \frac{101}{2.55} = 39.6 \text{ cc}$
Total weight=10,101.00gm	Total volume=10,039.60 cc

 $Y_s = \frac{\text{Total weight}}{\text{Total volume}} = \frac{10,101.00}{10,039.60} \text{ gm/cm}^3$ = 1.005 gm/cm³

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TABLE 3.1 - FLOW TEST RESULTS AND CALCULATIONS

	<u></u>			·····			t
Initial	Final	Time	Volume		Volume '	dE _s /dt'	T_{R}
level h	Lever u	in sec. t	in cc V	head Ah	per cc V	=1 71.32	=0.0016Y _s ∆h
(cm)	(cm)		•	(h_1+h_2)	$\frac{\lambda^2}{t}$		gm/cm ²
				(c ²)	1		
1	2	3	4	5	6	1 7	8
1% Pur	e Bentoni	te Slurr	<u>y</u> , Y	s = 1.00	5 gm/cm^3		
31.00	26.70	34.30	350	28.85	10.20	1750	0.0464
26.70	21.20	1+6.90	445	23.95	9•50	1630	0.0384
21.20	18,90	18.90	170	20.05	9.00	1540	0.0322
18.90	16.30	24.90	210	17.60	8.45	1450	0.0282
16.30	13.80	31.10	240	15.05	7.70	1320	0.0242
13.80	10.40	34.10	230	12.10	6.75	1157	0.0194
10.40	7.80	40.70	220	9.10	5.40	925	0.0146
7.80	5.10	51.40	210	6.45	4.10	702	0.0104
5.10	3.20	56.00	160	4.15	2.86	490	0.0067
I	Head for	no flow	= 0.00	cm , T _R	1 = 0 g	m ∕ cn ²	
1% Slui	rry with	25% Repl	acement	of Bent	onite by	black co	tton soil
			.			······	
s = 1	.005 gm/c	m					
34.70	31.20	35.90	2 9 0	32.95	11.20	1920	0.0530
31.20	27.10	31.40	380	29.15	12.10	2075	0.0468
27.10	23.20	32.90	325	25.15	9.80	1680	0.0404
23.20	19.50	33.30	300	21.35	9.00	1540	0.0342
19.50	16.20	31.00	260	17.85	8.40	1440	0.0286
16.20	12.90	34.30	264	15.55	7.70	1320	0.0250
12.90	10.10	33.10	222	11.50	6.70	1150	0.0185
10.10	7.70	37.00	206	8.90	5.57	955	0.0143
7.70	5.30	44.20	190	6.50	4.30	737	0.0104
5.30	3.70	43.30	137	与•50	3.16	541	0.0072
3.70	2.40	46.80	100	3.05	2.14	367	0.0049
				-			
Hea	ad for no	flow =	0.00cm,	$T_{R1} = ($	gm/cm^2		
					~	~ ~	
~ ~							
		10					

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	2	3	4	5	6	7	8
<u>1% Slu</u>	rry with	. 50% R	eplacen	ent of be	ntonite	by Black	cotton soi
$\Upsilon_{s} = 1$.005 gm/	cm ³			· · · ·		
31.70	29.10	19.60	205	30.1+0	10.43	179 0	0.0490
29.10	26.10	25.00	252	27.60	10.10	1730	0.0444
26.10	23.30	23.00	223	24.70	9.70	16 6 0	0.0397
23.30	20.1+0	25.50	233	21.85	9.15	1570	0.0352
20.40	17.00	31.30	270	18.70	8.64	1480	0,0300
17.00	13.80	23.40	264	15.40	7.90	135 3	0.0248
13,80	11.30	29.40	201	12.55	6.85	1172	0.0202
11.30	8.70	36.50	215	10.50	5.90	1010	0.0169
8.70	6.30	39.10	187	7.50	4.78	810	0.0120
6.30	ı _{+•} 50	43.90	157	5.40	3•58	615	0.0087
4.50	3.00	1 45 •7 0	118	3.75	2.58	<u>4</u> ,1	0.0060
-	He	ad for no	o flow		•		
					<pre>/ ``D1</pre>	- 6,	
1% Slu	 rry with	75% rep]	 Lacemen	411 May 418 Alg	***		
		•	Lacemen	411 May 418 Alg	***		otton soil
	1,005 g	•	Lacemen 302	411 May 418 Alg	***		
δ Y _S =	1,005 g	m/cm ³		t of Bent	onite by	black co	otton soil
δ Y _s = 35•50	1,005 g 31.70	m/cm ³ 23.50	302	t of Bent 33.60	onite by 11.90	black co 2040	0.0540
δ Y _s = 35.50 31.70	1,005 g 31.70 27.70	m/cm ³ 25.50 30.90	302 329	t of Bent 33.60 29.70	onite by 11.90 11.65	black cc 2040 2000	0.0540 0.04 7 7
δ Y _s = 35.50 31.70 27.70	1,005 gr 31.70 27.70 23.60	m/cm ³ 25.50 30.90 32.90	302 329 325	t of Bent 33.60 29.70 25.65	onite by 11.90 11.65 9.90	black co 2040 2000 1690 1540	0.0540 0.0477 0.0422
δ Y _s = 35.50 31.70 27.70 23.60	1,005 gr 31.70 27.70 23.60 20.20	m/cm ³ 25.50 30.90 32.90 30.50	302 329 325 275	t of Bent 33.60 29.70 25.65 21.90	onite by 11.90 11.65 9.90 9.00	black co 2040 2000 1690	0.0540 0.0477 0.0422 0.0352
δ Y _s = 35.50 31.70 27.70 23.60 20.20	1,005 gr 31.70 27.70 23.60 20.20 16.30	m/cm ³ 25.50 30.90 32.90 30.50 37.70	302 329 325 275 315	t of Bent 33.60 29.70 25.65 21.90 18.25	onite by 11.90 11.65 9.90 9.00 8.36	black co 2040 2000 1690 1540 14 3 5	0.0540 0.0477 0.0422 0.0352 0.0295
δ Y _s = 35.50 31.70 27.70 23.60 20.20 16.30	1,005 gr 31.70 27.70 23.60 20.20 16.30 13.00	m/cm ³ 25.50 30.90 32.90 30.50 37.70 34.50	302 329 325 275 315 262	t of Bent 33.60 29.70 25.65 21.90 18.25 14.65	onite by 11.90 11.65 9.90 9.00 8.36 7.60	black co 2040 2000 1690 1540 14 35 1300	0.0540 0.0477 0.0422 0.0352 0.0295 0.0236
δ Y _s = 35.50 31.70 27.70 23.60 20.20 16.30 13.00	1,005 gr 31.70 27.70 23.60 20.20 16.30 13.00 9.90	m/cm ³ 25.50 30.90 32.90 30.50 37.70 34.50 35.90	302 329 325 275 315 262 245	t of Bent 33.60 29.70 25.65 21.90 18.25 14.65 11.45	onite by 11.90 11.65 9.90 9.00 8.36 7.60 6.65	black co 2040 2000 1690 1540 14 35 1300 1140	0.0540 0.0477 0.0422 0.0352 0.0295 0.0236 0.0184
$\delta Y_{s} = 35.50$ 31.70 27.70 23.60 20.20 16.30 13.00 9.90	1,005 gr 31.70 27.70 23.60 20.20 16.30 13.00 9.90 7.40	m/cm ³ 25.50 30.90 32.90 30.50 37.70 34.50 35.90 36.50	302 329 325 275 315 262 245 210	t of Bent 33.60 29.70 25.65 21.90 18.25 14.65 11.45 8.65	onite by 11.90 11.65 9.90 9.00 8.36 7.60 6.65 5.75	black co 2040 2000 1690 1540 14 35 1300 1140 985	0.0540 0.0477 0.0422 0.0352 0.0295 0.0236 0.0184 0.0139

1	2	3	4	5	6	7	8
1% Slu	arry with	100% bla	ack cot	ton soil,	$\Upsilon_{s} = 1.0$	005 gm/c	; _m 3
37.10	32.80	29.30	237	34•95	8.10	1390	0.0560
32.80	29.00	28.00	307	30.90	10.98	1880	0.0496
29.00	25.10	30.80	315	27.05	10.22	1755	0.0435
25.10	21.30	33.50	315	23.20	9.40	1610	0.0373
21.30	17.80	32.00	275	19•55	8.60	1472	0.0314
17.80	14.90	30.10	240	16.35	7.97	1365	0.0263
14.90	12.00	31.90	235	13•45	7.38	1262	0.0216
12.00	9.30	35.20	230	10.65	6.54	1120	0.0171
9.30	6.90	36.00	194	7•95	5.40	925	0.0128
6.90	5.20	33.10	142	6.05	4.30	736	0,0097
5.20	3.70	38.50	125	4.45	3.25	556	0.0072
3.70	2.50	46.50	95	2.10	2.04	350	0.0034
	Hond for	n no flot	r = 0.0	~ ~	-	. 2	
			· = · •	O_{cm}, T_{R}	1 = 0. gm/	cm ²	
 2% slu				$C_{\rm cm}, T_{\rm R}$		/ cm ²	
2% slu 36.00						1945	0.0555
	erry (100%	% Bentoni	te) Y	s = 1.013	gm/cm ³		0.0555 0.0491
36.00	100% 32.60	8 Bentoni 23.80	lte) Y 270	s = 1.013 34.30	gm/cm ³ 11.34	1945	
36.00 32.60	urry (100% 32.60 28.50	8 Bentoni 23.80 30.60	te) Y 270 305	s = 1.013 34.30 30.55	gm/cm ³ 11.34 9.98	1945 1710	0.0491
36.00 32.60 28.50	urry (100% 32.60 28.50 24.80	% Bentoni 23.80 30.60 28.80	te) Y 270 305 290	s = 1.013 34.30 30.55 26.60	gm/cm ³ 11.34 9.98 10.08	1945 1710 1730	0.0491 0.0430
36.00 32.60 28.50 24.80	urry (100% 32.60 28.50 24.80 21.30	% Bentoni 23.80 30.60 28.80 29.50	ete) Y 270 305 290 280	s = 1.013 34.30 30.55 26.60 23.05	gm/cm ³ 11.34 9.98 10.08 9.50	1945 1710 1730 1630	0.0491 0.0430 0.0370
36.00 32.60 28.50 24.80 21.0	urry (100% 32.60 28.50 24.80 21.30 18.00	8 Bentoni 23.80 30.60 28.80 29.50 30.50	270 270 305 290 280 270	s = 1.013 34.30 30.55 26.60 23.05 19.65	gm/cm ³ 11.34 9.98 10.08 9.50 8.86	1945 1710 1730 1630 1520	0.0491 0.0430 0.0370 0.0318
36.00 32.60 28.50 24.80 21.50 18.00	urry (100% 32.60 28.50 24.80 21.30 18.00 15.10	 Bentoni 23.80 30.60 28.80 29.50 30.50 29.50 	270 305 290 280 270 230	s = 1.013 34.30 30.55 26.60 23.05 19.65 16.55	gm/cm ³ 11.34 9.98 10.08 9.50 8.86 7.80	1945 1710 1730 1630 1520 1338	0.0491 0.0430 0.0370 0.0318 0.0268
36.00 32.60 28.50 24.80 21.0 18.00 15.10	urry (100% 32.60 28.50 24.80 21.30 18.00 15.10 12.80	8 Bentoni 23.80 30.60 28.80 29.50 30.50 29.50 29.50 26.50	ete) Y 270 305 290 280 270 230 180	s = 1.013 34.30 30.55 26.60 23.05 19.65 16.55 13.95	gm/cm ³ 11.34 9.98 10.08 9.50 8.86 7.80 6.80	1945 1710 1730 1630 1520 1338 1165	0.0491 0.0430 0.0370 0.0318 0.0268 0.0226 0.0189
36.00 32.60 28.50 24.80 21.0 18.00 15.10 12.80	urry (100% 32.60 28.50 24.80 21.30 18.00 15.10 12.80 10.50	<pre>% Bentoni 23.80 30.60 28.80 29.50 30.50 29.50 26.50 30.90</pre>	ete) Y 270 305 290 280 270 230 180 185	s = 1.013 34.30 30.55 26.60 23.05 19.65 16.55 13.95 11.65	gm/cm ³ 11.34 9.98 10.08 9.50 8.86 7.80 6.80 6.00	1945 1710 1730 1630 1520 1338 1165 1030	0.0491 0.0430 0.0370 0.0318 0.0268 0.0226
36.00 32.60 28.50 24.80 21.0 18.00 15.10 12.80 10.50	urry (100% 32.60 28.50 24.80 21.30 18.00 15.10 12.80 10.50 8.50	8 Bentoni 23.80 30.60 28.80 29.50 30.50 29.50 26.50 30.90 32.10	ete) Y 270 305 290 280 270 230 180 185 165	s = 1.013 34.30 30.55 26.60 23.05 19.65 16.55 13.95 11.65 9.50	gm/cm ³ 11.34 9.98 10.08 9.50 8.86 7.80 6.80 6.00 5.15	1945 1710 1730 1630 1520 1338 1165 1030 883	0.0491 0.0430 0.0370 0.0318 0.0268 0.0226 0.0189 0.0154
36.00 32.60 28.50 24.80 21.0 18.00 15.10 12.80 10.50 8.50	urry (100% 32.60 28.50 24.80 21.30 18.00 15.10 12.80 10.50 8.50 6.70	<pre>% Bentoni 23.80 30.60 28.80 29.50 30.50 29.50 26.50 30.90 32.10 34.80</pre>	ete) Y 270 305 290 280 270 230 180 185 165 147	s = 1.013 34.30 30.55 26.60 23.05 19.65 16.55 13.95 11.65 9.50 7.60	gm/cm ³ 11.34 9.98 10.08 9.50 8.86 7.80 6.80 6.00 5.15 4.21	1945 1710 1730 1630 1520 1338 1165 1030 883 725	0.0491 0.0430 0.0370 0.0318 0.0268 0.0226 0.0189 0.0154 0.0123

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1	2	3	14	5	6	7	8
2% Slu	rry with	25% Rep]	Lacemen	t of Bent	onite by	Black	Cotton Soi
Y _s =	1.013 gm	$/cm^3$					
36.30	33.20	21.70	250	34•75	11.50	1 97 0	0.0561
33.20	29.50	27.00	293	31.35	10.84	1860	0.0506
29.50	25.10	34.70	355	27.30	10.18	1740	0.0491
25.10	21.50	30.00	286	23.30	9•54	1630	0.0378
21.50	18.10	31.00	275	19.80	8.90	1523	0.0320
18.10	14.70	33.10	265	16.40	8.00	1 370	0.0265
14.70	11,90	33.15	233	13.30	7.05	1210	0.0216
11.90	9.20	36.10	214	10.55	5.92	1015	0.0171
9.20	7.20	33.70	163	8.20	<u>}+•8</u> }+	827	0.0133
7.20	5•50	31+•00	133	6.35	3.91	670	0.0103
5 •50	4.00	37.50	114	4.75	3.04	520	0.0077
4.00 H	2.90 lead for	45.70 no flow =	100	3.145 cm, ^T R1	2.19 = 0 gm/cr	376 n ²	0.0056
H 2% Şlu	2.90 ead for	45.70 no flow =	100	3.145 cm, ^T R1	2.19 = 0 gm/cr	376 n ²	0.0056
4.00 H $2\% \text{ Slu}$ $\Upsilon_{\text{S}} = 1$	2.90 ead for rry with .013 gm/	45.70 no flow = 50% rep: rcm ³	100 = 0.00 lacemen	3.45 ^{cm} , ^T R1 t of Bent	2.19 = 0 gm/cr	376 n ² black	0.0056 cotton soi
4.00 H 2% $$lu\Upsilon_{s} = 137.70$	2.90 ead for rry with .013 gm/ 34.10	45.70 no flow = 50% rep: cm ³ 25.60	100 = 0.00 lacemen 295	3.45 ^{cm} , ^T R1 t of Bent 35.90	2.19 = 0 gm/cr conite by 11.50	376 n ² black 1970	0.0056 cotton soi 0.0580
4.00 H 2% \$1u $\Upsilon_{s} = 1$ 37.70 34.10	2.90 ead for rry with .013 gm/ 34.10 30.30	45.70 no flow = 50% rep: cm ³ 25.60 27.70	100 = 0.00 lacemen 295 305	3.45 cm , T _{R1} t of Bent 35.90 32.20	2.19 = 0 gm/cr conite by 11.50 11.00	376 n ² black 1970 1885	0.0056 cotton soi 0.0580 0.0520
4.00 H 2% \$1u $\Upsilon_{s} = 1$ 37.70 34.10 30.30	2.90 ead for rry with .013 gm/ 34.10 30.30 26.30	45.70 no flow = 50% rep: 7cm ³ 25.60 27.70 31.50	100 = 0.00 lacemen 295 305 326	3.45 cm , T _{R1} t of Bent 35.90 32.20 28.30	2.19 = 0 gm/cr conite by 11.50	376 n ² black 1970 1885 1770	0.0056 cotton soi 0.0580 0.0520 0.0458
4.00 H 2% \$1u $\Upsilon_{s} = 1$ 37.70 34.10 30.30 26.30	2.90 ead for rry with .013 gm/ 34.10 30.30	45.70 no flow = 50% rep: 7cm ³ 25.60 27.70 31.50 26.10	100 = 0.00 lacemen 295 305 326 250	3.45 cm , T _{R1} t of Bent 35.90 32.20	2.19 = 0 gm/cr conite by 11.50 11.00 10.30	376 n ² black 1970 1885	0.0056 cotton soi 0.0580 0.0520 0.0458 0.0400
4.00 H 2% \$1u $\Upsilon_{s} = 1$ 37.70 34.10 30.30 26.30 23.20	2.90 ead for rry with .013 gm/ 34.10 30.30 26.30 23.20	45.70 no flow = 50% rep: 7cm ³ 25.60 27.70 31.50 26.10 26.50	100 = 0.00 lacemen 295 305 326 250 242	3.45 cm , T _{R1} t of Bent 35.90 32.20 28.30 24.75	2.19 = 0 gm/cr conite by 11.50 11.00 10.30 9.60	376 n ² black 1970 1885 1770 1645	0.0056 cotton soi 0.0580 0.0520 0.0458 0.0400 0.0352
4.00 H 2% \$1u $\Upsilon_{s} = 1$ 37.70 34.10 30.30 26.30	2.90 ead for rry with .013 gm/ 34.10 30.30 26.30 23.20 20.20	45.70 no flow = 50% rep: 7cm ³ 25.60 27.70 31.50 26.10	100 = 0.00 lacemen 295 305 326 250	3.45 cm, T _{R1} t of Bent 35.90 32.20 28.30 24.75 21.70	2.19 = 0 gm/cr conite by 11.50 11.00 10.30 9.60 9.10	376 n ² black 1970 1885 1770 1645 1560	0.0056 cotton soi 0.0580 0.0520 0.0458 0.0400 0.0352 0.0302
4.00 H 2% \$1u $\Upsilon_{s} = 1$ 37.70 34.10 30.30 26.30 23.20 20.20	2.90 ead for rry with .013 gm/ 34.10 30.30 26.30 23.20 20.20 17.10	45.70 no flow = 50% rep: 7cm ³ 25.60 27.70 31.50 26.10 26.50 29.00	100 = 0.00 lacemen 295 305 326 250 242 250	3.45 cm, T _{R1} t of Bent 35.90 32.20 28.30 24.75 21.70 18.65	2.19 = 0 gm/cr conite by 11.50 11.00 10.30 9.60 9.10 8.60	376 n ² black 1970 1885 1770 1645 1560 1472	0.0056 cotton soi 0.0580 0.0520 0.0458 0.0400 0.0352 0.0302 0.0256
4.00 H 2% \$1u $\Upsilon_{s} = 1$ 37.70 34.10 30.30 26.30 23.20 20.20 17.10	2.90 ead for rry with .013 gm/ 34.10 30.30 26.30 23.20 20.20 17.10 14.20	45.70 no flow = 50% rep: cm ³ 25.60 27.70 31.50 26.10 26.50 29.00 30.10	100 = 0.00 lacemen 295 305 326 250 242 250 237	3.45 cm, T _{R1} t of Bent 35.90 32.20 28.30 24.75 21.70 18.65 15.65	2.19 = 0 gm/cm conite by 11.50 11.00 10.30 9.60 9.10 8.60 7.90 7.00	376 n ² black 1970 1885 1770 1645 1560 1472 1352	0.0056 cotton soi 0.0580 0.0520 0.0458 0.0400 0.0352 0.0302 0.0302 0.0256 0.0208
4.00 H 2% \$1u $\Upsilon_{s} = 1$ 37.70 34.10 30.30 26.30 23.20 20.20 17.10 14.20	2.90 ead for rry with .013 gm/ 34.10 30.30 26.30 23.20 20.20 17.10 14.20 11.50	45.70 no flow = 50% rep: cm ³ 25.60 27.70 31.50 26.10 26.50 29.00 30.10 31.50	100 = 0.00 lacemen 295 305 326 250 242 250 237 220	3.45 cm, T _{R1} t of Bent 35.90 32.20 28.30 24.75 21.70 18.65 15.65 12.85	2.19 = 0 gm/cm conite by 11.50 11.00 10.30 9.60 9.10 8.60 7.90 7.00	376 ² black 1970 1885 1770 1645 1560 1472 1352 1200	0.0056 cotton soi 0.0580 0.0520 0.0458 0.0400 0.0352 0.0302 0.0302 0.0256 0.0208 0.0168
4.00 H 2% §1u $\Upsilon_{s} = 1$ 37.70 34.10 30.30 26.30 23.20 20.20 17.10 14.20 11.50	2.90 ead for rry with .013 gm/ 34.10 30.30 26.30 23.20 20.20 17.10 14.20 11.50 9.20	45.70 no flow = 50% rep: 7cm ³ 25.60 27.70 31.50 26.10 26.50 29.00 30.10 31.50 33.00	100 = 0.00 lacemen 295 305 326 250 242 250 237 220 195	3.45 cm, T _{R1} t of Bent 35.90 32.20 28.30 24.75 21.70 18.65 15.65 12.85 10.35	2.19 = 0 gm/cm conite by 11.50 11.00 10.30 9.60 9.10 8.60 7.90 7.00 5.90	376 2 black 1970 1885 1770 1645 1560 1472 1352 1200 1010	0.0056 cotton soi 0.0580 0.0520 0.0458 0.0400 0.0352 0.0302 0.0302 0.0256 0.0208 0.0208 0.0168 0.0133

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1	2	3	4	5	6	7	8
2% Slu	rry with	1 75% Rep	lacemen	t of Bent	conite by	black	cotton scil
$\Upsilon_{s} = 1$.013 gm/	. _{cm} 3					
37.30	33.50	28.20	328	35.40	11.63	2000	0.0573
33.50	29.40	29 •50	324	31.145	11.00	1885	0.0510
29.40	25.30	33.50	344	27.35	10.27	1760	0.0443
25.30	21.70	31.50	297	23.50	9.42	1612	0.0380
21.70	18.60	27.60	245	20.15	8 .9 0	1525	0.0326
18.60	16.00	26.10	219	17.30	8.40	1440	0.0280
16.00	13.10	31.30	21+0	14.55	7.66	1312	0.0236
13.10	10.30	32.90	220	11.70	6.69	1142	0.0189
10 .30	7.80	35.30	1 9 6	9.05	555	950	0.0146
7.80	6.20	30.90	140	7.00	4.50	770	0.0113
6.20	4.40	40.60	142	5.30	3.48	596	0.0086
4.40	3.00	41.10	110	3.50	2.68	460	0.0057
		41.10 .0w = 00 (-	-	460	0.0057
Head f	or no fl	.ow = 00 d	cm T	$R_1 = 0 gr$	n∕cm²		
Head f	or no fl	.ow = 00 d	cm T	-	n∕cm²		
Head f	or no fl	.ow = 00 d	cm T	$R_1 = 0 gr$	n∕cm²		
Head f 2% Slu	or no fl	.ow = 00 (100% bla	em T	$R_1 = 0 gm$ ton soil	$\gamma_{\rm s} = 1.0$	013 gm/0	
Head f 2% Slu 37.80 33.80	for no fl arry with 33.80	.ow = 00 0 100% bla 25.30	em T ack cot 300	$R_1 = 0 gm$ ton soil 35.80	$Y_{s} = 1.0$ 11.87	013 gm/ 2035	^{cm³} 0.0580
Head f 2% Slu 37.80 33.80 29.90	or no fl arry with 33.80 29.90	.ow = 00 0 100% bla 25.30 28.70	em T ack cot 300 320	$R_1 = 0 gm$ ton soil 35.80 31.85	$Y_{s} = 1.0$ 11.87 11.12	013 gm/o 2035 1910	o.0580 0.0516
Head f 2% Slu 37.80 33.80 29.90 26.10	For no fl arry with 33.80 29.90 26.10	.ow = 00 0 100% bla 25.30 28.70 29.50	em T ack cot 300 320 304	R1 = 0 gm ton soil 35.80 31.85 28.00	$Y_s = 1.0$ 11.87 11.12 10.30	2035 2035 1910 1765	0.0580 0.0516 0.0453
Head f 2% Slu 37.80 33.80 29.90 26.10 21.60	For no fl arry with 33.80 29.90 26.10 21.60	.ow = 00 0 100% bla 25.30 28.70 29.50 38.30	em T ack cot 300 320 304 365	R1 = 0 gm ton soil 35.80 31.85 28.00 24.35	$Y_{s} = 1.0$ 11.87 11.12 10.30 9.53	2035 2035 1910 1765 1630	cm ³ 0.0580 0.0516 0.0453 0.0394
Head f 2% Slu 37.80 33.80 29.90 26.10 21.60 18.60	For no fl arry with 33.80 29.90 26.10 21.60 18.60	.ow = 00 0 100% bla 25.30 28.70 29.50 38.30 28.40	cm T ack cot 300 320 304 365 250	$R_{1} = 0 gm$ ton soil 35.80 31.85 28.00 24.35 20.10	$Y_{s} = 1.0$ 11.87 11.12 10.30 9.53 8.80	013 gm/ 2035 1910 1765 1630 1510	cm ³ 0.0580 0.0516 0.0453 0.0394 0.0326
Head f 2% Slu 37.80 33.80 29.90 26.10 21.60 18.60 15.50	For no fl arry with 33.80 29.90 26.10 21.60 18.60 15.50	.ow = 00 0 100% bla 25.30 28.70 29.50 38.30 28.40 31.00	em T ack cot 300 320 304 365 250 250	R1 = 0 gm ton soil 35.80 31.85 28.00 24.35 20.10 17.05	$Y_s = 1.0$ 11.87 11.12 10.30 9.53 8.80 8.07	2035 2035 1910 1765 1630 1510 1382	cm ³ 0.0580 0.0516 0.0453 0.0394 0.0326 0.0276
Head f 2% Slu 37.80 33.80 29.90 26.10 21.60 18.60 15.50 12.60	For no fl arry with 33.80 29.90 26.10 21.60 18.60 15.50 12.60	.ow = 00 4 100% bla 25.30 28.70 29.50 38.30 28.40 31.00 30.00	em T ack cot 300 320 304 365 250 250 230	R1 = 0 gm ton soil 35.80 31.85 28.00 24.35 20.10 17.05 14.05	$Y_s = 1.0$ $\frac{Y_s = 1.0}{11.87}$ 11.12 10.30 9.53 8.80 8.07 7.66	013 gm/ 2035 1910 1765 1630 1510 1382 1312	cm ³ 0.0580 0.0516 0.0453 0.0394 0.0326 0.0276 0.0228
Head f 2% Slu 37.80	For no fl arry with 33.80 29.90 26.10 21.60 18.60 15.50 12.60 10.00	.ow = 00 4 100% bla 25.30 28.70 29.50 38.30 28.40 31.00 30.00 31.00	em T ack cot 300 320 304 365 250 250 250 230 200	R1 = 0 gm ton soil 35.80 31.85 28.00 24.35 20.10 17.05 14.05 11.30	$Y_{s} = 1.0$ 11.87 11.12 10.30 9.53 8.80 8.07 7.66 6.45	013 gm/ 2035 1910 1765 1630 1510 1382 1312 1103	cm ³ 0.0580 0.0516 0.0453 0.0394 0.0326 0.0276 0.0228 0.0228 0.0183
Head f 2% Slu 37.80 33.80 29.90 26.10 21.60 18.60 15.50 12.60 10.00	For no fl arry with 33.80 29.90 26.10 21.60 18.60 15.50 12.60 10.00 7.60	$rac{100\% = 00}{25.30}$ 25.30 28.70 29.50 38.30 28.40 31.00 30.00 31.00 36.10	em T ack cot 300 320 304 365 250 250 250 230 200 197	R1 = 0 gm ton soil 35.80 31.85 28.00 24.35 20.10 17.05 14.05 11.30 8.80	$Y_{s} = 1.0$ 11.87 11.12 10.30 9.53 8.80 8.07 7.66 6.45 5.45	013 gm/ 2035 1910 1765 1630 1510 1382 1312 1103 958	cm ³ 0.0580 0.0516 0.0453 0.0394 0.0326 0.0276 0.0228 0.0183 0.0142

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1	2	3	4	5	6	7	8
3% S _{lu}	rry Pure	Bentoni	te,Y	s = 1.018	gm/cm ³		
35.90	32.00	27.90	319	. 33•95	11.65	2000	0.0552
32.00	27.90	30.10	335	29•95	11.10	1 9 00	0.0487
27.90	24.00	31.10	312	26.96	10.00	1713	0.0439
24.00	20.60	31.30	279	22.30	8.91	1530	0.0363
20.60	17.30	33.50	262	18.95	7.81	1340	0.0309
17.30	14.50	34.20	230	15.90	8.71	1150	0.0260
4.50	11.80	39.30	225	13.15	5.72	9 80	0.0214
11.80	9.50	38.00	185	10.65	4.86	834	0.0173
9.50	7.80	37.00	150	8•65	4.05	695	0.0141
7.80	6.20	39.60	130	7.00	3.28	561	0.0114
6.20	4 •7 0	47.00	118	5 • 45	2.51	430	0.0089
lead f	or no flo	w = 0.00) cm ,	$T_{R1} = 0$	gm/cm ²		
 3% Slu	rry with	25% Rep]	acemen	t of Bento	nite by	black o	otton soil
					SUT De DA		
re = 1	.018 gm/ c	_{cm} 3					
•	.018. gm/0 34.20		295	36.00		1990	0.0585
37.80		_{25.50} 25.50 28.60	295 315		11.58 11.00		
87.80 84.20	34.20	25.50		36.00	11.58	1990	0.0585
37.80 34.20 30.20	34•20 30•20	25.50 28.60	315	36.00 32.20	11.58 11.00	1 99 0 1888	0.0585 0.0525
37.80 34.20 30.20 26.40	34.20 30.20 26.40	25.50 28.60 29.90	315 310	36.00 32.20 28.30	11.58 11.00 10.37	1990 1888 1778	0.0585 0.0525 0.0460
$f_{s} = 1$ 37.80 34.20 30.20 26.40 21.40 8.20	34.20 30.20 26.40 21.40	25.50 28.60 29.90 41.40	315 310 405	36.00 32.20 28.30 23.90	11.58 11.00 10.37 9.80	1990 1888 1778 1678	0.0585 0.0525 0.0460 0.0389
37.80 34.20 30.20 26.40 21.40	34.20 30.20 26.40 21.40 18.20	25.50 28.60 29.90 41.40 28.10	315 310 405 250	36.00 32.20 28.30 23.90 19.80	11.58 11.00 10.37 9.80 8.90	1990 1888 1778 1678 1526	0.0585 0.0525 0.0460 0.0389 0.0322
37.80 34.20 30.20 26.40 21.40 8.20 5.30	34.20 30.20 26.40 21.40 18.20 15.30	25.50 28.60 29.90 41.40 28.10 29.50	315 310 405 250 235	36.00 32.20 28.30 23.90 19.80 16.75	11.58 11.00 10.37 9.80 8.90 7.96	1990 1888 1778 1678 1526 1363	0.0585 0.0525 0.0460 0.0389 0.0322 0.0273
57.80 34.20 30.20 26.40 21.40 8.20 5.30 2.70	34.20 30.20 26.40 21.40 18.20 15.30 12.70	25.50 28.60 29.90 41.40 28.10 29.50 31.00	315 310 405 250 235 215	36.00 32.20 28.30 23.90 19.80 16.75 14.00	11.58 11.00 10.37 9.80 8.90 7.96 6.94	1990 1888 1778 1678 1526 1363 1188	0.0585 0.0525 0.0460 0.0389 0.0322 0.0273 0.0228
37.80 34.20 30.20 26.40 21.40 8.20 5.30 2.70	34.20 30.20 26.40 21.40 18.20 15.30 12.70 10.40	25.50 28.60 29.90 41.40 28.10 29.50 31.00 30.80	315 310 405 250 235 215 185	36.00 32.20 28.30 23.90 19.80 16.75 14.00 11.55	11.58 11.00 10.37 9.80 8.90 7.96 6.94 6.00	1990 1888 1778 1678 1526 1363 1188 1030	0.0585 0.0525 0.0460 0.0389 0.0322 0.0273 0.0228 0.0188
37.80 34.20 30.20 26.40 21.40 8.20 5.30 2.70 0.40	34.20 30.20 26.40 21.40 18.20 15.30 12.70 10.40 8.40	25.50 28.60 29.90 41.40 28.10 29.50 31.00 30.80 32.60	315 310 405 250 235 215 185 169	36.00 32.20 28.30 23.90 19.80 16.75 14.00 11.55 9.40	11.58 11.00 10.37 9.80 8.90 7.96 6.94 6.00 5.19	1990 1888 1778 1678 1526 1363 1188 1030 890	0.0585 0.0525 0.0460 0.0389 0.0322 0.0273 0.0228 0.0188 0.0153
37.80 34.20 30.20 26.40 21.40 8.20 5.30 2.70 0.40 8.40	34.20 30.20 26.40 21.40 18.20 15.30 12.70 10.40 8.40 6.30	25.50 28.60 29.90 41.40 28.10 29.50 31.00 30.80 32.60 41.00	 315 310 405 250 235 215 185 169 170 	36.00 32.20 28.30 23.90 19.80 16.75 14.00 11.55 9.40 7.35	11.58 11.00 10.37 9.80 8.90 7.96 6.94 6.00 5.19 4.15	1990 1888 1778 1678 1526 1363 1188 1030 890 711	0.0585 0.0525 0.0460 0.0389 0.0322 0.0273 0.0228 0.0188 0.0153 0.0120

1	2	3	4,	5	6	7	8
3% Slu	rry with	1 50% Repi	lacemen	t of bent	conite by	black	cotton soil
$\Upsilon_{s} = 1$.018 gm/	r _{cm} 3					
36.60	32•50	29.40	335	34.55	11.40	1955	0.0561
32.50	28.50	30.30	330	30.50	10.90	1870	0.0496
28.50	24.40	32.80	325	26.45	9.1	1700	0.0430
24.40	20.50	34.10	320	22.45	9.40	1670	0.0366
20.50	16.70	34.80	295	18.60	8.50	1454	0.0303
16.70	13.40	36.80	2 7 0	15.05	7.34	.1255	0.0245
13.40	10.70	34.90	220	12.05	6.30	1080	0.0196
10.70	8.20	37.70	200	9.45	5 • 30	909	0.0154
8.20	6.20	39.60	167	7.20	4.21	723	0.0117
6.20	4.50	44.80	141	5•35	3.14	538	0.0087
n, n s				3.75 $I_{R1} = 0$ g		379	0.0061
Head f 3% Slu	or no fl	.ow = 0.00) cm , '	T _{R1} = 0 g	m / cm ²	• • • •	0.0061
Head f <u>3% Slu</u> Y _s = 1	or no fl rry with .018 gm/	.ow = 0.00 n 75 repla cm ³	C cm , ' 	TR1 = 0 g of bento	m / cm ²	olack o	cotton soi]
Head f 3% Slu Y _s = 1 38.10	or no fl	.ow = 0.00 n 75 repla cm ³ 30.50	C cm , ' acement 365	^T R1 = 0 g of bento 36.00	m / cm ²	2050	otton soi] 0.0585
Head f 3% Slu Y _s = 1 38.10 33.90	or no fl rry with .018 gm/ 33.90	.ow = 0.00 n 75 repla cm ³	C cm , ' 	TR1 = 0 g of bento	m / cm ²	olack o	cotton soi]
Head f 3% Slu	or no fl rry with .018 gm/ 33.90 29.10	.ow = 0.00 n 75 repla cm ³ 30.50 35.10	2 cm , ' acement 365 385	^T R1 = 0 g of bento 36.00 31.50	m / cm ² nite by h 11.96 10.98	2050 1890	0.0585 0.0513
Head f 3% Slu $Y_s = 1$ 38.10 33.90 29.10 24.70	or no fl rry with .018 gm/ 33.90 29.10 24.70	.ow = 0.00 n 75 repla cm ³ 30.50 35.10 35.20	2 cm , ' acement 365 385 355	T _{R1} = 0 g of bento 36.00 31.50 26.90	m / cm ² nite by b 11.96 10.98 10.10	2050 1890 1730	0.0585 0.0513 0.0438
Head f $\frac{3\% \text{ Slu}}{Y_{s}} = 1$ $\frac{38.10}{33.90}$ $\frac{29.10}{29.10}$	or no fl rry with .018 gm/ 33.90 29.10 24.70 20.90	cw = 0.00 75 replations $30.5035.1035.2033.50$	365 385 355 310	T _{R1} = 0 g of bento 36.00 31.50 26.90 22.80	m / cm ² nite by h 11.96 10.98 10.10 9.25 8.55	2050 1890 1730 1585 1465	0.0585 0.0513 0.0438 0.0371 0.0310
Head f 3% Slu $Y_s = 1$ 38.10 33.90 29.10 24.70 20.90 17.30	or no fl rry with .018 gm/ 33.90 29.10 24.70 20.90 17.30	cw = 0.00 $75 replating cm^3$ 30.50 35.10 35.20 33.50 33.70	2 cm , ' acement 365 385 355 310 228 283	T _{R1} = 0 g of bento 36.00 31.50 26.90 22.80 19.10	m / cm ² nite by k 11.96 10.98 10.10 9.25 8.55 7.85	2050 1890 1730 1585	0.0585 0.0513 0.0438 0.0371 0.0310 0.0253
Head f 3% Slu $Y_s = 1$ 38.10 33.90 29.10 24.70 20.90 17.30 13.80	or no fl rry with .018 gm/ 33.90 29.10 24.70 20.90 17.30 13.80	cw = 0.00 75 repla cm^3 30.50 35.10 35.20 33.50 33.70 36.10	2 cm , ' acement 365 385 355 310 228	T _{R1} = 0 g of bento 36.00 31.50 26.90 22.80 19.10 15.55	m / cm ² nite by h 11.96 10.98 10.10 9.25 8.55	2050 1890 1730 1585 1465 1345	0.0585 0.0513 0.0438 0.0371 0.0310
Head f $\frac{3\% \text{ Slu}}{Y_{s}} = 1$ $\frac{38.10}{33.90}$ $\frac{29.10}{24.70}$ 20.90	or no fl rry with .018 gm/ 33.90 29.10 24.70 20.90 17.30 13.80 10.70	cw = 0.00 $75 repla cm^330.5035.1035.2033.5033.7036.1035.40$	2 cm , ' acement 365 385 355 310 228 283 244	T _{R1} = 0 g of bento 36.00 31.50 26.90 22.80 19.10 15.55 12.25	m / cm ² nite by R 11.96 10.98 10.10 9.25 8.55 7.85 6.90	2050 1890 1730 1585 1465 1345 1182	0.0585 0.0513 0.0438 0.0371 0.0310 0.0253 0.0200
Head f 3% Slu $Y_s = 1$ 38.10 33.90 29.10 24.70 20.90 17.30 13.80 10.70	or no fl rry with .018 gm/ 33.90 29.10 24.70 20.90 17.30 13.80 10.70 8.20	cw = 0.00 75 replation 30.50 35.10 35.20 33.50 33.70 36.10 35.40 36.00	2 cm , ' acement 365 385 355 310 228 283 244 205	$T_{R1} = 0$ g of bento 36.00 31.50 26.90 22.80 19.10 15.55 12.25 9.45	m / cm ² nite by h 11.96 10.98 10.10 9.25 8.55 7.85 6.90 5.70	2050 1890 1730 1585 1465 1345 1182 978	0.0585 0.0513 0.0438 0.0371 0.0310 0.0253 0.0200 0.0154

$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	1	2	3	4	5	6.	7	8
32.10 27.90 30.60 330 30.00 10.80 1850 0.0489 27.90 24.20 29.80 295 26.05 9.90 1698 0.0425 34.20 20.50 32.00 295 22.35 9.22 1580 0.0364 20.50 16.90 34.00 290 18.70 8.55 1465 0.0304 16.90 13.40 34.30 274 15.15 8.00 1370 0.0247 13.40 10.50 33.00 230 11.95 6.96 1193 0.0195 10.50 8.10 32.80 190 9.30 5.80 995 0.0151 8.10 6.00 33.90 162 7.05 4.78 820 0.0115 6.00 4.50 32.80 120 5.25 3.66 626 0.0086 Head for no flow = 0.00 cm , $T_{R1} = 0 \text{ gm / cm}^2$ 4% slurry Pure Bentomite Y = 1.024 gm/cm ³ 38.30 34.80 29.00 288 36.55 9.93 1700 0.0600 34.80 31.30 31.70 284 33.05 8.96 1540 0.0541 31.30 28.10 33.90 266 29.70 7.85 1345 0.0486 28.10 25.30 33.70 223 26.70 6.63 1136 0.0438 25.30 21.30 96.10 329 23.30 5.86 1080 0.0382 21.30 18.60 43.90 225 19.95 5.13 880 0.0327 18.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 13.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 Head for no flow = 1.30 cm , $T_{R1} = \frac{R}{2} \frac{\Delta h}{\Delta L} \mathbf{v}_S$	3% S11	arry with	n Pure Bla	ack Cot	ton Soil	$\Upsilon_{s} = 1$.018 gm/	<u>em</u> 3
27.90 24.20 29.80 295 26.05 9.90 1698 0.0425 34.20 20.50 32.00 295 22.35 9.22 1580 0.0364 20.50 16.90 34.00 290 18.70 8.55 1465 0.6304 16.90 13.40 34.30 274 15.15 8.00 1370 0.0247 13.40 10.50 33.00 230 11.95 6.96 1193 0.0195 10.50 8.10 32.80 190 9.30 5.80 995 0.0151 8.10 6.00 33.90 162 7.05 4.78 820 0.0115 6.00 4.50 32.80 120 5.25 3.66 626 0.0086 Head for no flow = 0.00 cm , $T_{R1} = 0 \text{ gm/cm}^2$ 4% slurry Pure Bentomite $Y_s = 1.024 \text{ gm/cm}^3$ 38.30 34.80 29.00 288 36.55 9.93 1700 0.0600 34.80 31.30 31.70 284 33.05 8.96 1540 0.0541 31.30 28.10 33.90 266 29.70 7.85 1345 0.0486 28.10 25.30 33.70 223 26.70 6.63 1136 0.0438 25.30 21.30 §6.10 329 23.30 5.86 1080 0.0382 21.30 18.60 43.90 225 19.95 5.13 880 0.0327 18.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 30.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 10.0161 10.0161 10.0161 for no flow = 1.30 cm , $T_{R1} = \frac{R}{2} \frac{\Delta h}{\Delta L} \mathbf{x}_s$	35.70	32.10	25.00	284	33.9 0	11.35	1945	0.0551
34.20 20.50 32.00 295 22.35 9.22 1580 0.0364 20.50 16.90 34.00 290 18.70 8.55 1465 0.3304 16.90 13.40 34.30 274 15.15 8.00 1370 0.0247 13.40 10.50 33.00 230 11.95 6.96 1193 0.0195 10.50 8.10 32.80 190 9.30 5.80 995 0.0151 8.10 6.00 33.90 162 7.05 4.78 820 0.0115 6.00 4.50 32.80 120 5.25 3.66 626 0.0086 Head for no flow = 0.00 cm , $T_{R1} = 0 \text{ gm/cm}^3$ 38.30 34.80 29.00 288 36.55 9.93 1700 0.0600 34.80 31.30 31.70 284 33.05 8.96 1540 0.0541 31.30 28.10 33.90 266 29.70 7.85 1345 0.0486 28.10 25.30 33.70 223 26.70 6.63 1136 0.0438 25.30 21.30 56.10 329 23.30 5.86 1080 0.0382 21.30 18.60 43.90 225 19.95 5.13 880 0.0327 18.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 13.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 4.21 $\frac{R}{2} \frac{\Lambda}{\Lambda L} \mathbf{x}_S$	32.10	27.90	30.60	3 30	30.00	10.80	18 50	0.0489
20.50 16.90 34.00 290 18.70 8.55 1465 0.3304 16.90 13.40 34.30 274 15.15 8.00 1370 0.0247 13.40 10.50 33.00 230 11.95 6.96 1193 0.0195 10.50 8.10 32.80 190 9.30 5.80 995 0.0151 8.10 6.00 33.90 162 7.05 4.78 820 0.0115 6.00 4.50 32.80 120 5.25 3.66 626 0.0086 Head for no flow = 0.00 cm , $T_{R1} = 0 \text{ gm / cm}^2$ 4% slurry Pure Bentomite $Y_s = 1.024 \text{ gm/ cm}^3$ 38.30 34.80 29.00 288 36.55 9.93 1700 0.0600 34.80 31.30 31.70 284 33.05 8.96 1540 0.0541 31.30 28.10 33.90 266 29.70 7.85 1345 0.0486 28.10 25.30 33.70 223 26.70 6.63 1136 0.0438 25.30 21.30 \$6.10 329 23.30 5.86 1080 0.0382 21.30 18.60 43.90 225 19.95 5.13 880 0.0327 18.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 13.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 Head for no flow = 1.30 cm , $T_{R1} = \frac{R}{2} \frac{\Delta h}{\Delta L} s_s$	27.90	24.20	29.80	295	26.05	9 .9 0	1698	0.0425
16.90 13.40 34.30 274 15.15 8.00 1370 0.0247 13.40 10.50 33.00 230 11.95 6.96 1193 0.0195 10.50 8.10 32.80 190 9.30 5.80 995 0.0151 8.10 6.00 33.90 162 7.05 4.78 820 0.0115 6.00 4.50 32.80 120 5.25 3.66 626 0.0086 Head for no flow = 0.00 cm , $T_{R1} = 0 \text{ gm/cm}^2$ 4% slurry Pure Bentomite Y _S = 1.024 gm/cm ³ 38.30 34.80 29.00 288 36.55 9.93 1700 0.0600 34.80 31.30 31.70 284 33.05 8.96 1540 0.0541 31.30 28.10 33.90 266 29.70 7.85 1345 0.0486 28.10 25.30 33.70 223 26.70 6.63 1136 0.0438 25.30 21.30 \$6.10 329 23.30 5.86 1080 0.0382 21.30 18.60 43.90 225 19.95 5.13 880 0.0327 18.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 13.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 Head for no flow = 1.30 cm , $T_{R1} = \frac{R}{2} \frac{\Lambda h}{\Lambda L} \frac{R}{4L} \frac{R}{4L} = \frac{R \times 1.3 \times 1.024}{4L} = 0.00212 \text{ gm/cm}^2$	34.20	20.50	32.00	295	22.35	9.22	1580	0.0364
13.40 10.50 33.00 230 11.95 6.96 1193 0.0195 10.50 8.10 32.80 190 9.30 5.80 995 0.0151 8.10 6.00 33.90 162 7.05 4.78 820 0.0115 6.00 4.50 32.80 120 5.25 3.66 626 0.0086 Head for no flow = 0.00 cm , $T_{R1} = 0 \text{ gm}/\text{cm}^2$ 4% slurry Pure Bentomite Y = 1.024 gm/cm ³ 38.30 34.80 29.00 288 36.55 9.93 1700 0.0600 34.80 31.30 31.70 284 33.05 8.96 1540 0.0541 31.30 28.10 33.90 266 29.70 7.85 1345 0.0486 28.10 25.30 33.70 223 26.70 6.63 1136 0.0438 25.30 21.30 56.10 329 23.30 5.86 1080 0.0382 21.30 18.60 43.90 225 19.95 5.13 880 0.0327 18.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 13.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 1ead for no flow = 1.30 cm , $T_{R1} \doteq \frac{R}{2} 1_{1} = \frac{R}{2} \frac{\Delta h}{\Delta L} \mathbf{x}_{S}$	20.50	16.90	34.00	290	18.70	8 • 5 5	1465	0.3304
10.50 8.10 32.80 190 9.30 5.80 995 0.0151 8.10 6.00 33.90 162 7.05 4.78 820 0.0115 6.00 4.50 32.80 120 5.25 3.66 626 0.0086 Head for no flow = 0.00 cm , $T_{R1} = 0 \text{ gm / cm}^2$ 4% slurry Pure Bentomite Y = 1.024 gm/cm ³ 38.30 34.80 29.00 288 36.55 9.93 1700 0.0600 34.80 31.30 31.70 284 33.05 8.96 1540 0.0541 31.30 28.10 33.90 266 29.70 7.85 1345 0.0486 28.10 25.30 33.70 223 26.70 6.63 1136 0.0438 25.30 21.30 56.10 329 23.30 5.86 1080 0.0382 21.30 18.60 43.90 225 19.95 5.13 880 0.0327 18.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 13.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 Head for no flow = 1.30 cm , $T_{R1} = \frac{R}{2} \frac{\Delta h}{\Delta L} x_s$	16.90	13.40	34.30	274	15 • 15	8.00	1370	0.0247
8.10 6.00 33.90 162 7.05 4.78 820 0.0115 6.00 4.50 32.80 120 5.25 3.66 626 0.0086 Head for no flow = 0.00 cm , $T_{R1} = 0 \text{ gm/cm}^2$ 4% slurry Pure Bentomite Y _s = 1.024 gm/cm ³ 38.30 34.80 29.00 288 36.55 9.93 1700 0.0600 34.80 31.30 31.70 284 33.05 8.96 1540 0.0541 31.30 28.10 33.90 266 29.70 7.85 1345 0.0486 28.10 25.30 33.70 223 26.70 6.63 1136 0.0438 25.30 21.30 \$6.10 329 23.30 5.86 1080 0.0382 21.30 18.60 43.90 225 19.95 5.13 880 0.0327 18.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 13.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 Head for no flow = 1.30 cm , $T_{R1} = \frac{R}{2} \frac{\Delta h}{\Delta L} s_s$	13.40	10.50	33.00	230	11.95	6.96	1193	0.0195
6.00 4.50 32.80 120 5.25 3.66 626 0.0086 Head for no flow = 0.00 cm , $T_{R1} = 0 \text{ gm / cm}^2$ 4% slurry Pure Bentomite $Y_s = 1.024 \text{ gm/cm}^3$ 38.30 34.80 29.00 288 36.55 9.93 1700 0.0600 34.80 31.30 31.70 284 33.05 8.96 1540 0.0541 31.30 28.10 33.90 266 29.70 7.85 1345 0.0486 28.10 25.30 33.70 223 26.70 6.63 1136 0.0438 25.30 21.30 56.10 329 23.30 5.86 1080 0.0382 21.30 18.60 43.90 225 19.95 5.13 880 0.0327 18.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 13.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 Head for no flow = 1.30 cm , $T_{R1} = \frac{R}{2} 1_1 = \frac{R}{2} \frac{\Delta h}{\Delta L} \mathbf{x}_s$	10.50	8.10	32.80	19 0	9.30	5.80	995	0.0151
Head for no flow = 0.00 cm , $T_{R1} = 0 \text{ gm}/\text{cm}^2$ $\frac{4\%}{R} \text{ slurry Pure Bentomite } Y_s = 1.024 \text{ gm/cm}^3$ 38.30 34.80 29.00 288 36.55 9.93 1700 0.0600 34.80 31.30 31.70 284 33.05 8.96 1540 0.0541 31.30 28.10 33.90 266 29.70 7.85 1345 0.0486 28.10 25.30 33.70 223 26.70 6.63 1136 0.0438 $25.30 21.30 $ $\frac{5}{6}.10 329 23.30 5.86 1080 0.0382$ 21.30 18.60 43.90 225 19.95 5.13 880 0.0327 18.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 13.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 Head for no flow = 1.30 cm , $T_{R1} = \frac{R}{2} \frac{\Delta h}{\Delta L} s_s$ $r_{R1} = \frac{R \times 1.3 \times 1.024}{R \times 1.3 \times 1.024} = 0.00212 \text{ gm/cm}^2$	8.10	6.00	33.90	162	7.05	4•78	820	0.0115
4% slurry Pure Bentomite $Y_s = 1.024 \text{ gm/cm}^3$ 38.30 34.80 29.00 288 36.55 9.93 1700 0.0600 34.80 31.30 31.70 284 33.05 8.96 1540 0.0541 31.30 28.10 33.90 266 29.70 7.85 1345 0.0486 28.10 25.30 33.70 223 26.70 6.63 1136 0.0438 25.30 21.30 56.10 329 23.30 5.86 1080 0.0382 21.30 18.60 43.90 225 19.95 5.13 880 0.0327 18.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 13.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 Head for no flow = 1.30 cm $T_{R1} \stackrel{+}{=} \stackrel{R}{=} 1_1 = \stackrel{R}{=} \frac{\Delta h}{\Delta L} x_s$ x_s	6.00	4.50	32.80	120	5•25	3.66	626	0.008.6
34. 80 31.30 31.70 284 33.05 8.96 1540 0.0541 31. 30 28.10 33.90 266 29.70 7.85 1345 0.0486 28. 10 25.30 33.70 223 26.70 6.63 1136 0.0438 25. 30 21.30 46.10 329 23.30 5.86 1080 0.0382 21. 30 18.60 43.90 225 19.95 5.13 880 0.0327 18. 60 15.60 55.50 238 17.10 4.30 735 0.0280 15. 60 13.00 61.30 215 14.30 3.44 590 0.0234 13. 00 10.70 70.10 190 11.85 2.71 465 0.0194 10. 70 8.90 72.40 153 9.80 2.12 364 0.0161 Head for no flow = 1.30 cm , $T_{R1} = \frac{R}{2} \mathbf{i}_1 = \frac{R}{2} \frac{\Delta h}{\Delta L} \mathbf{x}_s$								
$\begin{array}{cccccccccccccccccccccccccccccccccccc$	+% slu	irry Pure	Bentonii	eY <u>s</u>	= 1.024 gi	m/cm ³		
$\begin{array}{cccccccccccccccccccccccccccccccccccc$					<u>, gan an an an an air air an </u>		1700	0.0600
25.30 21.30 $\frac{1}{5}6.10$ 329 23.30 5.86 1080 0.0382 21.30 18.60 43.90 225 19.95 5.13 880 0.0327 8.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 3.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 1ead for no flow = 1.30 cm , $T_{R1} \stackrel{+}{=} \frac{R}{2} \mathbf{i}_{1} = \frac{R}{2} \frac{\Delta h}{\Delta L} \mathbf{v}_{S}$	38.30	34.80	29.00	288	36•55	9.93		
21.30 18.60 43.90 225 19.95 5.13 880 0.0327 18.60 15.60 55.50 238 17.10 4.30 735 0.0280 15.60 13.00 61.30 215 14.30 3.44 590 0.0234 13.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 tead for no flow = 1.30 cm , $T_{R1} \stackrel{+}{=} \frac{R}{2} \mathbf{i}_{1} = \frac{R}{2} \frac{\Delta h}{\Delta L} \mathbf{v}_{S}$	38.30 34.80	34.80 31.30	29.00 31.70	288 284	36•55 33•05	9•93 8•96	1540	0.0541
$18.60 15.60 55.50 238 17.10 4.30 735 0.0280$ $15.60 13.00 61.30 215 14.30 3.44 590 0.0234$ $13.00 10.70 70.10 190 11.85 2.71 465 0.0194$ $10.70 8.90 72.40 153 9.80 2.12 364 0.0161$ $1ead \text{ for no flow} = 1.30 \text{ cm}, T_{R1} \stackrel{+}{=} \frac{R}{2} \mathbf{i}_{1} = \frac{R}{2} \frac{\Delta h}{\Delta L} \mathbf{v}_{S}$ $P_{1} = \frac{R \times 1.3 \times 1.024}{R \times 1.3 \times 1.024} = 0.00212 \text{ gm/cm}^{2}$	38.30 34.80 31.30	34.80 31.30 28.10	29.00 31.70 33.90	288 284 266	36•55 33•05 29•70	9.93 8.96 7.85	1540 1345	0.0541 0.0486
$\begin{array}{cccccccccccccccccccccccccccccccccccc$	38.30 34.80 31.30 28.10	34.80 31.30 28.10 25.30	29.00 31.70 33.90 33.70	288 284 266 223	36•55 33•05 29•70 26•70	9.93 8.96 7.85 6.63	1540 1345 1136	0.0541 0.0486 0.0438
3.00 10.70 70.10 190 11.85 2.71 465 0.0194 10.70 8.90 72.40 153 9.80 2.12 364 0.0161 lead for no flow = 1.30 cm , $T_{R1} \stackrel{+}{=} \frac{R}{2} i_1 = \frac{R}{2} \frac{\Delta h}{\Delta L} v_s$	38.30 34.80 31.30 28.10 25.30	34.80 31.30 28.10 25.30 21.30	29.00 31.70 33.90 33.70 56.10	288 284 266 223 329	36.55 33.05 29.70 26.70 23.30	9.93 8.96 7.85 6.63 5.86	1540 1345 1136 1080	0.0541 0.0486 0.0438 0.0382
10.70 8.90 72.40 153 9.80 2.12 364 0.0161 lead for no flow = 1.30 cm, $T_{R1} \stackrel{+}{=} \frac{R}{2} i_1 = \frac{R}{2} \frac{\Delta h}{\Delta L} \delta_s$ P1 = $\frac{R \times 1.3 \times 1.024}{R} = 0.00212 \text{ gm/cm}^2$	38.30 34.80 31.30 28.10 25.30 21.30	34.80 31.30 28.10 25.30 21.30 18.60	29.00 31.70 33.90 33.70 56.10 43.90	288 284 266 223 329 225	36.55 33.05 29.70 26.70 23.30 19.95	9.93 8.96 7.85 6.63 5.86 5.13	1540 1345 1136 1080 880	0.0541 0.0486 0.0438 0.0382 0.0327
lead for no flow = 1.30 cm , $T_{R1} \stackrel{+}{=} \frac{R}{2} i_1 = \frac{R}{2} \frac{\Delta h}{\Delta L} \delta_s$ $R \ge 1.3 \ge 1.024$ = 0.00212 gm/cm ²	38.30 34.80 31.30 28.10 25.30 21.30 18.60	34.80 31.30 28.10 25.30 21.30 18.60 15.60	29.00 31.70 33.90 33.70 56.10 43.90 55.50	288 284 266 223 329 225 238	36.55 33.05 29.70 26.70 23.30 19.95 17.10	9.93 8.96 7.85 6.63 5.86 5.13 4.30	1540 1345 1136 1080 880 735	0.0541 0.0486 0.0438 0.0382 0.0327 0.0280
$r_{1} = \frac{\pi x + 3 x + 024}{\pi 2} = 0.00212 \text{ gm/cm}^2$	38.30 34.80 31.30 28.10 25.30 21.30 18.60 15.60	34.80 31.30 28.10 25.30 21.30 18.60 15.60 13.00	29.00 31.70 33.90 33.70 \$6.10 43.90 55.50 61.30	288 284 266 223 329 225 238 215	36.55 33.05 29.70 26.70 23.30 19.95 17.10 14.30	9.93 8.96 7.85 6.63 5.86 5.13 4.30 3.44	1540 1345 1136 1080 880 735 590	0.0541 0.0486 0.0438 0.0382 0.0327 0.0280 0.0234
$r_{1} = \frac{\pi x + 3 x + 024}{\pi 2} = 0.00212 \text{ gm/cm}^2$	38.30 34.80 31.30 28.10 25.30 21.30 18.60 15.60 13.00	34.80 31.30 28.10 25.30 21.30 18.60 15.60 13.00 10.70	29.00 31.70 33.90 33.70 56.10 43.90 55.50 61.30 70.10	288 284 266 223 329 225 238 215 190 153	36.55 33.05 29.70 26.70 23.30 19.95 17.10 14.30 11.85 9.80	9.93 8.96 7.85 6.63 5.86 5.13 4.30 3.44 2.71 2.12	1540 1345 1136 1080 880 735 590 465 364	0.0541 0.0486 0.0438 0.0382 0.0327 0.0280 0.0234 0.0194
	38.30 34.80 31.30 28.10 25.30 21.30 18.60 15.60 13.00 10.70	34.80 31.30 28.10 25.30 21.30 18.60 15.60 13.00 10.70 8.90	29.00 31.70 33.90 33.70 56.10 43.90 55.50 61.30 70.10 72.40	288 284 266 223 329 225 238 215 190 153	36.55 33.05 29.70 26.70 23.30 19.95 17.10 14.30 11.85 9.80	9.93 8.96 7.85 6.63 5.86 5.13 4.30 3.44 2.71 2.12	1540 1345 1136 1080 880 735 590 465 364	0.0541 0.0486 0.0438 0.0382 0.0327 0.0280 0.0234 0.0194

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********			lacemer	t of ben	tonite by	black	cotton so
-	1.024 gm/		200		44.00		0 0-10
37.70	33.20	32.10	378	35•45	11.80	2020	0.0581
33.20	28.60	35.20	370	30.90	10.50	1800	0.0506
28.60	24.60	35.30	325	26.60	9.20	1578	0.0435
24.60	21.10	35.30	284	22•85	8.05	1380	0.0374
21.10	18.20	33.70	240	19.65	7.11	1220	0.0322
18.20	15.20	38.20	235	16.70	6.15	10543	0.0274
15.20	12.70	39.80	204	13.95	5.11	87 8	0.0228
12.70	10.20	46.70	205	11.45	4.40	75 5	0.0188
10.20	7.80	56.80	197	9.00	3.148	596	0.0148
7.80	5.90	61.40	158	6.85	2•59	445	0.0112
4% S11	urry with	50% Rep	lacemen	r' = 0.0 t. Y = 1		2	
27 00							• ,
37.80	33.20	35.00	398	35.50	11.75	2020	0.0560
37.80 33.20	33•20 29•00	35. 00 32.10					0.0560 0.0490
37.80 33.20 29.00	33.20 29.00 24.20	35.00 32.10 40.10	398 341 390	35.50	11.75 10.60 9.70	2020	-
37.80 33.20 29.00 24.20	33.20 29.00 24.20 20.20	35.00 32.10 40.10 36.00	398 341	35.50 31.10	11.75 10.60	2020 1820	0.0490
37.80 33.20 29.00 24.20 20.20	33.20 29.00 24.20 20.20 16.60	35.00 32.10 40.10 36.00 37.00	398 341 390 319 287	35.50 31.10 26.60	11.75 10.60 9.70	2020 1820 1663	0.0490 0.0420
37.80 33.20 29.00 24.20 20.20 16.60	33.20 29.00 24.20 20.20 16.60 13.20	35.00 32.10 40.10 36.00 37.00 40.50	398 341 390 319	35.50 31.10 26.60 22.20	11.75 10.60 9.70 8.85	2020 1820 1663 1520	0.0490 0.0420 0.0361 0.0291
37.80 33.20 29.00 24.20 20.20 16.60 13.20	33.20 29.00 24.20 20.20 16.60 13.20 10.20	35.00 32.10 40.10 36.00 37.00	398 341 390 319 287	35.50 31.10 26.60 22.20 18.40	11.75 10.60 9.70 8.85 7.75	2020 1820 1663 1520 1330	0.0490 0.0420 0.0361 0.0291
37.80 33.20 29.00 24.20 20.20 16.60	33.20 29.00 24.20 20.20 16.60 13.20	35.00 32.10 40.10 36.00 37.00 40.50	398 341 390 319 287 266	35.50 31.10 26.60 22.20 18.40 14.90	11.75 10.60 9.70 8.85 7.75 6.55	2020 1820 1663 1520 1330 1122	0.0490 0.0420 0.0361 0.0291 0.0236
37.80 33.20 29.00 24.20 20.20 16.60 13.20	33.20 29.00 24.20 20.20 16.60 13.20 10.20	35.00 32.10 40.10 36.00 37.00 40.50 44.10	398 341 390 319 287 266 240	35.50 31.10 26.60 22.20 18.40 14.90 11.70	11.75 10.60 9.70 8.85 7.75 6.55 5.43	2020 1820 1663 1520 1330 1122 930	0.0490 0.0420 0.0361 0.0291 0.0236 0.0185
37.80 33.20 29.00 24.20 20.20 16.60 13.20 10.20	33.20 29.00 24.20 20.20 16.60 13.20 10.20 7.40	35.00 32.10 40.10 36.00 37.00 40.50 44.10 49.40	398 341 390 319 287 266 240 214	35.50 31.10 26.60 22.20 18.40 14.90 11.70 8.80	11.75 10.60 9.70 8.85 7.75 6.55 5.43 4.34	2020 1820 1663 1520 1330 1122 930 743	0.0490 0.0420 0.0361 0.0291 0.0236 0.0185 0.0139
37.80 33.20 29.00 24.20 20.20 16.60 13.20 10.20 7.40	33.20 29.00 24.20 20.20 16.60 13.20 10.20 7.40 5.40	35.00 32.10 40.10 36.00 37.00 40.50 44.10 49.40 49.40	398 341 390 319 287 266 240 214 159	35.50 31.10 26.60 22.20 18.40 14.90 11.70 8.80 6.40	11.75 10.60 9.70 8.85 7.75 6.55 5.43 4.34 3.25	2020 1820 1663 1520 1330 1122 930 743 557	0.0490 0.0420 0.0361 0.0291 0.0236 0.0185 0.0139 0.0101

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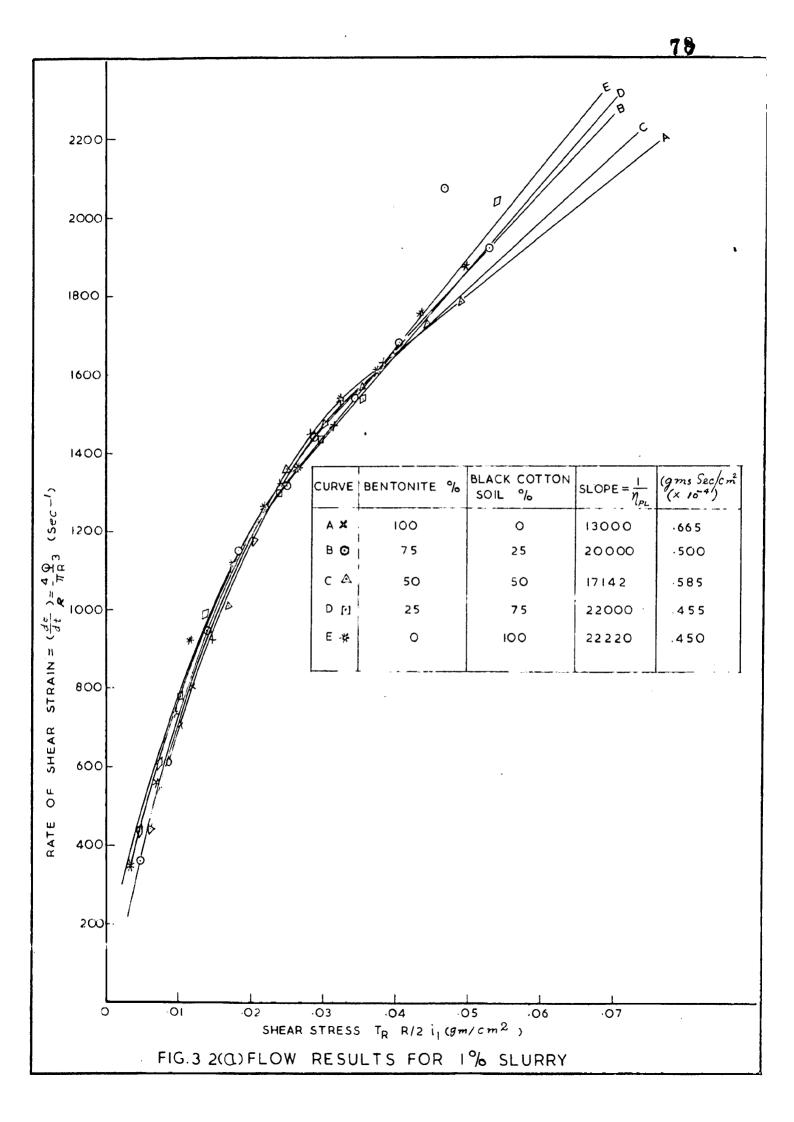
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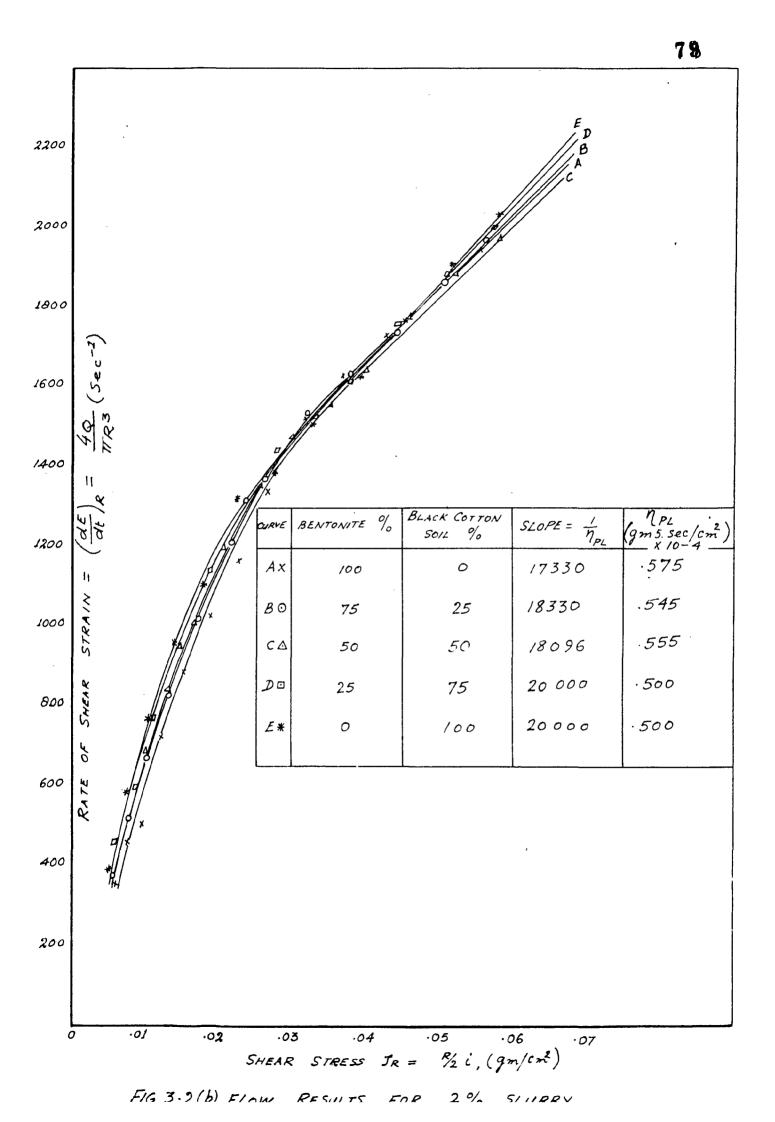
	2		3	4	5*	6.	7	8	
+% S1	urry	with	75 %	Replace	ement of 1	bentonite	by black	k cotton	soil
Υ _s =	1.02	+ gm/c	em3						
38.40			32.30	355	36.20	11.00	1885	0.0571	
34.00	29.	70	32.60	344	31.85	10.50	1800	0.0504	
29 . 70	25•	60	33.50	330	27.65	9•85	1 6 86	0.0436	
25.60	21.	10	39.30	365	23.35	9.30	1592	0.0369	
21.10	17.	60	34.50	280	19.35	8.15	1400	0.0306	
17.60	14.	20	39.00	275	15.90	7.05	1210	0.0251	
14.20	11.	50	39.60	220	12.85	5•55	952	0 . 020 3	
11.50	• 9•	30	36•20) 171	10.40	4.72	8 1 0	0.0164	
9.30	7.	50	43.00	160	8.40	3.72	637	0.0133	
	,	00	45.70	11 0	6.75	2.40	411 *	0.0107	
7.50 Head			w = 0.()0 cm ,	$T_{R1} = 0$	gm / cm^2	* 1 84 45 55 46	••••••	
Head	for n	o flo	w = 0.(RI	gm /cm ² ,Y _s = 1.02	24 gm/cm ³	; }	•
Head	for n	o flo with	w = 0.(RI		24 gm/cm ³	0.0575	••• •••
Head +% Sl	for n urry 34.	o flo with 20	w = 0.0	Lack cot	ton soil,	Y _s = 1.02		-	••• •••
Head +% S1 38.60	for n urry 34. 29.	o flo with 20 80	w = 0.0 pure b] 33.30	Lack cot 370	ton soil, 36.40	Y _s = 1.02 11.20	1920	0.0575	••• •••
Head +% S1 38.60 34.20	for n urry 34. 29. 25.	o flo with 20 80 50	w = 0.0 pure b] 33.30 33.90	Lack cot 370 355	ton soil, 36.40 32.00	Y _s = 1.02 11.20 10.50	1920. 1800	0•0575 0•0505	••• •••
Head +% S1: 38.60 34.20 29.80	for n urry 34. 29. 25. 21.	o flo with 20 80 50 40	w = 0.0 pure bl 33.30 33.90 36.40	Lack cot 370 355 357	ton soil, 36.40 32.00 27.65	$Y_{s} = 1.02$ 11.20 10.50 9.76 9.00	1920 1800 1675	0.0575 0.0505 0.0436 0.0370	** * **
Head +% S1 38.60 34.20 29.80 25.50	for n urry 34. 29. 25. 21. 17.	o flo with 20 80 50 40	w = 0.0 pure b] 33.30 33.90 36.40 37.00	Lack cot 370 355 357 333	ton soil, 36.40 32.00 27.65 23.45	$Y_s = 1.02$ 11.20 10.50 9.76 9.00 8.20	1920 1800 1675 1542	0.0575 0.0505 0.0436 0.0370	••• •••
Head +% S1 38.60 34.20 29.80 25.50 21.40	for n urry 34. 29. 25. 21. 17. 13.	0 flc with 20 80 50 40 00 50	w = 0.0 pure b 33.30 33.90 36.40 37.00 42.70	Lack cot 370 355 357 333 350	R1 36.40 32.00 27.65 23.45 19.20	$Y_s = 1.02$ 11.20 10.50 9.76 9.00 8.20 7.3^{1}	1920 1800 1675 1542 1405	0.0575 0.0505 0.0436 0.0370 0.0303	••••••••••••••••••••••••••••••••••••••
Head +% S1 38.60 34.20 29.80 25.50 21.40 17.00	for n urry 34. 29. 25. 21. 17. 13. 10.	0 flc with 20 80 50 40 00 50	w = 0.0 pure b] 33.30 33.90 36.40 37.00 42.70 38.20	Lack cot 370 355 357 333 350 280	R1 36.40 32.00 27.65 23.45 19.20 15.25	$Y_s = 1.02$ 11.20 10.50 9.76 9.00 8.20 7.3^{1}	1920 1800 1675 1542 1405 1257	0.0575 0.0505 0.0436 0.0370 0.0303 0.0241	••• •••
Head +% S1 38.60 34.20 29.80 25.50 21.40 17.00 13.50	for n urry 34. 29. 25. 21. 17. 13. 10. 7.	0 flc with 20 80 50 40 00 50 40	w = 0.0 pure b] 33.30 33.90 36.40 37.00 42.70 38.20 40.50	Lack cot 370 355 357 333 350 280 250	R1 36.40 32.00 27.65 23.45 19.20 15.25 11.95	$Y_s = 1.02$ 11.20 10.50 9.76 9.00 8.20 7.3^{1} 6.16	1920 1800 1675 1542 1405 1257 1057	0.0575 0.0505 0.0436 0.0370 0.0303 0.0241 0.0189	••••••••••••••••••••••••••••••••••••••
Head +% S1 38.60 34.20 29.80 25.50 21.40 17.00 13.50 10.40	for n urry 34. 29. 25. 21. 17. 13. 10. 7. 5.	o flc with 20 80 50 40 50 40 80 60	w = 0.0 pure b] 33.30 33.90 36.40 37.00 42.70 38.20 40.50 4.170	Lack cot 370 355 357 333 350 280 250 207	kion soil, 36.40 32.00 27.65 23.45 19.20 15.25 11.95 9.10	$Y_s = 1.02$ 11.20 10.50 9.76 9.00 8.20 7.3 6.16 4.96 3.90	1920 1800 1675 1542 1405 1257 1057 850	0.0575 0.0505 0.0436 0.0370 0.0303 0.0241 0.0189 0.0144	
Head +% S1 38.60 34.20 29.80 25.50 21.40 17.00 13.50 10.40 7.80	for n urry 34. 29. 25. 21. 17. 13. 10. 7. 5. 4.	o flc with 20 80 50 40 50 40 80 60	w = 0.0 pure b] 33.30 33.90 36.40 37.00 42.70 38.20 40.50 4.170 46.20	Lack cot 370 355 357 333 350 280 250 207 180	ki ston soil, 36.40 32.00 27.65 23.45 19.20 15.25 11.95 9.10 6.70	$Y_s = 1.02$ 11.20 10.50 9.76 9.00 8.20 7.3 6.16 4.96 3.90	1920 1800 1675 1542 1405 1257 1057 850 668	0.0575 0.0505 0.0436 0.0370 0.0303 0.0241 0.0189 0.0144 0.0106	

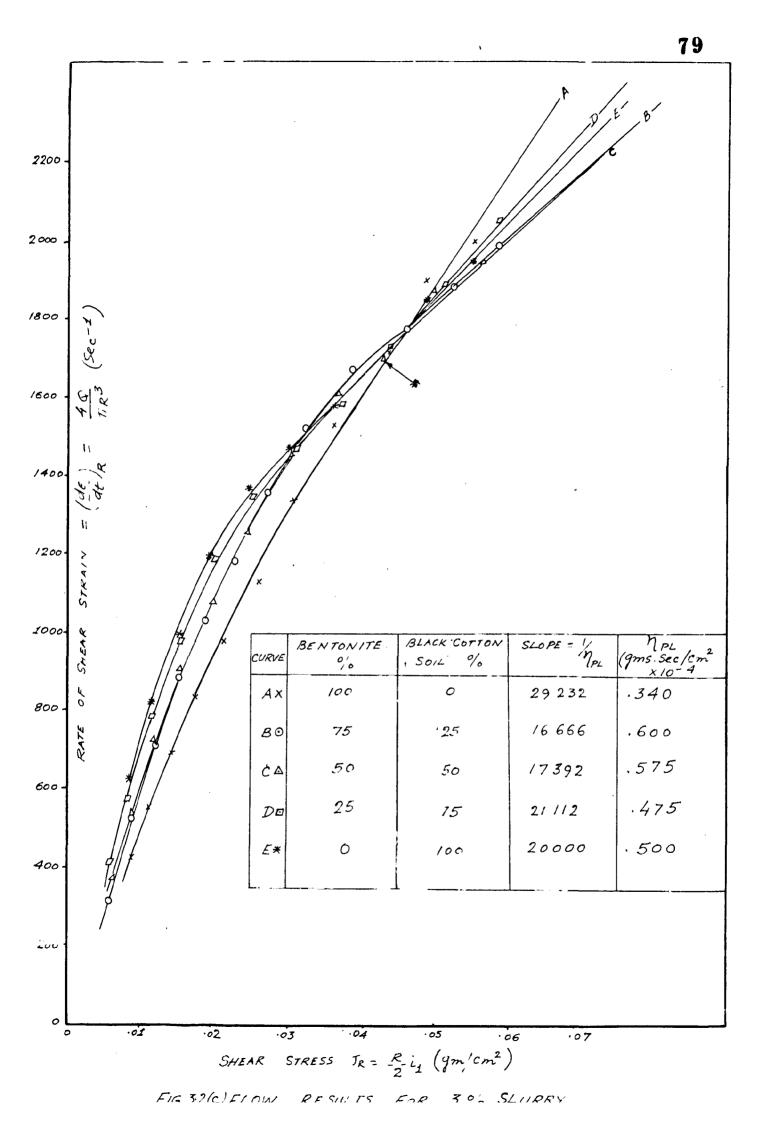
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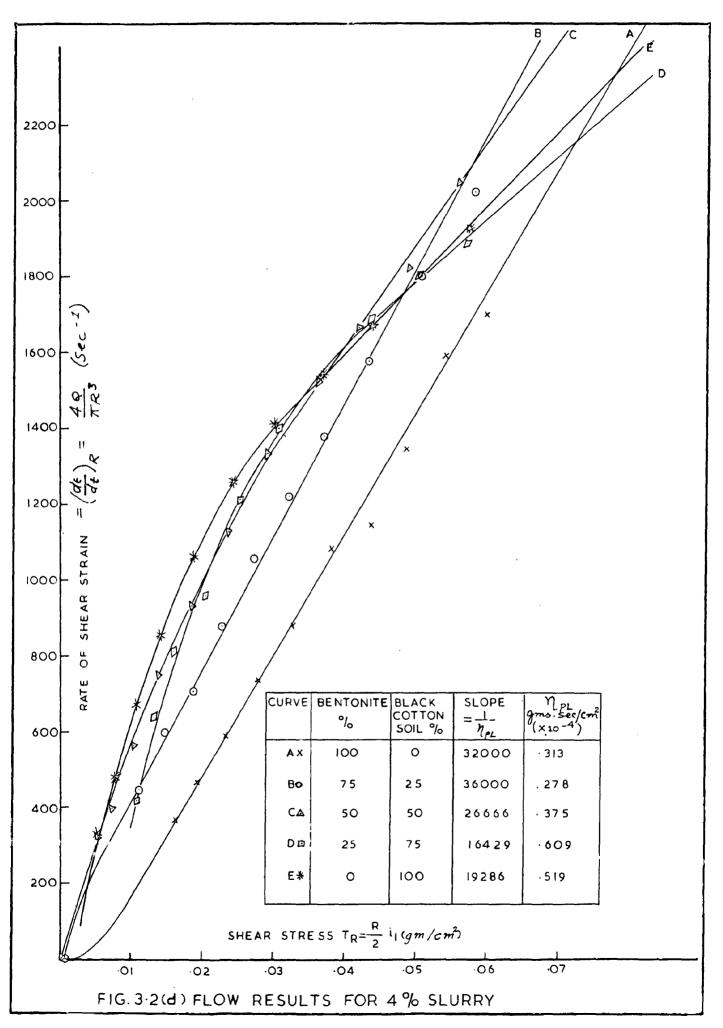
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3.5.2 Sieve Analysis of Sand

TABLE 3.2 SIEVE ANALYSIS RESULTS FOR THE SAND

Weight of Sand Taken 990.0 gm.

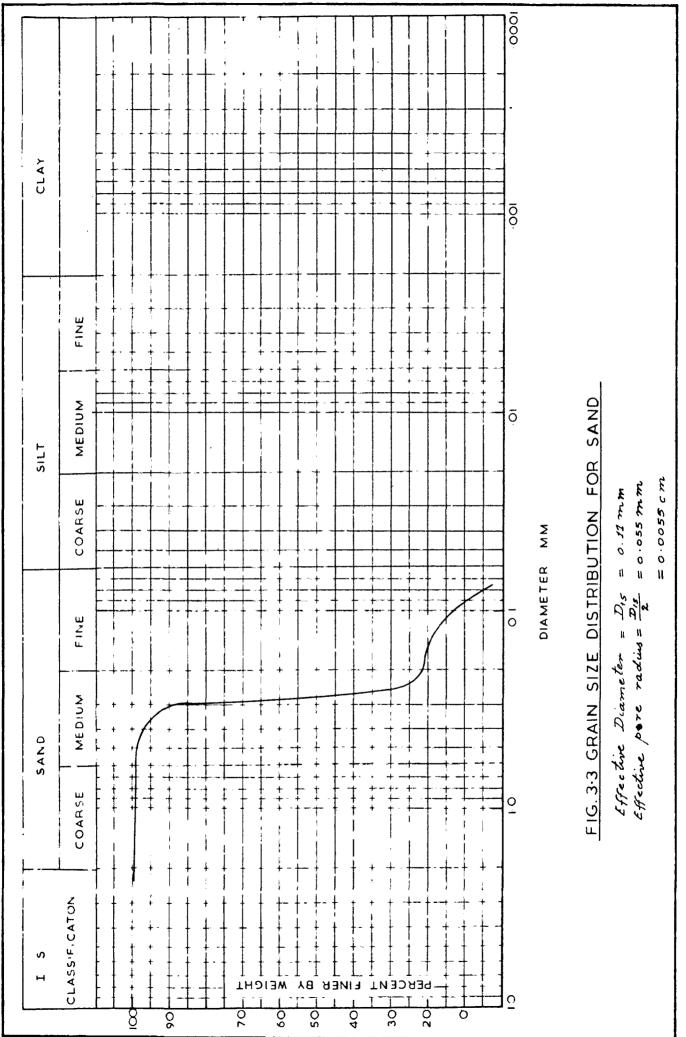
S . No	Sieve • No.	Particle size D (mm)	Weight of empty sieve (gm)	Weight of sieve and sand retained (gm)	of sand retained	% retained	tive %	Cumula- tive % finer
1	2.36	2.360	511.0	518.0	7.0	0.76	0.76	99.24
2	1.00	1.000	455.0	458.0	3.0	0.33	1.09	98.91
3	850#	0.850	385.0	386.0	1.0	0.11	1.20	98.80
4	600 µ	0.600	14148.0	449.0	1.0	0.11	1.31	98 •69
5	420 #	0 • 420	367.0	373.0	6.0	0.65	1.96	98.04
6	300µ	0.300	439.0	512.8	73.8	7.95	8.41	91.59
7	225 #	0.225	320.0	1000.6	680.6	68.74	77.15	22.85
8	150#	0.150	408.0	420.0	12.0	1.30	78.45	a1.55
9	75 H	0.075	376.0	559•6	183.6	18•54	96.99	3.01
10	Pa \$ 3	-	398.0	420 . 0	22.0 passing 754)			

3.5.3 Stability Results and Calculations

Stabilizing effect of slurry saturated zone

From equation 2.28 the stabilizing effect of the slurry saturated zone, P_G is given by $P_{s}+E_{p} = \frac{\cot\theta(\frac{1}{2}YH^{2}+ qH)(\sin\theta-\cot\theta\tan\phi) + (\frac{1}{2}Y_{w}m^{2}H^{2}\csc\theta\tan\phi) - P_{g}(\tan\phi)}{\cos\theta + \sin\theta} \tan\phi$ where the variables are as defined earlier. $P_{g} = \frac{2 \tau_{m} \ell^{2}}{(Equation 2.10 (a))}$

r cos O



r being the effective pore radius of the sand =
$$\frac{D_{15}}{2}$$

Also $P_s + = \frac{1}{2} Y_s H^2$
 $E_p = \frac{\tau_s H^2}{2a} + \frac{\pi \tau_s H}{2}$
and substituting above
 $\frac{1}{2} Y_s H^2 + \frac{\tau_s H^2}{2a} + \pi \tau_s H$
 $\frac{\cot \theta (\frac{1}{2} T H^2 + q H) (sin\theta + \cot \theta \theta sin \theta t an \theta)}{\cos \theta + \sin \theta \tan \theta}$
 τ_s , the structural shear strength of the slurry in the trench
is sufficiently small to be neglect $\frac{\theta d}{\theta s}$, thus,
 $\frac{1}{2} Y_s H^2$
 $= \frac{\cot \theta (\frac{1}{2} T H^2 + q H) (sin\theta - \cot \theta \sin \theta t) + (\frac{1}{2} Y_s M^2 H^2 \cos \theta \cos \theta \tan \theta t) - P_c) \tan \theta t}{\cos \theta + \sin \theta \tan \theta t}$
Taking θ for the sand as 30° , $\theta = 45 + \frac{\theta t}{2} = 60^\circ$. For the
loading tests carried out all the variables except P_c are
known (For dry sand $m = 0$). For each loading test, P_c is
calculated as follows -
Sample Calculation for P_c
 $H = 14.7$ cm,
Failure load = 45.74 kg.
Area of Loading plate = 316 sq.cm.

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$$\therefore \mathbf{q}' = \frac{45 \cdot 74}{316} = 0.145 \text{ kg/cm}^2 = 145 \text{ gm/cm}^2$$

$$2\ell = 3 \text{ cm}, \quad \ell = 1.50 \text{ cm}$$
For dry sand $\mathbf{m} = 0$, $Y = 1.578 \text{ gm/cm}^3$

$$\therefore \frac{1}{2} \times 1.024 \times (14.7)^2$$

$$= \frac{\cot 60^{\frac{1}{2}} \times 1.578 \times (14.7)^2 + 145 \times 14.7}{\cos 60^{\circ} + \sin 60^{\circ} - \cos 60^{\circ} \tan 30^{\circ}} + [0 - P_{\text{G}}] \tan 30^{\circ}}{\cos 60^{\circ} + \sin 60^{\circ} \tan 30^{\circ}}$$

$$\therefore \frac{110.9}{(-77)4(170+2134)} = \frac{0.5774(170+2134)(0.866-0.520.5774) + (-P_{\text{G}} \times 0.5774)}{0.5 + 0.866 \times 0.5774}$$

$$= 770 - 0.5774 P_{\text{G}}$$

$$\therefore P_{\text{G}} = \frac{770 - 110.9}{0.5774} = 1140 \text{ gm/cm}$$
From sieve analysis results (Fig. 3.3), $\mathbf{r} = \frac{D_{15}}{2} = 0.0055 \text{ cm}.$

$$\therefore 1140 = \frac{2 \tau_{\text{m}} \times (1.50)^2}{r \cos 6} = \frac{2 \tau_{\text{m}} \times 2.25}{0.00555 \times 0.55}$$

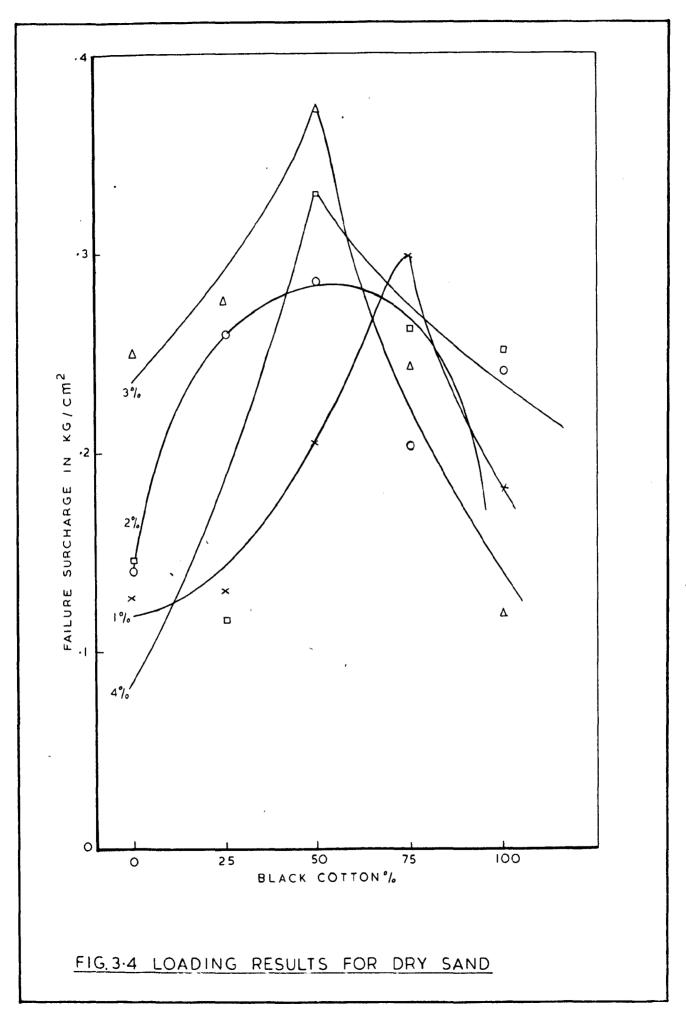
$$\therefore \tau_{\text{m}} = \frac{1140 \times 0.005 \times 0.5}{2 \times 2.25} \text{ gm/cm}^2$$

$$= 0.696 \text{ gm/cm}^2$$
say = 0.7 gm/cm²

All the calculations and results for dry sand are summarised on Table 3.3 . Results for loading on saturated sand are shown on Table 3.4 . Fig. 3.4 shows a plotting of the loading results for dry sand.

				85 -	
	T m (kg/cm ²) (x10 ⁻ 3)	3 1 1 1 1		0.394 0.168 0.084 0.054 0.054	0.536 0.536 0.0536 0.0575 0.0575 0.1055
ر م	Р _G (kg/cm)		2-117 2-160 2-160 2-17 2-970 2-970		1.133 0.870 2.185 2.070
- 10 ,	P _G tan g (kg/cm)		0.645 0.645 0.936	1.158 1.270 1.600 0.505	0.659
/Ω Bm/ cm−	Hydro- static Pres.of slurry 2 sH ² (kg/cm)	0 136 0 1136 0 113 0 111		0.107 0.105 0.106 0.100 0.100	0.110
$Q/\frac{1}{2} (-1) = 1$	Failure sur- charge (kg/cm	0.127 0.130 0.204 0.298 0.181	0.140 0.260 0.286 0.203 0.203	0.250 0.276 0.374 0.374 0.242 0.118	0.115 0.330 0.262 0.250
(gurbson	Total Failure load (kg)	40 24 41 24 64 24 98 24 67 24		79.21 82.24 119.24 76.24 37.24	15 74 101 24 83 04 79 24
ouec	Plastic viscosi- ty N _{PL} gm-sec/ e cm ² (x10-3)	0.0665 0.0500 0.0585 0.0485 0.0485	0.0575 0.0515 0.0555 0.0500	0.0340 0.0600 0.0575 0.0175 0.0175	0.0313 0.0278 0.0375 0.0509 0.0519
אזע) כנתו	f Double width of slurry satura- ted zone 2ℓ cm	1111		<i>N</i> œũ ũర u <i>V</i> V Q O	ww.o ⁴⁰ 0 000 <i>7.7</i>
CLTNCAN JO	Depth of slurry H (cm)	6 11116 10800 10800	1004ma	14-5 13-19 13-19 14-00 14-00	44647
THIMMOC	- Black cotton soil (%)	0 X2X2	1005 1005 005 005 005 005 005 005 005 00	0 202 002 002	00700 007000
J•J	Benton- ite (%)	00 20 20 20 20	100 2070 2070	100 27 07 20 20 20	100 707 07070
HUHUT.	Slurry densi- ty ^Y s (gm/cm3	1.005	1.013	1.018	1.024
	Slurry Conc- entra tion (%)	~	N	e	*

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Slurry con- centr- ation %	density	Bentonite %	Black cotton soil %	Depth of slurry H cm	Remarks
	2	3	1 4	5	6
1	1.005	100	0	15.0	No load . Failure occurred as soon as partition was removed
		25	75	14.0	No load . Failure occurred before partition was comple- tely removed i.e. failure at partial removal of partition
2	1.013	100	0	14.5	No load.Failure at partial removal of partition.
3	1.018	100	0 14.5 No load. Failure occurred imm ediately after complete remov of partition.		
4.	1.024	100	0	14.5	No load but just stable.

3.6 PROBLEMS ENCOUNTERED

A few problems were encountered in this study. The slurry made from the commercial bentonite and black cotton soil could not be kept free of segregation. O_{η} ly the 3% and the 4% pure bentonite slurries did not show much segregation. This segregation disrupted the the uniformity of the slurry and subsequently the shear strength of the slurry already created. In this respect it would be difficult to achieve the expected maximum gel strength, $\tau_{\rm m}$, of the slurry.

In order to avoid the segregation the slurry had to be kept in agitation throughout and the shear strength utilized is that under disturbed conditions.

Efforts to determine the shear strength of the slurry with a shear vane apparatus failed mainly due to the small amount of the shear strength as compared to the magnitude which could be determined by the available apparatus.

Initially trials to obtain failure load by placing different loads on the loading plate did not result either to success or practicability. The loading method as shown on plate 3.3 was found to be very unsafe. The loads could slip which was very likely at failure as the loading plate was tilted by the flow of sand into the slurry. At one occasion the loading box was greatly damaged and time had to be allowed for repair work. Fortunately no person or other apparatus was nearby otherwise greater damage would have occurred. It was also difficult to accurately record the failure load due to difficulty in gradual variation of the loading operation. At times it was necessary to repeat the loading due either to unreliable results or the necessity of checking the failure load. This method of loading was unsuitable for this checking as the slurry could not be recovered after failure. The sand was flown into the slurry by the sudden skidding of the weights and no undisturbed slurry could be recovered. The axial loading unit of the triaxial machine, however, was a remedy to all these pitfalls.

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CHAPTER IV

DISCUSSION OF RESULTS

1. Bentonite replacement by black cotton soil in bentonite slurries increases the fluidity of the slurry. This is obvious from the flow test results in Table 3.1 which show that for the same slurry concentration and mean head, more flow per unit time, Q, is obtained for the slurries where bentonite has been replaced by black cotton soil. Likewise from the loading test results, table 3.3, for same slurry concentration, the rate of permeation into the sand is more for the slurries where bentonite has been replaced by black cotton soil. The increase in the fluidity of the slurry is due to the lowering of the plastic viscosity , $\eta_{\rm PL}$, of the slurry (Figures 3.2(a), (b), (c) and (d)).

2. Decrease in plastic viscosity of the slurry with subsequent increase of permeation rate into soil means increase of the slurry saturated zone. This will increase the magnitude of the pore pressure, $P_{\rm G}$, (equation 2.10(a)), resulting in more stabilizing force of the trench (equation (2.28)). This is confirmed by the results of table 3.3. which show that more replacement of bentonite by black cotton soil means more slurry saturated zone and more surcharge that can be sustained.

3. For each slurry concentration, there is a particular proportion of bentonite which can be replaced to give a peak sustained surcharge. This proportion varies between 50% and 75%. (Table 3.3 and Figure 3.4). 4. For dry sand and for the slurry concentration range studied (1% to 4%), the major stabilizing force is that due to the shearing resistance of the slurry saturated zone; P_G tan ϕ . As shown on table 3.3, the stabilizing force due to P_G , is between 5 and 10 times the hydrostatic force.

In order to visualize the magnitude of this contribution in phototype (field) trenches the following calculations are necessary.

> From equation (2.10(a)), $P_{G} = \frac{2 \tau_{m} \ell^{2}}{r \cos \theta}$

Assuming that for the model and prototype, the same sand, and slurry are used, and that the inclination of the failure wedge to the horizontal is the same, then,

$$P_{\rm G} \propto x^2$$
 ...(4.1)

Further, from equation (2.6) the depth, 2ℓ , penetrated by the slurry is given by

$$2\ell = \frac{pgr}{2J_{f}}$$

i.e. $2\ell \propto p$...(4.2)

where p is the applied pressure differential

. . $\ell \propto (\gamma_{s}H - \gamma_{w} m H)$, for saturated sand, and

$$\ell \propto Y_{s}H$$

i.e. $\ell \propto H$ for dry sand (m=0) ...(4.2a)
 $\cdot P_{G} \propto \frac{2}{\ell}$
 $\propto H^{2}$ (from equations (4.1) and 4.2a) ...(4.3)

Thus if

Negative gel pressure in model P_{Gm} = = Negative gel pressure in prototype (field) P_{Gp} depth of slurry in model Hm depth of slurry in prototype. H Then, $\frac{P_{GP}}{P_{Cm}} = \frac{H_P^2}{H^2}$ $P_{GP} = \left(\frac{H_P}{H_m}\right)^2 P_{Gm}$...(4.4) From equation (2.28), if q_{p} is the unit surcharge in the prototype then, $\frac{1}{2} \Upsilon_{s} H_{p}^{2} = \frac{\cos \left(\frac{1}{2} \Upsilon_{p} H_{p}^{2} + q_{p} H_{p}^{2}\right) (\sin \theta - \cos \theta \sin \phi') - P_{GP} \tan \phi'}{\cos \theta + \sin \theta \tan \phi'}$ (Since m = 0 for dry sand) $\frac{1}{2} \Upsilon_{\rm s} H_{\rm p}^2 = 0.334 \left(\frac{1}{2} \Upsilon H_{\rm p}^2 + q_{\rm p} H_{\rm p}\right) - 0.5773 P_{\rm GP} \qquad \dots (4.5)$ for $\phi' = 30^{\circ}$ and $\theta = 60^{\circ}$

From a known value of H_p , using the values of H_m , Y,Ys and P_{Gm} from table 3.3, values of P_{Gp} can be calculated from equation (4.4) and subsequently values of the first term on the right hand side of equation (4.5) Then

(i) Total disturbing force = 0.334 (¹/₂ Y H²_p + q_p H_p)
(ii) Stabilizing forces are

(a) Hydrostatic pressure = ¹/₂ Y_sH²_p
(b) Horizontal component of P_{GP} = P_{GP} tan \$\phi\$!=0.5773P_{GP}

neglecting other forces which are small in comparison. Calculations based on a 10 meter deep trench are summarised in Table 4.1

Table 4.1 shows that for the full scale trench, the stabilizing force due to the slurry saturated zone is still predominant. However, the neglecting of the stabilizing forces due to the filter cake, the pressure resistance of the slurry and the other small stabilizing forces will have the effect of making $P_{\rm G}$ tan ϕ appear slightly larger than it should actually be. From table 2.4 and 2.5 these neglected forces account for less than 10-20% of the disturbing force. Even with this allowance the contribution of the slurry saturated zone in the full scale trench is still more than 60%. This would mean that the slurry saturated zone contribution appears to be more than three times that of the hydrostatic force, in the particular conditions assumed.

5. For the few tests carried on saturated sand, stability was only achieved for 4% pure bentonite slurry. This seems to indicate that denser slurries are required for stability of trenches in saturated soils. Experiments carried out by RAJU and SASTRY¹² on saturated sand showed that denser slurries (3% and above) are required for stability of trenches in saturated sands. The trench was 50 cm deep and the sand was saturated from the bottom until filure occurred. The sand was saturated after the filter cake and the slurry saturated zone had formed. All the trenches failed when the water level in the sand was between 39 and 48 cm above bottom except for 7% pure bentonite slurry. The sand used had D_{10} of 0.3 mm. There was no surcharge. These

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TABLE 4.1 APPLICATION OF LOADING RESULTS TO FULL SCALE TRENCH.

 $\frac{1}{2}$ Y H²_p = 789 kg/cm $H_p = 10.0 \text{ metres}$ $Y = 1.578 \text{ gm/cm}^3$

	Disturbing	• P _G tanø±	12	82 +0 82 +0 82 35 82 60 84 60 84 60	84.60 91.55 92.10 91.95	91-50 92-50 92-680 93-355 83-355	85.60 82.10 93.85 91.85 91.20
	% of D. Force	Hydrost atic Pres.of	11	177 60 177 60 177 40 177 20	7 8 9 9 9 9 9 0 7 7 8 9 0 7 9 0 7 0 7 0 7 0 7 0 7 7 8 9 9 0 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7	8.50 9.70 16.65	77 88 81 77 0 0 7 7 7 0
	izing ces.	sta+ P _G tanø sres <mark>.</mark> 5773P _G v' (kg/cm)	10	0000 000 000 000 00 00 00 00 00 00 00 0	2850 7500 7800 7800	57+80 6190 6300 2550	30 50 23 50 7773 5300 73773 73773 7300
	· ·	P tic r a) of surr	(kg/cm)	502.5	506.5	509 • 0	512.0
	Di stur force		8	2852.5 2843.0 4832.5 3302.55 3302.55	3376 6006 6406 5306 306 306 306	5989 0 6609 0 6609 0 54449 0 3059 0	3562.0 2862.0 8342.0 6285.0 5812.0
	P _{GP}	(KB/ CIII)	7	4025 4560 7500 11000 1850	10300 10300 13600 13600	9.500 10.700 16.200 8.560 1,44.50	5300 13600 13600 9300
			9	1.100 1.028 1.650 2.490 1.572	1.117 2.160 2.440 2.970	2.000 2.210 2.760 1.870 0.873	1.133 0.876 2.485 2.100 2.070
	Depth slurry		Ъ	14111 6074706 008000	2007.00 007.00	770277 08-150	4404 <i>0</i> 000000
	Slurry Density S (gm/cm ³		*	1.005	1.013	1•018	1.024
	Black Cotton . soil %		Э	107200 070070		0 N N N O 0 N O N O L	4 07070 07070
	Ben nit			20070 20070 20070	4 07070 07070	00 00 00 00 00 00 00 00 00 00 00 00 00	270 270 070 070
	Slurry	2000	-	, -	N	m .	<u>ب</u> + د

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results reveal less stabilizing effect of slurry saturated zone than that in the present series. (The 4% pure bentonite test was just stable). One factor would be that the sand used by Raju and Sastry being coarser (the one used here had D_{15} of 0.055 mm) would, from equation(2.10a), decrease the magnitude of the negative pure pressure P_{G} , which is as shown is inversely proportional to r and r = $D_{15}/2$ (or $D_{10}/2$). P_{G} tan ϕ is the contribution of the slurry saturated zone to stability. However, the main aspect seems to be the nature of shear strength in the slurry saturated zone and themode of loading. The penetration of slurry creates large capillary forces leading to a large apparent cohesion in the dry sand. This apparent cohesion is lost on saturation leading to loss of strength, even in cases where the slurry has been allowed to penetrate, as by Raju and Sastry, under dry conditions. Actually in the field, below water table the sand is already saturated at the time of excavation, and the penetration itself is much less than for dry sands. Thus, the large stabilising force of slurry saturated zone, can only be relied upon above the water table.

The equilibrium achieved for the 4% pure bentonite slurry can be viewed from Rankine's Earth pressure theory approach. The activating force, P_a , is made up of (a) Lateral pressure due to submerged sand (of submerged density Y') and (b) lateral pressure due to water (there was no surcharge) i.e. $P_a = \frac{1}{2} K_a Y' H^2 + \frac{1}{2} Y_W H^2$, $K_a = \frac{1}{3}$ for $p' = 30^\circ$, the state of the

$$P_{a} = \frac{1}{6} (1.960 - Y_{w}) + \frac{1}{2}Y_{w}H^{2} = \frac{1}{6}x \ 0.96H^{2} + \frac{1}{2}H^{2+} (Y_{w} = 1.00 \text{ gm/cm}^{3})$$

= 0.66 H²
H was 14.5 cm.
 $P_{a} = 0.66 (14.5)^{2}$
= 140 gm/cm.
Hydrostatic pressure of the slurry = $\frac{1}{2}Y_{s}H^{2}$
= $\frac{1}{2}x \ 1.024 x (14.5)^{2} \text{ gm/cm}$
= 108 gm/cm.

All other stabilizing forces sum up to

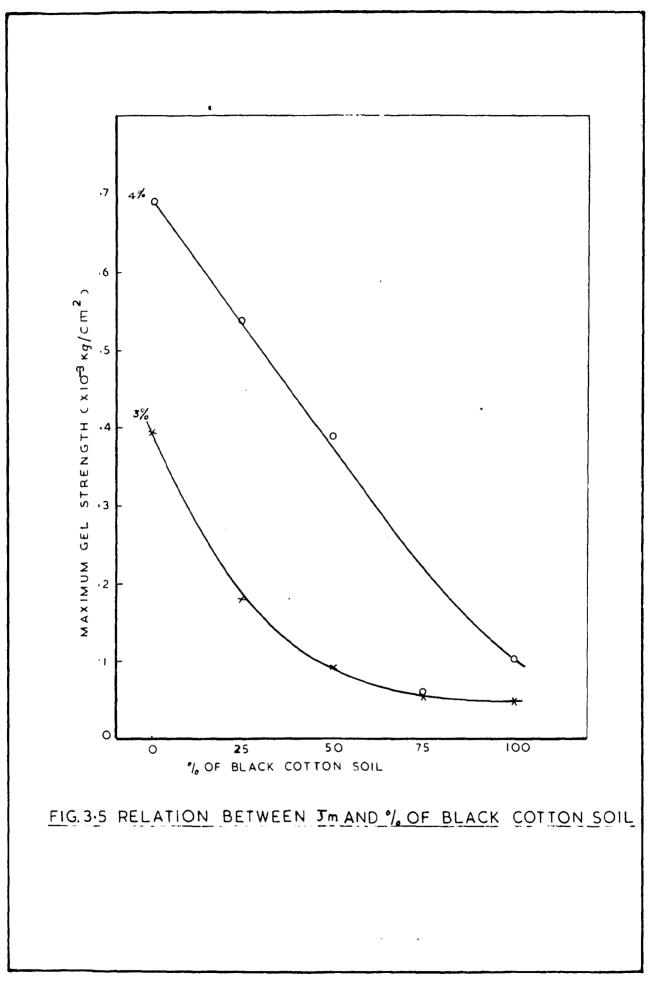
(140 - 108) gm/cm = 32 gm/cm.

The hydrostatic pressure of the slurry contributes a resistance, 77.3% of the disturbing force.

5 **?**

This indicates that the hydrostatic pressure of the slurry provides the major contribution to stability in saturated sands. The permeation of the saturated sand by the slurry is not effected unless a substantial differential head between the slurry and the water level in the sand is maintained (equation 2.6). This limits the depth of the slurry saturated zone, further as remarked earlier, capillary tensions cannot be sustained in saturated sand.

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CHAPTER V

CONCLUSIONS

From the observed results and the discussions of Chapter IV, the following conclusions can be drawn.

- 1. Lean bentonite slurries have less plastic viscosity than dense ones. This reduce plastic viscosity increases their fluidity and their ability to flow and permeate soil pores. Replacement of bentonite by black cotton soil also reduces the plastic viscosity of the slurry and thus increases the permeation rate. The reduction of plastic viscosity with increase in percentage of black cotton soil does not, however show a regular trend. This is probably due to the limited accuracy of apparatus used.
- 2. In stability of slurry trenches the three main stabilizing factors are the hydrostatic pressure of the slurry, the shearing resistance of the slurry saturated zone and the passive resistance of the slurry. In the lean slurries studied, the structural shear strength of the slurry is very small. This will also make the contribution of the . passive resistance of the slurry to stability small. Thus in the slurries studied the first two were the main stabilizing factors.
- 3. In dry sand trenches and for the sand size used, the major stabilizing force is that due to the shearing resistance of the slurry saturated zone. The rate of penetration and the depth of the slurry saturated zone can be increased by

replacing some proportion of bentonite by black cotton soil.

- 4. For each slurry concentration there is a particular proportion of bentonite which can be replaced by black cotton soil to achieve optimum stabilizing effect.
- 5. For saturated sands the rate of permeation of bentonite slurry into the sand is greatly reduced . In this case, unless the permeation rate can be increased by creating a sufficient pressure differential, the effect of the slurry saturated zone is not significant. Stability will depend mainly on the hydrostatic pressure of the slurry. The effect of using bentonite s lurries with black cotton soil as admixture for stability of saturated sand trenches will thus need maintaining a pressure differential to effect formation of a sizable slurry saturated zone. Even so due to full saturation of pores the effect would be relatively less. This needs further exploratory work.

Scope for Further Work

1. The results of loading tests on dry sand indicated the importance of the slurry saturated zone formed by permeation of slurry into the sand to improve the physical properties of the zone by gelling in soil pores under quiescent conditions. Further study is necessary to investigate the extent to which the shear strength of the zone is improved by the slurry. Triaxial shear tests on specimens sampled from the zone would give qualitative results.

- 2. The effect of additives as sodium carbonate need also to be thoroughly studied.
- 3. There are many variables controlling the stabilizing effect. Some of these, like the structural shear strength of the slurry, resistance to deformation of the filter cake and the electro-osmotic forces, though small, have been neglected.A qualitative contribution of these factors needs investigation and refined apparatus is needed. This will also enable more control of the variables.
- 4. The results obtained are based on commercial bentonite, which as discussed earlier had, among other things, the problem of segregation. Usage of better quality bentonite may have some effect on the results.
- 5. The range of slurry concentration studied is only from 1% to 4%. A wider range, preferably upto 8% is suggested.
- 6. Most river valley projects involve s ites of saturated sand deposits. The few tests carried on saturated sand indicate that the properties of the slurry required for stability (e.g. density, black cotton soil etc) are different from those of slurries for stabilizing dry sand. A thorough exploratory work is needed in this field which in most cases is of more practical application to the hydraulic engineering profession.

As a concluding note, it is worth mentioning that the experimental work has been planned and carried out with the

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purpose of investigating the usefulness of black cotton soil as an admixture to bentonite slurry in a qualitative way. This aim has been fulfilled to a certain extent. The tests were mainly conducted on dry sand and more work is necessary on other types of soils which occur in fields of engineering practice. All the variables and factors likely to affect the study should, where possible, be evaluated as accurately as possible with their quantitative and qualitative contribution fully determined.

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