This is to certify that the

thesis entitled

"MECHANISM OF FRICTION AND COHESION IN CLAYS"

presented by

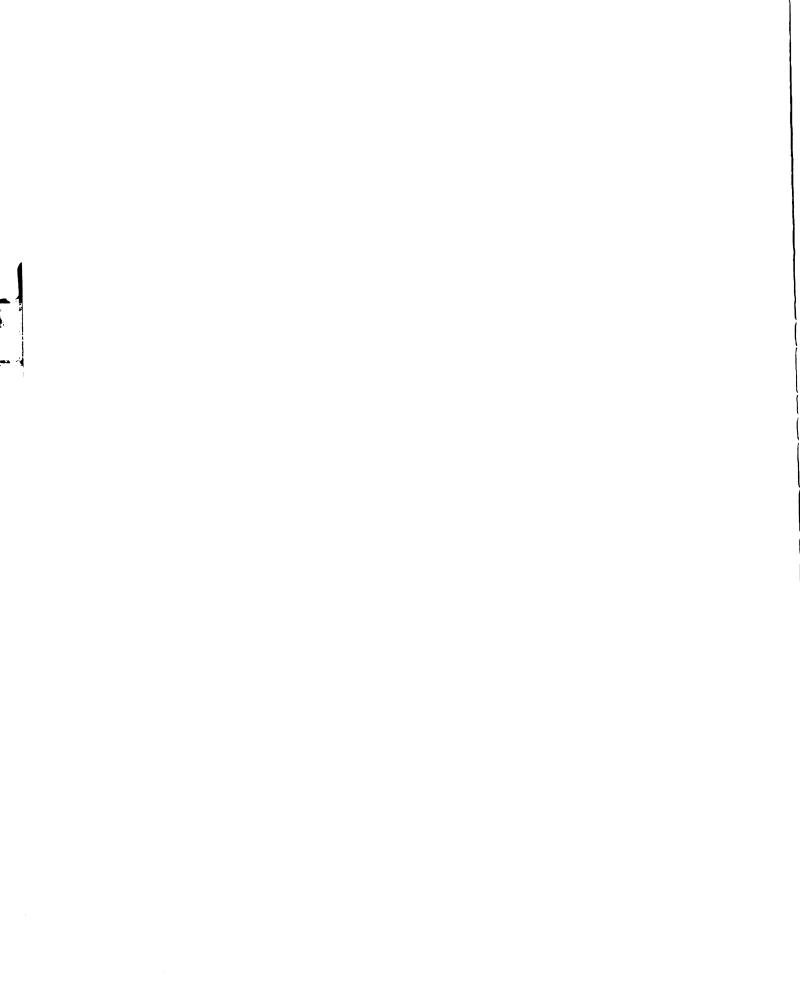
Abraham K. Loh

has been accepted towards fulfillment of the requirements for

Ph. D. degree in Civil Engineering

Major professor

Date 10, 1964



ABSTRACT

MECHANISM OF FRICTION AND COHESION IN CLAYS

by Abraham K. Loh

A particle model is proposed for the behavior of friction and cohesion in clays. In this model, the resistance at the contacts of the particles is assumed to consist of a mineral friction and an adhesion. It varies with the orientation of the particle relative to the direction of shear.

The angle of internal friction is the result of the resistance due to mineral friction at the contacts. The cohesion consists of a non-viscous part and a viscous part. The non-viscous part of cohesion is the result of the resistance due to adhesion at the contacts. The behavior of the viscous part of cohesion follows the flow phenomenon of the clay particle structure as a rate process.

The experimental behaviors of friction and cohesion were investigated by means of the triaxial test. Creep, Creep-CFS and CFS tests were carried out on five different clays that include laboratory-prepared and undisturbed clays.

It is found that the experimental behavior of friction and cohesion under constant strain rate and creep can be satisfactorily correlated with the predicted behavior according to the proposed model.

MECHANISM OF FRICTION AND COHESION IN CLAYS

By

Abraham K. Loh

A THESIS

submitted to

Michigan State University
in partial fulfillment of the requirements

for the degree of

DOCTOR OF PHILOSOPHY

Department of Civil Engineering

1964

33886

ACKNOWLEDGEMENTS

The author wishes to express his sincere gratitude to his major professor, Dr. T. H. Wu for his guidance and encouragement in the preparation of this thesis as well as throughout the author's doctoral study. He is also indebted to his guidance Committee members, Dr. O. B. Andersland, Department of Civil Engineering, Dr. L. E. Malvern, Department of Metallurgy, Mechanics and Materials Science, Dr. D. J. Montgomery, Department of Physics and Astronomy and Dr. J. L. Dye, Department of Chemistry for their guidance throughout his doctoral study. F. J. Holliday and W. L. Warder are thanked for their assistance in the laboratory.

The financial support of the National Science Foundation is gratefully acknowledged.

TABLE OF CONTENTS

Chapter				Page
1.	INTR	ODUCTIO	N	1
2.	THEO	RETICAL	CONSIDERATIONS	4
	2.1	Partic	le Model	4
		2.1.1	Particle Movement and Resistance	5
			a. Sliding Type of Movement and Resistance	6
			b. Displacement Type of Movement and Resistance	7
		2.1.2	Contact Resistance	8
			a. Frictional Resistance against Sliding	9
			b. Frictional Resistance against Displacement	13
			c. Adhesional Resistance against Sliding	21
			d. Adhesional Resistance against Displacement	23
		2.1.3	Redistribution of Contacts	27
			a. Partial Remoldingb. Complete Remoldingc. Partial Re-orientationd. Complete Re-orientation	28 29 29 30
		2.1.4		30 35
	2.2	The Me	chanism of Cohesion in Clays	40
3.	EXPE	RIMENTA	L PROGRAM	45
		Materi	ive als Used Preparation	45 45 4 6
		3.3.1 3.3.2 3.3.3 3.3.4 3.3.5	Remolded Sault Clay Grundite	46 46 48 48

Chapter				Page
	3.4	Triaxia	al Tests	49
		3.4.2	CFS Test Creep-CFS Test Creep Test	50 53 56
4.	ANAL	YSIS OF	EXPERIMENTAL RESULTS	51
	4.1	Present	tation of Test Data	61
			CFS Tests Creep-CFS Tests	61 62
	4.2	The Bel	navior of Friction	63
		4.2.1	The Experimental Behavior of Friction	63
			a. CFS Testsb. Creep-CFS Testsc. Discussion	63 64 66
		4.2.2	Analysis of the Behavior of Friction	66
			a. Behavior of D' under Constant Strain Rate b. Behavior of D' under Creep	66 3 7
		4.2.3	Summary	72
	4.3	The Bel	navior of Cohesion	73
		4.3.1	The Experimental Behavior of Cohesion	73
			a. CFS Testsb. Creep-CFS Testsc. Discussion	73 73 74
		4.3.2	Analysis of the Behavior of Cohesion	74
			 Behavior of c' under Constant Strain Rate Behavior of c' under Creep (Creep-CFS tests) 	· 74
		4.3.3	•	87
5.	CONC	LUSIONS		87
BIRLING	DŊDĽIV			0.1

LIST OF TABLES

Table		Page
1.	<pre>Index properties of clays studied</pre>	47
2.	Summary of CF3 tests	51
3.	Summary of Creep-CFS tests	54
4.	Summary of Creep tests	57
5a.	Summary of model parameters (after R. W. Christensen)	58
5b.	Summary of model parameters from Creep tests	59
5.	Strain and time characteristics of friction CFS tests	65
7.	Friction-time characteristicsCreep-CFS test on Sault clay	66
8.	Estimated errors between c' and c'oct	81

LIST OF FIGURES

Figure		Page
la.	Section through particles	93
lb.	Definition of $A_{\hat{\mathcal{C}}}$	93
2.	Sliding movement and resistance	94
3.	Displacement movement and resistance	94
4.	Frictional resistance against sliding	95
5.	Frictional resistance against displacement	95
5.	Comparison of t_{2g}	9 6
7.	Spectrum of friction resistance: $\rho = 0.3$	97
8.	Spectrum of friction resistance: $\rho = 0.5$	98
9.	Comparison of friction resistance for $\alpha = 50^{\circ}$ at 60°	99
10.	Adhesional resistance against sliding	100
11.	Adhesional resistance against displacement	100
12.	Spectrum of adhesion resistance	101
13.	Stress-displacement curves for ρ = 0.3	102
14.	Average stress-displacement curves for ρ = 0.3.	102
15.	Stress-displacement curves for $\rho = 0.5$	103
16.	Average stress-displacement curves for ρ = 0.5.	103
17.	Behavior of ϕ according to particle model	104
18.	Rheologic model for clay particle structure	105
19.	<pre>Φ'-strain curves-CF3 tests on compacted Sault clay</pre>	105
20.	<pre>\$\display -strain curves-CFS tests on remolded Sault clay</pre>	103
21.	<pre></pre>	107
22.	<pre>0'-strain curves-CFS tests on Grundite</pre>	107

Figure		Page
23.	<pre>↑'-strain curvesCFS tests on undisturbed Willow Run clay</pre>	108
24.	on compacted Sault clay	109
25.	<pre>↑'-strain rate and ↑'-time curvesCFS tests on over-consolidated Sault clay</pre>	110
25.	<pre>↑'-strain rate and ⊅'-time curvesCFS tests on Grundite</pre>	111
27.	on undisturbed Willow Run clay	112
28.	Behavior of friction in Creep-CFS tests on Sault clay	113
29.	p'-N characteristicsCFS tests on Sault clay	114
30.	Comparison of behaviors of friction in Creep- CFS tests on consolidated Sault clay	115
31.	Comparison of behaviors of friction in Creep-CFS tests on compacted Sault clay	116
32.	c'-strain curvesCFS tests on compacted Sault clay	117
33.	c'-strain curvesCFS tests on remolded Sault clay	117
34.	c'-strain curvesCFS tests on over- consolidated Sault clay	118
35.	c'-strain curvesCFS tests on Grundite	118
36.	c'-strain curvesCFS tests on undisturbed Willow Run clay	119
37.	c'-strain rate curvesCFS tests on compacted Sault clay	120
38.	c'-strain rate curvesCFS tests on Grundite .	120
39.	c'-strain rate curvesCFS tests on over- consolidated Sault clay	121
40.	c'-strain rate curvesCFS tests on undisturbed Willow Run clay	

Figure		Page
41. Behavior of cohesion in Creep-CFS tests on Sault clay		122
42. Comparison of behaviors of cohesion: compacted Sault clay	٠.٠	123
43. Comparison of behaviors of cohesion: consolidated Sault clay		124
44. Comparison of behaviors of cohesion: undisturbed Marine City clay		124
45. Comparison of behaviors of cohesion: over-consolidated Sault clay		125
46. Comparison of behaviors of cohesion: Grundite		126
47. Comparison of bheaviors of cohesion in Creek CFS tests on Sault clay	-	127
48. Comparison of c' and c' : Creep-CFS tests on compacted Sault clay		128
49. Comparison of c* and f*: Creep-CFS tests on Sault clay		129

LIST OF APPENDICES

Appendix	I	Page
I.	COMPUTATIONS OF CONTACT RESISTANCE	130
Tables		
A-1	Comparison of $t_{2\theta}$	131
A-2	Frictional resistance against sliding, t_1 .	132
A-3	Frictional resistance against displacement, t ₂	134
A-4	Frictional resistance against displacement, $t_2 = 0.3 \dots \dots \dots \dots$	135
A-5	Frictional resistance against displacement, t ₂	138
A- 6	Frictional resistance against displacement, t ₂ , = 0.5	140
A-7(a)	Adhesional resistance against sliding, t_{3} .	143
A-7(b)	Adhesional resistance against displacement, t ₄	143
II.	BEHAVIOR OF THE PARTICLE MODEL	144
	A. SymbolsB. Procedure of computations of stress- displacement characteristics accord-	145
	ing to the modes of:	145
	 (a) Partial remolding (b) Complete remolding (c) Partial re-orientation (d) Complete re-orientation 	147 149 152 157
	C. Illustrations	162
	Fig. A-1(a) Partial remotding Fig. A-2(b) Complete remolding Fig. A-3(c) Partial re-orientation Fig. A-4(d) Complete re-orientation	164 165 167
	D. Results of computations	169

Appendix		Page
	Table A-8(a) (a) Partial remolding, $r = 0.3$	170
	Table A-8(b) (b) Complete remolding, $= 0.3$	171
	Table A-8(c) (c) Partial re-orientation, = 0.3	172
	Table A-8(d) (d) Complete re-orientation, $i^{-} = 0.3$	173
	Table A-9(a) (a) Partial remolding, $_{i} = 0.5$	⊥74
	Table A-9(b) (b) Complete remolding, $f = 0.5$	175
	Table A-9(c) (c) Partial re-orientation, $\epsilon = 0.5$	176
	Table A-9(d) (d) Complete re-orientation, = 0.5	
	Table A-10 Calculation of average	178
	stress-displacement curves	179
III.	TEST DATA	180
Α.	Figures	180
A-5	Typical Creep-CFS test data for Sault clay .	181
A-6	Creep data for F-C-1-1 (Spec. No.)	182
A-7	Creep data for F-C-1-2 (Spec. No.)	183
A-8	Creep data for F-C-1-3 (Spec. No.)	184
A-9	Creep data for F-C-1-4 (Spec. No.)	185
A-10	Creep data for F-OC-1-2 (Spec. No.)	186
A-11	Croep data for F-OC-1-3 (Spec. No.)	187
A-12	Creep data for F-OC-1-4 (Spec. No.)	188
A-13	Results of c' and f' for compacted Sault clay	189
A-14	Results of c' and '' for compacted Sault clay	190
A-15	Results of c' and p' for remolded Sault clay	191
A-1 6	Results of c' and \$' for consolidated	192

Appendix		Page
A-17	Results of c' and t' for over-consolidated Sault clay	193
A-18	Results of c' and ¢' for Grundite	195
A-19	Results of c' and \$' for undistributed Willow Run clay	196
A-20	Results of c' and f' for undisturbed Marine City clay	197
В. 7	Tables	198
A-11	Results of CFS tests on compacted Sault clay	199
A-12	Results of CFS tests on over-consolidated Sault clay	200
A-13	Results of CFS tests on Grundite	201
A-14	Results of CFS tests on undisturbed Willow Run clay	202
IV.	SAMPLE CALCULATIONDETERMINATION OF PARAMETERS IN RHEOLOGIC MODEL	203
V.	ANALYSIS OF T' AND C'	205
Tables		
A-15	Friction-displacement rate characteristics at failureCFS tests on Sault clay	206
A-16	Predicted † in Creep-CFS tests on Sault clay	207
	Validity of Equation (25')	208
A-17	Predicted values of c'(f ₁)compacted Sault clay	209
A-18	Predicted values of c'(f ₁)over- consolidated Sault clay	210
A- 19	Predicted values of $c'(f_1)$ Grundite	211
A-20	Predicted values of c'(f ₁)over- consolidated Sault clay	212

Appendix		Page
A-21	Predicted values of c'(f ₁)undisturbed Marine City clay	213
A-22	Predicted c' in Creep-CFS tests on Sault clay	214
A-23	Influence of $^{\cdot}$ on c_{C}^{\prime} Creep-CFS tests	215
A-24	Values of c* in Creep-CFS tests on Sault clay	216

NOTATION

a
$$= \frac{n_5 A_c}{A_c}$$
 area of a contact along θ total soil area parallel to θ along section $A-A$ (Fig. 1b)
$$\frac{\theta-\pi}{2}$$

$$\frac{\theta-\pi}{2}$$

$$\frac{\theta-\pi}{2}$$

$$\frac{\theta-\pi}{2}$$

$$\frac{\theta-\pi}{2}$$

$$\frac{\theta-\pi}{2}$$

$$\frac{\theta-\pi}{2}$$

$$\frac{\theta-\pi}{2}$$

$$\frac{\theta-\pi}{2}$$

$$\frac{\pi}{2}$$

$$\frac{\pi$$

	axial strain
	initial axial strain in creep
∞	final axial strain in creep
C	@ any elapsed time of creep
f	@ peak ('1 - '3)
u	$(3.7)^{-1}_{3}$ = constant
um	minimal of urequired to attain tum
• "	rate of axial strain
• € C	strain rate @ or near end of creep in
	Creep-CFS test
е	displacement along - due to one contact
	failure
f _l	shear stress that causes flow in viscous
	element of rheologic model for clay particle
	structure = predicted viscous cohesion
f _o	f _l @ zero time
f*	predicted cohesion function = f_1/f_0
∴F	Activation Energy
	total shear strain
1	shear strain of the viscous element in
	rheologic model
•	rate of total shear strain
; 1	rate of shear strain of viscous element in
	rheologic model
· 'oct	rate of octahedral shear strain
h	Plank's constant
k	Boltzmann's constant

k	effective principal stress ratio = $\frac{1}{1}$
^k 1, ^k 2	Hookean springs in rheologic model
*	vertical distance between successive
	interparticle equilibrium positions
7 1	distance between flow units in direction of
	flow
tan	coefficient of mineral friction
٦	number of bonds per unit area over the
	plane of shear
m	number of slipping contacts associated with
	the displacement of particle 2
М	range in (degrees) in which contact
	failure occurs under k
n	number of contacts within a gross soil area
	A_{α} along the $pprox$ -plane
•••	number of contacts with the angle
n _m	number of contact failure in M
N	= n _m .e, contact displacement at any strain
N _u	$N \cdot t_u$, contact displacement at $\frac{1}{3} =$
	constant
$^{ m N}_{ m C}$	contact displacement at any elapsed time of
•	creep
· ·	= N/t, rate of contact displacement
N _C	= N_c/t_c , rate of displacement under creep
N _k	N_c/t_k , rate of displacement under creep
	required to attain ${}^{\downarrow}_{k}$

· N _m	= N/t_m , rate of contact displacement
	required to attain ϕ_{m}^{I}
d _p	probability of particle orientation
\mathbf{P}_{θ}	normal force to the θ -plane of a
	particle
P _F	displacement resistance due to friction
P _A	displacement resistance due to adhesion
Φ'	angle of internal friction at any
	strain €
Φ' _f	Φ at peak ($\sigma_1 - \sigma_3$)
φ ' u	Φ' at $\overline{\sigma}_1/\overline{\sigma}_3$ = constant
Φ' m	maximum value of Φ^{\bullet} at any strain ϵ
ф ' С	<pre>o' at any elapsed time of creep</pre>
Φ	maximum value of $^{\bullet}_{\text{C}}$ attainable at the
	stress level k in creep
Φ¹ um	maximum value of Φ^{\bullet} attainable under
	$\dot{N}_{\rm m}$ (@ $\overline{\sigma}_{\rm i}/\overline{\sigma}_{\rm 3}$ = constant)
R	universal gas constant
$R_{\mathfrak{S}}$	sliding resistance due to friction
ρ	a coefficient in Frictional Resistance
	against Displacement
ρ_{m}	a coefficient in Adhesional Resistance
	against Displacement
$\overline{\sigma}_{\theta}$	Effective normal stress to the
	plane θ

- S	normal stress at a contact with an angle $ heta$
	(force per unit contact area)
C	consolidation pressure
^σ 1', ^σ 3	total major and minor principal stresses
$\frac{1}{1}, \frac{\pi}{2}, \frac{\pi}{3}$	effective majo r , intermediate and minor
	principal stresses
ر آ.s' ³ 3s	major and minor principal stresses at a contact
t	time; also time required to reach N
tc	elapsed time of creep
t _m	minimal of time required to attain in
t _k	time required to attain $\mathfrak{t}_{k}^{\bullet}$
t _u	time required to reach $N_{\rm u}$
t _{um}	Minimal of time required to attain time
. t _{1:}	frictional resistance against sliding
	contributed by a contact with an angle
	(shear force per unit contact area)
t ₂ .	frictional resistance against displacement
	contributed by a contact with an
	angle : (shear force per unit contact area)
t ₃	adhesional resistance against sliding
	contributed by a contact with an
	angle (shear force per unit contact area)
t ₄	adnesional resistance against displacement
	contributed by a contact with an
	angle : (shear force per unit contact area)
Т	absolute temperature

tangential force on the --plane of a particle Т applied shear stress octahedral shear stress oct applied shear stress along the --plane shearing resistance on the --plane due to the frictional resistance against sliding at the n contacts within a soil area A_{α} (force per unit soil area) shearing resistance on the --plane due to the frictional resistance against displacement at the n contacts within a soil area A_{γ} (force per unit soil area) shearing resistance on the -plane due to the adhesional resistance against sliding at the n contacts within a soil area A_{α} (force per unit soil area) shearing resistance on the --plane due to the adhesional resistance against displacement at the n contacts within a soil area A (force per unit soil area) shear stress along a plane 9 shear stress at a contact with an angle θ (force per unit contact area) that portion of τ_0^{\bullet} due to n_{ω} contacts with angle θ (force per unit soil area).

that portion of $\tau_{\alpha}^{\bullet,\bullet}$ due to $\mathbf{n}_{\tilde{\upsilon}}$ contacts with angle θ (force per unit soil area) rate of applied shear stress redistribution of contacts θ angle of orientation of a particle with the horizontal axial deformation in creep = Δ (length) u initial deformation in creep = Δ (length) ultimate deformation in creep = $\Delta(length)_f$ uf $creep function = \frac{u - u_o}{u_f - u_o}$ U* initial water content wi final water content = $1/2 \alpha \beta \frac{k_1 k_2}{k_1 + k_2}$ t in Equation (27) Z(t)

CHAPTER I

INTRODUCTION

The shearing resistance of clays is not well understood. The lack of understanding arises from the complexity of the physical properties of clays.

The conventional concept of expressing the shearing resistance of clays in terms of a cohesion and a friction is attributed to Coulomb. While this concept is relatively simple, it does not explain the mechanism of shearing resistance in clays.

Hvorslev (1936, 1960) expressed the Coulomb criterion in terms of a true cohesion (c_e) and a true angle of friction (ϕ_e). Experimental studies have shown, however, that the parameters c_e and ϕ_e are not unique for a given soil but depend upon the structure and stress history of the soil. Conceivably, other factors may also affect c_e and ϕ_e .

The Coulomb-Hvorslev criterion is a failure theory and is not concerned with the deformation phenomena at stresses below failure. Schmertmann and Osterberg (1960) studied the variation of the cohesion and friction components of the shearing resistance with strain prior to failure. Their results show that the cohesion is mobilized at small strains while considerable deformation is needed to mobilize the full frictional resistance. Similar observations are

also reported by Wu, Douglas and Goughnour (1962).

For an understanding of the shearing resistance of cohesive soils, an examination of the clay structure appears to be a logical starting point. Tan (1957) presented a schematic picture of clay mineral network with edge-to-face contacts of clay particles. Tan's concept of the structure of clay as a solid skeleton of plate-like particles in a "card-house" structure was verified by Rosenqvist (1959) by means of the electron microscope. Mitchell (1956) and Lambe (1960) have demonstrated that deformations tend to cause particle orientation in clays. By means of the petrographic micrograph, Wu, Douglas and Goughnour (1962) illustrated the tendency of the particles to align themselves parallel to a slip plane.

The movement of the clay particles under load is considered as a rate process (Glasstone, Laidler and Eyring, 1941) by Murayama and Shibata (1961), Christensen (1964) and Mitchell (1964). These studies have shown that the shearing resistance can be related to the bonds at the contacts between the clay particles.

In this study, friction and cohesion are defined respectively as the components of shearing resistance dependent and independent of the effective stress as proposed by Schmertmann and Osterberg (1960). The general objective is to correlate the behavior of friction and cohesion with the movement and resistance of the clay particles. A particle model is proposed. In this model,

the angle of internal friction and the non-viscous part of cohesion are assumed to be the result of the resistance at the particle contacts. The viscous part of cohesion is treated as the result of the flow characteristics of the contacts and is assumed to obey the rate process theory.

The region within which two adjacent particles come closest together is termed a "contact." Actual mineral-to-mineral contact of the surfaces is not required.

CHAPTER 2

THEORETICAL CONSIDERATIONS

In this chapter, friction and cohesion are considered from the viewpoint of the clay particle structure. A particle model is developed. In this model, the resistance at the contacts of the particles is assumed to consist of a mineral friction and an adhesion. The angle of internal friction is assumed to be derived from mineral friction at the contacts. The resistance due to adhesion at the contacts is considered to be a non-viscous cohesion. The behavior of the viscous cohesion is assumed to be governed by the flow characteristics of the contacts as a rate process. A rheologic model is used to represent the combined action of the contact resistance and the viscous flow.

2.1 Particle Model

The deformation and thereby the deformational resistance of clays involve, in essence, the interaction of the clay particles. The arrangement and contact geometry of the particles in the clay mineral network, therefore, play an important role in the strength and deformation behavior of clays. In the following, the shearing resistance is treated as the result of the contact resistance of the particles in the clay matrix. Following the derivation by Skempton (1960), the contact resistance is assumed to consist of a mineral friction and an adhesion. It is further assumed that the mineral friction and adhesion at the contacts

constitute the resistance to sliding and to displacement of the contacts.

2.1.2 Particle Movement and Resistance

The mechanism assumes that the strength of the clay matrix is derived from consolidation, during which the clay particles arrange themselves in some manner such that they form a network. Due to different contact geometry, the strength of the contacts may vary widely. When the matrix is under stress, some contacts fail while others remain intact.

In order to make an analysis of the mechanism, a model is proposed in the following sections based on several simplifying assumptions. It is not intended to imply that these assumptions truly represent the local conditions in an actual clay structure. However, it is believed that the analysis based on the model nevertheless contributes to the understanding of the phenomenon and provides a systematic basis for correlating experimental observations.

The simplifying assumptions of the model are:

- 1) deformations are the result of particle movements and may be represented by the displacements at the contacts;
- 2) stresses on the contact may be calculated by the equations for the stresses in a continuum; 3) the major and minor principal stresses at the contact, σ_{1s} and σ_{3s} act in the same directions as the effective major and minor principal stresses on a specimen, $\overline{\sigma}_1$ and $\overline{\sigma}_3$ respectively; 4) all contacts have the same area $\mathbf{A}_{\mathbf{C}}$; 5) normal and tangential forces are the same at all contacts with the same angle θ ; 6) all contacts with the same angle θ have equal strengths;

7) applied stresses are uniformly distributed over the soil area under consideration; 8) applied loads are equally distributed among the contacts; 9) the model is two-dimensional only.

In an assembly of particles, various types of particle movement can be expected. For example, a particle can rotate, translate or slide over another. One may even envision a particle to slide over another while causing the other to rotate or translate. The movements therefore are combinations of sliding, rotation, translation and others. Since the particles are interconnected through the contacts, the particle movements must necessarily follow the movements at the contacts. Therefore, it may be assumed that the particle movements can be represented by the movements at the contacts.

In Fig. la is shown a section A-A at an inclination of α with the horizontal cut through the particles in a specimen under the effective major and minor principal stresses $\overline{\sigma}_1$ and $\overline{\sigma}_3$. Along the α -direction, the shear stresses τ whose magnitudes are equal but of opposite direction are also shown in Fig. la.

If we consider relative displacement between the two halves along A-A, we can distinguish two kinds of contact displacement. At a, the upper half may slip over the lower half by simple sliding of particle 2 over particle 1. At b, the upper half can slip over the lower half only if particle 2 is displaced. This would involve the displacement of particles 3 and 4.

The two types of movement and the corresponding resistance are as follows.

a. Sliding Type of Movement and Resistance

The simplest type of contact displacement is that produced by one particle sliding over another. Consider the contact between particles 1 and 2 at a in Fig. 1a; the forces acting on it are shown in Fig. 2.

With reference to Fig. 2,

 \mathbf{P}_{θ} is the normal force to the face of particle 1 on which slip occurs

 \mathbf{T}_{θ} is the tangential force along the face of particle l $\mathbf{C}_{\mathbf{a}}$ is the resisting force due to adhesion \mathbf{P}_{θ} tank is the resisting force due to mineral friction along θ tank is the coefficient of mineral friction

R is the total resisting force along θ at the contact. The sliding movement of the contact is analogous to the sliding phenomenon of solid bodies (Bowden and Tabor, 1954). When the contact is acted upon by the forces P_{θ} and T_{θ} , P_{θ} tan μ and C_a are brought into action. The force R is given by

Sliding occurs when

Equation (1) defines the resistance to sliding of the contact along θ . In 2.1.2, P_{θ} tan μ and C_{a} are considered separately. P_{θ} tan μ is considered under "Frictional Resistance against Sliding" in 2.1.2.a. C_{a} is considered under "Adhesional Resistance against Sliding" in 2.1.2.c.

b. Displacement Type of Movement and Resistance

Consider the contact between particles 2 and 5 at b in Fig. la. Sliding is unlikely since the contact is required to undergo an "uphill" displacement along the face of particle 2. However, even if this contact at b remains intact, it may still undergo displacement—if particle 2 is unable to resist the forces transmitted to it by the contact. Particle 2 is in contact with particles 1 and 4; particle 4 is in turn in contact with particle 3. Thus, to displace particle 2 (and thereby the contact at b), the resisting forces due to the contacts with particles 1 and 4 have to be overcome. Accordingly, we define the displacement resistance of particle 2 (and thereby the contact at b) as the force perpendicular to the particle that the contacts with particles 1 and 4 are able to withstand.

In Fig. 3, P is the displacement resistance of particle 2. It is the total resistance against the displacement of particle 2. It acts at an angle of $(\theta - 90^{\circ})$ with the horizontal. P consists of two parts: one part due to mineral friction, denoted by P_F and the other part due to adhesion, denoted by P_A . P_F is considered under "Frictional Resistance against Displacement" in 2.1.2.b. P_A is considered under "Adhesional Resistance against Displacement" in 2.1.2.d.

2.1.2 Contact Resistance

The contact resistance consists of: a) Frictional Resistance against Sliding; b) Frictional Resistance against

Displacement; c) Adhesional Resistance against Sliding; d) Adhesional Resistance against Displacement.

a. Frictional Resistance against Sliding

Fig. 4(a) shows the forces acting on the contact at a; it differs from Fig. 2 in that only the sliding resistance due to friction is shown. P_{θ} and T_{θ} are, as before, the normal and tangential forces on the face of particle 1. $t_{1\theta}$ is that part of the contact resistance against sliding which is due to friction. It acts at an angle α , the inclincation of the failure plane, with the horizontal.

The sliding resistance due to friction is \mathbf{P}_{θ} tan μ . Sliding occurs when

$$tan\mu = T_{\theta}/P_{\theta}$$
 (3)

Assuming that each contact is treated as a continuum, the stress condition at incipient sliding as defined by (3) can be represented by a Mohr circle. Such a representation is shown in Fig. 4(b).

Let

 $n_{\mathcal{G}}$ = the number of contacts with an angle θ along section A-A in Fig. la

$$\begin{array}{rcl}
\mathbf{n} & & \theta = \pi \\
\mathbf{n} & & \Sigma & \mathbf{n}_{\theta} \\
\theta = \mathbf{0}
\end{array}$$

 $A_{\mathcal{G}}$ = the total soil area parallel to \mathcal{G} along section A-A (Fig. 1b)

 ${\bf A}_{\alpha}$ = the gross soil area along the α -plane (Fig. 1b)

Hence

$$\mathbf{A}_{\alpha} = \sum_{\theta=0}^{\theta=\pi} \mathbf{A}_{\theta} \cos (\theta - \alpha)$$

In Fig. 4(b), σ_{1s} and σ_{3s} (in force per unit contact area) are the major and minor principal stresses at the contact. According to assumptions (2) to (5) on p. 5,

$$n_{\theta} A_{c} \cos \theta \sigma_{1s} = A_{\theta} \cos \theta \overline{\sigma}_{1}$$

$$n_{\theta} A_{c} \cos \theta \sigma_{3s} = A_{\theta} \sin \theta \overline{\sigma}_{3}$$

or

$$\sigma_{1s} = \overline{\sigma}_{1} \frac{A_{\theta}}{n_{\theta} A_{c}}$$

$$\sigma_{3s} = \overline{\sigma}_{3} \frac{A_{\theta}}{n_{\theta} A_{c}}$$
(4)

where

 $A_{_{\mbox{\scriptsize C}}}$ is the area of a contact along θ

 ${\rm A}_{\theta}$ cos θ and ${\rm A}_{\rm C}$ cos θ are respectively the projections of . ${\rm A}_{\theta} \mbox{ and } {\rm A}_{\rm C} \mbox{ on the horizontal plane}$

 \mathbf{A}_{θ} sin $\boldsymbol{\theta}$ and $\mathbf{A}_{\mathbf{C}}$ sin $\boldsymbol{\theta}$ are respectively the projections of

 \mathbf{A}_{θ} and $\mathbf{A}_{\mathbf{C}}$ on the vertical plane

 $\overline{\sigma}_1$ and $\overline{\sigma}_3$ are assumed uniformly distributed

 ${\rm A_c}$ cos ${\rm \theta}$ ${\rm \sigma_{ls}}$ and ${\rm A_c}$ sin ${\rm \theta}$ ${\rm \sigma_{3s}}$ are assumed to be the same at all contacts with angle ${\rm \theta}$

For forces normal and parallel to $\boldsymbol{\theta}\text{,}$

$$A_{\theta} \overline{\sigma}_{\theta} = n_{\theta} A_{c} \sigma_{s\theta}$$

$$A_{\theta} \tau_{\theta} = n_{\theta} A_{c} \tau_{s\theta}$$

where

is the effective normal stress to the plane θ (force per unit soil area)

is the shear stress along the plane θ (force per unit soil area)

is the shear stress at the contacts with angle θ (force per unit contact area)

 $^{A}_{C}$ $^{\sigma}_{S\theta}$ and $^{A}_{C}$ $^{\tau}_{S\theta}$ are assumed to be the same at all contacts with angle θ

For a contact with an angle θ :

$$T_{\theta} = A_{C} \tau_{S\theta} = \frac{A_{\theta} \tau_{\theta}}{n_{\theta}}$$

$$P_{\theta} = A_{C} \sigma_{S\theta} = \frac{A_{\theta} \sigma_{\theta}}{n_{\theta}}$$

Equation (3) becomes

$$tan\mu = \frac{T_{\theta}}{P_{\theta}} = \frac{\tau_{\theta}}{\overline{c}_{\theta}} = \frac{\tau_{s\theta}}{\sigma_{s\theta}}$$

Hence sliding at the n_θ contacts occurs at the contact stress $c_{s\theta}$ and $\tau_{s\theta}$ in Fig. 4(b).

Along the α -plane,

$$A_{\alpha}\tau_{\theta}^{\dagger}\theta = n_{\theta}A_{c}t_{1\theta}$$
and
$$\theta = \pi$$

$$A_{\alpha}\sum_{\theta=0}^{\infty}\tau_{\alpha\pi}^{\dagger} = A_{\alpha}\tau_{\alpha}^{\dagger} = \sum_{\theta=0}^{\infty}n_{\theta}A_{c}t_{1\theta}$$

$$\theta = \pi$$

where

is the frictional resistance against sliding contributed by a contact with an angle θ (force per unit contact area)

is the shearing resistance on the α -plane due to the frictional resistance against sliding at the n contacts; it is assumed uniformly distributed over A_{α} (force per unit soil area)

 $au_{lpha heta}^{ullet}$ is that portion of au_{lpha}^{ullet} due to n_{eta} contacts with angle heta (force per unit soil area)

In Fig. 4(b), the vector OB has components $\tau_{s\theta}$ and $\sigma_{s\theta}$. It is inclined at an angle μ to the σ_s -axis.

From the Mohr diagram in Fig. 4(b), the radius of the circle is $(\sigma_{s\theta} - \frac{\sigma_{1s} + \sigma_{3s}}{2}) \frac{1}{\cos 2\theta}$.

Hence

$$t_{1\theta} = (\sigma_{s\theta} - \frac{\sigma_{1s} + \sigma_{3s}}{2}) \frac{\sin(\pi - 2\sigma)}{\cos 2\theta})$$
 (6a)

Also

$$\sigma_{s\theta} \tan \mu = (\sigma_{s\theta} - \frac{\sigma_{1s} + \sigma_{3s}}{2}) \tan 2\theta$$

Whence

$$\sigma_{s\theta} = (\frac{\sigma_{1s} + \sigma_{3s}}{2}) \frac{\sin 2\theta}{\sin 2\theta - \tan\mu \cos 2\theta}$$

Substituting this value for $\sigma_{s\dot{\phi}}$ in (6a),

$$t_{1\theta} = 1/2(1+k) \quad \sigma_{3s} \sin 2\alpha \left(\frac{\tan\mu}{\sin 2\theta - \tan\mu \cos 2\theta} \right) \quad (6b)$$

where

$$k = \frac{\overline{\sigma}}{\frac{1}{\sigma}} = \frac{0.5}{\frac{1s}{\sigma}}$$

Substituting for c_{3s} from (4) in (6b),

$$t_{1\theta} = 1/2(1+k) \overline{c}_{3} \sin 2\pi \left(\frac{\tan \mu}{\sin 2\theta - \tan \mu \cos 2\theta} \right) .$$

$$\frac{A_{\theta}}{n_{\theta} A_{C}}$$
(6)

The applied shear stress along α is given by

$$\tau_{\alpha} = 1/2 (\overline{o}_{1} - \overline{o}_{3}) \sin 2\alpha$$

$$= 1/2 (k-1) \overline{o}_{3} \sin 2\alpha$$

When τ_α is imposed on a soil area A_α within which there are n contacts of area A_c each and $\theta = \pi$

$$n = \sum_{\tilde{U}=0}^{U=w} n_{\tilde{G}} ,$$

$$n A_{\mathbf{c}} t_{\dot{\alpha}} = A_{\dot{\alpha}} \tau_{\dot{\alpha}}$$

$$n_{\theta} A_{\mathbf{c}} t_{\dot{\alpha}} = A_{\dot{\alpha}} \tau_{\dot{\alpha}\theta}$$

$$(7)$$

where

 $\tau_{_{\mbox{$\mathcal{Q}$}}}$ is the applied shear stress assumed uniformly distributed over $\mathbf{A}_{_{\mbox{$\mathcal{Q}$}}}$

is that portion of τ_{α} to be carried by $n_{\tilde{\theta}}$ contacts with angle 3 (force per unit soil area)

is the shear stress along α at a contact with an angle θ . It is assumed to be the same at all contacts (force per unit contact area).

From Fig. 4(b),

$$t_{\alpha} = 1/2 (\sigma_{1s} - c_{3s}) \sin 2\alpha$$

$$= 1/2 (k-1) \sigma_{3s} \sin 2\alpha$$

Substituting for σ_{3s} from (4),

$$t_{\alpha} = 1/2(k-1) \overline{\sigma}_{3} \sin 2\alpha \frac{A_{\theta}}{n_{\theta} A_{C}}$$
 (7a)

It can be seen that the ratio $\frac{A_{\theta}}{n_{\theta}} \ A_{\mathbf{C}}$ appears

both in equation (7a) for t_{α} and equation (6) for $t_{1\theta}$.

Let this ratio be $a_{\tilde{\theta}} = \frac{n_{\theta} A_{C}}{A_{O}}$. We write (6) as

$$t_{1\theta} = 1/2 (1+k) \overline{\sigma}_3 \sin 2\alpha \left(\frac{\tan \mu}{\sin 2\theta - \tan \mu \cos 2\epsilon} \right) \frac{1}{a_{4\theta}}$$

It is to be noted that unless the particle distribution is uniform with respect to θ , A_{θ} and n_{θ} vary with θ . The absolute magnitude of $t_{1\theta}$ in equation (6) therefore also varies with A_{θ} and n_{θ} .

b. Frictional Resistance against Displacement

Fig. 5(a) shows the displacement resistance of particle 2 due to friction. It is assumed that the displacement of the particle is perpendicular to the direction θ . The resistance offered by friction is denoted by P_F ; it is assumed to act in a direction perpendicular to the particle. $A_C t_{2\theta}$ is the shear force along the α -plane that is required to overcome P_F . α is the angle of the inclination of the failure plane with the horizontal.

The particles in the network are interconnected. Hence, to displace particle 2, a number of particles must be displaced. Let m be the number of slipping contacts associated with the displacement of particle 2.

For the displacement of particle 2 to take place, the frictional resistance at each of the m contacts must be exceeded. $P_{\rm F}$ is defined as a resistance contributed by the m slipping contacts. It is assumed to be the same for all contacts. For simplicity, we assume,

$$P_{\mathbf{F}} = \frac{A_{\theta}}{n_{\theta}} P_{\mathbf{F}} = \frac{A_{\theta}}{n_{\theta}} \overline{\sigma}_{\mathbf{n}} m \tan \mu$$
 (8)

where

- $P_F = \overline{\sigma}_n$ m tan μ . It is assumed to act in the same direction as P_F and uniformly distributed (force per unit soil area).
- is the mean effective normal stress (force per unit soil area)
 - m is the number of slipping contacts associated with the displacement of particle 2.

In this study, only the experimental behaviors under triaxial compression with $\frac{1}{2} = \frac{1}{3}$ are investigated.

Hence

$$\overline{\sigma}_{n} = (\frac{2 + k}{3}) \overline{\sigma}_{3}$$

Substituting this value for $\overline{\sigma}_n$ in (8),

$$P_{F} = (\frac{2 + k}{3}) \overline{\sigma}_{3} \frac{A_{\theta}}{n_{\theta}}. m tan \mu$$
 (8')

Let there be n contacts of area $\mathbf{A}_{\mathbf{C}}$ each within $\mathbf{A}_{i,i}$

and

$$n = \begin{array}{c} \theta = \pi \\ \Xi \\ \theta = 0 \end{array}$$

Along the α -plane,

$$P_{F} \sin(\theta - a) = A_{C} t_{2\theta}$$

or

$$t_{2\theta} = \frac{P_F \sin (\theta - \phi)}{A_C}$$
 (9)

and

where

- is the frictional resistance against displacement contributed by a contact with an angle θ (force per unit contact area).
- is the shearing resistance on the α -plane due to the frictional resistance against displacement at the n contacts. It is assumed uniformly distributed over A_{α} . (force pre unit soil area) Substituting for P_F from (8') in (9a),

$$t_2 = m \tanh \left(\frac{2+k}{3}\right) \overline{c}_3 \sin \left(\theta - \alpha\right) \frac{A_{\theta}}{n_{\theta} A_{c}}$$
 (9)

Alternatively, we assume p_F to be the average normal stress, or $p_F=\overline{\sigma}_n$. Assuming that P_F is the same for all contacts, we write

$$P_{F} = \frac{A_{\theta}}{n_{\theta}} \quad P_{F} = \left(\frac{2 + k}{3}\right) \overline{c}_{3} \frac{A_{\theta}}{n_{\theta}}$$
 (10a)

Substituting for \overline{c}_3 from (4) in (10a),

$$P_{F} = (\frac{2 + k}{3}) \sigma_{3s} A_{C}$$

or

$$p_s = p_F \frac{1}{a_\theta} = (\frac{2 + k}{3}) \sigma_{3s}$$
 (10b)

The stress condition at the contact b as defined by (10b) is represented by the Mohr circle shown in Fig. 5(b).

In Fig. 5(b), σ_{1s} and σ_{3s} are the major and minor principal stresses at the contact. The major principal plane is in the horizontal direction. According to assumptions (2) to (5) on p. 5, p_s and τ_s (in force per unit contact area) are respectively the normal and shear stresses at the contact. α is the angle of the inclination of the failure plane. The objective is to calculate $t_{2\theta}$.

From the Mohr diagram in Fig. 5(b),

$$t_{2\theta} = (p_s - \sigma_{3s}) \tan(\pi - \theta) \frac{1}{\sin(2\theta - \pi)} \sin(\pi - 2\alpha)$$

or

$$t_{2\theta} = (p_s - \sigma_{3s}) \frac{\sin 2\alpha}{1 + \cos 2\theta}$$

Substituting for p_s from (10b) and ϵ_{3s} from (4),

$$t_{2\theta} = (\frac{k-1}{3}) \overline{\sigma}_3 \frac{\sin 2\alpha}{1 + \cos 2\theta} \cdot \frac{A_{\theta}}{n_{\theta} A_{C}}$$

Using $a_{\theta} = \frac{n_{\theta} A_{C}}{A_{0}}$,

$$t_{2\theta} = (\frac{k-1}{3})\overline{\sigma}_3 \frac{\sin 2\alpha}{1 + \cos 2\theta} \cdot \frac{1}{a_2}$$
 (10)

Using equation (9), $t_{2\theta}$ can be obtained with given values of m, k, α and μ for a range in θ . As an approximation, we assume m = 6. This means that the displacement of particle 2 involves the slip of 6 contacts. The variation of $t_{2\theta}$ with respect to θ for m = 6, k = 1.3, α = 50° and $tan\mu$ = 0.1 is shown in Fig. 6. Results of the computations are tabulated in Table A-1 in Appendix I.

Using equation (10), the variation of $t_{2\theta}$ with respect to θ for k=1.3, $\alpha=50^{\circ}$ and $tan\mu=0.1$ is shown in Fig. 6. Results of the computations are tabulated in Table A-1 in Appendix I.

Since the assumptions used in the derivation of equations (9) and (10) are rather severe, the variation of $t_{2\theta}$ with θ can only be considered as estimates of the range of $P_{\mathbf{F}}$. If we take an average of the 2 equations, we may represent this average by

$$t_{2\theta} = \rho \left(\frac{2+k}{3}\right) \overline{c}_3 \frac{1}{\sin(\theta-\alpha)} \cdot \frac{1}{a_{\theta}}$$
 (11)

where ρ is a coefficient.

Using equation (11), the variation of $t_{2\theta}$ with respect to θ is shown for $\rho=0.3$ and 0.5 in Fig. 6. For comparison, the values of k=1.3, $\alpha=50^{\circ}$ and $tan\mu=0.1$ are used. Results of the computations are tabulated in Table A-1 in Appendix I.

It can be seen from Fig. 6 that equation (11) is a reasonable average of equations (9) and (10). Throughout this study, (11) is used for $t_{2\theta}$ for convenience.

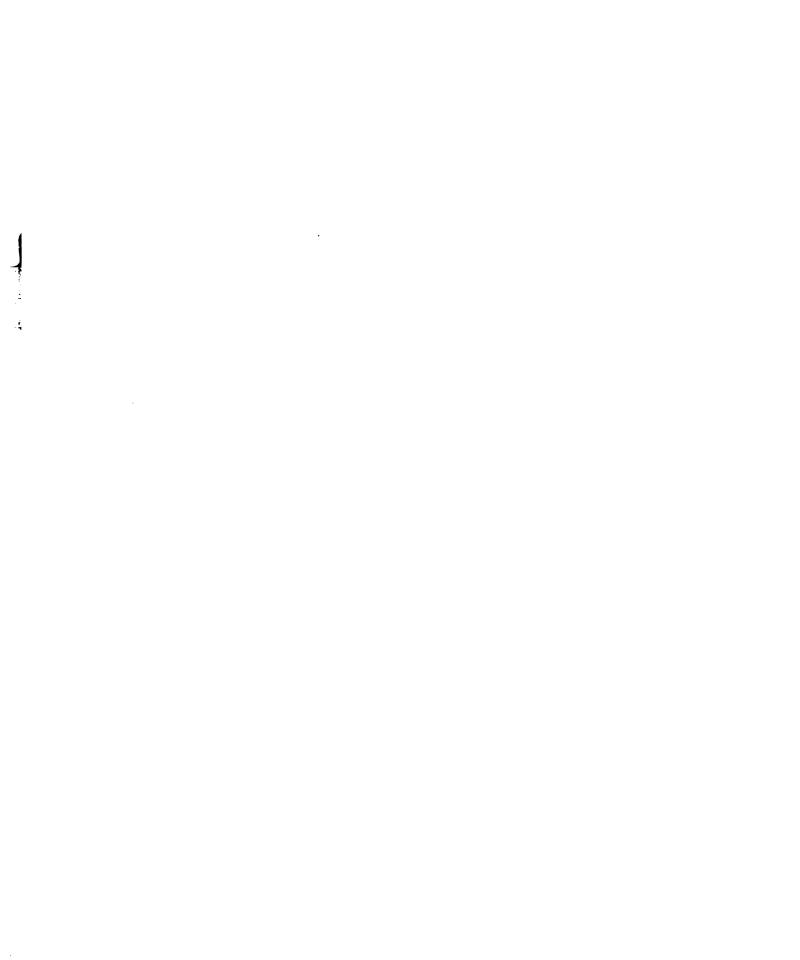
Using equations (6) and (11), the contact resistance can be obtained with given values of k, α , μ and ρ for a range in θ , which represents the direction of contact movement.

According to Horn and Deere (1962), for most clay minerals, tanu is of the order of 0.1 and appears to vary only within limited ranges. In this study, tanu is assumed to be a constant equal to 0.1.

For most clays, the angle of internal friction ϕ ranges between 10 - 30 degrees. Since $\alpha = 45^{\circ} + \phi/2$, α ranges between 50 - 60 degrees.

An examination of equations (6) and (11) indicates that for the above values of $tan\mu$, α and ρ

$$\begin{array}{l} t_{1\theta} > t_{2\theta} & \text{when } 0^{\circ} \langle \theta \leqslant 6^{\circ} , \\ \\ t_{1\theta} \doteq t_{2\theta} & \text{when } 6^{\circ} \langle \theta \leqslant 10^{\circ} , \\ \\ t_{1\theta} < t_{2\theta} & \text{when } 10^{\circ} \leqslant \theta \leqslant 85^{\circ} , \end{array}$$



and

$$t_{1\theta} \doteq t_{2\theta}$$
 when $85^{\circ} < \theta < 90^{\circ}$.

The contact resistance due to friction is, therefore, given by

$$t_{1\theta} = 1/2(1+k) \overline{s}_3 \sin 2 \left(\frac{\tan \mu}{\sin 2\theta - \tan \mu \cos 2\theta} \right) \frac{1}{a_{\theta}}$$
 (6)

for
$$10^{\circ} \langle \theta \langle 85^{\circ} \rangle$$

and

$$t_{2\theta} = \rho(\frac{2+k}{3}) \overline{\sigma}_{3} \frac{1}{\sin(\theta - \alpha)} \frac{1}{a_{\theta}}$$

$$for 85^{\circ} < \theta < 180^{\circ}$$

$$t_{2\theta} = \rho(\frac{2+k}{3}) \overline{\sigma}_{3} \frac{1}{\sin(\alpha - \theta)} \frac{1}{a_{\theta}}$$

$$for 0^{\circ} < \theta < 10^{\circ}$$

By using various values of k in (6) and (11), $\alpha=50^{\circ}$ and $\tan\mu=0.1$, the variation of contact resistance with respect to 5 is computed for $\rho=0.2$, 0.3, 0.4 and 0.5. Results of the computations are tabulated in Tables A-2, A-3, A-4, A-5 and A-6 in Appendix I. Figs. 7 and 8 show the calculated spectrum for $\rho=0.3$ and 0.5 respectively.

Fig. 9 shows the comparison of contact resistance due to friction for the same value of $\tan\mu=0.1$ and $\rho=0.3$ but different values of $\alpha=50^{\circ}$ and 60° . It can be seen from Fig. 9 that for k between 1.5 and 2.9, the difference is rather small. Hence, throughout this study, the value of $\alpha=50^{\circ}$ is used.

c. Adhesional Resistance against Sliding

Fig. 10(a) shows the forces acting on the contact at a; it differs from Fig. 2 in that only the sliding resistance due to adhesion is shown. P_{θ} and T_{θ} are, as before, the normal and tangential forces on the face of particle 1. $t_{3\theta}$ is that part of the contact resistance atainst sliding which is due to adhesion. It acts at an angle 0, the inclination of the failure plane, with the horizontal.

The sliding resistance due to adhesion is $\boldsymbol{C}_{\boldsymbol{a}}.$ Sliding occurs when

$$T_{\theta} = C_{a} \tag{12}$$

The stress condition at incipient sliding as defined by (12) is represented by the Mohr circle shown in Fig. 10(b).

In Fig. 10(b), σ_{1s} and σ_{3s} are the major and minor principal stresses at the contact. The major principal plane is in the horizontal direction. OB is the resistance due to adhesion at the contact. It is a constant equal to c_s . c_s is defined by

$$n_{\theta} A_{c} c_{s} = A_{\theta} c_{a} \text{ or } c_{s} = c_{a} \frac{1}{a_{\theta}}$$

and

$$\sum_{\theta=0}^{\theta=\pi} n_{\theta} A_{c} c_{s} = A_{\alpha} c_{a}$$

where

- is the adhesion at a contact with angle 5; it is assumed to be the same for all contacts (force per unit contact area)
- c_a is the adhesion along θ ; it is assumed to be a constant along any θ and independent of stress (force per unit soil area)

For a contact with an angle θ :

$$T_{G} = A_{C} \tau_{SO}$$
 and $C_{A} = A_{C} c_{S}$

Equation (12) becomes $\tau_{s\theta} = c_s$

Hence sliding at the $n_{\hat{\theta}}$ contacts occurs at the stress $\sigma_{\mathbf{s}\hat{\theta}}$ and $\tau_{\mathbf{s}\hat{\theta}}$ in Fig. 10(b).

Along the α -plane,

$$A_{\alpha} \tau_{\alpha\theta}^{\dagger\dagger} = n_{\theta} A_{c} t_{3\theta}$$
and
$$\theta = \pi$$

$$A_{\alpha} \Sigma \tau_{\alpha\theta}^{\dagger\dagger} = A_{\alpha} \tau_{\alpha}^{\dagger\dagger} = \Sigma n_{\theta} A_{c} t_{3\theta}$$

$$\theta = \pi$$

$$\Sigma \tau_{\alpha\theta}^{\dagger\dagger} = A_{\alpha} \tau_{\alpha}^{\dagger\dagger} = \Sigma n_{\theta} A_{c} t_{3\theta}$$

$$\theta = \pi$$

$$\Sigma = 0$$

where

- $t_{3\theta}$ is the adhesional resistance against sliding contributed by a contact with an angle θ (force per unit contact area)
- $\tau_{\alpha}^{\text{ii}}$ is the shearing resistance on the α -plane due to the adhesional resistance against sliding at the n contacts. It is assumed uniformly distributed over \mathbf{A}_{α} . (force per unit soil area)
- $\tau_{\alpha\theta}^{\bullet,\bullet}$ is that portion of $\tau_{\alpha}^{\bullet,\bullet}$ due to n_{θ} contacts with angle θ (force per unit soil area)

From the Mohr diagram in Fig. 10(b),

 $t_{3\hat{\sigma}} = (\sigma_{1s} - \sigma_{3s})1/2 \sin(\pi - 2\alpha) = 1/2 (\sigma_{1s} - \sigma_{3s})\sin 2\alpha$ Also

$$c_s = 1/2 (\sigma_{1s} - \sigma_{3s}) \sin 2\theta$$

Whence

$$t_{3\theta} = \frac{\sin 2\alpha}{\sin 2\theta} c_{s} \tag{14a}$$

Substituting for $c_s = c_a \frac{1}{a_{ij}}$ in (14a),

$$t_{3\theta} = \frac{\sin 2\alpha}{\sin 2\theta} c_a \cdot \frac{1}{a_{\theta}}$$
 (14)

d. Adhesional Resistance against Displacement

Fig. 11 shows the displacement resistance of particle 2 due to adhesion. This resistance is denoted by P_A and is assumed to act in a direction perpendicular to the particle. $A_c t_{4\theta}$ is the shear force along the α -plane that is required to overcome P_A . α is the angle of the inclination of the failure plane with the horizontal.

In accordance with the assumptions in 2.1.2.b, we write

$$P_A = m c_s A_c$$

$$P_A \sin (\theta - \alpha) = A_C t_{40}$$

or

$$t_{4\theta} = m c_s \sin (\theta - a)$$
 (15)

and

where

t_{4\theta} is the adhesional resistance against displacement contributed by a contact with an angle θ (force per unit contact area)

is the shearing resistance on the α -plane due to the adhesional resistance against displacement at the n contacts. It is assumed uniformly distributed over A_{α} .

As before, we represent (15) by

$$t_{40} = \rho_{\rm m} \frac{c_{\rm s}}{\sin (\theta - \alpha)}$$

where ρ_{m} is a coefficient ranging approximately between 2-5.

Substituting for $c_s = c_a \frac{1}{a_{ij}}$,

$$t_{4\theta} = \rho_{m} \frac{c_{a}}{\sin (\theta - \alpha)} \cdot \frac{1}{a_{\theta}}$$
 (16)

Since contact displacement would occur by whichever means possible, equations (14) and (16) are subject to the same restrictions on θ as equations (6) and (11). Accordingly, the contact resistance due to adhesion is given by

$$t_{3\theta} = c_a \frac{\sin 2\alpha}{\sin 2\theta} \cdot \frac{1}{a_{\theta}}$$
for $10^{\circ} < \theta < 85^{\circ}$

and

$$t_{4\theta} = \rho_{m} \frac{c_{a}}{\sin (\theta - \alpha)} \cdot \frac{1}{a_{\theta}}$$

$$for 85^{\circ} < \theta < 180^{\circ}$$

$$t_{4\theta} = \rho_{m} \frac{c_{a}}{\sin (\alpha - \theta)} \cdot \frac{1}{a_{\theta}}$$

$$for 0^{\circ} < \theta < 10^{\circ}$$
(16)

Using $\alpha=50^{\circ}$ in (14) and (16), the variation of the contact resistance with respect to θ is computed for $\rho_{m}=2,3,4$ and 5. Results of the computations are tabulated in Tables A-7(a) and A-7(b) in Appendix I.

Fig. 12 shows the calculated spectrum for $\rho_{\rm m}$ = 2,3,4 and 5.

To estimate the values of $\mathbf{t_{3}}$ and $\mathbf{t_{4}}$, we find from Fig. 12:

t_{3.2}

$$\begin{cases}
\text{minimum} = 1 & c_a \frac{1}{a_0} \\
\text{maximum} = 5.7 & c_a \frac{1}{a_0} \\
\text{average} = 3.35 & c_a \frac{1}{a_0}
\end{cases}$$

(
$$\rho_{m=3}^{t_4}$$
)
$$\begin{cases} \text{minimum} = 3 \text{ c}_{a} \frac{1}{a_{m}} \\ \text{maximum} = 5.5 \text{ c}_{a} \frac{1}{a_{m}} \\ \text{average} = 4.25 \text{ c}_{a} \frac{1}{a_{m}} \end{cases}$$

Consider t_3 and t_4 as contributing directly to the non-viscous cohesion c'. As an approximation, take

average
$$\begin{pmatrix} t_3 \\ t_4 \end{pmatrix} = c' \frac{1}{a}$$

For $\frac{1}{3}$ = 2.0 kg/cm², the average value of non-viscous c' is at most 0.15 kg/cm² for the slow CFS tests (Table 2).

Thus

average
$$\begin{vmatrix} t_{3} \\ t_{4} \end{vmatrix} = 0.15 \frac{1}{a_{0}} \text{ kg/cm}^{2}$$

Average $t_{3} = 3.35 c_{a} \frac{1}{a} = 0.15 \frac{1}{a}$, $c_{a} = 0.0448 \text{ kg/cm}^{2}$

... max.
$$t_{3} = 5.7 c_a \frac{1}{n_0} = 5.7 \times 0.0448 \frac{1}{a_0} = 0.255 \frac{1}{a_0} \text{ kg/cm}^2$$

Average
$$t_{4} = 4.25 \text{ c}_{a} \frac{1}{a_{0}} = 0.15 \frac{1}{a_{0}}$$
, $c_{a} = 0.0353 \text{ kg/cm}^{2}$
... max. $t_{4} = 5.5 \text{ c}_{aa} = 5.5 \text{ x} \cdot 0.0353 \frac{1}{a_{0}} = 0.194 \frac{1}{a_{0}} \text{ kg/cm}^{2}$

For comparison, we take k = 1.3 and $\frac{1}{3} = 2.0 \text{ kg/cm}^2$. From Fig. 7 ($f^2 = 0.3$):

$$t_{1} = \begin{cases} \min m = 0.1 \frac{1}{a} \times \frac{1}{3} = 0.2 \frac{1}{a} & \text{kg/cm}^2 \\ \max m = 0.4 \frac{1}{a} \times \frac{1}{3} = 0.8 \frac{1}{a} & \text{kg/cm}^2 \end{cases}$$

$$t_{2} = \begin{cases} minimum = 0.35 \frac{1}{a} \times \frac{1}{3} = 0.7 \frac{1}{a} \\ maximum = 0.5 \frac{1}{a} \times \frac{1}{3} = 1.0 \frac{1}{a} \end{cases}$$
 kg/cm²

Comparing the average value of t_3 and t_4 (0.255 and 0.194 x $\frac{1}{a}$ kg/cm²) with the range of values for t_1 and t_2 , we find that t_3 and t_4 will attain their maximum value long before t_1 and t_2 become fully developed.

2.1.3 Redistribution of Contacts

The contact resistance expressed by equations (6), (11), (14), and (16) is inadequate to describe the shear behavior of a matrix of clay particles in that the re-distribution of the displaced particles resulting from failure of the contacts has not been considered. Taking up new positions in the network, the displaced particles may or may not give rise to the reforming of contacts. If no contacts are reformed, a reduction in overall strength results. Even if contacts are reformed, the

overall strength can increase, decrease or remain the same. For example, if a contact is reformed as a result of the displacement of a contact that has failed, the strength of this reformed contact depends on the new position taken up by the displaced particle. The strength of this reformed contact, therefore, can be equal to, greater or smaller than that of the displaced contact. It follows that various mew states of particle arrangements are possible as a result of the various modes of redistribution of the displaced particles. In the following, four possible modes of redistribution are presented.

As shown in 2.1.2 on p. 27, when the matrix is under stress, the adhesion resistance is overcome before the friction resistance becomes fully developed. Since the range in adhesion resistance is much smaller than that in friction resistance, the various modes of redistribution would have only a small effect on the adhesion resistance. For simplicity, only the friction resistance is considered in the analysis of contact redistribution.

The four modes of redistribution are defined as follows: a) Partial Remolding; b) Complete Remolding; c) Partial Re-orientation; d) Complete Re-orientation.

a. Partial Remolding

Partial Remolding is a mode of redistribution in which when the resistance of a contact is exceeded, the movement of this displaced contact is such that the contact maintains the same resistance. This contact, therefore, can carry no additional shear stress.

b. Complete Remolding

Complete Remolding is a mode of redistribution in which when the resistance of a contact is exceeded, the movement of this displaced contact is such that no contact is reformed. Once its resistance is exceeded, this displaced contact, therefore, transfers all of its share of shear stress to the other contacts.

c. Partial Re-orientation

Partial Re-orientation is a mode of redistribution in which when the resistance of a contact is exceeded, the movement of this displaced contact is such that a contact having a greater strength than that of the displaced contact is reformed. Since the reformed contacts can carry additional shear stress, the overall strength of the matrix increases. All particle contacts whose resistance are governed by equation (6) belong to the sliding group; all particle contacts whose resistance are governed by equation (11) belong to the displacement group.

If a contact is reformed as a result of the movement of a displaced contact in the sliding group, the position taken up by the reformed contact is assumed to be restricted to the range of the sliding group. Likewise, if a contact is reformed as a result of the movement of a displaced contact in the displacement group, the position taken up by the reformed contact is assumed to be restricted to the range of the displacement group. Uniform distribution of the displaced contacts is assumed within the two groups.

Since the sliding group has a lower maximum resistance than the displacement group, contacts in the sliding group will be exhausted first. Once the sliding group is exhausted, <u>all</u> displaced contacts redistribute themselves into the displacement group.

d. Complete Re-orientation

Complete Re-orientation is a mode of redistribution in which when the resistance of a contact is exceeded, the movement of this displaced contact is such that a contact having a greater strength than that of the displaced contact is reformed. It is assumed that this reformed contact can take all possible position of orientations. While both groups are in existence, the position taken up by the reformed contact is not restricted to be within the range of either group. Uniform distribution of the displaced contacts is assumed. Because of the smaller maximum strength of the sliding group, this group is exhausted first. After this, all contacts are in the displacement group. Since the reformed contacts can carry additional shear stress, the overall strength of the matrix increases.

The behavior of the real material most probably is a combination of the four modes presented.

2.1.4 Behavior of the Particle Model

In this section, the stress-displacement characteristics of the contacts according to the various modes of redistribution are calculated. The spectrum of contact

strengths due to friction are given graphically in Figs. 7 and 8. The modes of redistribution are defined in 2.1.3.

For convenience, the following definitions and symbols have been introduced:

Stress	All stresses applied to the matrix are		
	expressed in terms of k, the effective		
	principal stress ratio.		
Matrix Failure	That state of stress at which every		
	point on the failure plane satisfies		
	the condition of maximum obliquity.		
<u>n</u>	Number of contacts within a soil area		
	Agalong anplane.		
<u>M</u>	Range in θ (degrees) in which contact		
	failure occurs under k.		
n _m	Number of contact failure in M.		
<u>-</u> 	Applied shear stress along α -plane		
	(failure plane) in force per gross		
stress-factor	area of soil.		
	A magnification factor on τ_{χ} resulting		
	from redistribution of displaced		

Assuming a random arrangement of particles and using $\tan \alpha = 0.1$, $\alpha = 50^{\circ}$ and $\beta = 0.3$, the variation of n_{m} with respect to k is computed for the four modes of redistribution. The detailed procedure of computations for each mode of redistribution is presented in Appendix II.

contacts.

Figs. A-1, A-2, A-3 and A-4 in Appendix II serve as illustrations to the procedure of computations. Results of the computations for the various modes are summarized in Tables A-8(a), A-8(b), A-8(c) and A-8(d) in Appendix II.

In the same manner, computations for the $k-n_m$ characteristics are also carried out for $\rho=0.5$. They are summarized in Tables A-9(a), A-9(b), A-9(c) and A-9(d) in Appendix II.

Assuming that each contact failure contributes a displacement of e along the 3-plane, the total displacement due to n_m contacts is given by $N = n_m e$. The results obtained in Tables A-8(a), A-8(b), A-8(c), A-8(d), A-9(a), A-9(b), A-9(c) and A-9(d) are used to plot the stress-displacement (k-N) curves for the various modes of redistribution. Figs. 13 and 15 show the stress-displacement curves for $\rho = 0.3$ and 0.5 respectively.

In Figs. 13 and 15, curve (a) shows the k-N characteristics for partial remolding. The characteristic of this curve is such that when a stress is applied to the matrix, contacts with the smallest resistance along a failure plane slip first. Upon failure of these contacts, displacements take place. As displacements are taking place, these contacts maintain the same strengths. Therefore these contacts can carry no additional stress. When the next load increment is applied, the remaining contacts are required to carry a bigger share of the increment than that in the first increment. Thus as displacements

proceed under additional applied stresses, more contacts fail and more contacts with the same strengths as those displaced are reformed. The remaining intact contacts are required to carry a progressively bigger share of the stress. A point is reached at which there are no longer any intact contacts left along the failure planes of the matrix. At this point, the matrix has reached failure. Thus when failure is reached, the matrix deforms indefinitely under the failure stress. The behavior described by curve (a) is similar to that of a remolded clay whose overall strength may be explained in terms of the "exhaustions" of intact contacts as stresses are applied to the matrix.

In Figs. 13, and 15, curve (b) shows the k-N characteristic for complete remolding. The characteristic of this curve is such that when a stress is applied to the matrix, contacts with the smallest resistance along a failure plane slip first. Upon failure of these contacts, displacements take place. As displacements are taking place, contacts are not reformed. As displacements proceed, the contacts that have failed transfer all of their share of stress to the other intact contacts thus causing more contacts to fail. When all the stresses carried by the failed contacts are transferred to the stronger contacts, displacements cease. Thus the intact contacts are required to carry a bigger share of the load as displacements proceed under the same load. Under additional stresses,

more contacts fail and the remaining contacts are required to carry a progressively bigger share. A point is reached at which there are no longer any intact contacts left along the failure planes of the matrix. At this point, the matrix has reached failure. Thus when the failure stress is reached, the failure of all the contacts along the failure planes results in a reduction in strength of the matrix to zero. The behavior of curve (b) is therefore similar to that of a quick clay, which is known to have a sudden reduction in strength at failure.

In Figs. 13 and 15, curves (c) and (d) show the k-N characteristics for partial re-orientation and complete re-orientation respectively. The characteristics of these curves are such that when a stress is applied to the matrix, contacts with the smallest resistance along a failure plane slip first. Upon failure of these contacts, displacements take place. As displacements are taking place, contacts with greater strengths are reformed. After these contacts are reformed, displacements cease. displacements proceed under additional applied stresses, more contacts fail and more contacts with greater strengths are reformed. A point is reached at which the load on the contacts reaches the maximum value attainable. this point, the matrix has reached failure. Thus when failure is reached, the matrix deforms indefinitely. The behavior described by curves (c) and (d) is thus similar to that of a consolidated clay whose overall

strength requires a considerable amount of deformation to be fully mobilized.

Each of the four stress-displacement curves is admittedly a gross simplification of the behavior of a real material. To approximate the behavior of a real material in which all four processes take place, an average is taken of the four cases. Figs. 14 and 16 show the average k-N curves for $_{i}$ = 0.3 and 0.5 respectively. In Figs. 14 and 16, curve A is obtained by averaging all the four curves (a), (b), (c) and (d) at N = 0 to n e and by averaging only curves (a), (c) and (d) at N > n e. Curve B is obtained by averaging all the four curves at all values of N. Since complete remolding is probably only valid for a quick clay, curve A may be a more realistic representation. Results of the computations for Curves A and B for $_{i}$ = 0.3 and 0.5 are tabulated in Table A-10 in Appendix II.

2.1.5 The Mechanism of Friction in Clays

The behavior of the particle model represented by the stress-displacement characteristics of curve A in Figs. 14 and 16 may be summarized in the following hypothesis for the mechanism of friction.

The resistance at the contact consists of an adhesion and a mineral friction; the adhesion and mineral friction constitute the resistance against sliding and that against displacement. The contact resistance is

dependent on the geometric orientations and contact geometry of the particles. Along the direction of shear, the variation of the contact resistance with respect to the particle orientation is expressed by a spectrum. The contact resistance due to adhesion is estimated to be small in comparison with that due to friction and is considered to have little effect on the stress-strain properties.

When a shear stress is imposed on a specimen, contacts with the smallest resistance along the direction of shear slip first. Since the resistance due to adhesion is small in comparison with that due to friction, it is fully mobilized at all the contacts instantaneously. This quantity is independent of the normal stress and is considered to be a non-viscous cohesion. The part due to mineral friction is directly proportional to the normal stress and contributes to the angle of internal friction o'. Following these contact failures, particles are displaced from one position to another and new contacts may be reformed. Even if the position taken up by a displaced particle is favorable to the reforming of contacts, this process requires a certain time interval. However, even if this requirement of time is met, for a reformed contact to develop its maximum resistance to shear, a certain minimal of time may be required. Furthermore, due to the different orientations and contact geometry of the displaced particles in their new positions, the strengths of the reformed contacts vary. Thus, the resistance of a

reformed contact depends on the orientation and contact geometry of the displaced particle as well as the time available or elapsed time between displacements.

Consider the behavior of the angle of internal friction of when a specimen is subjected to a constant strain rate test. When shear stresses are applied to the specimen, contacts fail and displacements take place. As displacements are taking place, contacts with greater strengths are being developed. The overall strength of the matrix therefore increases gradually with the deformation. Accordingly, of increases gradually with the deformation. As deformation proceeds further, all the contacts are loaded to the maximum contact strength attainable. Then the reformed contacts have the same strengths as those displaced. Thus, when failure is reached, of reaches a maximum and remains at this value with further deformation.

For simplicity, we may assume that the rate of contact displacement is proportional to the strain rate. If the strain rate that corresponds to a rate of contact displacement N_m is such that sufficient time is available for the reformed contacts to develop their maximum resistance, the maximum value of Φ' at any given strain is attained. This is denoted by Φ'_m . If a faster strain rate that corresponds to a rate of contact displacement N_1 is used, then there is insufficient time for the reformed contacts to develop their maximum resistance. The value of Φ' at

any given strain would then be smaller than $^{\circ}_{m}$. Accordingly, the value of $^{\diamond}$ ' at failure $(^{\diamond}_{u})$ is also smaller. If the strain rate is less than that required to create the rate of contact displacement N_{m} , the same $^{\diamond}$ '-strain characteristic as for that of N_{m} is obtained.

Thus, according to the hypothesis, the behavior of the angle of internal friction under constant strain rate is as follows:

At any strain
$$($$
, if $N > N_m$, $\Phi' < \Phi_m'$

$$N \leqslant N_m, \Phi' = \Phi'_m$$
At failure $(\overline{1}/\overline{3} = \text{const.})$,
if $N > N_m, \Phi'_u < \Phi'_u$

$$N \leqslant N_m, \Phi'_u = \Phi'_u$$

where N = N/t, is the rate of contact displacement

N is the contact displacement at any strain \in

t is the time required to reach N

 $N_{m} = N/t_{m}$, is the rate of contact displacement required to attain ϕ_{m}^{*}

t m is the minimal of time required to attain $\mathfrak{P}_{m}^{\bullet}$

is the angle of internal friction at any
 strain =

 Φ_{m}^{\bullet} $\,$ is the maximum value of Φ^{\bullet} at any strain \in

 $N_u = N \times t_u$, is the contact displacement at failure

is the time required to reach N_{11}

 $^{\circ}$ is the value of $^{\circ}$ at N (at failure)

t um is the minimal of time required to attain $\tilde{\nu}_{\text{um}}$ = N/N_{m}

and $\phi^{\text{!}}_{\text{um}}$ is the maximum value of $\phi^{\text{!}}$ attainable under N_{m} (at failure)

The variation of Φ' with respect to strain ϵ at various rates of contact displacement \mathring{N} is illustrated in Fig. 17(a). It can be seen from Fig. 17(a) that

at any strain ϵ ,

since
$$\dot{N}_2 > \dot{N}_1 > \dot{N}_m$$
 , $\Phi_2' < \Phi_1' < \Phi_m'$

and at failure,

since
$$\dot{N}_2 > \dot{N}_1 > \dot{N}_m$$
 , $\Phi_{u2}' < \Phi_{u1}' < \Phi_{um}'$

Fig. 17(b) shows qualitatively the variation of Φ' with the rate of contact displacement N at various strains ϵ .

The variation of Φ' under creep with respect to the displacement rate under creep is assumed to be similar to that of Φ' with respect to N. When a creep load K is applied, the weaker contacts along a failure plane slip first. Following these contact failures, displacements take place and contacts with greater strengths are reformed. After these contacts are reformed, displacements cease. At any creep strain ϵ_{C} , in order to attain a value of Φ' denoted by Φ'_{C} , a certain number of contacts must be displaced. This gives rise to a displacement N_{C} which requires a time t_{C} to reach. Thus, Φ'_{C} increases with t_{C} and approaches a maximum value Φ'_{K} at a time t_{K} beyond which it remains virtually constant. Alternatively, Φ'_{C} increases with

decreasing displacement rate under creep N_c and approaches a maximum value ${}^{\diamond}_k$ at a rate N_k below which it remains virtually constant. According to the hypothesis, however, only the value of ${}^{\diamond}_k$ at t_k is known. Values of ${}^{\diamond}_c$ at any creep time t_c may be obtained as follows.

The $^{\circ}_{c}$ - $^{\circ}_{c}$ curve as shown in Fig. 17(c) is similar to the $^{\dagger}_{u}$ - $^{\circ}_{u}$ - $^{\circ}_{c}$ characteristics under constant strain rate in Fig. 17(b). We may obtain values of $^{\circ}_{c}$ at any creep time $^{\circ}_{c}$ by comparing $^{\circ}_{c}$ = $^{\circ}_{c}$ / $^{\circ}_{c}$ and $^{\circ}_{u}$ = $^{\circ}_{u}$ / $^{\circ}_{u}$ in the following manner:

If
$$N_{C} \langle N_{m}, T_{C} \rangle = \Phi_{k}^{\dagger}$$

$$N_{C} \rangle N_{m}, \Phi_{C}^{\dagger} = \frac{\Phi_{k}^{\dagger}}{\Phi_{um}^{\dagger}} \times \Phi_{u}^{\dagger}$$
(18)

where the value of $\mathfrak{D}_{\mathbf{u}}^{\mathsf{I}}$ corresponds to $N = N_{\mathbf{C}}^{\mathsf{I}}$.

2.2. The Mechanism of Cohesion in Clays

To fully describe the shearing resistance, the mechanism of friction advanced in 2.1.5 must necessarily be supplemented by the hypothesis proposed for the mechanism of cohesion.

In the development of the mechanism of friction in 2.1.5, the resistance encountered by a particle when it is displaced from one position to another has not been taken into consideration. The clay particle is known to be surrounded by a layer of adsorbed water of high viscosity. When a particle is displaced from one position to another, it would encounter viscous resistance. It is the hypothesis

that this viscous resistance constitutes a large part of the cohesion in clays. Accordingly, the flow phenomenon is treated by a viscous flow model, after the initial yield value given by the contact resistance due to friction and adhesion has been exceeded. If the applied stress on the contact is less than this yield value, no flow occurs.

As shown by Christensen (1964), the deformational properties of the clay particle structure can be represented schematically by a rheologic model shown in Fig. 18. In Fig. 18,

k₁ represents the initial elastic response of the
flowing contacts,

k₂ corresponds to the resistance due to primard and non-viscous cohesion as calculated by the particle model,

and f correspond to viscous cohesion.

According to the rate theory (Glasstone, Laidler and Eyring, 1941), the flow equation of the contacts is given by

$$f_1 = 3 \sinh x f_1 \tag{19}$$

where

 $\dot{r}_1 = \text{rate of shear strain in the viscous element}$ $= \frac{2}{1} \frac{kT}{n} \exp \left(-\frac{\Delta F}{RT}\right)$

= vertical distance between successive interparticle equilibrium positions

= distance between flow units in the direction of flow

k = Boltzmann's constant

h = Plank's constant

T = the absolute temperature

 $\triangle F$ = activation energy

R = universal gas constant

 $= \frac{1}{23 kT}$

a = number of contacts per unit area over the plane of shear

f' = shear stress in the viscous element

 f_y = yield value or yield strength of the contacts, given by the contact resistance due to adhesion and friction

 $f_1 = f_1' - f_y$ is the shear stress that causes flow in the viscous element.

Thus the right hand side of the rheologic model represents the effects of the contact stresses that cause flow. When a shear stress $\bar{}$ is applied, if the stresses on the contacts due to this shear stress exceed the yield value f_y of the contacts, viscous flow occurs. As deformation proceeds, the stresses tending to cause flow are transferred to the stronger contacts represented by k_2 until all flow stops, giving rise to a new yield value of the contacts. This new yield value is then given by the increased resistance of the contacts due to the reforming of stronger contacts.

When a shear stress t is applied, we have

$$t = k_2 \gamma + f_1 \tag{20}$$

$$f_1 = k_1 \left(- \frac{1}{1} \right) \tag{21}$$

where ; is the total shear strain.

Combining (19), (20) and (21), the governing differential equation for this model can be shown to be

$$\frac{d}{dt} \left[(k_1 + k_2) / - \tau \right] = k_1 / \sinh / f_1$$
 (22)

For constant strain rate, \$\foata = constant

we have

$$\frac{\mathrm{df}_1}{\mathrm{dt}} = k_1 \left(\frac{\mathrm{d}_2}{\mathrm{dt}} - \beta \sinh \beta f_1 \right) \tag{23}$$

Letting

$$\vec{z}^{\,\dagger} = \frac{\dot{\vec{z}}}{\dot{\vec{z}}}$$

we have

$$\frac{d(\mathbf{f}_1)}{d(\mathbf{f}_1)} = \mathbf{f}_1 - \sinh(\mathbf{f}_1)$$

substituting for $\sinh (ef_1) = \frac{e^{-2}f_1}{2}$

we have

$$\frac{d(e^{f_1})}{1/2 - 1/2e^{f_1} + e^{f_1}} = d(e^{k_1}t)$$

Integrating,

$$\frac{2}{\sqrt{1 + \frac{1}{2}}} \times \tanh^{-1} \left(\frac{e^{\frac{t}{1}} - \frac{1}{2}}{\sqrt{1 + \frac{1}{2}}} \right) = \frac{1}{2} k_1 t - C$$

$$f_1 = \frac{1}{2} \log_e \left[\frac{1}{2} + \sqrt{1 + \frac{1}{2}} + \frac{1}{2} \tanh^{\frac{1}{2}} \left(\frac{1}{2} k_1 t - C \right) \right]$$
 (24)

Applying the Boundary Condition : t = 0, $f_1 = 0$

$$f_{1} = \frac{1}{7} \log_{e} \left[\frac{1}{2} + \sqrt{1 + \frac{1}{2}}^{2} \tanh \left[\frac{\sqrt{1 + \frac{1}{2}}^{2}}{2} \cosh_{1} t + \tanh^{-1} \frac{1 - \frac{1}{2}}{\sqrt{1 + \frac{1}{2}}^{2}} \right] \right]$$

or

$$f_1 = \frac{1}{\pi} \log_e \left[z' + \sqrt{1 + \beta'^2} \tanh \left[\frac{\sqrt{1 + \beta'^2}}{2} \frac{x k_1}{\beta'} y + \tanh^{-1} \frac{1 - \beta'}{\sqrt{1 + \beta'^2}} 2 \right] \right] (25)$$

For Creep loading, i = 0 (see Christensen, 1964)

From (22)
$$\frac{k_1 + k_2}{k_1 k_2} \frac{df_1}{dt} = - f \sinh f_1$$
 (26)

Solving (26) and applying the Boundary condition:

$$t = 0$$
, $f_1 = \frac{k_1}{k_1 + k_2}$

we have

$$f_1 = -\frac{1}{4} \log_e \tanh \left[\frac{1}{2} - \frac{k_1 k_2}{k_1 + k_2} + \tanh^{-1} \exp(-\frac{k_1 k_2}{k_1 + k_2}) \right]$$
 (27)

Equation (25) expresses the variation of the resistance to viscous flow of the clay particle structure or viscous cohesion with respect to shear deformation under the condition of constant strain rate. Equation (27) expresses the variation of the resistance to viscous flow of the clay particle structure or viscous cohesion with respect to time under the condition of constant shear stress or creep. After the evaluation of the parameters and k, and k, equations (25) and (27) can be used to analyze the behavior of the viscous part of cohesion in clays.

CHAPTER 3

EXPERIMENTAL PROGRAM

3.1 Objective

The purpose of the experimental program is to examine if the hypotheses advanced for the mechanism of friction and cohesion in Chapter 2 can adequately simulate the actual behavior of friction and cohesion in clays.

In order to achieve this objective, the experimental behaviors of friction and cohesion are obtained by carrying out a number of CFS and Creep-CFS triaxial tests on five different clays. The CFS-triaxial tests measure the variation of friction and cohesion with respect to axial strain under different strain rates. The Creep-CFS-triaxial tests measure the variation of friction and cohesion with respect to time under creep. Since the predicted values of cohesion require the parameters in the rheologic model of clay particle structure to be known, a number of creep tests are also carried out in order that these parameters can be evaluated.

3.2 Materials Used

The clays used in this study include laboratory consolidated, compacted and remolded clays, undisturbed clays and Grundite, a commercial illite. The majority of the tests are performed on laboratory consolidated,

compacted and remolded specimens of a red clay from Sault Ste. Marie, Michigan. Undisturbed samples in the form of thin-walled shelby tubes were obtained from Willow Run and Marine City, Michigan. The index properties of the clays tested are given in Table 1.

3.3 Sample Preparation

3.3.1 Consolidated Sault Clay

Bulk samples of air-dried Sault Ste. Marie clay were soaked in distilled water for a few days before they were thoroughly remolded at water contents slightly above the liquid limit. The thick slurry was then placed in a 6-inch consolidometer and allowed to consolidate. After a standing period of several weeks, consolidation loads were added to the clay in increments until a pressure of 0.36 kg/cm² was attained. For the normally-consolidated specimens, the consolidated cake was then extruded, sliced into pieces, sealed in wax and stored. For the overconsolidated specimens, the cake was loaded to a pressure of 9.0 kg/cm². The load was then removed. The unloaded cake was allowed to rebound for several weeks before it was extruded.

3.3.2 Compacted Sault Clay

Bulk samples of air-dried clay were thoroughly remolded and mixed with distilled water to a consistency corresponding to a water content of about 40%. The clay

Table 1 Index Properties of Clays Studied

CLAY	L.L.	P.L.	P.I.	Clay Fraction
Sault Ste. Marie	60.5	23.6	36.2	60
Willow Run	29.5	19.0	10.5	40
Marine City 1	46.8	24.1	23.7	
Marine City 2	41.4	21.7	19.7	68
Grund it e	78.0	25.0	53.0	65

was then left in the moist room for at least a week in order to ensure an even water content distribution. A cake was then made out of the remolded clay by placing it in a static compaction mold. Hand kneading was employed during the placement of the clay in order to remove as much air as possible. In general, a pressure of 1 kg/cm² was applied to the cake by the loading ram of a hydraulic testing machine. The pressure was maintained for one hour and released. The cake was then sliced, sealed in wax and stored.

3.3.3 Remolded Sault Clay

Bulk samples of air-dried clay were thoroughly remolded and mixed with distilled water to a consistency corresponding to a water content of about 40%. The clay was then placed in a crock and stored in the moist room for future use. It was molded by hand into a cylinder slightly larger than a triaxial specimen and trimmed to the desired size.

3.3.4 Grundite

The Grundite specimens were prepared in the same manner as was outlined for the compacted Sault clay.

3.3.5 Undisturbed Clays

Undisturbed samples were extruded from 3" O.D. thin-walled shelby tubes and trimmed to the desired specimen size of 3 inches in length and 1.4 inches in diameter.

3.4 Triaxial Tests

All specimens are 3 inches in length and 1.4 inches in diameter. In order to accelerate drainage and porewater pressure response, filter paper side drains and internal wool drains were used on all specimens. Two "Trojan" rubber membranes were used on all specimens. For the long term creep tests on the undisturbed and laboratory clays, silicon grease was applied between the membranes as an attempt to prevent leakage. In some long term creep tests on undisturbed Marine City clay, a mercury jacket (for details see Bishop and Henkel, 1957) was placed around the specimen in order to prevent osmosis through the membranes.

All samples were consolidated under an all-round pressure. Volume changes were measured throughout the consolidation process. Water contents were measured before consolidation and after completion of the triaxial tests. After consolidation, all samples were back-pressured for at least 10 hours to saturate the sample and the system.

Area corrections have been applied in the evaluation of all the test data (see e.g., Bishop and Henkel, 1957).

No attempt was made to correct for the effects of membranes, filter strip restraint and internal drains.

Details of the triaxial testing equipments have been reported elsewhere (see Wu, et al., 1962) and are not repeated here.

3.4.1 CFS Test

The objective of this test is to obtain the development of cohesion (c') and friction (b') with strain under constant strain rate by imposing two different stress states on the specimen through pore-water control. Thus in essence, the CFS test is a drained test under controlled effective stresses. The procedure adopted follows the steps outlined by Schmertmann and Osterberg (1960) with some slight modifications. For a detailed description of the testing procedure adopted in the Soil Mechanics Laboratory at Michigan State University, see Holliday (1963).

The variations of c' and &' with respect to axial strain under different strain rates for the clays testes are presented in graphical form in Figs. A-13, A-14, A-15, A-16, A-17, A-18, A-19, and A-20 in Appendix III. Pertinent details of all the tests are summarized in Table 2.

In Table 2, w_i and w_f are the initial and final water contents. c is the consolidation pressure. is the constant strain rate at which the test was carried out; it ranges from 0.04 %/hr to 6.7 %/hr. The values of c and c at the peak deviator stress, $(c_1 - c_3)_{max}$ are denoted by the subscript f. All specimens were strained until the effective principal stress ratio, c and c at this state are denoted by the subscript f. The mode of failure, observed from visual inspection of the specimen after failure is also included in Table 2.

Bulge Brittle Brittle Bulge Bulge Bulge Fallure Mode Bulge Bulge Bulge Bulge Bulge Bulge 1.23 2.20 20 20 1.20 $\cdot \Psi$ 12.70 6.50 14.00 14.60 16.40 = constant ė, deg. end of test kg/cm² 0.460 250 150 100 100 Ū√3₃ 0004 9.01 10.01 10.01 9.0 ų, L 0 BB 21.20 deg g. 1 1 $(G_i - G_i)_{max}$ peak Stress $\frac{c_{\mathbf{f}}}{kg/cm^2}$ 0.190 1 1 1 1 . . 4.50 7.00 -<u>(1)</u> J 356 1 1 1 1 1 1 % 4.0 0.0 0.0 2.0 0 0.9 $\kappa \epsilon \sim \kappa$ 6 29.85 30.70 30.20 27.40 32.50 ₽6 45.60 45.60 45.60 50.00 50.00 49.80 팔 BE Sault Sault Clay Type Sault Consol Remolded Compacted Spec. SHR-1 SHR-5 HR-1 HR-2 F-1

Tests

CFS

Summary of

N

Table

Bulge Bulge Bulge Bulge

Brittle Brittle

Fallure Mode Brittle Brittle Brittle

Brittle

Brittle

1.120 0.122 3.280 0.057 4.150 0.234 %/hr 57.25 57.25 1.00 ·W 33.50 37.80 24.10 20.70 20.70 26.00 13.58 824.03 534.03 20.60 deg, **6** = constant test $c_{\rm u}^{\bullet}$ 0.520 0.100 0.394 0.364 0.430 0.450 0.317 0.110 0.00 0.308 0.180 0.332 of \bar{G}_1/\bar{G}_3 end 0.00 10.01 0.00 11.00 00,00 2.0 w 2 86 050 000 0 16.90 25.40 16.90 33.50 17.50 17.40 23.90 ø. deg. - C3)max peak Stress 0.410 0.220 0.343 0.110 0.02 0.590 0.345 0.318 Gr. 1 1 1 1 6 **K**8/ Tests 80 80 80 80 80 2.50 M. 26 5.0 1 1 1 1 CFS kg/2 0000 0000 0000 2.0 of 2.0 6 Summary 18.43 18.22 22.10 24.16 29.00 28.65 29.30 25.00 98 28.60 28.60 28.15 45.90 47.30 41.90 19.00 17.88 23.39 24.94 25.20 ¥ BE 2(cont'd) CIFA uny Sault Clay Type Marine Consolida. crudite MITTOM Over-.stbnU .atbaU F-00-1 Spec. Number WR-12 WR-13 WR-7 WR-8 Table 7-2-0 9999 9000

For discussions on the validity of the CFS test, the reader is referred to Schmertmann (1962) and Wu (1963).

3.4.2 Creep-CFS Test

This test is, in essence, a two-phase test consisting of a creep test as the first phase and a CFS test as the second phase. The objective of this test is to obtain the values of cohesion (c') and friction ($^{+}$) after various times of creep. To do this, it is necessary to extrapolate the c'-1 and $^{+}$ -2 curves of the CFS phase to the axial strain at the end of the creep phase. For a detailed account of the method of extrapolation, see Holliday (1963).

Pertinent details of all the test results are summarized in Table 3. In Table 3, \mathbf{w}_i and \mathbf{w}_f are the initial and final water contents. \mathbf{c}_c is the consolidation pressure. \mathbf{c}_c is the axial strain at the end of the creep phase. \mathbf{c}_c is the rate of creep strain evaluated at or near the end of the strain-time curve of the creep phase. \mathbf{c}_c is the constant strain rate at which the CFS phase was carried out. $\mathbf{k} = \frac{1}{1}\sqrt{2}\mathbf{s}_3$ is the effective principal stress ratio at the end of the creep phase. \mathbf{c}_c^i and \mathbf{s}_c^i are the values of \mathbf{c}_c^i and \mathbf{s}_c^i at end of creep at various times of creep \mathbf{t}_c .

A creep stress of $1.15~\rm kg/cm^2$ with creep time varying from 30 to 7200 minutes was used on the consolidated Sault clay. A creep stress of $1.0~\rm kg/cm^2$ with creep time varying from 120 to 4320 minutes was used on the compacted Sault clay.

Table 3 Summary of Creep-CFS Tests

) ,	at end creep	1.96	2.17	2.25	2.14	2.10	1.96	2.085	2.85	3.26	
av. CFS É	%/hr	1.1.	=	=	=	=	=	2	0.85	0.85	
ø end Creep	•ਲ ਼ ਲਵਾਈ	00.00	07.0	6.20	3.75	5.10	2.40	6.20	9.00	12.21	
o end Creep	kg/cm ²	.575	.564	.420	.473	.432	.350	.403	•393	.379	
ėc © end Creep	%/hr	84000	.43650	.22000	.0825	.0450	.0300	.0200	.01395	\$4600.	
Creep Strain ϵ_{c}	<i>P</i> \$	1.96	2.14	2.31	2.72	2.56	2.90	2.04	2,565	4.010	
Creep Tine	min.	30	09	240	240	390	084	720	0441	7200	
Creep Stress $(\vec{J}_1 - \vec{G}_2)$	kg/cm ²	1.15		•	=	2	p= @	:	5	\$	
<u>1</u> 0	kg/cm ²	2.0	=	=	=	=	Ε	=	=	=	
Z M	PS.	32.20	32.40	32.60	32.50	32.00	33.20	32.40	33.10	33.70	
با *	₽6	40.70	148.90	47.90	40.80	02.04	40.50	40.65	00.74	47.00	
LAbe	CJay			1Tne	s p	əte	pţŢc	suo	5		
Soec. Lunber		F-C-5	F-C-4	F-C-2	F-C-6	F-C-9	F-C-8	F-C-7	F-C-3	F-C-1	

M

 $= \frac{\tilde{G}_1/\tilde{G}_3}{\text{at end}}$ 1.85 1.92 2.56 2.43 2,02 av.cfs É %/hr 0.90 1.00 0.80 0.90 1.00 ø end Creep deg. 2.85 5.00 6.00 7.40 kg/cm² c. @ end Creep .422 .376 .356 694. .372 .05335 00100 .00500 *0030t .12700 @ end Creep %/hr Creep Strain 0.866 0.712 1,385 0.861 1.417 Summary of Creep-CFS Tests Creep Time min. 120 1480 960 1440 4320 kg/cm² 0-03 Creep Stress 1.00 = = = kg/ 42.30 31.40 2.0 41.80 31.10 2.0 6 42.60 31.50 42.90 32.60 41.20 32.20 Mf Ьв Sault CJsl Llbe Spec. Number R-C-10 R-C-11 R-C-9 R-C-7 R-C-8

Table 3 (cont'd)

A typical Creep-CFS test data in the graphical form is shown in Fig. A-5 in Appendix III.

3.4.3 Creep Test

As mentioned in 3.1, the objective of the creep test is to obtain the time-deformational characteristics of a clay under creep in order to evaluate the parameters in the rheologic model. With the values of the parameters given, the value of c' can be calculated by means of equations (25) and (27).

In this test, upon completion of consolidation and back-pressure, the sample was loaded with dead weights under undrained condition. Both incremental and single creep loads were used. Axial deformations and pore-pressures were measured and recorded at intervals throughout the entire test. Each creep load was maintained until the pore-pressure changes were negligible and the deformation had almost ceased. The accuracy of the deformation and pore-pressure measurement was \pm 0.0001" and \pm 0.01 kg/cm² respectively.

A complete listing of the specimens tested is provided in Table 4. Specimen No. C-C-1, C-C-7 and F-C-9 are quoted from Christensen (1964); Specimen No. IC-1, IC-2, and D-2-4-2 are quoted from Christensen and Wu (1964). A summary of the pertinent details on these tests is given in Tables 4, 5(a) and 5(b).

The method of evaluation of the parameters of the rheologic model has been outlined in detail by

29.10 29.80 31.80 31.60 27.90 34.40 26.10 Final W.C. 93 Initial W.C. 43.80 54.15 41.00 27.60 44.30 41.50 26.10 42.70 ьв Stress History N.C N.C. N.C N.C ບໍ່ N.C. 0°C Consolida. Pressure kg/cm² 2.0 2.0 2.0 2.0 2.0 2.0 2.0 Compacted Sault Undisturbed Marine City Clay Type & Origin Summary of Creep Tests Compacted Sault Sault Sault Consolida. Consolida. Consolida. Sault Grund1 te Specimen No. Table 4 D-2-4-2 F-00-1 C-C-2 F-C-9 F-C-1 C-C-1 1C-21

Summary of Model Parameters(After R.W. Christensen) Table 5(a)

Sample No.	Load Incre.	Initial (G, - G ₃)	Final (G, - G ₃) Δ (G, - G ₃)	∆(C, - G	Ä,	, z	k ₁ +k ₂	R	β 10-7	Éo	ng.
		kg/cm ²	kg/cm ²	kg/cm ²	kg/cm ²	kg/cm ²		cm ² /kg	min-1	BR	<i>9</i> €
C-C-1	н	00000	0.500	0.500	7660.00		154.00 0.9830	21.50	1,895	000000	0.1081
	8	0.500	0.765	0.265	559.00	65.30	65.30 0.8950	28.80	1.503	0,1081	0.2440
	<u>س</u>	0.765	666.0	0.234	1093.00	19.15	0.9820	21.30	3.850	0,2440	0.6580
	7	0.999	1.240	0.241	482.00	10.30	10.30 0.9790	20.80	13.24	0.6580	1.4300
	70	1.240	1.490	0.250	606.00	7.43	0.9880	13.84	19.48	1.4300	2.5500
	9	1.490	1.708	0.218	283.00	5.70	0.9820	22.50	15.95	2.550	3.850
	2	1.708	1.890	0,182	248.00	1.08	1.08 0.9950	50.90	41.60	3.850	9.370
2-0-0	r-l	00000	0.9870	0.9870	516.00	23.20	0.9570	13.96	61.50	00000	1.420
F-C-9	н	00000	1.103	1.103	46.70		8,47 0.848	22.70	5.980	00000	4.350

٢			<i>N</i> 000	22 80 50	0 00 0		40
	8	86	2.666	0.082	0 0 0 0 4	2.51	2.93
	ů	<i>P6</i>	0.0000 0.5350 1.0600 1.6600	0.0000 0.0822 0.3580	0.0000 0.0703 2.2030 3.0680	00000	0.000
	β 10-7	m1n-1	5.440 8.730 5.680 0.214	2.970 1.545 11.200	0.900 1.300 2.000	7.40	32.00 16.00
	8	cm ² /kg	14.00 15.50 14.60 67.35	23.80 19.45 17.70	65.00 12.00 30.00 26.00	20.70	12.20
Tests	7+ ¹ 7	1	0.975 0.908 0.928 0.7965	0.798 0.920 0.995	0.762 0.889 0.806 0.938	0.910	0.940
Creep	к 2	kg/om ²	29.30 27.10 13.00 8.32	104.30 37.20 2.12	445.00 77.00 41.00	9.30	8.30 10.50
ers from	г я	kg/cm ²	1580.7 268.9 167.0 32.78	411.70 423.30 453.88	305.00 355.00 323.00 629.00	90.70	117.0
Parameters	D(0' - 0	kg/cm ²	0.5000 0.2430 0.2535 0.2450	0.2565 0.3085 0.1950	0.2000	0.700	0.500
[Model	Final (0, - 03)	kg/om ²	0.5000 0.7415 0.9950 1.2400	0.7515 1.0600 1.2550	0.4700 0.6700 0.8700 1.0700	0.700	0.500
Summary of	Initial (G, - G ₃)(kg/cm ²	0.0000 0.4985 0.7415 0.9950	0.4950 0.7515 1.0600	0.2700 0.4700 0.6700 0.8700	00000	0.000
5(b) Su	Load Incre.		からで	なるま	ここり	~	40
Table 5(Sample No.		F C - 1	F-00-1	D-2-4-2	IC-1	10-2

Christensen (1964) and is not repeated here. A typical sample calculation of the determination of the model parameters is included in Appendix IV.

All relevant data, upon which the determination of the parameters of the rheologic model is based, are presented in Figures A-6, A-7, A-8, A-9, A-10, A-11 and A-12 in Appendix III. In these figures,

u is the deformation at zero time

u_f is the ultimate deformation

is the deformation rate

U* = $(u-u_0)/(u_f-u_0)$ is the dimensionless creep function

$$Z(t) = 1/2 = \frac{k_1 k_2}{k_1 + k_2} t \text{ in Equation (27)}$$

A =
$$\tanh^{-1} \exp \left(-\frac{\frac{k_1}{k_1}}{\frac{k_1}{k_2}}\right)$$
 in Equation (27)

and D = $\frac{1}{1} - \frac{1}{3}$ is the creep stress

In Fig. A-12, the deformation is still progressing when the oil in the triaxial cell became depleted. There followed the temporary release of cell pressure and creep load that accompanied the replacement of the oil supply. The second part of the curve was measured after the cell pressure and creep load were resumed.

A summary of the parameters of the rheologic model is given in Tables 5(a) and 5(b). In Tables 5(a) and 5(b), $\frac{1}{0}$ and $\frac{1}{0}$ are the initial and final axial strain in % for the load increment shown. All the other notations are defined as before in 2.2.

CHAPTER 4

ANALYSIS OF EXPERIMENTAL RESULTS

In this chapter, the experimental results are analyzed in terms of the hypotheses presented in Chapter 2.

The behaviors of friction and cohesion are analyzed separately in 4.2 and 4.3.

4.1 Presentation of Test Data

4.1.1 CFS Tests

The variations of f' with respect to axial strain under different constant strain rates for the series of tests on each of the clays (compacted Sault, remolded Sault, over-consolidated Sault, Grundite and undistrubed Willow Run) are shown in Figs. 19, 20, 21, 22 and 23 respectively.

The variations of \$' with respect to strain rate at axial strains of 1%, 2%, 3%, etc. for the series of tests on each of the clays (compacted Sault, overconsolidated Sault, Grundite and undisturbed Willow Run) are shown in Figs. 24(a), 25(a), 26(a) and 27(b) respectively. These values of \$'\$ were obtained from the \$'-\$ curves. They are tabulated in Tables A-11, A-12, A-13 and A-14 in Appendix III.

The variations of \mathfrak{d}^{\bullet} with respect to the time required to reach the axial strains of 1%, 2%, 3%, etc.

under different constant strain rates for the series of tests on each of the clays (compacted Sault, over-consolidated Sault, Grundite, and undisturbed Willow Run) are given in Tables A-11, A-12, A-13 and A-14 in Appendix III. The value of Φ' as a function of elapsed time is plotted in Figs. 24(b), 25(b), 26(b) and 27(a).

The variations of c' with respect to axial strain under different constant strain rates for the series of tests on each of the clays (compacted Sault, remolded Sault, overconsolidated Sault, Grundite and undisturbed Willow Run) are shown in Figs. 32, 33, 34, 35 and 36 respectively.

The variations of c' with respect to strain rate at axial strains of 1%, 2%, 3%, etc. for the series of tests on each of the clays (compacted Sault, Grundite, overconsolidated Sault and undisturbed Willow Run) are shown in Figs. 37, 38, 39 and 40 respectively. These values of c' were obtained from the c'- curves. They are tabulated in Tables A-11, A-12, A-13 and A-14 in Appendix III.

Due to insufficient test data, curves are not shown for the consolidated Sault, remolded Sault and undisturbed Marine City clays.

4.1.2 Creep-CFS Tests

Pertinent details of all the test results have been summarized in Table 3.

The variation of $^{\circ}_{\mathbf{C}}$ with respect to creep strain rate for the consolidated and compacted Sault Clays is shown in Fig. 28(a).

The variation of $^{\circ}_{C}$ with respect to creep time for the consolidated and compacted Sault clays is shown in Fig. 28(b).

The variation of c_{C}^{\prime} with respect to creep strain rate for the consolidated and compacted Sault clays is shown in Fig. 41(a).

The variation of c_C^{\bullet} with respect to creep time for the consolidated and compacted Sault clays is shown in Fig. 41(b).

4.2 The Behavior of Friction

4.2.1 The Experimental Behavior of Friction

a. CFS Tests

The experimental behavior of friction in the CFS test may be summarized as follows.

For all the clays tested, p' increases with decreasing strain rate and approaches an ultimate value at a small value of strain rate below which it remains virtually constant. This can be seen from Figs. 19, 20, 21, 22, 23, 24(a), 25(a), 26(a) and 27(b).

Alternatively, it can be said that Φ^{\bullet} increases with increasing testing time (or decreasing strain rate) and approaches an ultimate value after a certain minimal

of time t_{um} beyond which it remains virtually constant. This can also be seen from Table 2, Figs. 24(b), 25(b), 26(b) and 27(a). For the compacted Sault clay, Grundite and Willow Run clay, it appears that ϕ_u^{\dagger} would increase appreciably with further reduction in strain rate. The minimal of time required to attain the ultimate value ϕ_{um}^{\dagger} varies with the clay structure. For each of the clays tested, the estimated values are given in Table 6.

For all the clays except Grundite, the minimal of axial strain $\frac{1}{1}$ required to attain the fully mobilized value $\frac{1}{1}$ ($\frac{1}{1}$ at $\frac{1}{1}$ = constant) decreases with decreasing strain rate. This is in full agreement with the findings above. The value of $\frac{1}{1}$ at the various strain rates for each of the clays (compacted Sault, overconsolidated Sault, Grundite and undisturbed Willow Run) is also given in Table 6.

b. Creep-CFS Tests

The experimental behavior of friction in the Creep-CFS test may be summarized as follows.

As shown in Fig. 28(b), $^{\circ}_{\mathbf{c}}$ increases with increasing creep time and approaches an ultimate value $^{\circ}_{\mathbf{k}}$ at a certain time $\mathbf{t}_{\mathbf{k}}$ beyond which it remains virtually constant. The value of $\mathbf{t}_{\mathbf{k}}$ for the consolidated and compacted Sault Clays is given in Table 7.

The findings of the Creep-CFS tests are in full qualitative agreement with that of the CFS tests.

Strain and time characteristics of friction-CFS tests. Table 6.

Clay	% mn	. %/hr	p¹ deg.	t _{um} min.	um degrees
Compacted Sault	10.00 9.40 7.40 6.40	5.150 1.200 0.360 0.156	7.20 14.40 20.00 24.30	3,000	25.50
Over-Consolidated Sault	5.00 4.50 4.00	3.28 1.120 0.122	13.58 20.70 26.00	2,300	28.80
Grundite	5.00 7.00 7.00 9.00	4.150 1.160 0.234 0.057	2.70 5.40 8.40 9.50	9, 500	9.40
Undisturbed Willow Run	5.50 3.00 2.25	1.500 0.170 0.04	20.70 37.80 33.50	4,500	33.50

Table 7. Friction-time characteristics--Creep-CFS test on Sault clay.

Clay	t _k min.	⊅¦ deg.
Consolidated Sault	7,500	12.50
Compacted Sault	5,000	7.50

c. Discussion

It may be concluded from the experimental findings that the behavior of friction is time-dependent. The nature of friction is such that to attain full mobilization at any stress level below failure or at failure, a certain minimal of time is required.

It is of interest to note that a value of the Hvorslev's true angle of friction ($^{\circ}_{e}$) as high as 33-44 degrees had been reported on triaxial tests of Mexico clay under an average constant strain rate of some 0.083%-strain per hour by Lo (1962). A value of $^{\circ}$ as high as 33.5-37.5 degrees has also been obtained on undisturbed Willow Run clay at an average strain rate of 0.105 %-strain per hour. Although the nature of Mexico clay is expected to be different from that of the undisturbed Willow Run clay, the trend that "friction" is time-dependent is significant.

4.2.2 Analysis of the Bhavior of Friction

a. Behavior of 1 under Constant Strain Rate

The experimental findings summarized in 4.2.1.a are in qualitative agreement with the hypothesis for the

behavior of friction under constant strain rate expressed by equation (17) in 2.1.5. This provides an insight into the actual mechanism of friction in clays. It may be stated that the nature of friction is such that while the slip at the contacts appears to be the governing mechanism, the resistance to shear at these contacts are also governed by a time element. This time element may be related to the states of interparticle force fields at the contacts when the contacts are failed and reformed.

b. Behavior of ¢' under Creep

If one accepts the validity of equation (17), the behavior of ${\bf 1}'$ under creep (${\bf 1}'$) may be analyzed according to Equation (18) in which

$$N_{C} \langle N_{m} \rangle$$
, $\Phi_{C}^{\dagger} = \Phi_{k}^{\dagger}$

if

$$\dot{N}_{C} > \dot{N}_{m}$$
, $\Phi_{C}^{I} = \frac{\Phi_{k}^{I}}{\Phi_{um}^{I}} \times \Phi_{u}^{I}$

where the value of v_u corresponds to $N = N_C$.

To evaluate Φ_{C}^{\prime} in (18), the following procedure is used.

Values of Φ_u^* (Φ^* @ $\overline{c_1}/\overline{c_3}$ = constant) can be obtained from results of the CFS tests under different constant strain rates for the compacted and consolidated Sault clays in Table 2. With $\overline{c_u}$ and $\overline{c_v}$ for each test given in Table 2, the time $\overline{c_u}$ to reach $\overline{c_u}$ for each test is obtained as $\overline{c_u} = \overline{c_u}/\overline{c_v}$. The displacement at failure, $\overline{N_u}$ can be

;
•
•
,
::
i
•

obtained from Curve A in Figs. 14 and 16. Since failure has taken place when N_u is reached, N_u is taken as 6.n.e for all the specimens under consideration. Thus, for each CFS test, $N = 6 \cdot n \cdot e/t_u$. As an example, we calculate N for Specimen R-6 of the compacted Sault clay in Table 2. From Table 2, $n_u = 9 \%$ $n_u = 0.12 \%/hr$

Table 2,
$$t_u = 9 \% = 0.12 \%/hr$$

$$t_u = u_u/\hat{\epsilon} = 9/0.12 = 75 \text{ hours}$$

N = 6.n.e/75 = 0.08 n.e/hr

In the same manner, other values of N for the series of CFS tests on the consolidated, remolded, and compacted Sault clays are computed and summarized in Table A-15 in Appendix V. The corresponding values of $\frac{1}{u}$ for these specimens are also shown in Table A-15 in Appendix V.

The variation of \mathfrak{I}'_u with respect to N for the compacted Sault, remolded Sault and consolidated Sault clays is shown in Fig. 29. It can be seen from Fig. 29 that the \mathfrak{I}'_u -N characteristics of these clays can all be represented by one straight line. Accordingly, Fig. 29 is used for the prediction of \mathfrak{I}'_c in the Creep-CFS tests of both the consolidated and compacted Sault clays.

For each of the Creep-CFS tests on the consolidated and compacted Sault clays, the value of k at end of creep is given in Table 3. With this value of k, $N_{\rm C}$ (displacement under creep) can be obtained from Curve A in Figs. 14 or 16. $N_{\rm C}$ is then obtained as $N_{\rm C}/t_{\rm C}$. $t_{\rm C}$, the creep time for each test is given in Table 3. As an example, we calculate $N_{\rm C}$ for Specimen F-C-5 of the consolidated Sault clay in Table 3.

		,
		,
		•
		į

From Table 3: $k = 1.96 t_{C} = 0.5 hour$

For r = 0.3: From Fig. 14, for k = 1.96, $N_c = 2.05$ n.e $N_c = N_c/t_c = 2.05/0.5 = 4.10$ n.e/hr

For $N_{C} = 0.5$: From Fig. 16, for k = 1.96, $N_{C} = 0.9$ n.e $N_{C} = N_{C}/t_{C} = 0.9/0.5 = 1.8$ n.e/hr

In the same manner, other values of $\dot{N}_{\rm C}$ for the specimens of the consolidated and compacted Sault clays are computed for $\dot{r}=0.3$ and 0.5. They are summarized in Table A-16 in Appendix V.

From Table 7:

is assumed.

 $\mathfrak{p}_{k}^{\bullet}$ = 12.5 degrees for the consolidated Sault clay $\mathfrak{p}_{k}^{\bullet}$ = 7.5 degrees for the compacted Sault clay

Due to insufficient CFS test data on the consolidated Sault clay, a value of $\hat{\nu}_{um}^{\dagger}$ for this clay is not available. However, since the $\hat{\nu}_{u}^{\dagger}$ - \hat{N} characteristic of this clay can be satisfactorily represented by that of the compacted Sault clay, the same value of $\hat{\nu}_{um}^{\dagger}$ for the compacted Sault clay

Thus, according to Table 6, $t_{um}^{1} = 25.5^{\circ}$. From Figure 29, for $t_{um}^{1} = 25.5^{\circ}$, $t_{m}^{1} = 0.05$ n.e/hr

As an example, we calculate the predicted values of $^{\prime}$ for Specimen F-C-5 of the consolidated Sault clay in Table 3.

From Table A-15 in Appendix V, $\dot{N}_{C} = 4.1$ n.e/hr Since $\dot{N}_{m} = 0.05$ n.e/hr, $\dot{N}_{C} > \dot{N}_{m}$

From (18), $\text{predicted } \Phi_{\mathbf{C}}^{\bullet} = \frac{\Phi_{\mathbf{k}}^{\bullet}}{\Phi_{\mathbf{l},\mathbf{m}}^{\bullet}} \times \Phi_{\mathbf{u}}^{\bullet}$

		·
		•
		;
		;
		;

From Fig. 29, for $N = N_C = 4.1$ n.e/hr, $\phi_{11}^* = 6.2$ deg.

Substituting in (18),

predicted $\Phi_{c}' = \frac{12.5}{25.5} \times 6.2 = 3.04 \text{ deg.}$

= 0.5: From Table A-16 in Appendix V, $N_{\rm C}$ = 0.9 n.e/hr Since $N_{\rm C}$ $N_{\rm m}$

From (18), predicted $\Phi_{\mathbf{C}}^{\mathbf{I}} = \frac{\Phi_{\mathbf{C}}^{\mathbf{I}}}{\Phi_{\mathbf{U}\mathbf{m}}^{\mathbf{I}}} \times \Phi_{\mathbf{U}}^{\mathbf{I}}$

From Fig. 29, for $N = N_C = 0.9$ n.e/hr $v_U = 10.0$ deg.

... predicted $\Phi_{C}^{1} = \frac{12.5}{25.5} \times 10 = 4.9 \text{ deg.}$

In the same manner, other values of predicted $^{\circ}_{C}$ for the specimens of the consolidated and compacted Sault clays are evaluated at the various creep time $^{\circ}_{C}$ in Table 3. Results of the computations of the values of predicted $^{\circ}_{C}$ for $^{\circ}_{C}$ = 0.3 and 0.5 are summarized in Table A-16 in Appendix V.

The above calculations do not take into account the resistance due to adhesion. Estimates show that the adhesion resistance is approximately equivalent to a k of 1.3 (2.1.2, p.27). This means that k < 1.3, N = 0 If this is included in the calculations, the resulting value of N would be smaller and $^{\dagger}_{c}$ would be larger. The difference in $^{\dagger}_{c}$ would be around 25%. Since the calculations for $^{\dagger}_{c}$ are approximate and are made to check the principle of the particle model, it is felt that the additional refinement of calculating the adhesion resistance is not warranted.

The predicted values of $t_{\rm c}^{\prime}$ from Table A-15 in Appendix V are plotted against the creep time $t_{\rm c}$ in Figs. 30 and 31 for the consolidated and compacted Sault clays respectively.

For comparison, the observed values of ' are shown in Figs. 30 and 31 for the two clays. It can be seen from these figures, that the shape of the predicted behavior is in agreement with that of the observed behavior. The magnitude between predicted and observed, however, differs.

It should be noted that values of ${\uparrow}^{\bullet}_{C}$ are measured in the CFS test at a strain rate ${}^{\bullet}=0.3-1.1$ %/hr. This strain-rate is in most cases different from the values of ${}^{\bullet}_{C}$ at the end of the creep period. This change in strain-rates may be expected to effect the measured value of ${}^{\dagger}_{C}$ as it has already been shown that ${}^{\bullet}_{C}$ is strain-rate sensitive. Since the ${}^{\bullet}_{C}$ in the CFS phase is greater than ${}^{\bullet}_{C}$, the measured values of ${}^{\dagger}_{C}$ cannot be taken as the true values that exist at the end of the creep period. If one applies a correction with the relationship in Fig. 24, the agreement between measured and calculated (Figs. 30 and 31) would be improved somewhat.

Alternatively, we may compare $\frac{1}{k}$ with the value of $\frac{1}{m}$ at the same strain $\frac{1}{c}$ in a CFS test. If we take the slowest CFS test for the compacted Sault clay for $\frac{1}{m}$, we get 15° (Spec. RA-3, at = 1.4%, Fig. 19). This is considerably greater than 7.5° . However, the CFS test is

run at a strain-rate of about 1 %/hr. Hence the difference may be due to the strain rate effect.

4.2.3 Summary

A particle model is proposed for the mechanism of friction in clays. In this model, the resistance at the contacts of the particles is assumed to consist of a mineral friction and an adhesion. It varies with the orientation of the particle relative to the direction of shear.

The angle of internal friction is the result of the resistance due to mineral friction at the contacts. The resistance due to adhesion is estimated to be small in comparison with that due to friction. When the matrix is under stress, the adhesion resistance is fully mobilized at all the contacts instantaneously. This quantity is independent of the normal stress and is considered to be a non-viscous cohesion.

When the clay mineral network is under a shear stress, contacts with the smallest resistance along the direction of shear slip first. Following these contact failures, particles are displaced from one position to another. As displacements are taking place, contacts are being reformed. As displacements proceed, contacts along the direction of shear are reformed. The resistance of these reformed contacts depends on the positions taken up by the displaced particles and the elapsed time between displacements. At any stress level below failure or at

failure, the maximum value of the angle of internal friction is attained only if the elapsed time between displacements is sufficient for the reformed contacts to develop their maximum resistance to shear.

The model is used to explain the behavior of the angle of internal friction under a constant strain rate test. By means of equations (17) and (18), the data from constant strain-rate tests are used to calculate the angle of internal friction in the creep test.

The experimental behaviors of the angle of internal friction under constant strain rate and creep are in qualitative agreement with the model behaviors.

4.3 The Behavior of Cohesion

4.3.1 The Experimental Behavior of Cohesion

a. CFS Tests

The experimental behavior of cohesion in the CFS test may be summarized as follows.

For all the clays tested, c' is strongly dependent on strain rate. It decreases with decreasing strain rate and approaches an ultimate value at a very small value of strain rate beyond which it remains virtually constant. This can be seen from Figs. 32, 33, 34, 35, 36, 37, 38, 39, and 40.

b. Creep-CFS Tests

The experimental behavior of cohesion in the Creep-CFS test may be summarized as follows.

For the consolidated and compacted Sault clays tested, $c_{\mathtt{C}}^{\prime}$ decreases with decreasing creep strain rate and

approaches an ultimate value at a certain value of creep strain rate beyond which it remains virtually constant; it does not approach zero when the strain rate approaches zero. This is readily seen in Fig. 41(a).

Alternatively, as shown in Fig. 41(b), $c_{\rm C}^{\prime}$ decreases with increasing creep time and approaches a constant value rather than zero when the creep time approaches a large value.

c. Discussion

The experimental findings on the behavior of cohesion from the CFS tests indicate that cohesion is strongly
dependent on strain rate. The trend indicates that c'
does not approach zero as the strain rate continues to
decrease. The findings from the Creep-CFS tests indicate
clearly that cohesion decreases with increasing time and
approaches a constant value rather than zero when the time
approaches a large value. It may, therefore, be concluded
that viscous flow is only partially responsible for the
nature of cohesion. It appears that the nature of cohesion
is partly viscous and partly non-viscous.

4.3.2 Analysis of the Behavior of Cohesion

a. Behavior of c' under Constant Strain Rate

The behavior of c' is analyzed in terms of the viscous resistance of the clay particle structure expressed by equation (25) in which

$$f_1 = \frac{1}{2} \log_e \left[\frac{2!}{2!} + \sqrt{1 + \frac{2!}{2!}}^2 \tanh \left(\frac{\sqrt{1 + \frac{2!}{2!}}^2}{2} + \frac{k_1}{2!} \gamma + \tanh^{-1} \frac{1 - \frac{k_1}{2!}}{\sqrt{1 + \frac{2!}{2!}}^2} \right) \right]$$

The hyperbolic tangent approaches unity if

$$\frac{\sqrt{1 + \frac{\pi}{2}}^2}{2} \frac{9k_1}{4!} + \tanh^{-1} \frac{1 - \frac{\pi}{2}!}{\sqrt{1 + \frac{\pi}{2}!}^2} > 3$$

It is shown in Appendix V under "Validity of Equation (25')" that the values of k_1 , and β at all stress levels of the clays tested are such that the hyperbolic tangent does approach unity. Equation (25) is then reduced to

$$f_1 = \frac{1}{2} \log_e (z' + \sqrt{1 + z'^2}) \div \frac{1}{2} \log_e 2z'$$
since 2' is large compared with 1.

To evaluate f_1 for the compacted Sault clay in a CFS test under a constant strain rate *, values of * and * are required. These parameters are not constant for a single test specimen but vary with the stress level. Hence the values of and f from creep tests on the compacted Sault clay are only valid within the range of stresses and strains from which the parameters are obtained. example, in Table 5(a), under the first load increment of Specimen C-C-l (C-C-l-l) of the compacted Sault clay, the values of 21.5 cm $^2/\text{kg}$ and 1.895 x 10^{-7} min $^{-1}$ obtained for and f respectively are only valid for the stress range of D = $(\frac{1}{1} - \frac{3}{3}) = 0$ to 0.5 kg/cm², and the strain range of $=\frac{1}{2}$ to $\frac{1}{2}$ = 0 to 0.1081 %. These values of $\frac{1}{2}$ and $\frac{1}{2}$, in addition, are only valid for the compacted Sault clay. Accordingly, the value of f_1 evaluated at a value of ctthat corresponds to a value of in a CFS test of the compacted Sault clay is only valid for = 0 to 0.1081 %.

Likewise, the limitations on the validity of \mathbf{f}_1 apply to other load increments on the compacted Sault clay and on the other clays.

As an example, we calculate the values of f_1 for the CFS test on Specimen R-A3 of the compacted Sault clay. From Table 2, i = 0.156 %/hr = 2.6 x 10^{-5} min⁻¹. In an undrained triaxial compression test, $f_{\text{oct}} = \sqrt{2}$ in which is the major principal strain and hence the axial strain and f_{oct} is the octahedral shear strain. Therefore, $f_{\text{oct}} = \sqrt{2}$ is $f_{\text{oct}} = \sqrt{2}$ in $f_{\text{oct}} = \sqrt{2}$ in $f_{\text{oct}} = \sqrt{2}$ in $f_{\text{oct}} = \sqrt{2}$ is $f_{\text{oct}} = \sqrt{2}$ in $f_{\text{oc$

From Table 5(a), under the first load increment of Spec. C-C-1:

$$D = (\frac{1}{1} - \frac{1}{3}) = 0.0 \text{ to } 0.5 \text{ kg/cm}^2$$

$$= 0 \text{ to } 0.1081 \%$$

$$= 21.5 \text{ cm}^2/\text{kg and}$$

$$= 1.895 \times 10^{-7} \text{ min}^{-1}$$

Therefore,

$$f_1 = \frac{1}{1000} \log_e (20^{\circ}) = \frac{1}{21.5} \times \log_e (2 \times 194.5) = 0.278 \text{ kg/cm}^2$$

From Table 5(a), under the second load increment of Spec. C-C-1:

$$D = (\frac{1}{1} - \frac{1}{3}) = 0.5 \text{ to } 0.765 \text{ kg/cm}^2$$

$$= 0.1081 - 0.244 \%$$

$$= 28.8 \text{ cm}^2/\text{kg and}$$

$$= 1.503 \times 10^{-7} \text{ min}^{-1}$$

Therefore,

$$f_1 = \frac{1}{2} \log_e (2\pi^2) = 3.68 \times 10^{-5} / 1.503 \times 10^{-7} = 245$$

 $f_1 = \frac{1}{2} \log_e (2\pi^2) = \frac{1}{28.8} \times \log_e (2 \times 245) = 0.215 \text{ kg/cm}^2.$

In the same manner, values of f₁ for the CFS tests on compacted Sault, over-consolidated Sault, Grundite, consolidated Sault and undisturbed Marine City clays are computed. Results of the computations are summarized in Tables A-17, A-18, A-19, A-20 and A-21 in Appendix V.

Results of f_1 values from Table A-17 in Appendix V are plotted against the final axial strain ∞ obtained from the incremental creep loading tests from which the values of α and α are determined. For example, for Spec. R-A3 in Table A-17:

 $f_1 = 0.28 \text{ kg/cm}^2 \text{ corresponds to } = 0.0 - 0.1081 \%$ $f_1 = 0.217 \text{ kg/cm}^2 \text{ corresponds to } = 0.0 - 0.1081 - 0.244 \%$ $f_1 = 0.249 \text{ kg/cm}^2 \text{ corresponds to } = 0.0 - 0.1081 - 0.244 \%$ Thus for Spec. R-A3, values of f_1 of 0.28, 0.215 and 0.249 kg/cm² are plotted against of 0.1081, 0.244 and 0.658 % respectively.

In the same manner, other values of f_1 and the corresponding values of ∞ are plotted for Spec. R-A3. Figure 42(a) shows the variation of f_1 with respect to the final axial strain ∞ for the different constant strain rates used in the CFS tests. The value of f_1 for ∞ of

		1
		! !

9.37% in Table A-17 is not plotted in Fig. 42(a). This is because under this load increment, Spec. C-C-1-7 underwent large deformations and the ∞ of 9.37% was determined by extrapolation. Hence the parameters may not be very accurate.

For comparison, the observed c'-strain characteristics at different constant strain rates are shown in Fig. 42(b). From Figs. 42(a) and 42(b), it can be seen that the the shape of c' vs ϵ is in good agreement with that of f_1 vs ϵ . At the same strain rate, the predicted values (f_1) are consistently smaller than the observed values (c'). If a value of some 0.125 kg/cm² were added to f_1 , a quantitative agreement between the predicted and observed values results.

It is noted in 4.3.1.c that the nature of cohesion is partly viscous and partly non-viscous. The theoretical treatment here appears to explain satisfactorily that part of cohesion that is of viscous nature as borne out by the experimental findings on the compacted Sault clay.

Consolidated Sault Clay and Undisturbed Marine City Clay-The f_1 values from Tables A-20 and A-21 in Appendix V are plotted against $o^{-1}\infty$ in Figs. 43 and 44 for these two clays. For comparison, the c'values are shown in the same figures.

In Fig. 43, with the expection of the f_1 value at t=1.065-1.693 %, the relative values of t_1 and c' are the same as that obtained for the compacted Sault clay—the predicted value (t_1) is consistently smaller than

that of the observed value (c'). The theoretical treatment also explains satisfactorily the behavior of the viscous part of cohesion for the consolidated Sault clay.

For the undisturbed Marine City clay in Fig. 44, the shape of c'vs ϵ is in reasonable agreement with that of f_1 vs ϵ .

Over-consolidated Sault Clay and Grundite--Because of the small range of $\epsilon = \epsilon_0 - \epsilon_\infty$ in the creep tests from which α and β are determined, the f_1 values from Tables A-18 and A-19 in Appendix V are plotted against the strain rates. Figs. 45(b) and 46(b) show the variation of f_1 with respect to the constant strain rate at different values of $\epsilon = \epsilon_0 - \epsilon_\infty$ for the over-consolidated Sault clay and Grundite respectively.

For comparison, the observed c'-strain rate characteristics for these two clays are shown in Figs. 45(a) and 46(a). It can be seen from these figures that the shape of c' vs $\dot{\epsilon}$ is in reasonable agreement with that of f_1 vs $\dot{\epsilon}$ for both clays. It is of interest to note that for both clays, the predicted value (f_1) is again consistently smaller than the observed value (c').

<u>Discussion</u>--It should be noted that the value of c'refers to an α -plane while that of f_1 refers to the octahedral plane. The inclinations of the α -plane and the octahedral plane are respectively at angles of $\alpha = 45^{\circ} + \frac{\phi}{2}$

The rate process is represented by a viscous element in a rheologic model. In this model, the angle of internal friction and the non-viscous part of cohesion represent the yield value or yield strength of the contacts. Viscous flow of the particles takes place when this yield value of the contacts is exceeded. The behaviors of the viscous part of cohesion under constant strain rate and creep are expressed by equations (25) and (27).

The experimental findings on the behavior of cohesion show that it is partly non-viscous and partly viscous. The experimental behaviors of the viscous part of cohesion under constant strain rate and creep are in qualitative agreement with the predicted behaviors according to equations (25) and (27).

		,
		!
		!
		į
		,
		,
		,
		ì
		:

and 54.7° with the horizontal. Strictly speaking, f_1 should only be compared with c_{oct}^{\bullet} where

$$c_{\text{oct}}' = c' \cos (\alpha - 54.7^{\circ}) = c' \cos (\phi'/2 - 9.7^{\circ})$$

Using an average value of † ' for the range of strains within which f_1 is valid, the estimated errors are given in Table 8. It can be seen from Table 8 that the error of using c' instead of c_{oct}^{\dagger} is less than 1%. Hence, the errors are negligible.

b. Behavior of c' under Creep (Creep-CFS tests)

The behavior of $c_{\rm C}^{\bullet}$ is analyzed in terms of the viscous resistance of the clay particle structure under creep expressed by Equation (27) in which

$$f_1 = -\frac{1}{k} \log_e \tanh \left[\frac{1}{2} + \frac{k_1 k_2}{k_1 + k_2} + \tanh^{-1} \exp(\frac{k_1 k_1}{k_1 + k_2}) \right]$$

To evaluate f_1 at a time t_c in a Creep-CFS test on the consolidated or compacted Sault Clay, values of k_1 , k_2 , and f obtained from single increment creep test are required. Creep tests under $(\frac{1}{1}-\frac{1}{3})$ same as that in the Creep-CFS tests have been carried out on the consolidated and compacted Sault clays. Values of k_1 , k_2 , and f determined from these creep tests are summarized under Spec. F-C-9 for the consolidated Sault clay and under Spec. C-C-7 for the compacted Sault clay in Table 5(a). With these values of k_1 , k_2 , and f is evaluated as follows.

Consolidated Sault Clay--From Spec. F-C-9 in Table 5(a),

		•
		!
		1

Table 8. Estimated errors between c' and c'_{oct} .

Clay	Äve. ۰'	$c'\cos(\frac{x}{2}'-9.7^{\circ})$	$error = \frac{c' - c'_{oct}}{c'_{oct}}$
Compact. Sault	8 ⁰	0.995 c'	0.502
Consoli. Sault	8 ⁰	0.995 c'	0.502
Over-Con. Sault	8°	0.995 c'	0.502
Und. Marine City	18 ⁰	0.999 c'	0.010
Grundite	5 ⁰	0.992 c'	0.0805

$$= 22.70 \text{ cm}^2/\text{kg}, = 5.98 \times 10^{-7} \text{ min}^{-1}$$

$$\frac{k_1}{k_1 + k_2} = 0.848$$

$$k_2 = 8.47 \text{ kg/cm}^2$$

Therefore,

$$1/2 = \frac{k_1 k_2}{k_1 + k_2} = 1/2 \times 22.7 \times 5.98 \times 10^{-7} \times 0.848 \times 8.47$$
$$= 4.87 \times 10^{-5} \text{ min}^{-1}$$

In a triaxial compression test,

the octahedral shear stress
$$\frac{1}{1} = \frac{1}{2} (\frac{2}{3} (\frac{1}{3} - \frac{1}{3})$$

In the Creep-CFS test series on consolidated Sault clay in Table 3,

$$(x_1-x_3) = 1.15 \text{ kg/cm}^2$$
 ... $x_{\text{oct}} = \frac{\sqrt{2}}{3} (1.15) = 0.542 \text{ kg/cm}^2$

Substituting and using oct for -,

$$\exp\left(-\frac{k_1}{k_1+k_2}\right) = \exp\left(-22.7 \times 0.848 \times 0.542\right) = 4.65 \times 10^{-5}$$

...
$$\tanh^{-1} \exp \left(-\frac{\frac{k_1}{k_1 + k_2}}\right) = 4.65 \times 10^{-5}$$

Substituting in (27) and using t_c for t,

$$f_1 = -\frac{1}{22.7} \log_e \tanh (4.87 \times 10^{-5} t_c + 4.65 \times 10^{-5})$$
 (27a)

Using (27a), f_1 can be evaluated at the various creep times t_c of the series of Creep-CFS tests on the consolidated Sault clay in Table 3.

As an example, we calculate f_1 for Spec. F-C-5 in Table 3. From Table 3, $t_c = 30$ min.

Substituting for $t_c = 30$ in (27a), we have

$$f_1 = -\frac{1}{22.7} \log_e \tanh(4.87 \times 10^{-5} \times 30 + 4.65 \times 10^{-5})$$

$$= -\frac{1}{22.7} \log_e \tanh 1.5 - 5 \times 10^{-3}$$

$$= +\frac{1}{22.7} \times 6.5 = 0.287 \text{ kg/cm}^2$$

In the same manner, values of f_1 at the other values of t_c of the series of Creep-CFS tests on the consolidated Sault clay in Table 3 are computed and summarized in Table A-22 in Appendix V

Compacted Sault Clay--From Spec. C-C-7 in Table 5(a), $= 13.96 \text{ cm}^2/\text{kg}, \qquad = 61.5 \times 10^{-7} \text{ min}^{-1}$ $\frac{k_1}{k_1 + k_2} = 0.957 \qquad k_2 = 23.20 \text{ kg/cm}^2$

The value of f obtained for Spec. C-C-7 is 61.5×10^{-7} and that obtained for the 4th load increment of Spec. C-C-1 is 13.24×10^{-7} in Table 5(a). Since the two tests are approximately at the same stress level, a single creep loading test should give values comparable to that from the incremental loading test. In view of the fact that the values of f obtained from all the load increments in Spec. C-C-1 are themselves reasonably consistent, a value of 20×10^{-7} min⁻¹ for f has been assumed as a compromise. In fact, the value of f_1 is relatively insensitive to f. If $f_1 = 61.5 \times 10^{-7}$ is used in the calculations, the value of f_1 is only about 30% smaller.

Using $z = 20.0 \times 10^{-7} \text{ min}^{-1}$, substituting for the values of $\frac{k_1}{k_1 + k_2}$ and k_2 , we have

$$1/2 = \frac{k_1 k_2}{k_1 + k_2} = 1/2 \times 13.96 \times 20 \times 10^{-7} \times 23.2 \times 0.957$$
$$= 3.11 \times 10^{-4} \text{ min}^{-1}$$

In the Creep-CFS test series on the compacted Sault clay

in Table 3, $(\frac{1}{1} - \frac{1}{3}) = 1.0 \text{ kg/cm}^2$

the octahedral shear stress $\frac{1}{2}$ oct $=\frac{12}{3}$ (1.0) = 0.465 kg/cm² Substituting and using $\frac{1}{2}$ for $\frac{1}{2}$,

$$\exp\left(-\frac{\frac{k_1}{k_1 + k_2}}\right) = \exp\left(-13.95 \times 0.957 \times 0.455\right) = 0.002$$

$$\therefore \tanh^{-1} \exp\left(-\frac{\frac{k_1}{k_1 + k_2}}\right) = 0.002.$$

Substituting in (27) and using t_c for t

$$f_1 = -\frac{1}{2} \log_e \tanh(3.11 \times 10^{-4} t + 0.002)$$
 (27b)

Using (27b), f_1 can be evaluated at the various creep times t_c of the series of Creep-CFS tests on the compacted Sault clay.

As an example, we calculate f_1 for Spec. R-C-8 in Table 3. From Table 3, $t_{\rm C}$ = 120 min.

Substituting for t_c in (27b), we have

$$f_1 = -\frac{1}{13.96} \log_e \tanh (3.11 \times 10^{-4} \times 120 + 0.002)$$

= $-\frac{1}{13.96} \log_e \tanh 0.0393 = +\frac{1}{13.96} \times 3.23$
= 0.232 kg/cm^2

In the same manner, values of f_1 at the other values of $t_{\rm C}$ of the series of Creep-CFS tests on the compacted Sault clay in Table 3 are computed and summarized in Table A-22 in Appendix V.

Results of the computations on the variation of f_1 with respect to the creep time $t_{_{\rm C}}$ for the consolidated and compacted Sault clays are plotted in Fig. 47(a). For comparison, the observed behaviors of $c_{_{\rm C}}^{\dagger}$ for these two clays

are shown in Fig. 47(b). From these figures, it can be seen that although there is qualitative agreement between the calculated and measured c', the absolute magnitudes of the two quantities differ by about 0.3 kg/cm².

An examination of Table 3 reveals that the strain rates at the end of the creep phase of some of these tests are several hundred times smaller than that of the CFS phase. It appears desirable to estimate the effect of this change in strain rates on the measured c'c. Assuming that c'c is solely dependent on the creep strain rate c'c, we may obtain values of c'c corresponding to the various values of c'c from the c'versus strain rate characteristics.

Because of the completeness of CFS test data, the compacted Sault clay is investigated.

For the compacted Sault clay, values of c' corresponding to the values of $\frac{1}{c}$ in the Creep-CFS tests can be obtained from Fig. 37. For example, for $\frac{1}{c} = 0.127$ %/hr in Spec. R-C-8 (Table 3), the corresponding value of c' obtained from Fig. 37 is 0.18 - 0.26 kg/cm².

In the same manner, values of c' corresponding to the other values of \dot{c}_c in the Creep-CFS test series of the compacted Sault clay are obtained from Fig. 37. For convenience, we designate these values of c' as c_d' . They are compiled in Table A-23 in Appendix V. Fig. 48 shows the comparison of the values of c_c' and c_d' .

It is felt that although this change in strain rates could affect the absolute magnitudes of the measured

values of c_c^\prime , the general trend as well as the relative magnitude of c_c^\prime should be valid. In view of this, the measured and calculated values of c_c^\prime are compared by converting them into dimensionless "cohesion" as follows.

For the measured values

the dimensionless "cohesion"
$$c^* = \frac{c_c^{\prime} - c_{\infty}^{\prime}}{c_0^{\prime} - c_{\infty}^{\prime}}$$

where

 c_{C}^{\bullet} is the measured value

 $c_{\infty}^{"}$ is the constant value of $c_{c}^{"}$ when the creep time approaches a large value

and

 $c_{_{\scriptsize{O}}}^{"}$ is the constant value of $c_{_{\scriptsize{C}}}^{"}$ when the creep time approaches zero.

From Fig. 41(b), for both the compacted and consolidated Sault clays, c_∞' and c_0' are taken as 0.355 kg/cm² and 0.575 kg/cm² respectively.

For the calculated values

the dimensionless "cohesion"
$$f^* = \frac{f_1 - f_\infty}{f_0 = f_\infty}$$

where

- is the calculated value in equations (27a) and (27b) for the consolidated and compacted Sault clays respectively
- f_{∞} and f_{o} are the calculated values of f_{1} at infinity and zero time in equations (27a) and (27b) $for \ the \ consolidated \ and \ compacted \ Sault \\ clays \ respectively.$

It can be seen from equations (27a) and (27b) that $f_{\infty}=0$. Therefore, $f^{*}=f_{1}/f_{o}$

For the consolidated Sault clay, $f_0 = -\frac{1}{22.7} \log_e 4.65 \times 10^{-5}$ = 0.439 kg/cm²

For the compacted Sault clay, $f_0 = \frac{1}{13.96} \log_e \tanh 0.002$ = 0.445 kg/cm²

The variations of f^* and c^* with respect to the creep time t_C for the series of Creep-CFS tests on the consolidated Sault and compacted Sault clays are computed and tabulated in Tables A-22 and A-24 in Appendix V.

Fig. 49 shows the comparison of the variations of f^* and c^* with respect to the creep time t_c for the two clays. It can be seen from Fig. 49 that the agreement is reasonably satisfactory. It is noted in 4.3.1.c that the nature of cohesion is partly viscous and partly non-viscous. The theoretical treatment here appears to explain satisfactorily the behavior of the viscous part of cohesion under creep for the consolidated and compacted Sault clays.

4.3.3 Summary

A hypothesis is proposed for the behavior of cohesion in clays. The cohesion is assumed to consist of a non-viscous part and a viscous part. The non-viscous part is the result of the adhesion at the contacts of the particles in the particle model. The behavior of the viscous part follows the flow phenomenon of the clay particle structure as a rate process.

CHAPTER 5

CONCLUSIONS

A particle model is proposed for the behavior of friction and cohesion in clays. In this model, the resistance at the particle contacts is assumed to consist of a mineral friction and an adhesion. It varies with the orientation of the particle relative to the direction of shear.

The angle of internal friction is the result of the resistance due to mineral friction at the contacts. The cohesion consists of a non-viscous part and a viscous part. The non-viscous part of cohesion is the result of the resistance due to adhesion at the contacts. The behavior of the viscous part of cohesion is governed by the flow phenomenon of the clay particle structure and is considered as a rate process.

The model predicts that when a shear stress is imposed on the clay mineral network, contacts with the smallest resistance along the direction of shear slip first. Following these contact failures, particles are displaced from one position to another. The resistance against displacement is due to the viscous part of cohesion whose behavior follows the rate process. As deformation is taking place, stronger contacts are being reformed and the stresses causing flow are transferred to the reformed contacts. The maximum value of the angle of internal

friction at any strain is attained only if the elapsed time between displacements is sufficient for the reformed contacts to develop their maximum resistance to shear.

If the clay mineral network is stressed under a fast strain rate, the rate of contact displacement is fast and the elapsed time between displacements is short. The value of the angle of internal friction attained at any strain or at failure is therefore low. If the strain rate is slow, the rate of contact displacement is slow and the value of the angle of internal friction is high. If the strain rate is such that the elapsed time between displacements is sufficient for the reformed contacts to develop their maximum resistance, a maximum value of the angle of internal friction is attained at any given strain. The viscous part of cohesion decreases with decreasing strain rate.

Under a creep load, the displacements also lead to contacts with greater strengths. After these contacts are reformed, displacements cease. At any creep strain $\mathbb{F}_{\mathbf{C}}$, to attain a $\mathbb{F}_{\mathbf{C}}$, a displacement $\mathbb{N}_{\mathbf{C}}$ must take place. To reach $\mathbb{N}_{\mathbf{C}}$, a time $\mathbb{T}_{\mathbf{C}}$ is required. The angle of internal friction $\mathbb{F}_{\mathbf{C}}$ therefore increases with the creep time $\mathbb{T}_{\mathbf{C}}$ and approaches a maximum value $\mathbb{F}_{\mathbf{K}}$ when $\mathbb{T}_{\mathbf{C}}$ approaches a large value $\mathbb{T}_{\mathbf{K}}$. The viscous part of cohesion decreases with increasing creep time.

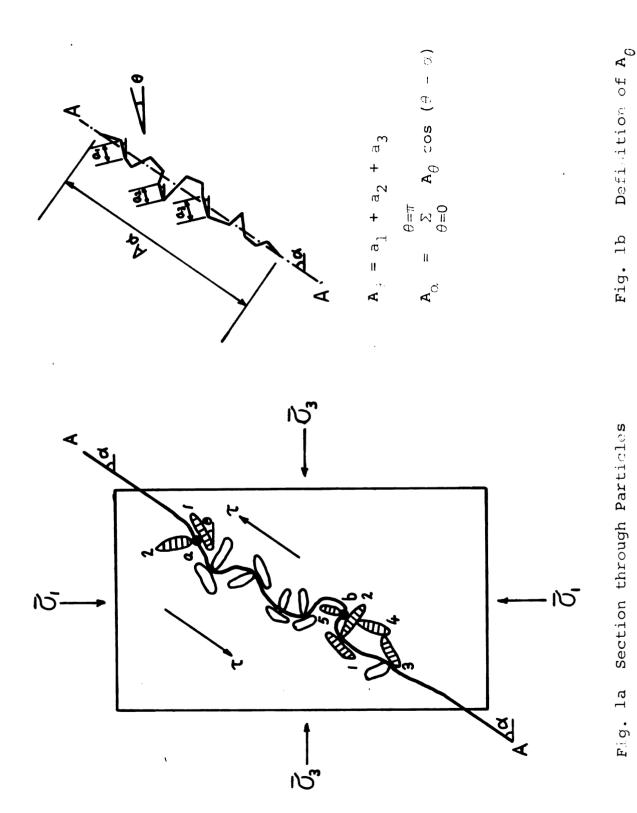
Approximate calculations with the particle model give results that are in reasonable agreement with the experimental data.

BIBLIOGRAPHY

- Bishop, A. W. and Henkel, K. J. (1957). The Measurement of Soil Properties in the Triaxial Test. Edward Arnold, London.
- Bowden, F. P. and Tabor, D. (1954). Friction and Lubrication of Solids. Clarendon Press, Oxford.
- Christensen, R. W. (1964). Analysis of Clay Deformation by Rate Process Theory. Ph.D. thesis, Michigan State University.
- Christensen, R. W. and Wu, T. H. (1964). Analysis of Clay Deformation by Rate Process Theory. Journ. Soil Mech. & Found., Proc. ASCE (forthcoming).
- Glasstone, S., Laidler, K and Eyring, H. (1941). The theory of Rate Processes. McGraw-Hill, New York.
- Holliday, F. J. (1963). Stress Transfer due to Creep in a Saturated Clay. M.S. thesis, Michigan State University.
- Horn, H. M. and Deere, D. U. (1952). Frictional Characteristics of Minerals. Geotechnique, Vol. 12, 4: pp. 319.
- Hvorslev, M. J. (1950). Physical Components of the Shear Strength of Saturated Clays, ASCE Research Conference on Shear Strength of Cohesive Soils, pp. 159.
- Lambe, T. W. (1960). A Mechanistic Picture of Shear Strength in Clays. ASCE Research Conference on Shear Strength of Cohesive Soils, pp. 555.
- Lo, K. Y. (1962). Shear Strength Properties of a Sample of Volcanic Material of the Valley of Mexico. Geotechnique, Vol. 12: 4: 303.
- Mitchell, J. K. (1956). The Fabrics of Natural Clays and its Relation to Engineering Properties. Proc. HWY. Res. Board, Vol. 35.
- Mitchell, J. K. (1964). Shearing Resistance of Soils as a Rate Process. Journ. Soil Mech. & Found., Proc. ASCE, Vol. 90, SMl, pp

- Murayama, S. and Shibrata, T. (1961). Rheological Properties of Clays. Proc. 5th Int. Conf. Soil Mech. & Found. Eng., Vol. 1: 269-273.
- Rosenqvist, I. Th. (1959). Mechanical Properties of Soil-Water System. Journ. Soil Mech. & Found., Proc. ASCE, Vol. 85, SM2, pp. 31.
- Schmertmann, J. H. and Osterberg, J. O. (1960). An Experimental study of the Development of Cohesion and Friction with Axial Strain in Saturated Cohesive Soils. ASCE Research Conf. on Shear Strength of Cohesive Soils, pp. 643.
- Schmertmann, J. H. (1962). Comparison of One and Two-Specimen CFS Tests. Journ. Soil Mech. & Found., Proc. ASCE, Vol. 88, SM6.
- Skempton, A. W. (1960). Effective Stress in Soils, Concrete and Rocks. Pore Pressure and Suction in Soils.

 Butterworths, London.
- Tan, T. K. (1957). Discussion, Proc. 4th Int. Conf. Soil Mech. & Found Eng., Vol. 3: pp. 87-89.
- Wu, T. H. Douglas, A. G. and Goughnour, R. D. (1962). Friction and Cohesion of Saturated Clays. Journ. Soil Mech. & Found., Proc. ASCE, Vol. 88, SM3, pp. 1-23.
- Wu, T. H. (1963). Discussion on "Comparison of One and Two-Specimen CFS Tests." Journ. Soil Mech. & Found., Proc. ASCE, Vol. 89, SM4, pp. 229.



Section through Particles F.g. la

Fig. 1b

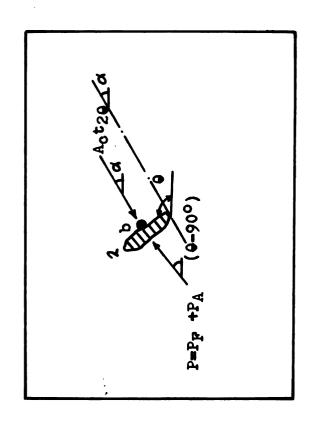


Fig. 3 Displacement Movement and Resistance

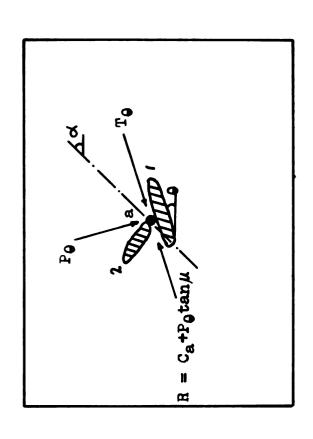
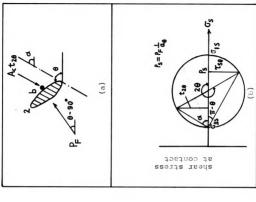


Fig. 2 Sliding Movement and Resistance



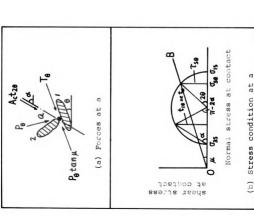


Fig. 4. F. Grional resistance ega..st sliding.

Prictional resistance against displacement.

Fig. 5.

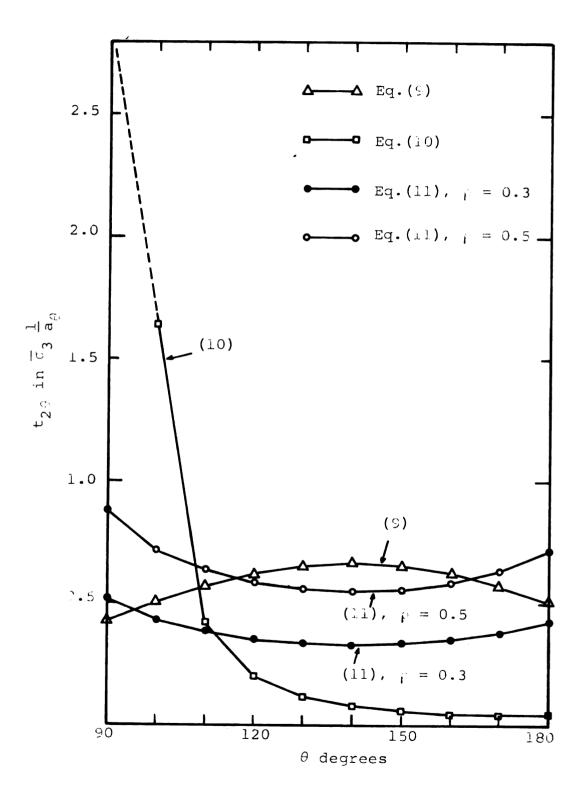


Fig. 5. Comparison of t₂₀.

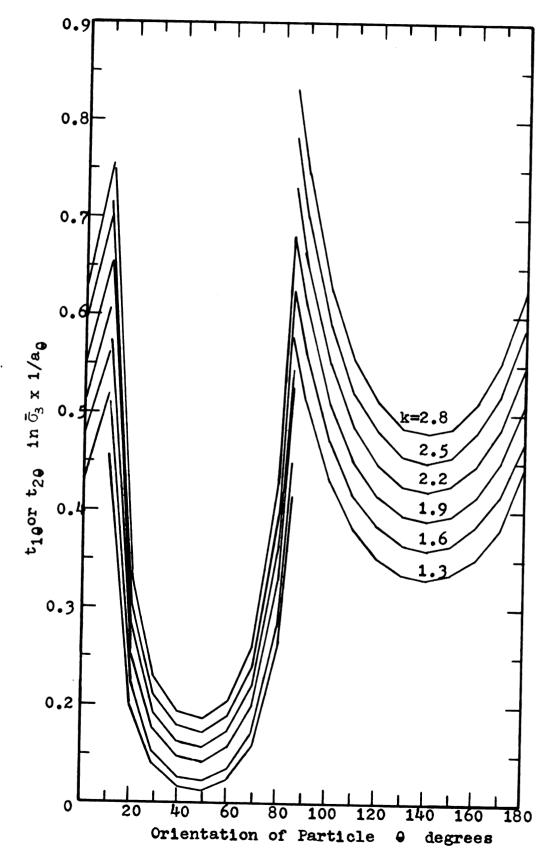


Fig. 7 Spectrum of Friction Resistance- $\rho = 0.3$

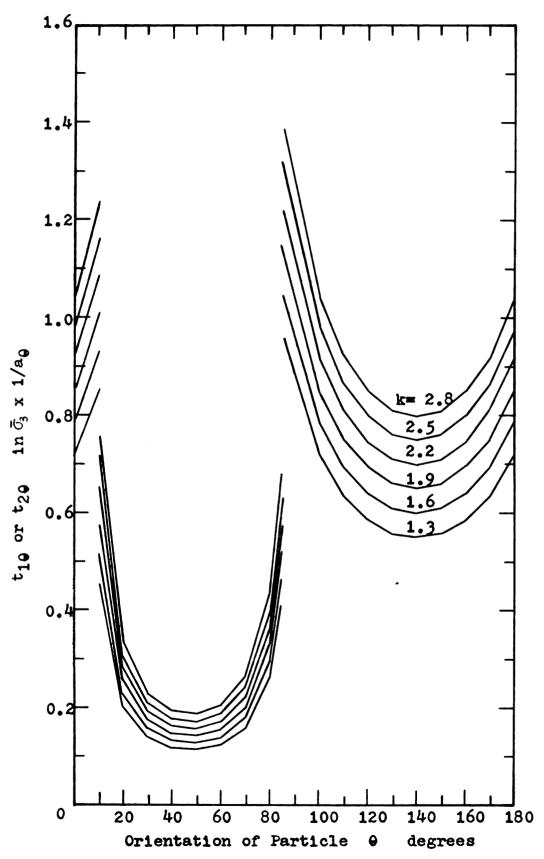
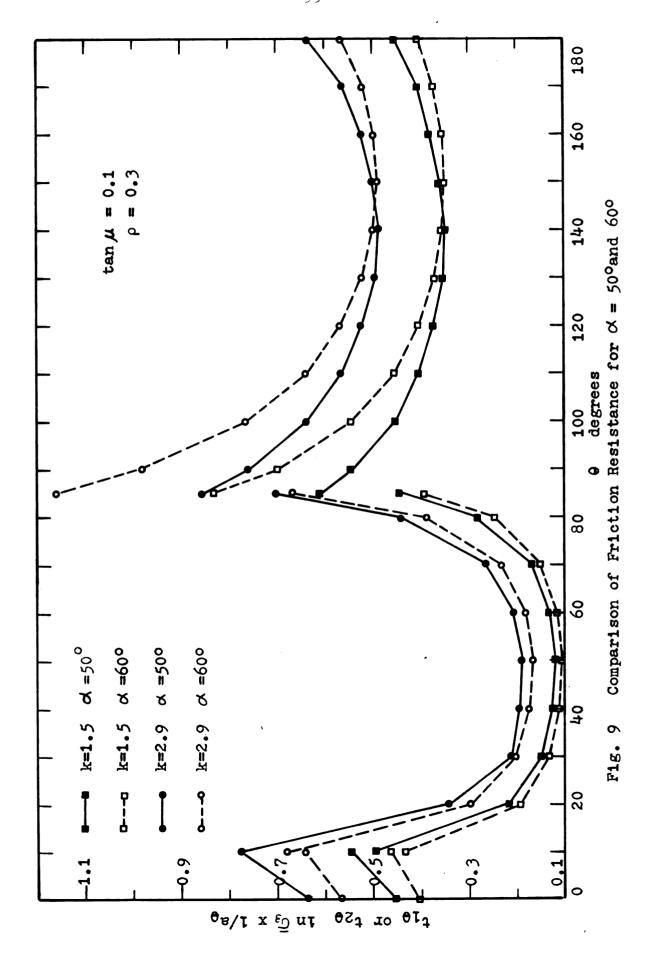


Fig. 8 Spectrum of Friction Resistance— $\rho = 0.5$



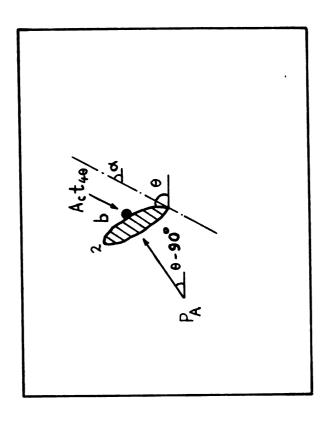


Fig. 11. Adnesional resistince against displacement.

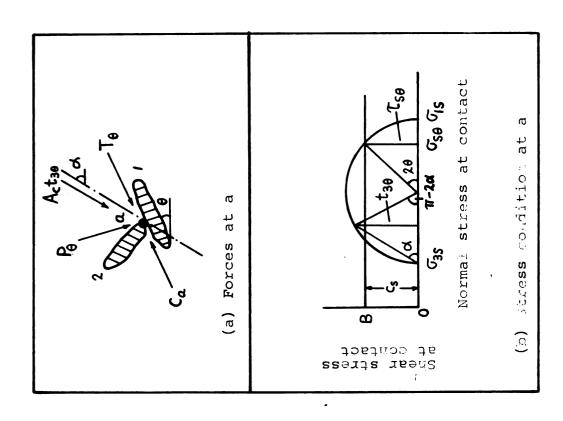


Fig. 10. Admssional nesistants against sliding.

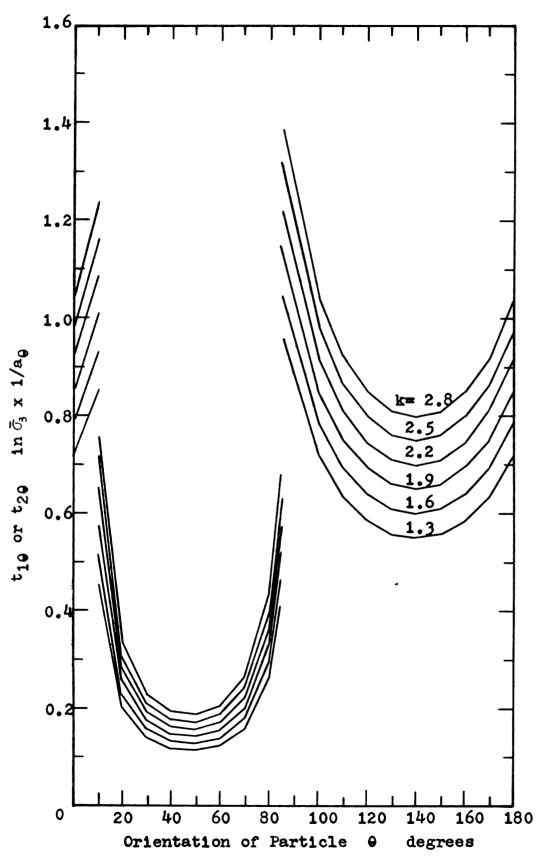
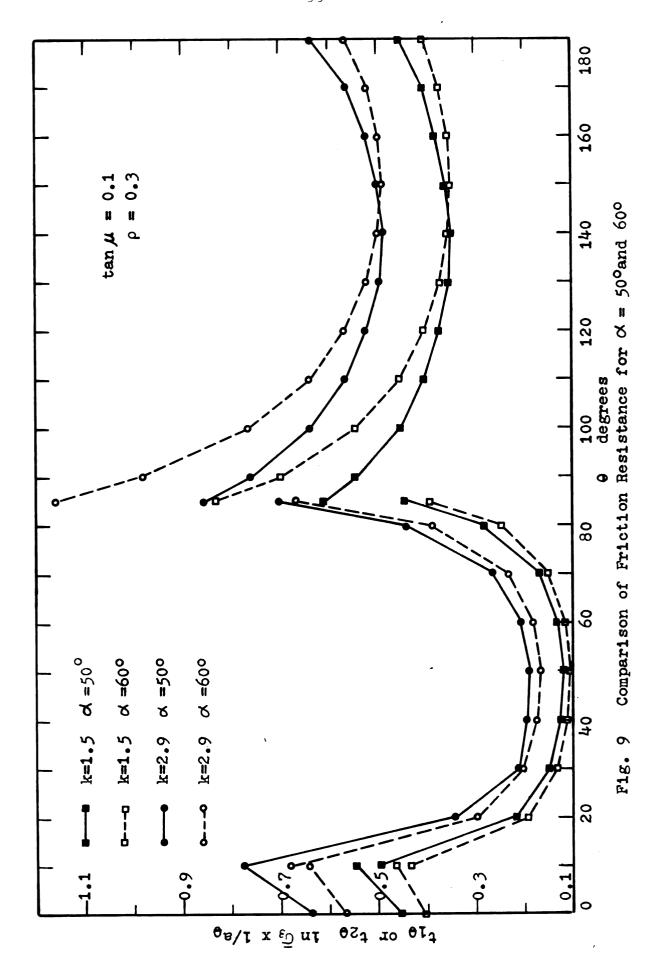


Fig. 8 Spectrum of Friction Resistance— $\rho = 0.5$



;	1
:	-
;	
i	
,	
, •	1
, !	
1	

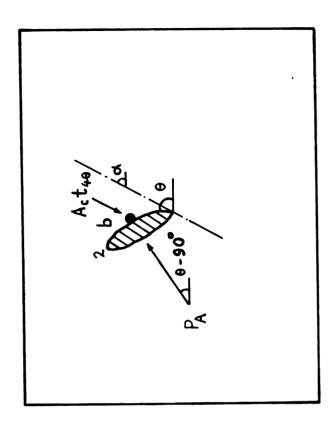


Fig. 11. Adhesional resisterce against displacement.

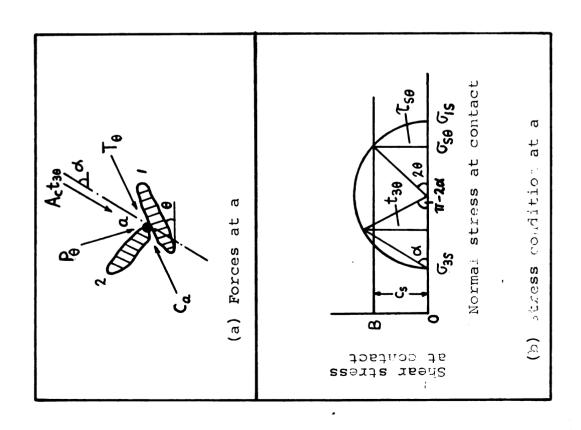


Fig. 10. Adhasional resistaros agailist stiding.

		· -

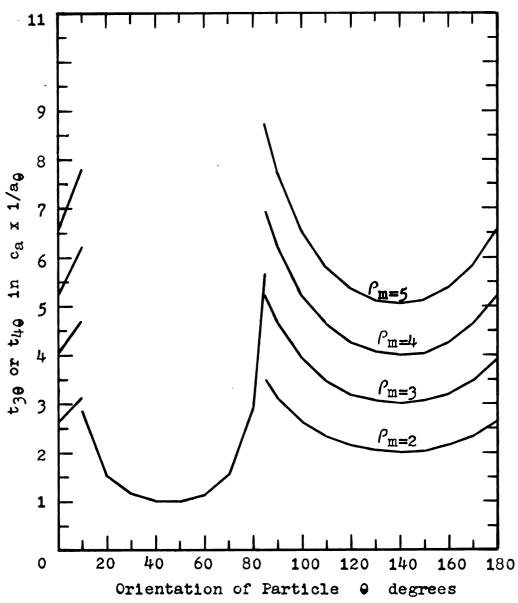
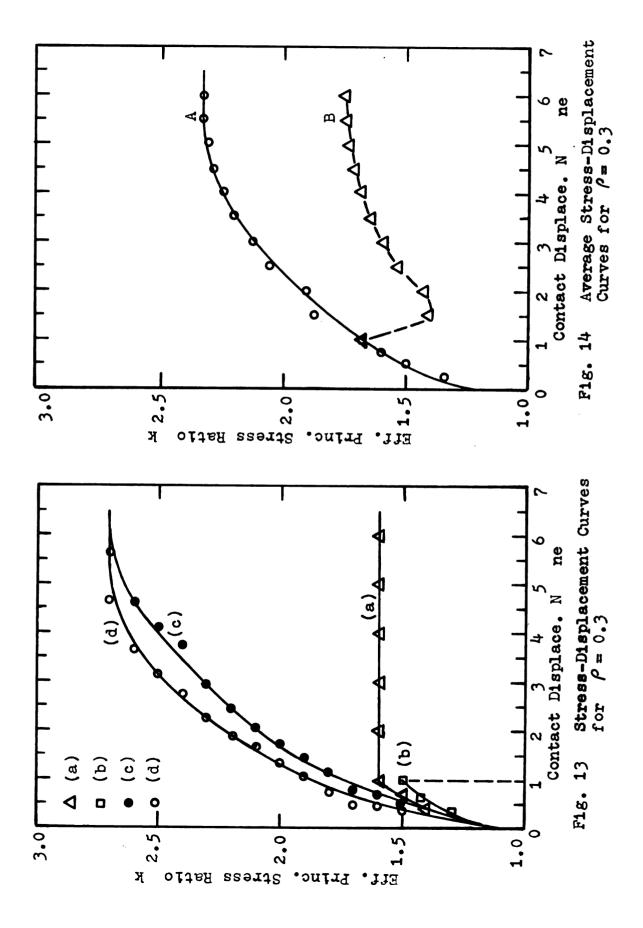
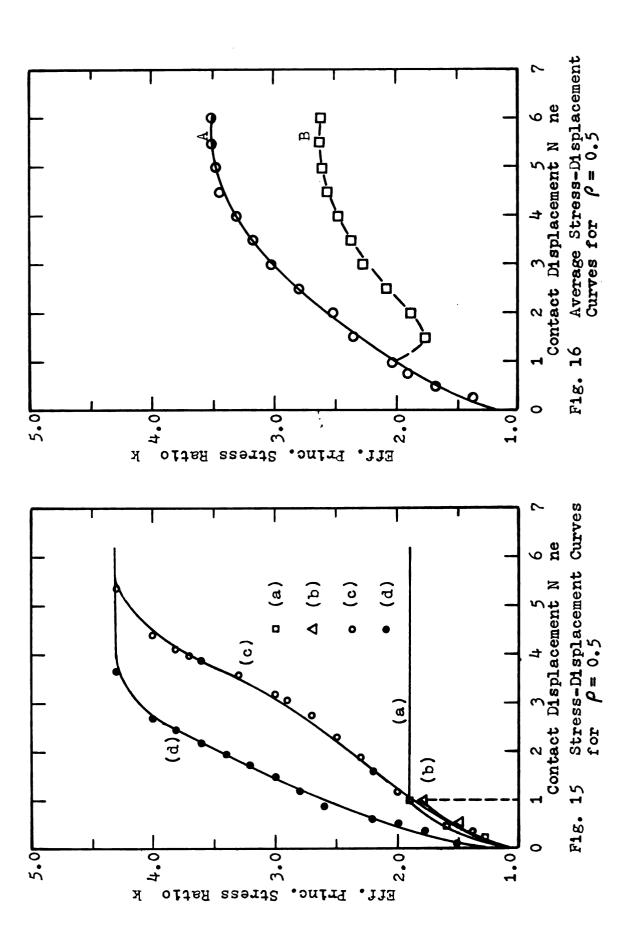
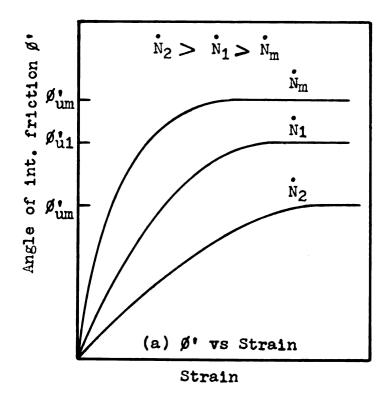


Fig. 12 Spectrum of Adhesion Resistance







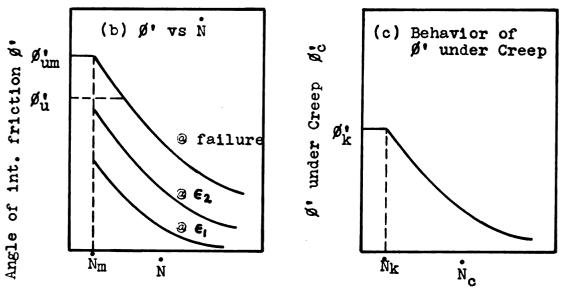


Fig. 17 Behavior of \emptyset according to Particle Model

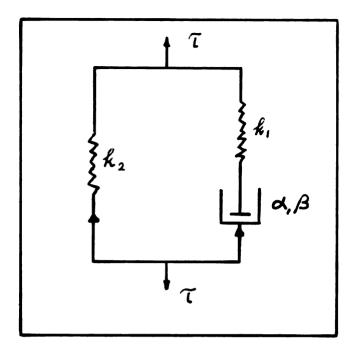
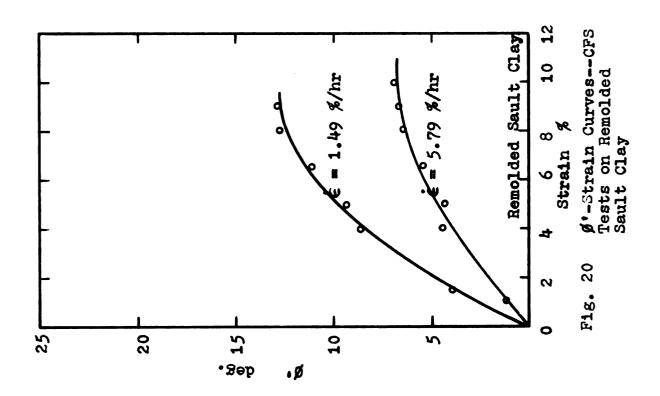
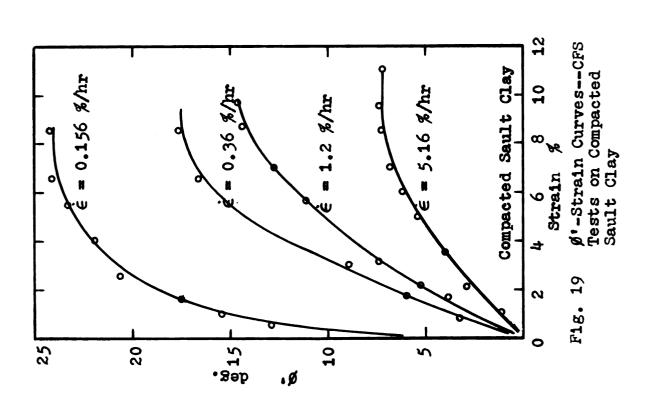
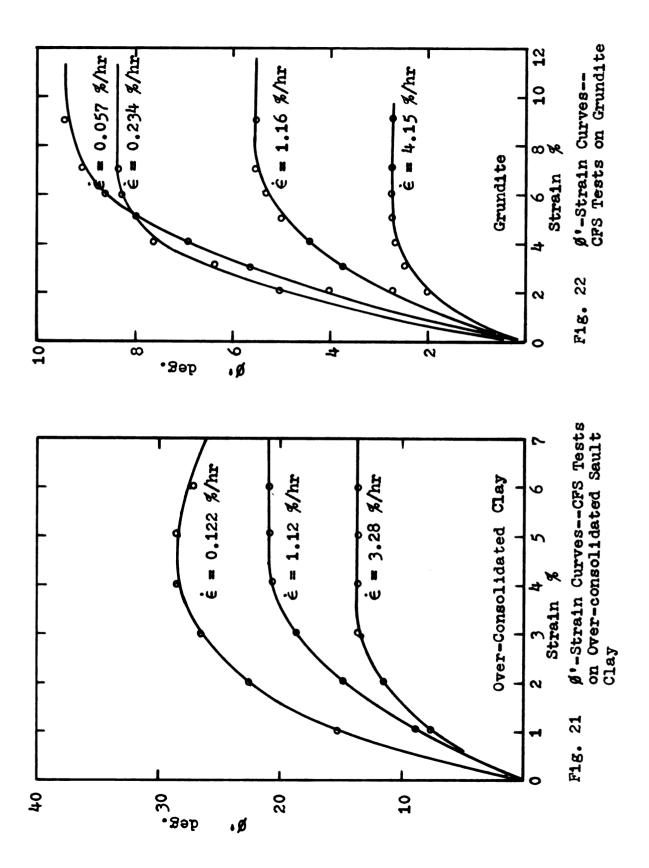


Fig. 18 Rheologic Model for Clay Particle Structure







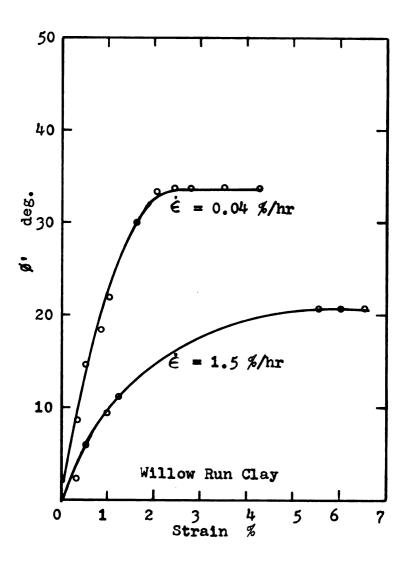
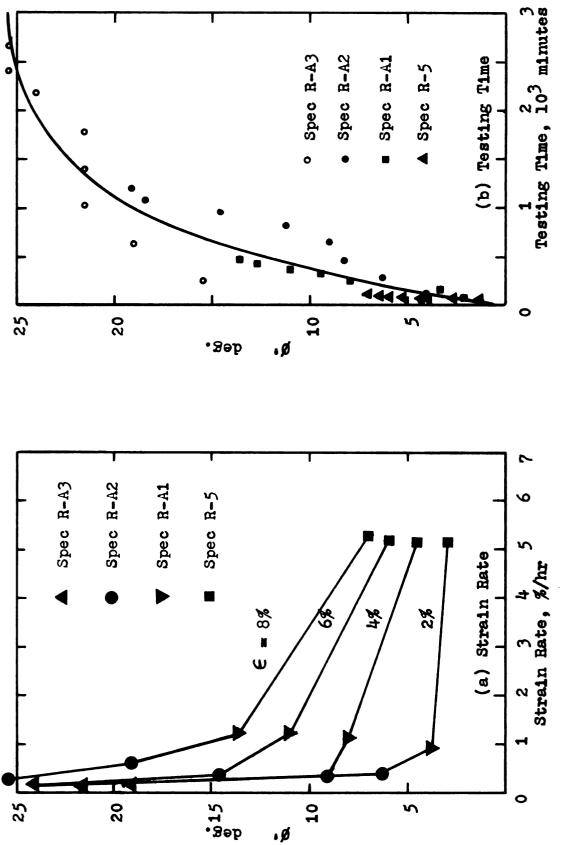
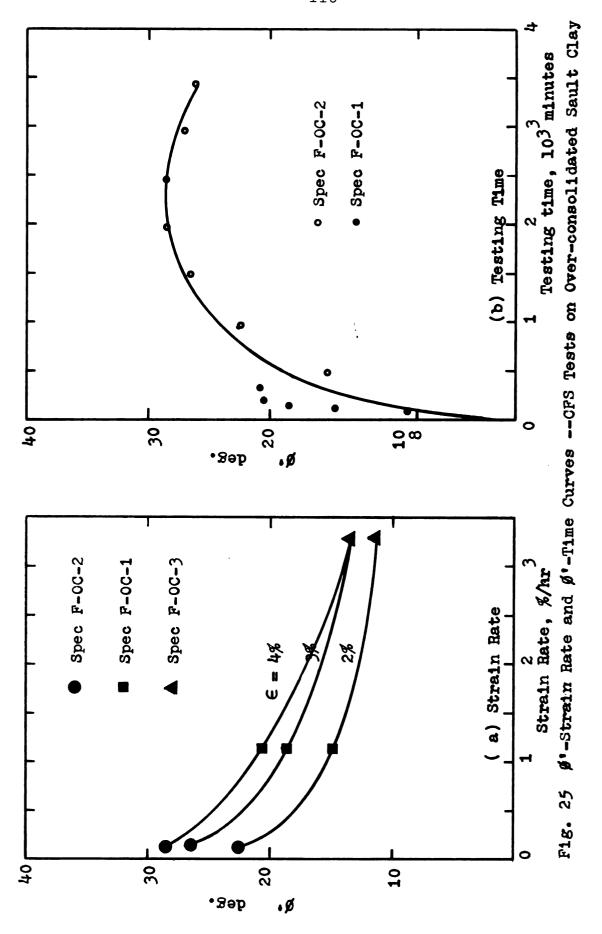
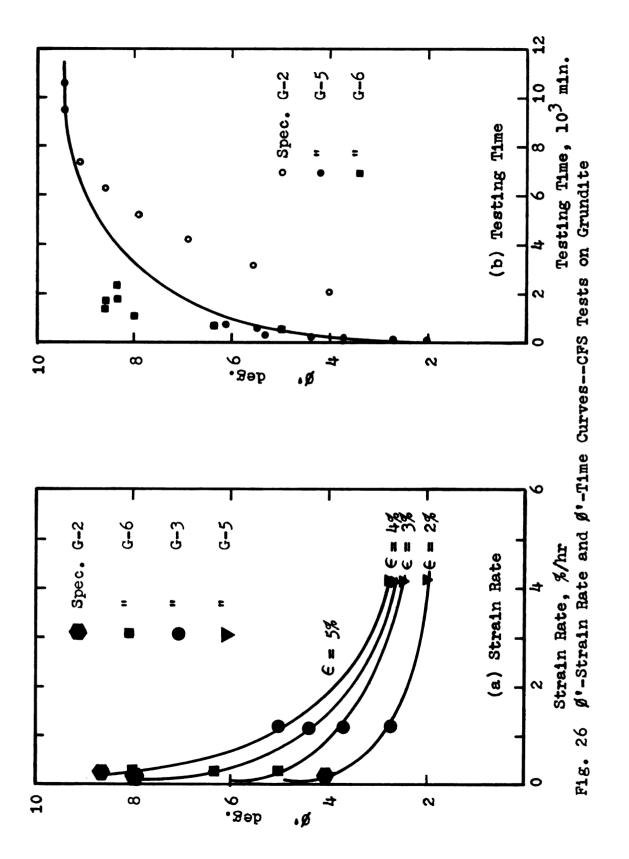


Fig. 23 Ø'-Strain Curves--CFS Tests on Undisturbed Willow Run Clay



\$ -- Strain Rate and \$ -- Time Curves -- CFS Tests on Compacted Sault Clay F18. 24





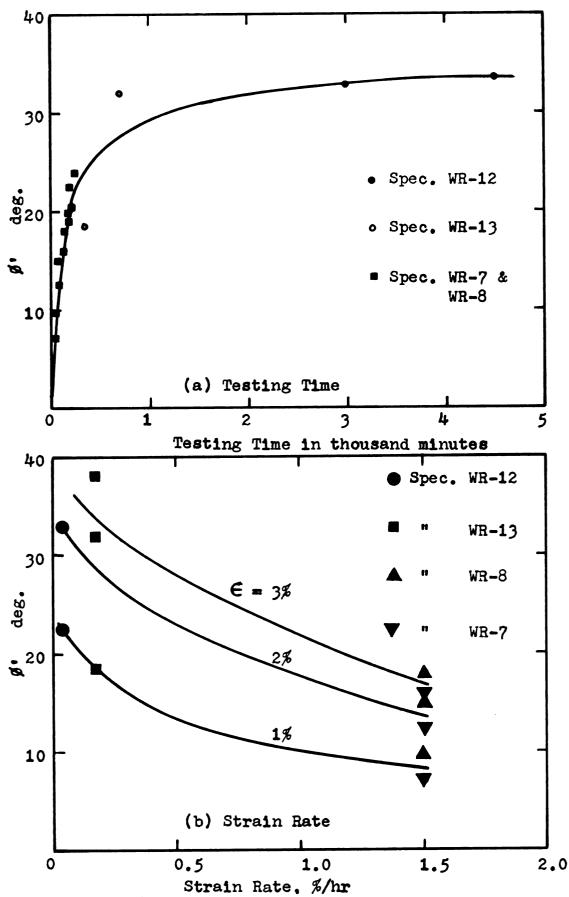
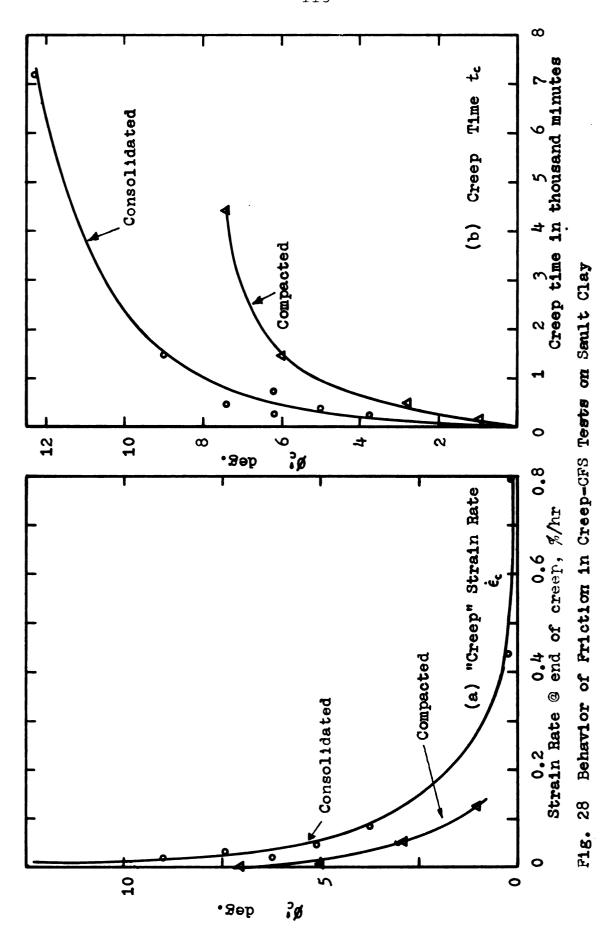
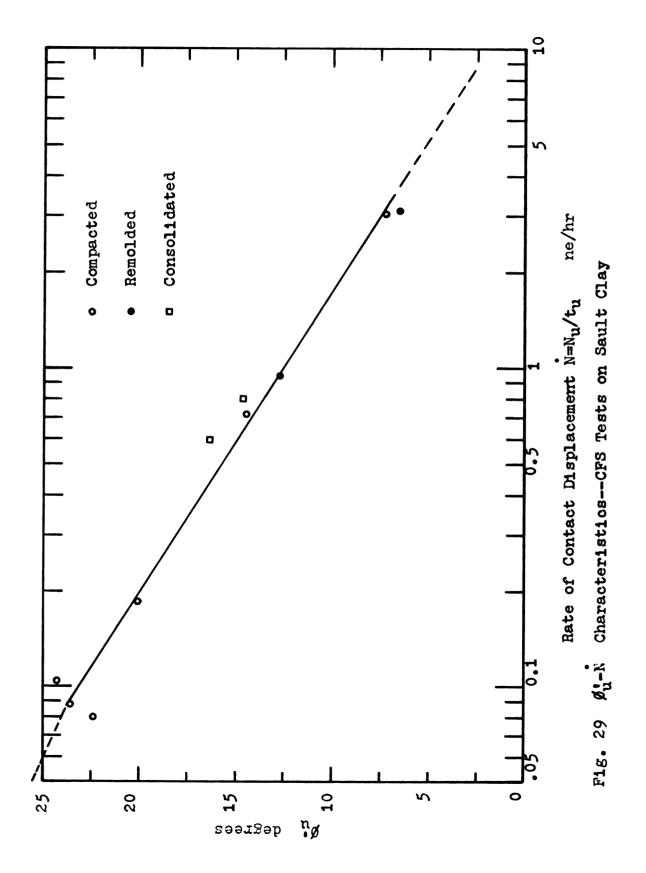
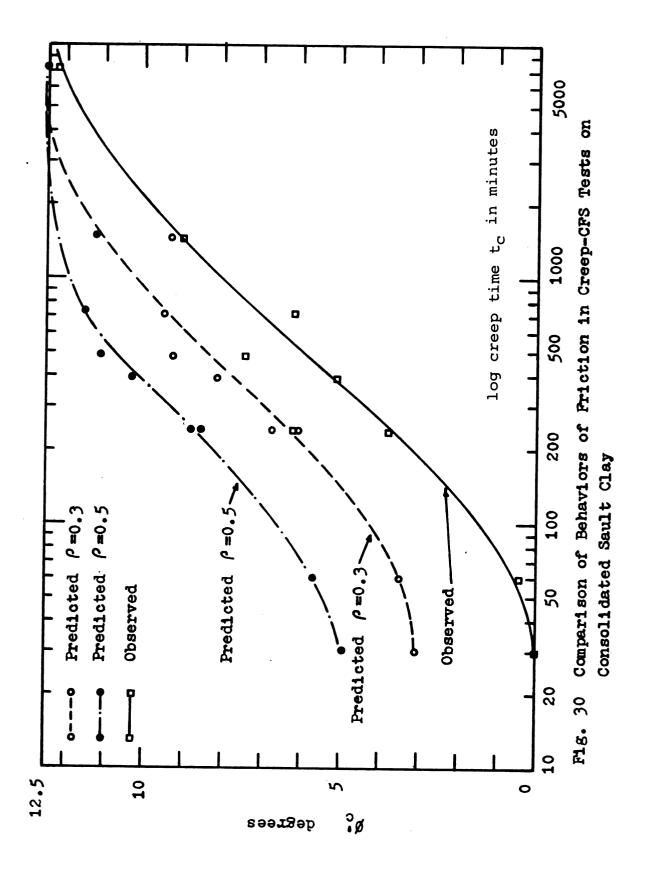
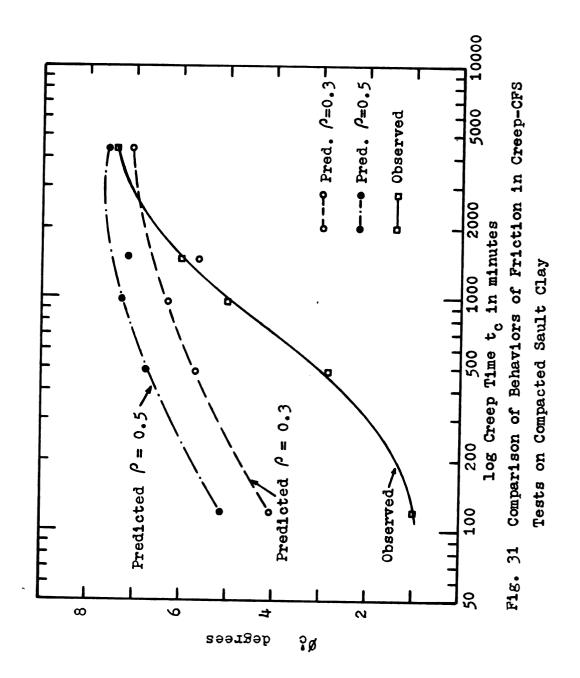


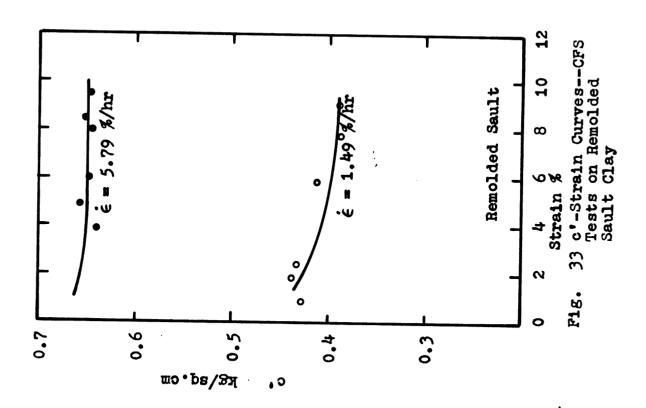
Fig. 27 Ø'-Time and Ø'-Strain Rate Curves--CFS Tests on Undisturbed Willow Run Clay

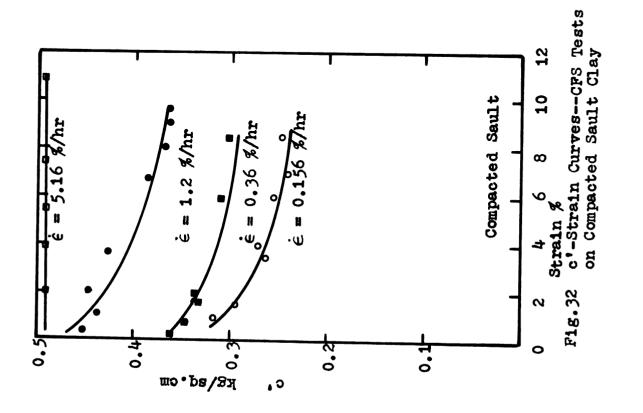


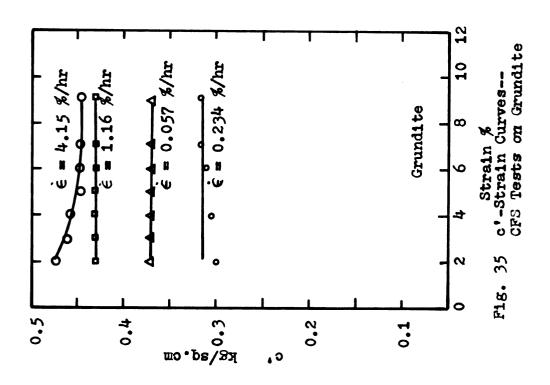


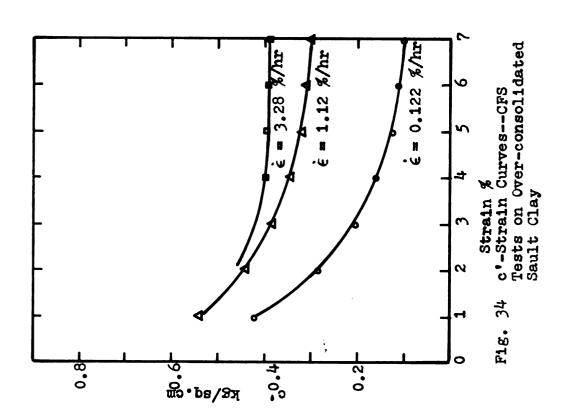












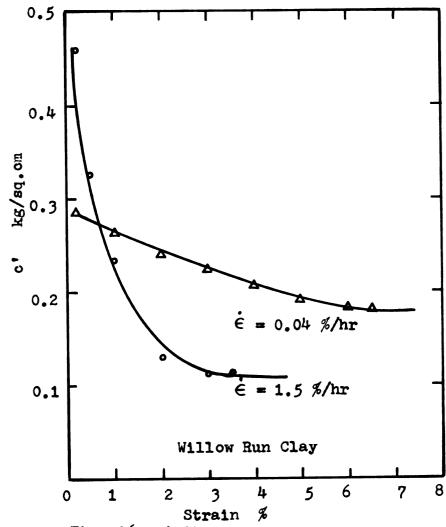
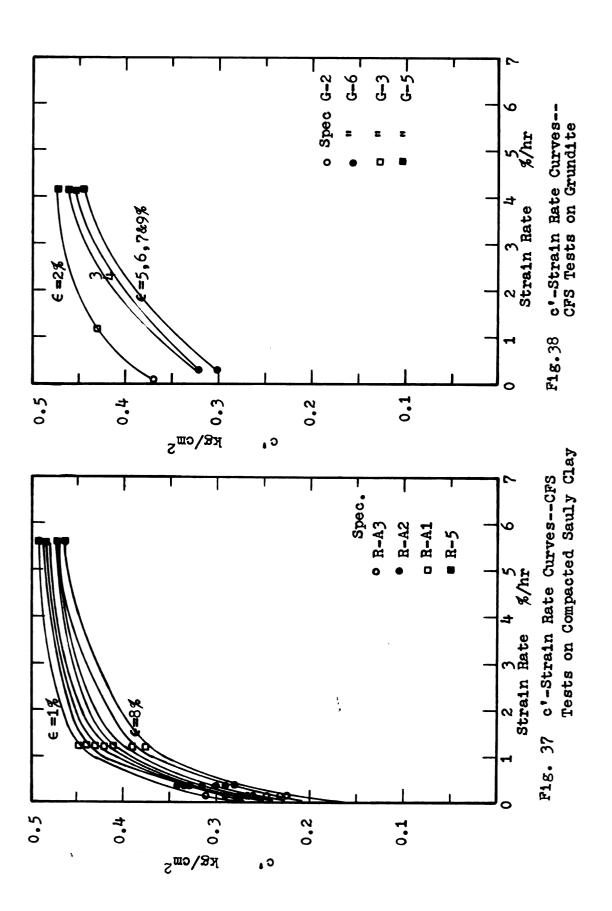
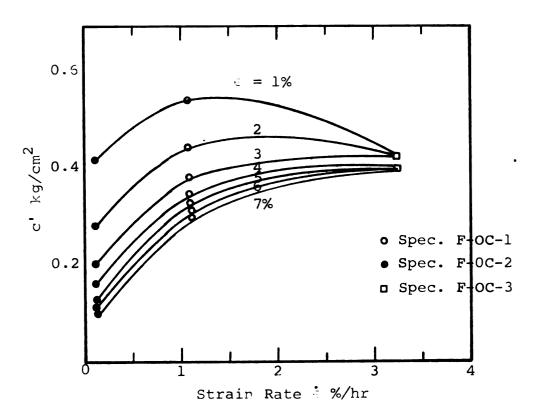


Fig. 36 c'-Strain Curves--CFS Tests on Undisturbed Willow Run Clay





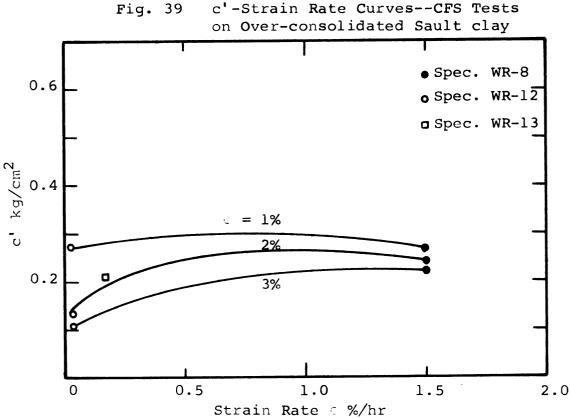
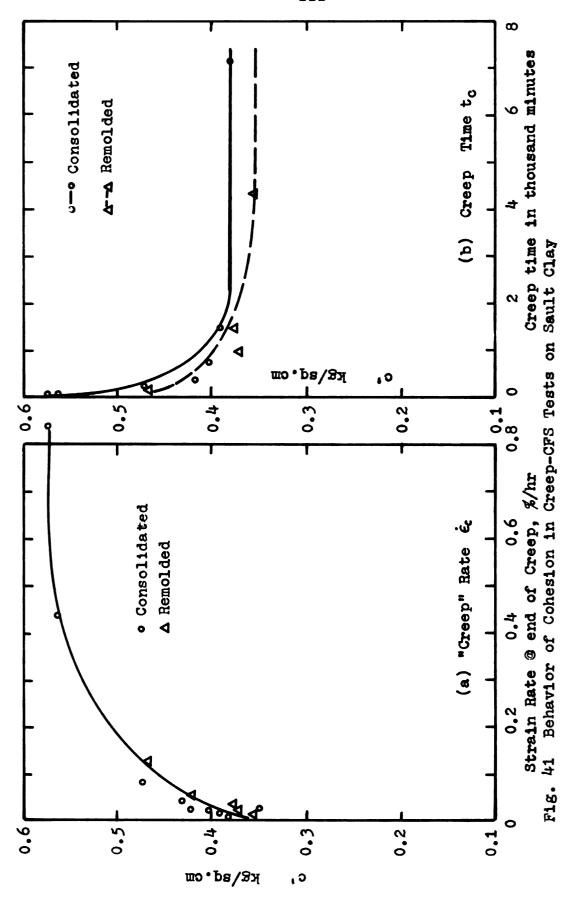
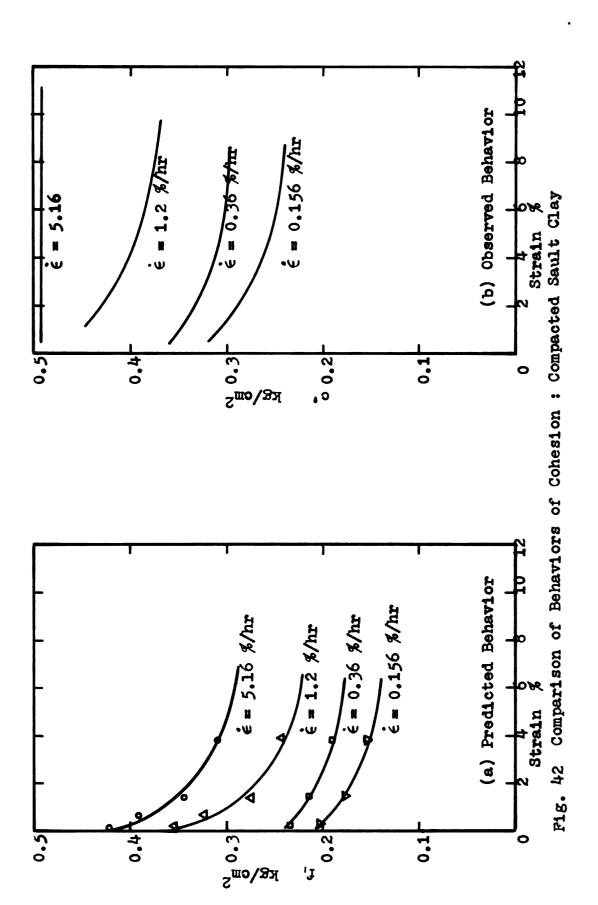
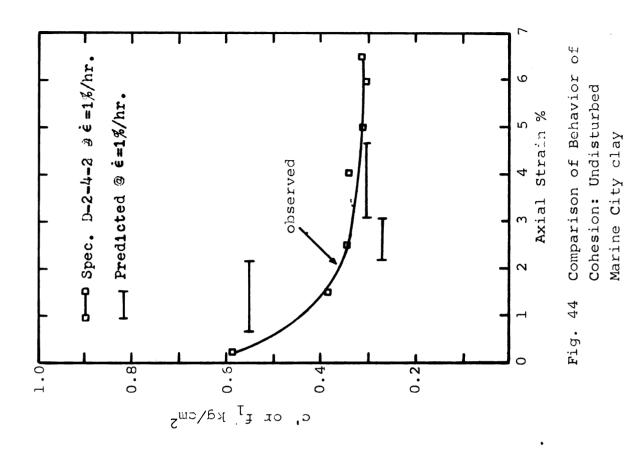
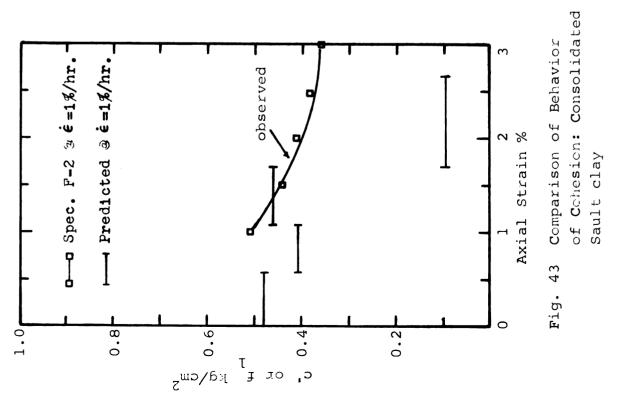


Fig. 40 c'-Strain Rate Curves -- CFS Tests on Undisturbed Willow Run Clay









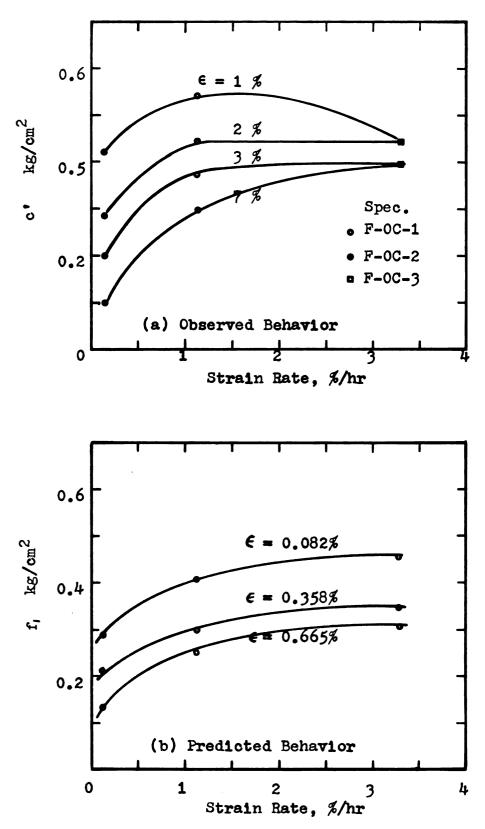
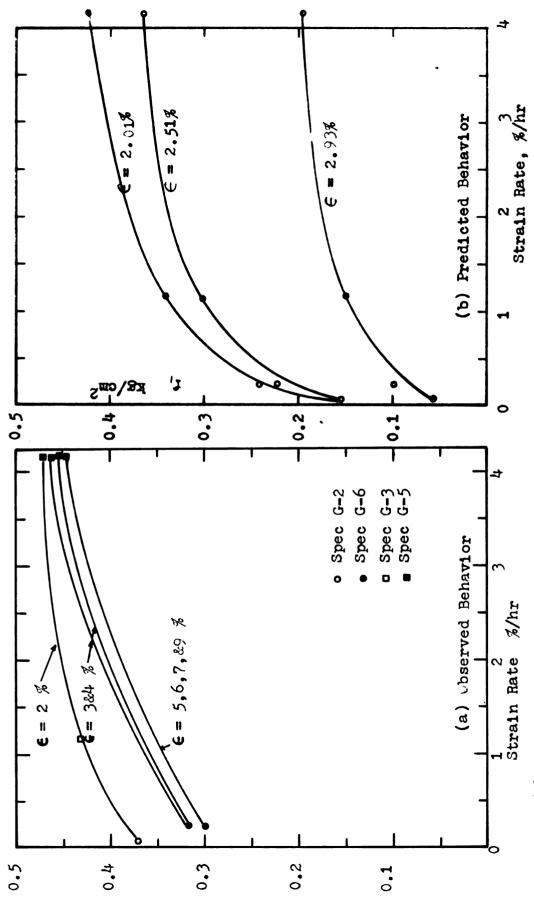
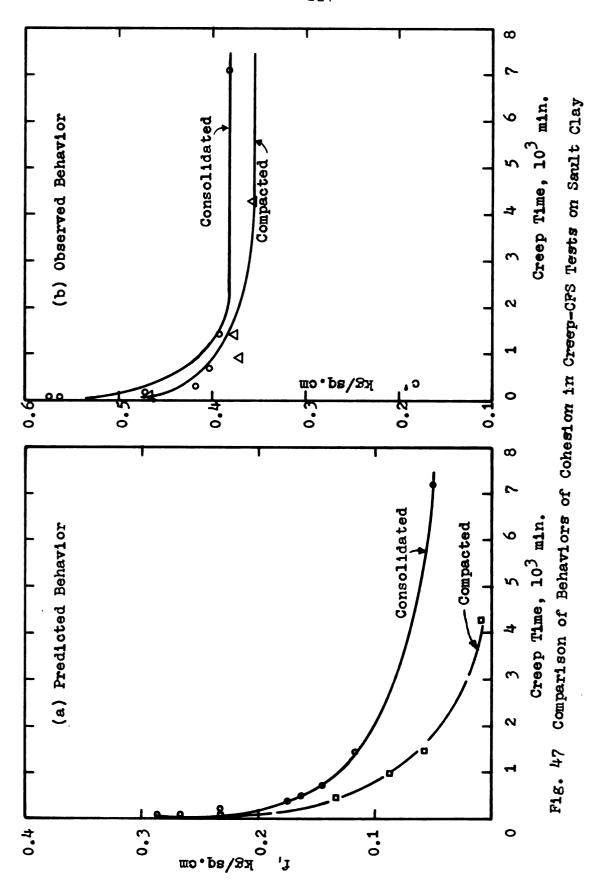
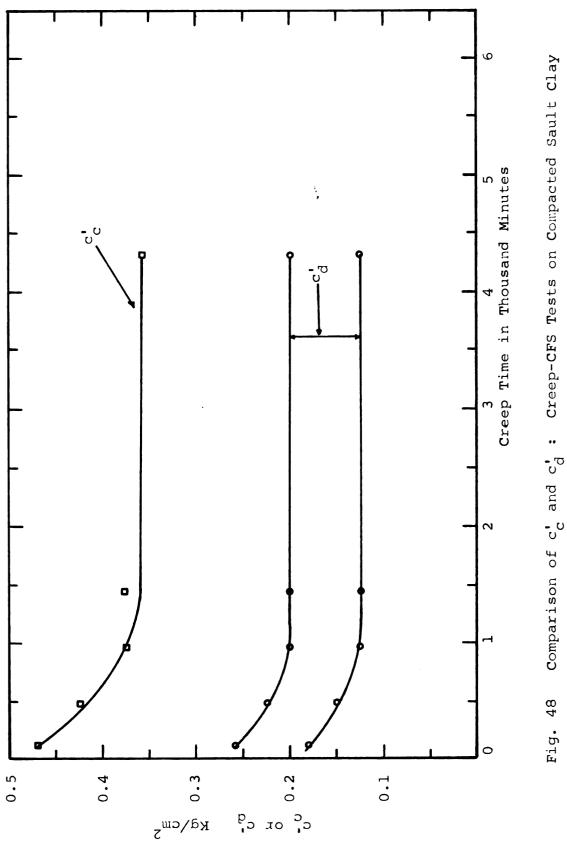


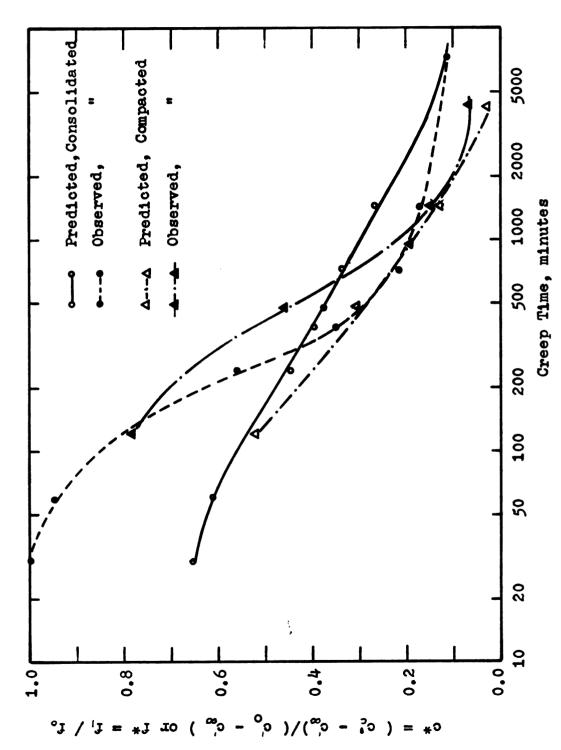
Fig. 45 Comparison of Behaviors of Cohesion: Over-consolidated Sault Clay



Comparison of Behaviors of Cohesion : Grundite F13. 46







Comparison of c* and f*: Creep-CFS Tests on Sault Clay Fig. 49

APPENDIX I

COMPUTATIONS OF CONTACT RESISTANCE

Table A-1. Comparison of $t_{2\theta}$

θ		t_{2} in $\overline{z}_3 \frac{1}{a_{}}$				
deg.	Equation (9)	Equation (10)	Equation (11) ρ = 0.3	Equation (11) $\rho = 0.5$		
90	0.423		0.513	0.855		
100	0.505	1.542	0.430	0.718		
110	0.571	0.420	0.381	0.635		
120	0.618	0.197	0.351	0.585		
130	0.549	0.119	0.336	0.559		
140	0.660	0.083	0.330	0.550		
150	0.649	0.066	0.336	0.559		
160	0.518	0.056	0.351	0.585		
170	0.571	0.051	0.381	0.635		
180	0.505	0.049	0.430	0.718		

Frictional Resistance against Sliding t19 Table A-2

	•	2.0	22211112235 111112236 11644820 12610
*	ರ	1.9	002230 0020 0020 0020 0020 0020 0020 0020 0020 0020 0020 0020 0020 0020 0020 0020 00
tan µ	tap A co	1.8	3355 3355
	n 20 -	1.7	2360 113240 113250 13325 13325 13325 1355 1355 1355 1355 1
)) (24	31	1.6	25 20 20 20 20 20 20 20 20 20 20 20 20 20
where		1.5	2000 2000 2000 2000 2000 2000 2000 200
20 x F		1.4	2860 12860 12860 12860 12860 12860 12860 12860 12860 12860 12860 12860 18960 18960 1
<u>G</u> (1+k) }stn		1.3	2565 2565 2565 2565 2565 2565 2565 2565
t,p= 1/2(1		1.2	2620 11950 11330 11195 11180 1180 12520 4070
·		k= 1.1	2380 2380 11268 11168 11168 11168 12380 3760 12380
	Ē4		4035 11225 11033 11033 1204 1300 2300 3640
	•	dog	8872655555555555555555555555555555555555

Frictional Resistance against Sliding Table A-2(cont'd)

	× - q		
	cos 20	2.9	7.3.6.4.4.4.4.4.4.4.4.4.4.4.4.4.4.4.4.4.4
tan μ	- tan u	2.8	74 64 64 64 64 64 64 64 64 64 64 64 64 64
	sin 20	2.7	64 64 64 64 64 64 64 64 64 64 64 64 64 6
11 Cr.	4	2.6	74 6 6 6 6 6 6 6 6 6 6 6 6 6 6 6 6 6 6 6
자 Whe Y		2.5	74 74 74 74 74 74 74 74 74 74 74 74 74 7
8	3	2.4	64 44 44 44 44 44 44 44 44 44 44 44 44 4
<u>G</u> (1+k)≩stn	28/414/5	2.3	60000000000000000000000000000000000000
l n		2.2	66000000000000000000000000000000000000
		k= 2.1	60000000000000000000000000000000000000
	Ŀ		2420 112420 112420 112090 112090 12090 12090 12090 12090
	•	deg	4444444466668 040404040404040

Frictional Resistance against Displacement Table A-3

 $\rho = 0.2$

£4.		t20= 2	$\frac{+ k}{3} \rho_{\vec{0}_3 x} F$	F where	11 PT	1/81n($1/\sin(\theta - \alpha)^{\kappa \frac{L}{d_{\theta}}}$)× <u>†</u>		
	K= 1.1	1.2	1.3	1.4	1.5	1.6	1.7	1.8	1.9	
1.305	.2700	.2780	.2870	.2960	.3045	.3130	.3220	.3300	.3390	
1.555	.3215	.3315	.3420	.3520	.3630	.3730	.3840	.3940	.4050	
1.740	.3600	.3710	.3830	.3940	0904.	.4180	.4300	0044.	.4530	
1.555	.3215	.3315	.3420	.3520	.3630	.3730	.3840	.3940	.4050	
1.305	.2700	.2780	.2870	.2960	.3045	.3130	.3220	.3300	.3390	-
1.155	.2390	.2460	.2540	.2620	.2700	.2770	.2850	.2920	.3000	
1.065	.2200	.2270	.2340	.2420	.2485	.2560	.2630	.2700	.2770	
1.050	.2100	.2160	.2235	.2300	.2370	.2440	.2505	.2570	.2670	
1.000	.2065	.2130	.2200	.2270	.2335	.2400	.2470	.2530	.2600	
1.015	.2100	.2160	.2235	.2300	.2370	0442.	.2505	.2570	.2640	
1.065	.2200	.2270	.2340	.2420	.2485	.2560	.2630	.2700	.2770	
1.155	.2390	.2460	.2540	.2620	.2700	.2770	.2850	.2920	.3000	
1.305	.2700	.2780	.2870	.2960	.3045	.3130	.3220	:3300	.3390	

Frictional Resistance against Displacement Table A-3(cont'd)

 $\rho = 0.2$

		t 2	t20= PO 2	+ K X F	where	íz,	= 1/sin(0 ·	- X) × L	. 66		
• 5	ር _ት	k= 2.0	2.1	2.2	2.3	7.2	2.5	2.6	2.7	2.8	2.9
0	1.305	.3480	.3570	.3660	.3740	.3820	.2920	0004.	.4100	0214.	.4250
10	1.555	.4150	.4250	.4350	0544.	.4550	0994*	.4770	.4870	4965	.5070
85	1.740	.4650	.4750	.4880	.5000	.5100	.5200	.5350	.5450	.5555	. 5690
06	1.555	.4150	.4250	.4350	.4450	.4550	.4660	.4770	.4870	\$964.	.5070
100	1.305	.3480	.3570	.3660	.3740	.3820	.3920	0007	.4100	02T4°	.4250
110	1.155	.3080	.3160	.3240	.3310	.3385	.3460	.3540	.3620	.3690	.3770
120	1.065	.2840	.2910	.2980	.3160	.3120	.3195	.3270	.3340	.3400	.3480
130	1.015	.2710	.2775	.2840	.2910	.2980	.3040	.3115	.3180	.3240	.3320
140	1.000	.2670	.2735	.2800	.2870	.2930	.3000	.3070	.3135	.3200	.3265
150	1.015	0175.	.2775	.2840	.2910	.2980	.3140	3115.	.3180	.3240	.3320
160	1.065	.2840	.2910	.2980	.3060	.3120	.3195	.3270	.3340	.3400	.3480
170	1.155	.3080	.3160	.3240	.3310	.3385	.3460	.3540	.3620	.3690	.3770
180	1.305	.3480	.3570	.3660	.3740	.3820	.3920	0007.	0014.	۰4170	.4250

t20 Frictional Resistance against Displacement Table A-4

			t20=PC	ρ _G (2+k)/3 x	3 ×	where F	= 1/sin(•	- × × × − × × × − × × − ×		
O	ር _ቀ	k= 1.1	1.2	1.3	1.4	1.5	1.6	1.7	1.8	1.9	2.0
0	1,305	0514°	0214*	.4300	011111	.4560	0024*	0684*	0564.	. 5090	.5220
10	1,555	.4820	0264°	.5130	.5280	.5450	0095•	9925.	.5910	0809°	.6220
20	2,000	.6200	0049*	.6600	.6800	0669°	.7200	.7400	.7600	.7800	.8000
80	2,000	.6200	0686*	.6600	.6800	0669°	.7200	0074.	.7600	.7800	.8000
85	1.740	.4500	.5560	.5750	. 5900	0019.	.6260	.6450	0099.	.6800	0960.
96	1.555	.4820	.4965	.5130	.5280	.5450	.5600	.5750	.5900	• 6080	.6220
100	1,305	.4050	0214*	.4300	0444.	.4560	.4700	.4830	.4950	.5090	. 5220
110	1,155	.3580	0696.	.3810	.3930	0504.	.4150	.4280	.4380	.4500	.4620
120	1.065	.3300	.3410	.3510	.3630	.3725	.3840	.3940	4050	.4150	.4260
130	1.015	.3150	0426.	.3355	.3450	.3560	.3660	.3760	.3860	.3960	0904
140	1,000	0016.	.3200	.3300	3405	.3500	0096.	0026.	.3800	.3900	0004*
150	1.015	.3150	0426.	.3355	.3450	.3560	.3660	.3760	.3860	.3960	.4060
160	1.065	.3300	.3410	.3510	.3630	.3725	.3840	0466.	.4050	.4150	.4260
170	1,155	.3580	0696	.3810	.3930	0504°	.4150	.4280	.4380	.4500	.4620
180	1.305	.4050	.4170	.4300	0444.	.4560	.4700	.4830	0564.	. 5090	. 5220

Table A-4(cont'd)

			t20= P	PO3 (2+k)/3 x F	/3 x F	where	F = 1/81	where F = 1/sin(θ - \varnothing)× $\frac{1}{\Delta_{\theta}}$	ام مام مام		
•	દ ્ય	1	0	6	-		2 6	0	۵	0 0	
689		k= 2.1	2.2	2.3	۲•4	۲•۶	0.2	7.7	٥•٧	۲.,۶	
0	1,305	.5350	• 5500	.5610	.5730	.5890	0009.	.6150	.6250	.6380	
10	1.555	.6380	.6530	.6680	.6830	.7000	.7160	.7310	.7450	0092•	
20	2,000	.8200	.8400	.8600	.8800	0006	.9200	0076	0096•	.9800	
80	2,000	.8200	0048	.8600	.8800	0006.	.9200	.9400	0096	.9800	
85	1.740	.7130	.7310	.7500	.7650	.7830	.8020	.8170	.8330	.8540	
06	1.555	.6370	.6520	.6680	.6820	. 7000	.7150	.7300	.7450	.7600	
100	1.305	.5350	.5500	.5610	.5730	. 5890	0009	.6150	.6250	.6380	
110	1.155	.4750	.4850	0964*	.5085	. 5200	.5310	.5435	.5540	.5650	
120	1,065	.4360	0244.	.4600	.4680	.4800	0064.	.5010	.5100	.5220	
130	1.015	.4165	.4260	.4360	0244	.4560	0494	.4270	.4850	.4980	
140	1,000	·4100	.4200	.4300	0044.	.4500	0094.	4200	.4800	0064.	
150	1.010	.4165	.4260	.4360	0244	.4560	0294	.4770	.4850	.4980	
160	1.065	.4360	0244.	0094.	0894.	0084.	0064.	.5010	.5100	. 5220	
170	1.155	.4750	.4850	.4965	.5085	. 5200	.5310	.5435	.5540	. 5650	
180	1.305	.5350	.5500	.5610	.5730	.5890	•6000	.6150	.6250	.6380	

Frictional Resistance against Displacement t20 Table A-5

			t20=/	$t_{2\theta} = \rho \bar{O}_3(2+k)/3 x$	/3 x F	where	F = 1/sin	•	- \(\) x \(\) = \(\) a \(\) a \(\)	. •	
• 5	፲	k= 1.1	1.2	1.3	1.4	1.5	1.6	1.7	1.8	1.9	2.0
o	1.305	.5400	.5560	.5740	.5920	0609°	.6260	0079*	0099*	.6780	0969.
10	1.555	.6430	0699°	0489*	.7040	.7260	0442	.7680	.7980	.8100	.8300
50	2,000	.8280	.8520	.8800	.9080	.9320	0096•	.9860	1.013	1.040	1.068
80	2,000	.8280	.8520	.8800	.9080	.9320	0096•	.9860	1.013	1.040	1.068
85	1.740	.7200	.7420	.7660	.7880	.8120	.8360	.8600	.8800	0906•	.9300
06	1.555	.6430	.6630	0489	.7040	.7260	.7460	.7680	.7980	.8100	.8300
100	1.305	.5400	.5560	.5740	.5920	0609°	.6260	0449.	0099*	•6780	0969•
110	1,155	.4780	.4520	.5080	.5240	.5400	.5540	.5700	.5840	0009.	0919.
120	1.065	0044.	04540	.4680	0484°	0264*	.5120	. 860	.5400	.5540	. 5680
130	1,015	.4200	.4320	0244.	0094.	0424°	0884*	.5010	.5140	.5280	.5420
140	1.005	.4130	.4260	001/11	.4540	.4670	.4800	0464.	.5060	. 5200	.5340
150	1.015	.4200	.4320	0244.	0094*	04240	0884.	.5010	.5140	.5280	.5420
160	1.065	0077	.4540	.4680	0484°	.4970	.5120	.5260	.5400	.5540	. 5680
170 180	1.155	.5400	. 4930	.5085	. 5240	.5390	.5550	. 5700 . 6440	.5850	.6000	.6150

Table A-5(cont'd)

<u></u>		
90	2.9	.8500 1.308 1.308 1.138 1.138 6640 6640 66530 66530 8500
- × × - × s ₉	2.8	.8340 1.280 1.280 .9930 .8340 .6480 .6480 .6480 .8340
1/sin(0	2.7	8200 1.254 1.254 1.254 1.254 1.254 1.250 1.250 1.250 1.250 1.250 1.250 1.250 1.250
E4	2.6	.8000 1.228 1.228 0.070 .9540 .6230 .6230 .6230 .8000
where	2.5	1.220 1.2320 1.200
:)/3 x F	2.4	1.9640 1.9100 1.
P 5/(2+k)/3	2.3	1.874880 1.664880 2468000 24680000 2468000000000000000000000000000000000000
t28=	2.2	7.8. 6.8.7.2.7.2.2.2.2.2.2.2.2.2.2.2.2.2.2.2.2
	k= 2.1	
Çt.	•	446644444444444 64006464000046 64006464646 7000077777004670
a	deg	00000000000000000000000000000000000000

 $\rho = 0.5$ t20 Frictional Resistance against Displacement Table A-6

			t20= P	P 5₃(2+k)/3 x F	/3 x F	where 1	F = 1/s1	where $F = 1/\sin(\theta - \alpha) \times \frac{l}{\alpha_{\theta}}$	∠ × (∠) × (β		
o bap	ſ Σ 4	k= 1.1	1.2	1.3	1.4	1.5	1.6	1.7	1.8	1.9	2.0
0	1.305	.6750	0569°	.7180	.7400	.7610	.7840	.8050	.8250	0248*	.8700
10	1.555	.8040	.8290	.8550	.8800	.9075	.9320	0096•	.9850	1.012	1.038
20	2.000	1.035	1.0650	1.1000	1.1350	1.1650	1.2000	1.2300	1.2660	1,3000	1.335
80	2.000	2	8	:	8	E	8	:	E	8	=
85	1.740	0006.	,9280	.9570	.9850	1.0150	1.0450	1.075	1.1000	1.1300	1,162
100	1.305	.6750	0569.	.7180	0072.	.7610	.7840	.8050	.8250	.8470	.8700
110	1.155	.5970	.6150	.6350	.6550	.6750	0669.	.7130	.7300	.7500	.7700
120	1.065	.5500	.5680	.5850	.6050	.6210	0049.	.6580	.6750	.6930	.7100
130	1.015	.5250	.5400	.5590	.5750	.5930	0019.	.6250	.6420	0099*	.6780
140	1.000	.5160	.5320	.5500	.5680	.5840	0009.	.6180	.6320	.6500	0899*
150	1.0150	.5250	.5400	.5590	.5750	.5930	0019*	.6250	.6420	0099*	.6780
160	1.065	.5500	.5680	.5850	.6050	.6210	0049.	.6580	.6750	0669°	.7100
170	1.155	.5970	.6150	.6350	.6550	.6750	0669°	.7130	.7300	.7500	.7700
180	1.305	.6750	.6950	.7180	.7400	.7610	.7840	.8050	.8250	.8470	.8700

Table A-6(cont'd)

0 F k= 2.1 2.2 2.3 2.4 2.5 2.5 2.6 0 1.305 .8920 .9150 .9550 .9800 1.0000 10 1.555 1.0620 1.0880 1.1120 1.1580 1.1640 1.1990 20 2.000 1.3670 1.400 1.4350 1.470 1.500 1.5350 80 2.000 1.3670 1.400 1.4350 1.470 1.500 1.5350 100 1.000 1.3670 1.400 1.4350 1.470 1.5305 1.340 110 1.155 .7900 .8100 .8280 .8460 .8650 .7850 120 1.065 .7270 .7450 .7450 .7500 .7600 140 1.005 .6840 .7000 .7180 .7450 .7500 .7500 150 1.015 .7270 .7450 .7860 .8000 .8180 170 .7280 .7860 <t< th=""><th></th><th></th><th></th><th>t20= P</th><th>$\rho \overline{C}_3(2+k)/3 \times F$</th><th>/3 x F</th><th>where</th><th>where $F = 1/\sin(\varphi - \alpha) \times \frac{1}{\alpha\theta}$</th><th>- 0)u</th><th>X) x L</th><th></th><th></th></t<>				t20= P	$\rho \overline{C}_3(2+k)/3 \times F$	/3 x F	where	where $F = 1/\sin(\varphi - \alpha) \times \frac{1}{\alpha\theta}$	- 0)u	X) x L		
L= 2.1 2.2 2.4 2.5 2.5 1.36 1.365 1.305 1.305 1.0620 1.0880 1.1120 1.1380 1.1640 1.555 1.0620 1.0880 1.1120 1.1380 1.1640 1.3670 1.400 1.4350 1.470 1.5000 1.740 1.188 1.220 1.250 1.275 1.305 1.305 1.305 1.250 1.250 1.275 1.305 1.305 1.305 1.305 1.250 1.250 1.275 1.305 1.055 1.250 1.250 1.250 1.250 1.250 1.005 1.005 1.000 1.000 1.000 1.000 1.000 1.000 1.000 1.000 1.000 1.000 1.000 1.000 1.000 1.000 1.000 1.000 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.005 1.000 1.005 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000 1.005 1.000	٩	ß										
1.305 .8920 .9150 .9550 .9800 .9800 1.555 1.0620 1.0880 1.1120 1.1380 1.1640 2.000 1.3670 1.400 1.4350 1.470 1.5000 2.000 " " " " 1.740 1.188 1.220 1.250 1.275 1.305 1.305 .8920 .9150 .9550 .9800 1.155 .7900 .8100 .8280 .8460 .8650 1.015 .6940 .7100 .7280 .7450 .7600 1.015 .6940 .7100 .7280 .7450 .7600 1.015 .6940 .7100 .7280 .7450 .7600 1.015 .7270 .7450 .7600 .7600 1.065 .7270 .7450 .7600 .8000 1.065 .7270 .7450 .7650 .7650 .7600 1.065 .7270 .7450 .7650 .7650 .7600 .7600 1.065 .7270 .7450	, sap	4		2.2	2.3	7.2	2.5	2.6	2.7	2.8	5.9	
1.555 1.0620 1.0880 1.1120 1.1380 1.1640 2.000 1.3670 1.400 1.4350 1.470 1.5000 2.000 " " " " 1.740 1.188 1.220 1.250 1.275 1.305 1.305 .8920 .9150 .9550 .9800 1.305 1.155 .7900 .8100 .8280 .8460 .8650 1.015 .6940 .7100 .7280 .7500 1.015 .6940 .7100 .7280 .7450 1.015 .6940 .7100 .7280 .7450 1.015 .7270 .7450 .7800 .8000 1.015 .7270 .7800 .8000 1.065 .7270 .8280 .8460 .8000 1.155 .7900 .8100 .8280 .8460 .8650	0	1.305	.8920	.9150	.9350	.9550	.9800	1.000	1.025	1.042	1.062	
2.000 1.3670 1.400 1.4350 1.470 1.5000 2.000 " " " 1.740 1.188 1.220 1.250 1.275 1.305 1.305 .8920 .9150 .9350 .9800 1.305 1.155 .7900 .8100 .8280 .8460 .8650 1.065 .7270 .7450 .7600 .8000 1.015 .6940 .7100 .7280 .7450 .7500 1.015 .6940 .7100 .7280 .7450 .7600 1.015 .7270 .7450 .7450 .7600 1.015 .7270 .7450 .7600 .7600 1.015 .7270 .7450 .7600 .7600 1.015 .7270 .7450 .7600 .7600 1.065 .7270 .7650 .7600 .8000 1.065 .7270 .7650 .7650 .7650 .8600 1.155 .7900 .8100 .8280 .8460 .8650	10	1.555		1.0880		1.1380	1.1640	1.190	1.218	1.242	1.270	
2.000 " " " 1.740 1.188 1.220 1.275 1.305 1.305 .8920 .9150 .9550 .9800 1.155 .7900 .8100 .8280 .8460 .8650 1.065 .7270 .7450 .7800 .8000 1.015 .6940 .7100 .7280 .7500 1.000 .6840 .7100 .7280 .7500 1.015 .6940 .7100 .7280 .7450 .7500 1.015 .6940 .7100 .7280 .7450 .7500 1.015 .7270 .7450 .7800 .8000 1.065 .7270 .7850 .7800 .8000 1.155 .7900 .8100 .8280 .8460 .8650	20	2.000	1.3670	1.400	1.4350	1.470	1.5000	1.5350	1.5700 1.600	1.600	1.635	
1.740 1.188 1.220 1.255 1.275 1.305 1.305 .8920 .9150 .9550 .9800 1.155 .7900 .8100 .8280 .8460 .8650 1.065 .7270 .7450 .7650 .7800 .8000 1.015 .6940 .7100 .7280 .7450 .7500 1.015 .6940 .7100 .7280 .7450 .7600 1.015 .7270 .7450 .7800 .8000 1.065 .7270 .7450 .7800 .8000 1.065 .7270 .8280 .8460 .8650	80	2.000	8	:	:	8	8	•	=	2	=	
1.305 .8920 .9150 .9550 .9800 1.155 .7900 .8100 .8280 .8460 .8650 1.065 .7270 .7450 .7800 .8000 1.015 .6940 .7100 .7280 .7450 .7500 1.000 .6840 .7100 .7180 .7500 .7500 1.015 .6940 .7100 .7280 .7450 .7600 1.065 .7270 .7450 .7800 .8000 1.155 .7900 .8100 .8280 .8460 .8650	85	1.740	1.188	1,220	1,250	1.275		1.340	1,360	1,388	1.420	
1.155 .7900 .8100 .8280 .8460 .8650 1.065 .7270 .7450 .7800 .8000 1.015 .6940 .7100 .7280 .7450 .7600 1.000 .6840 .7000 .7180 .7320 .7500 1.015 .6940 .7100 .7280 .7450 .7600 1.065 .7270 .7450 .7800 .8000 1.155 .7900 .8100 .8280 .8460 .8650	100	1.305	.8920	.9150	.9350	.9550	.9800	1.000	1.025	1.042	1.062	
1.065 .7270 .7450 .7650 .7800 .8000 1.015 .6940 .7100 .7280 .7450 .7600 1.000 .6840 .7000 .7180 .7320 .7500 1.015 .6940 .7100 .7280 .7450 .7600 1.065 .7270 .7450 .7800 .8000 1.155 .7900 .8100 .8280 .8460 .8650	110	1,155	.7900	.8100	.8280	.8460	.8650	.8850	.9050	.9220	0646.	
1.015 .6940 .7100 .7280 .7450 .7600 1.000 .6840 .7000 .7180 .7320 .7500 1.015 .6940 .7100 .7280 .7450 .7600 1.065 .7270 .7450 .7800 .8000 1.155 .7900 .8100 .8280 .8460 .8650	120	1.065	.7270	.7450	.7650	.7800	.8000	.8180	.8350	.8500	.8700	
1.000 .6840 .7000 .7180 .7320 .7500 1.015 .6940 .7100 .7280 .7450 .7600 1.065 .7270 .7450 .7800 .8000 1.155 .7900 .8100 .8280 .8460 .8650	130	1.015	0769.	.7100	.7280	.7450	.7600	.7790	.7950	.8100	.8300	
1.015 .6940 .7100 .7280 .7450 .7650 1.065 .7270 .7450 .7800 .8000 1.155 .7900 .8100 .8280 .8460 .8650	140	1.000	0489.	. 7000	.7180	.7320	.7500	.7680	.7840	.8000	.8160	
1.065 .7270 .7450 .7650 .7800 .8000 .1.155 .7900 .8100 .8280 .8460 .8650	150	1.015	0469.	.7100	.7280	.7450	.7600	.7790	.7950	.8100	.8300	
1.155 .7900 .8100 .8280 .8460 .8650	160	1.065	.7270	.7450	.7650	.7800	.8000	.8180	.8350	.8500	.8700	
	170	1,155	.7900	.8100	.8280	0948°.	.8650	.8850	.9050	.9220	.9430	
180 1.305 .8920 .9150 .9350 .9550 1.0000	180	1.305	.8920	.9150	.9350	.9550	.9800	1.0000	1.0250 1.0420	1.0420	1.0620	

Table A-6(cont'd)

			t20= /	P G3(2+k)/3)/3 x F	where	F = 1/sin(•	- 0x)×1/2		
O Sp	ſĿ _i	k= 3.0	3.1	3.2	3.3	3.4	3.5	3.6	3.7	3.8	0.4
0	1.305	1.090	1,107	1.130	1,152	1.174	1.198	1.218	1.240	1,260	1.305
10	1.555	1.295	1.320	1.348	1.372	1.400	1.427	1.450	1.475	1.502	1.555
20	2,000	1.667	1.700	1.730	1.768	1.800	1.835	1.867	1.900	1.930	2.000
80	2,000	=	=	=	=	•	*	2	=	E	=
85	1.740	1.450	1.478	1.510	1.540	1.565	1.596	1.623	1.650	1.680	1.740
100	1,305	1.088	1.107	1.130	1,152	1.174	1,198	1.218	1.240	1,260	1.305
110	1.155	.961	.9800	1.000	1.020	1.040	1.060	1.078	1.095	1.115	1,155
120	1.065	.887	.9050	.9230	0046.	.9580	.9760	.9930	1.010	1.030	1,065
130	1.015	.8450	.8600	.8800	.8950	0416.	.9300	.9450	0496.	.9800	1.015
140	1.000	0468.	.8500	.8660	0488*	0006	.9170	.9330	.9500	0996.	1,000
150	1.015	0548°	.8600	.8800	.8950	.9140	.9300	.9450	0496.	.9800	1.015
160	1.065	.8870	.9050	.9230	0076.	.9580	0926.	.9930	1.0100	1.030	1,065
170	1.155	.961	.9800	1.000	1.0200	1.040	1.060	1.0780	1.095	1,115	1.155
180	1.305	1.088	1.107	1,130	1,152	1.174	1,198	1,218	1,240	1,260	1.305

Table A-7(a) Adhesional Resistance against Sliding t30

O(degrees)	sin 29	t ₃₀ in c _a x 1/a ₀
10	0.342 0.642	2.88 1.53
30 40	0.865 0.985	1.53 1.14 1.00
50 60	0.985	1.00 1.14
70 80	0.642 0.342	1.53 2.88
85	0.174	5.66

Table A-7(b) Adhesional Resist. against Displacement t40

•	1/sin(9 - d)		t40 11	n ca x 1/	ag
deg.		Pm=2	$P_{m=3}$	$\rho_{m=4}$	$\rho_{m=5}$
0 10 85 90 100 110 120 130 140 150 160 170 180	1.305 1.555 1.740 1.555 1.305 1.155 1.065 1.015 1.065 1.155 1.305	2.61 3.11 3.48 3.11 2.61 2.31 2.03 2.00 2.03 2.13 2.31 2.61	3.915 4.665 5.220 4.665 3.915 3.465 3.045 3.045 3.195 3.465 3.915	5.22 6.96 6.22 5.22 4.62 4.06 4.06 4.06 4.62 5.22	6.53 7.77 8.70 7.70 6.53 5.77 5.00 5.07 5.07 5.77 6.53

APPENDIX II

BEHAVIOR OF THE PARTICLE MODEL

A. Symbols

A. Symbols

Stress	All stresses applied to the matrix are
	expressed in terms of k, the effective
	principal stress ratio.
Matrix Failure	That state of stress at which every
	point on the failure plane satisfies the
	condition of maximum obliquity.
<u>n</u>	Number of contacts within a soil area $A_{C\!A}$
	along an -plane.
М	Range in 3 (degrees) in which contact
	failure occurs under k.
n _m	Number of contact failure in M
~	Applied shear stress along a-plane
	(failure plane) in force per gross area of
	soil
stress-factor	A magnification factor on τ_j resulting
	from redistribution of displaced contacts.

APPENDIX II

BEHAVIOR OF THE PARTICLE MODEL

- B. Procedure of Computations of Stress-displacement characteristics according to the modes of:
 - (a) Partial Remolding
 - (b) Complete Remolding
 - (c) Partial Re-orientation
 - (d) Complete Re-orientation

(a) Partial Remolding.

For simplicity, the distribution of particles is assumed to be uniform throughout the entire range of $\theta = 0^{\circ} - 180^{\circ}$ as shown in Fig. A-1-A.

We have accordingly

the distribution function $y = \frac{1}{\pi} \times n = 0.318$ n No. of contacts in sliding = $\frac{85-10}{180}$ n = 0.43 n

No. of contacts in displacement

$$= \frac{(10-0) + (180-85)}{180}$$
 n = 0.57 n

All values of τ_{α} are computed by the expression $\tau_{\alpha} = \bar{\sigma}_{3} \frac{(k-1)}{2} \sin 2\alpha x \text{ stress-factor}$ $t_{\alpha} = (\text{force per unit contact area}) = \tau_{\beta} \times \frac{1}{a_{\theta}}$

1)
$$k = 1.3$$

$$= 0.1475 \frac{1}{3} \qquad t = 0.1475 \frac{1}{3} \times \frac{1}{a}$$

From Fig. A-1-B-1, M = 28-68

$$n_{\rm m} = \frac{(68-28)}{(85-10)} \times 0.43 \text{ n} = 0.235 \text{ n}$$

By definition, 0.235 n contacts carry no additional shear stress. Thus, at the end of k=1.3,

stress-factor = 1/(1-0.235) = 1.31

All the remaining 0.765 n contacts are in the range of = 10-28 and 68-85 in the sliding group and = 0-10 and 85-180 in the displacement group

Fig. A-1-B-2 shows the range of contact failure for k=1.3.

2) k = 1.4

$$= 1.31 \times 0.197 = 0.258 = 0.2$$

From Fig. A-1-C-1,

$$M = 16.75 - 79$$

$$\eta_{\rm m} = \frac{(79-16.75)}{(85-10)} \times 0.43 \text{ n} = 0.36 \text{ n}$$

Thus, at the end of k = 1.4,

stress-factor = 1/(1-0.36) = 1.56

All the remaining 0.64 n contacts are in the range of

 τ = 10-15.75 and 79-85 in the sliding group

and $\phi = 0-10$ and 85-180 in the displacement group

Fig. A-1-C-2 shows the range of contact failure for k = 1.4.

3) k = 1.5

$$t_{1} = 1.56 \times 0.246$$
 $\frac{1}{3} = 0.384$ $\frac{1}{3}$ $t_{1} = 0.384$ $\frac{1}{3} \times \frac{1}{a}$

From Fig. A-1-D-1:

In sliding,

M = 12.75 - 83

In displacement

M = 115 - 154

In sliding,

$$n_{m} = \frac{(83-12.75)}{(85-10)} \times 0.43 \text{ n} = 0.405 \text{n}$$

In displacement,

$$n_{\rm m} = \frac{(134-115)}{(10-0) + (180-85)} \times 0.57n = 0.264n$$

Total $n_m = (0.405 + 0.254)$ n = 0.669 n

0.599 n contacts carry no additional shear stress. Thus, at the end of k = 1.5,

stress-factor = 1/(1-0.699) = 3.02

All the remaining 0.331 n contacts are in the range of

 \sim = 10-12.75 and 83-85 in the sliding group

and -=0-10, 85-113 and 164-180 in the displacement group.

Fig. A-1-D-2 shows the range of contact failure for k = 1.5.

4) k = 1.6

$$= 3.02 \times 0.295 \frac{1}{3} = 0.89 \frac{1}{3}$$
 t $= 0.89 \frac{1}{3} \times \frac{1}{a}$

From the spectrum of resistance for k=1.6, M=0-180. This implies that all contacts have failed and the matrix has reached failure.

(b) Complete Remolding

The initial particle distribution is assumed to follow the same uniform distribution that was used in (a). It is shown in Fig. A-2-A. Accordingly, we have as before:

In Sliding, no. of contacts = 0.43 n, y = 0.328 n

In displacement, no. of contacts = 0.57 n, y = 0.311 n.

All values of are again computed by the expression $= -\frac{(k-1)}{2} \times \sin 2 \times \text{stress-factor}$

t =
$$x \frac{1}{a}$$
 in force per unit contact area.

1) k = 1.3

$$= 0.1475 \frac{-}{3}$$
 $t = 0.1475 \frac{-}{3} \times \frac{1}{a}$

From Fig. A-2-B-1, M = 28-68

$$n_{\rm m} = \frac{(68-28)}{(85-10)} \times 0.43 \text{ n} = 0.235 \text{ n}$$

By definition, 0.235 n contacts transfer all of their share of shear stress to the remaining 0.765 n contacts. Thus,

the first stress-factor = 1/(1-0.235) = 1.31.

The first redistribution t = 1.31 x 0.1475 $\frac{A}{3}$ $x_n A_c = 0.193 \frac{A}{3}$ $\frac{1}{a}$

From Fig. A-2-B-1, M = 21-75

$$n_{m} = \frac{(75-21)}{(85-10)} \times 0.43 \text{ n} = 0.315 \text{ n}$$

0.315 n contacts again transfer all of their share of shear stress to the other contacts. Thus,

the second stress-factor = 1/(1-0.315) = 1.45

The second redistribution t = 1.46 x 0.1475 $\frac{-}{3}$ x $\frac{A}{n}$ A_c = 0.2155 $\frac{-}{3}$ x $\frac{1}{a}$

From Fig. A-2-B-1 $n_{m} = \frac{(75.8-19)}{(85-10)} \times 0.43 \text{ n} = 0.336 \text{ n}$

The successive transfer of shear stress due to contact failure is continued until the additional stress to be carried by the remaining contacts is negligible. Thus, at the end of k=1.3

the final stress-factor = 1/(1-0.344) = 1.52.

All the remaining 0.656 n contacts are in the range of

= 10-13.25 and 77.5-85 in the sliding group

and = 0-10 and 85-180 in the displacement group.

Fig. A-2-B-2 shows the successive range of contact failure for k = 1.3.

2) k = 1.4

$$= 1.52 \times 0.197 = 0.30$$

From Fig. A-2-C-1, M = 14.6-81

$$n_{\rm m} = \frac{(81-14.6)}{(85-10)} \times 0.43 \text{ n} = 0.384 \text{ n}$$

0.384 n contacts transfer all of their share of shear stress to the remaining 0.516 n contacts. Thus,

the first stress-factor = 1/(1-0.384) = 1.620

The first redistribution = 1.620 x 0.197 $\frac{1}{3}$ = 0.319 $\frac{1}{3}$

From Fig. A-2-C-1, M = 14-81.5

$$n_{\rm m} = \frac{(81.5-14)}{(85-10)} \times 0.43 \text{ n} = 0.389 \text{ n}$$

The final stress-factor = 1/(1-0.389) = 1.635 Thus, at the end of k = 1.4, all the remaining 0.611 n

contacts are in the range of = 10-14 and 81.5-85 in the

sliding group

and = 0-10 and 85-180 in the displacement group

Fig. A-2-C-2 shows the gross range of contact failure for k=1.4.

3) k = 1.5

$$= 1.635 \times 0.246 = 0.4025 = 0$$

From Fig. A-2-D-1:

In sliding,

M = 12.4 - 82.8

In displacement

M = 111-170

In sliding

 $n_{\rm m} = \frac{(82.4-12.4)}{(85-10)} \times 0.43 \text{ n} = 0.405 \text{ n}$

In displacement,

 $n_{\rm m} = \frac{(170-111)}{(10-0) + (180-85)} \times 0.57 \text{ n} = 0.326 \text{ n}$

Total $n_{in} = (0.405 + 0.326)$ n = 0.731 n

0.731 n contacts transfer all of their share of shear stress to the remaining 0.269 n contacts. Thus,

the first stress-factor = 1/(1-0.731) = 3.54

The first redistribution t = 3.54 x 0.246 $\frac{1}{3}$ x $\frac{1}{a}$ = 0.87 $\frac{1}{3}$ x $\frac{1}{a}$

From Fig. A-2-D-2, contacts in the entire range of M = 0-180 have now failed and the matrix has reached failure.

Fig. A-2-D-2 shows the successive range of contact failure for k = 1.5.

(c) Partial Re-orientation

The initial particle distribution is assumed to follow the same uniform distribution that was used in (a) and (b). It is shown in Fig. A-3-A.

We have, as before:

In sliding, No. of contacts = 0.43 n, y = 0.318 n

In displacement, No. of contacts = 0.57 n, y = 0.318 n

Since the stress-factor is always unity, all values of T are given by $\frac{-1}{3} \frac{(k-1)}{2} \sin 2\pi$ without modification.

 $t_{i} = T_{i} \times \frac{1}{a}$ (in force per contact area)

1)
$$k = 1.3$$

$$\tau_{x} = 0.1475 \frac{\pi}{3}$$
 $t_{x} = 0.1475 \frac{\pi}{3} \times \frac{1}{a_{x}}$

From Fig. A-3-B-1, M = 28-68

$$n_{\rm m} = \frac{(68-28)}{(85-10)} \times 0.43 \text{ n} = 0.235 \text{ n}$$

By definition, 0.235 n contacts redistribute themselves over the range of = 10-28 and 68-85 in the sliding group. Thus at the end of k = 1.3, we have:

- 0.43 n contacts in = = 10-28 and 68-85 in the sliding group
- 0.57 n contacts in $\beta = 0-10$, and 85-180 in the displace. group.

Assuming a uniform redistribution with respect to -,

$$y \times (28-10 + 85-68) = 75 \times 0.328 \text{ n}$$

or y = 0.702 n in x = 10-28 and 68-85 in sliding y = 0.311 n in x = 0-10 and 85-180 in displace.

Fig. A-3-B-2 shows the range of contact failure for k = 1.3.

2) k = 1.4

$$t_{y} = 0.197 \frac{1}{3}$$
 $t_{y} = 0.197 \frac{1}{3} \times \frac{1}{a}$

From Fig. A-3-C-1, M = 21.5-28 and 68-74.5

$$n_{m} = \frac{(28-21.5) + (74.5-68)}{(28-10) + (85-68)} \times 0.43 \text{ n} = 0.16 \text{ n}$$

Total
$$n_m = (0.235 + 0.16) n = 0.395 n$$

0.16 n contacts redistribute themselves over the range of = 10-21.5 and 74.5-85 in the sliding group.

Thus, at the end of k = 1.4, we have

0.43 n contacts in = 10-21.5 and 74.5-85 in the sliding and 0.57 n contacts in = 0-10 and 85-180 in the displace.

Assuming a uniform redistribution with respect to make the second of the

$$y \times (21.5-10 + 85-74.5) = 75 \times 0.328 \text{ n}$$

or y = 1.17 n in f = 10-21.5 and 74.5-85 in sliding y = 0.311 n in f = 0-10 and 85-180 in displacement

Fig. A-3-C-2 shows the range of contact failure for k = 1.4.

3) k = 1.9

At the end of k = 1.8, we have

0.43 n contacts in 6 = 10-13.6 and 82-85 in sliding and 0.57 n contacts in 9 = 0-10 and 85-125 and 154-180 in displacement.

Assuming a uniform redistribution with respect to θ

$$y \times (13.6-10 + 85-82) = 75 \times 0.328 \text{ n}$$

or y = 3.82 n in = 10-13.6 and 82-85 in sliding

$$y \times (10-0 + 125-85 + 180-154) = 105 \times 0.311$$

or y = 0.43 n in = 0-10, 85-125 and 154-180 in displacement

Total n_m at the end of k = 1.8 is 1.1088 n

For
$$k = 1.9$$
, $= 0.443 \frac{1}{3}$ $t = 0.443 \frac{1}{3} \times \frac{1}{a}$

From Fig. A-3-D-1:

In sliding, M = 13-13.6 and 82-83

In displacement M = 110-125 and 154-169

In sliding
$$n_m = \frac{(13.6-13) + (83-82)}{(13-10) + (85-82)} \times 0.43 n = 0.1042 n$$

In displacement,

$$n_{\rm m} = \frac{(125 - 110) + (169 - 154)}{(10 - 0) + (125 - 85) + (180 - 154)} \times 0.57 \text{ n} = 0.225 \text{ n}$$
 Total $n_{\rm m}$ at k = 1.9 equals $(0.1042 + 0.225)$ n or 0.3293 n.

Total $n_m = (1.1088 + 0.3293) n = 1.438 n$

In the sliding group, 0.1042 n contacts redistribute themselves over the range of z=10-13 and 83-85.

In the displacement group, 0.225 n contacts redistribute themselves over the range of = 0-10, 85-110 and 169-180. We have, at the end of k = 1.9,

0.43 n contacts in $^{\circ}$ = 10-13 and 83-85 in sliding

and 0.57 n contacts in \cdot = 0-10, 85-110 and 169-180 in displace.

Assuming a uniform redistribution with respect to -,

$$y \times (13-10 + 85-83) = 75 \times 0.328$$

or $y = 4.91 \text{ n in}^{-1} = 10-13 \text{ and } 83-85 \text{ in sliding}$

$$y \times (10-0 + 110-85 + 180-169) = 105 \times 0.311$$

or y = 0.711 n in y = 0-10, 85-110 and 169-180 in displace.

Fig. A-3-D-2 shows the range of contact failure for k = 1.9.

4) k = 2.4

At the end of k = 2.3, we have

0.43 n contacts in = 10.25-10.8 in sliding

0.57 n contacts in = 7-10, 85-93 in displacement

Assuming a uniform redistribution with respect to ...

$$y \times (10.8-10.25) = 75 \times 0.328 \text{ n}$$

or y = 43.6 n in = 10-10.25 in sliding

$$y \times (10-7 + 193-85) = 105 \times 0.311 \text{ n}$$

or y = 2.97 n in = 7-10 and 85-93 in displacement.

Total n_m at end of k = 2.3 equals 2.9647 n.

For
$$k = 2.4$$
, $= 0.69 \frac{1}{3}$ $t = 0.69 \frac{1}{3}$ $\times \frac{1}{a}$

From Fig. A-3-E-1:

In sliding, M = 10.25-10.8

In displacement, M = 7-10 and 90-93.

In sliding, $n_m = \frac{(10.8-10.25)}{(10.8-10.25)} \times 0.43 \text{ n} = 0.43 \text{ n}$

In displacement $n_m = \frac{(10-7) + (93-90)}{(10-7) + (93-85)} \times 0.57 \text{ n} = 0.311 \text{ n}$

Total n_m at k = 2.4 equals (0.43 + 0.311) n or 0.741 n Total n_m = (2.9647 + 0.741) n = 3.7057 n

Thus, the entire sliding group is now exhausted.

0.741 n contacts redistribute themselves over the range of 0.85-90 in the displacement group. We have, at the end of 0.85-90 in contacts in 0.85-90.

Assuming a uniform redistribution with respect to $\hat{\tau}$,

 $y \times (90-85) = 180 \times 0.318 \text{ n},$

or $y = 36 \times 0.318$ n or 11.45 n in x = 85-90

Fig. A-3-E-2 shows the range of contact failure for k=2.4.

5) k = 2.5

$$t_{a} = 0.738 \frac{\pi}{3}$$
 $t_{a} = 0.738 \frac{\pi}{3} \times \frac{1}{a_{p}}$

From Fig. A-3-F-1, M = 88-90

$$n_{\rm m} = \frac{(90-88)}{(90-85)} \times n = 0.4 n$$

0.4 n contacts redistribute themselves over the range of 0.4 = 85-88. Thus, at the end of k = 2.5, we have 0.4 x n contacts in 0.4 = 85-88.

Assuming a uniform redistribution with respect to \dot{v} , y x (88-85) = 180 x 0.318 n or y = 19.1 n in \dot{v} = 85-88.

Fig. A-3-F-1 shows the range of contact failure for k=2.5.

The computations are carried out in the same manner for k=2.6 and 2.7. At k=2.7, all contacts are exhausted and the matrix has reached failure.

(d) Complete Re-orientation

The initial particle distribution is assumed to follow a uniform distribution throughout the entire range of $= 0-180^{\circ}$ as shown in Fig. A-4-A. We have, accordingly

$$y = \frac{1}{x} \times n = 0.318 \text{ n}$$

Initial no. of contacts in sliding = $\frac{85-10}{180}$ n = 0.4165 n Initial no. of contacts in displacement = $\frac{(10-0)+(180-85)}{180}$ x n = 0.5835 n

Since the stress-factor is always unity, all values of a given by $-\frac{(k-1)}{3}\sin 2$ without modification the force per unit contact area $= \frac{1}{2} \times \frac{1}{a}$

$$t_{i} = 0.1475 \, \overline{t}_{3}$$
 $t_{i} = 0.1475 \, \overline{t}_{3} \times \frac{1}{a}$

From Fig. A-4-B-1, M = 28-68 $n_{m} = \frac{68-28}{180} \times n = 0.222 \text{ n}$

By definition, 0.222 n contacts redistribute themselves over the range of x=0.28 and 68-180. Thus, at the end of x=1.3, we have 1 x n contacts in x=0.28 and 68-180.

Assuming a uniform redistribution with respect to 0,

$$y \times (28-0 + 180-68) = 180 \times 0.318 \text{ n}$$

or y = 0.41 n over the range of 0 = 0-28 and 68-180.

Fig. A-4-B-2 shows the range of contact failure for k = 1.3.

2) k = 1.4

$$\tau_{a} = 0.197 \, \overline{t}_{3} \, \text{t}_{a} = 0.197 \, \overline{t}_{3} \, \text{x} \, \frac{1}{a_{34}}$$

From Fig. A-4-C-1, M = 21.5-68 and 68-74.5

$$n_{\rm m} = \frac{(28-21.5)+(74.5-68)}{(28-0)+(180-68)} \times n = 0.093$$
 n

Total
$$n_m = ().222 + 0.093)$$
 $n = 0.315$ n

0.093 n contacts redistribute themselves over the range of = 0-21.5 and 74.5-180. Thus, at the end of k = 1.4 we have 1 x n contacts in = 0-21.5 and 74.5=180.

Assuming a uniform redistribution with respect to

 $y \times (21.5-0 + 180-74.5) = 180 \times 0.318 \text{ n}$

or y = 0.451 n over the range of = 0-21.5 and 74.5=180

Fig. A-4-C-2 shows the range of contact failure for k = 1.4.

3) k = 1.9

At the end of k = 1.8, we have

 $1 \times n \text{ contacts in } = 0-13.6, 82-125 \text{ and } 154-180.$

Assuming a uniform redistribution with respect to -,

 $y \times (13.6-0 + 125-82 + 180-154) = 180 \times 0.318 \text{ n}$

or y = 0.693 n over f = 0-13.6, 82-125 and 154-180

Total $n_m = 0.699 \text{ n}$

For k = 1.9, $= 0.443 \frac{1}{3}$ $t = 0.443 \frac{1}{3}$ $\times \frac{1}{a}$

From Fig. A-4-D-1:

In sliding, M = 13-13.6 and 82-83

In displacement, M = 110-125 and 154-169

In sliding, $r_m = \frac{(13.6-13)+(83-82)}{(13.6-0)+(125-82)+(180-154)} \times r_m = 0.0194r_m$

In displacement, $n_m = \frac{(125-110)+(169-154)}{(13.6-0)+(125-82)+(180-154)} \times n = 0.363 \text{ n}$

Total n_m at k = 1.9 equals (0.019 + 0.363)n or 0.3824 n Total $n_m = (0.699 + 0.3824)$ n = 1.08 n

0.3824 n contacts redistribute themselves over the range of x = 0-13, 83-110 and 169-180. Thus, at the end of k = 1.9, we have 1 x n contacts in x = 0-13, 83-110 and 169-180.

Assuming a uniform redistribution with respect to :,

 $y \times (13-0 + 110-83 + 180-169) = 180 \times 0.318 \text{ n}$

or y = 1.12 n in = 0-13, 83-110 and 169-180.

Fig. A-4-D-2 shows the range of contact failure for k = 1.9.

4) k = 2.4

At the end of k = 2.3, we have

1 x n contacts in x = 10.25-10.8, 7-10 and 85-93.

Assuming a uniform redistribution with respect to v,

 $y \times (10.8-10.25 + 10-7 + 93-85) = 180 \times 0.318 \text{ n}$

or y = 4.9 n in = 10.25-10.8, 7-10 and 85-93

Total $n_m = 2.231 n$

For k = 2.4, $= 0.69 \frac{1}{3}$ $t_{a} = 0.69 \frac{1}{3}$ $\times \frac{1}{a}$

From Fig. A-4-E-1:

In sliding, M = 10.25-10.8

In displacement M = 7-10 and 90-93

In sliding, $n_m = \frac{(10.8-10.25)}{(10.8-10.25)+(10-7)+(93-85)} \times n = 0.048 \text{ n}$

In displacement,

$$n_{m} = \frac{(10-7)+(93-90)}{(10.8-10.25)+(10-7)+(93-85)} \quad x \quad n = 0.52 \quad n$$

Total n_m at k = 2.4 equals (0.048+0.52) n or 0.568 n

Total $n_m = (2.231 + 0.568)$ n = 2.799 n

0.568 n contacts redistribute themselves over the range of t = 85-90. Thus, at the end of k = 2.4,

we have $1 \times n$ contacts in $\theta = 85-90$

Assuming a uniform redistribution with respect to θ ,

$$y \times (90-80) = 180 \times 0.318 \text{ n}$$

or $y = 36 \times 0.318 \text{ n} = 11.45 \text{ n}$

Fig. A-4-E-2 shows the range of contact failure for k=2.4.

5) k = 2.5

$$t_{x} = 0.738 \frac{1}{3}$$
 $t_{y} = 0.738 \frac{1}{3} \times \frac{1}{a}$

From Fig. A-4-F-1, M = 88-90

$$n_{\rm m} = \frac{(90-88)}{(90-85)} \times n = 0.4 \text{ n}$$

Total
$$n_m = (2.799 + 0.4) n = 3.199 n$$

0.4 n contacts redistribute themselves over the range of

= 85-88. Thus, at the end of k = 2.5,

we have $1 \times n$ contacts in $\theta = 85-88$

Assuming a uniform redistribution with respect to .,

 $y \times (88-85) = 180 \times 0.318 \text{ n}$

or y = 19.1 n

Fig. A-4-F-1 shows the range of contact failure

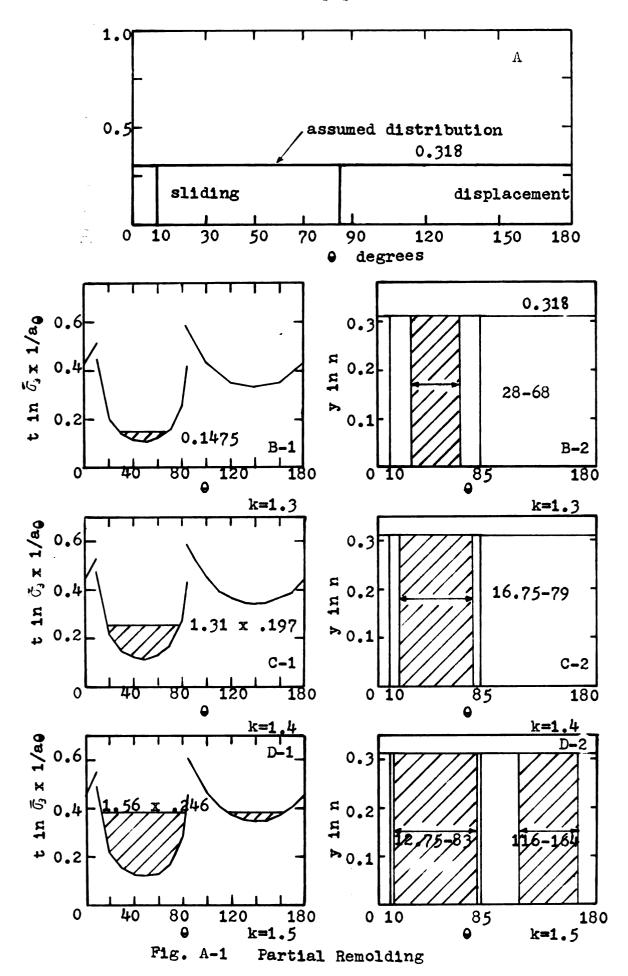
for k = 2.5.

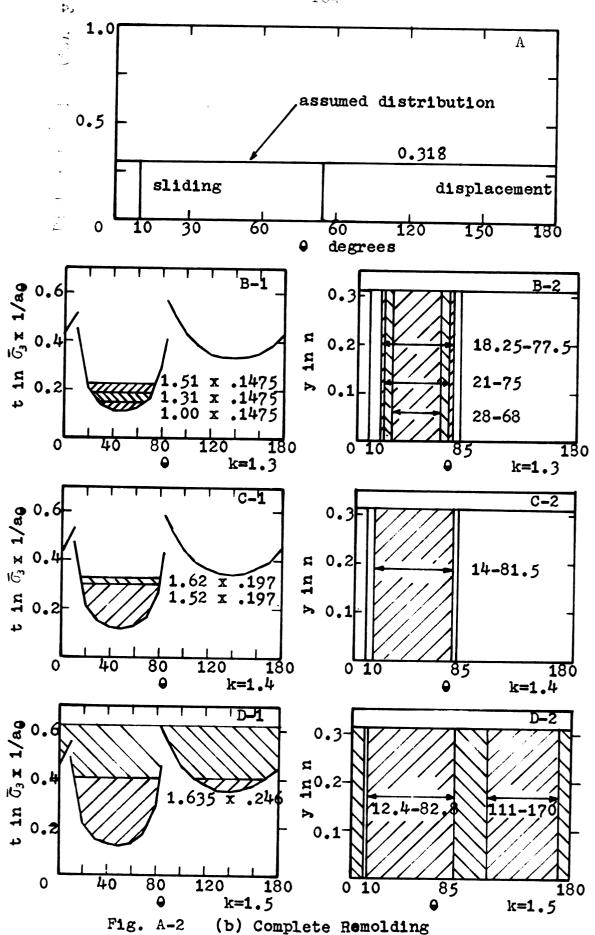
The computations are carried out in the same manner for k=2.6 and k=2.7. At k=2.7, the matrix has reached failure.

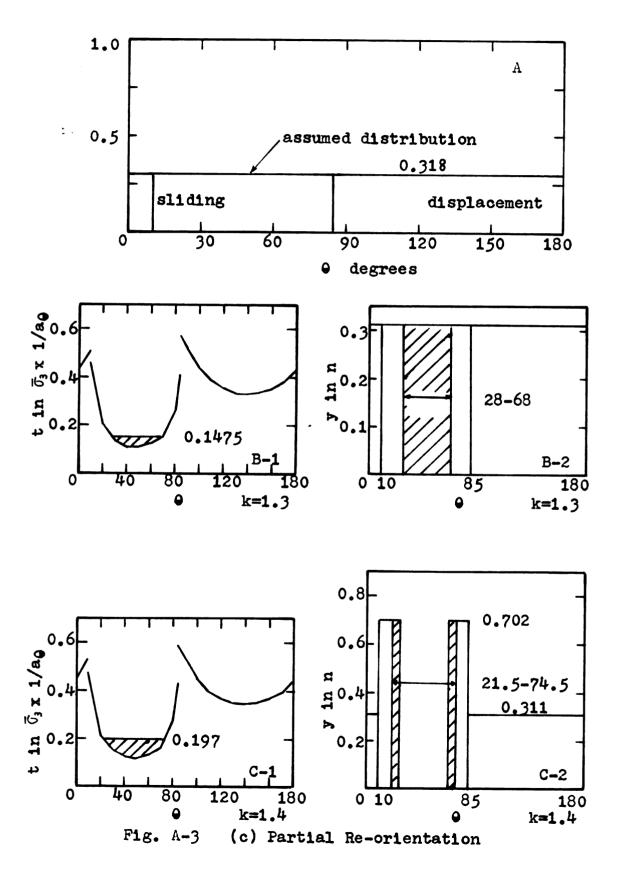
APPENDIX II

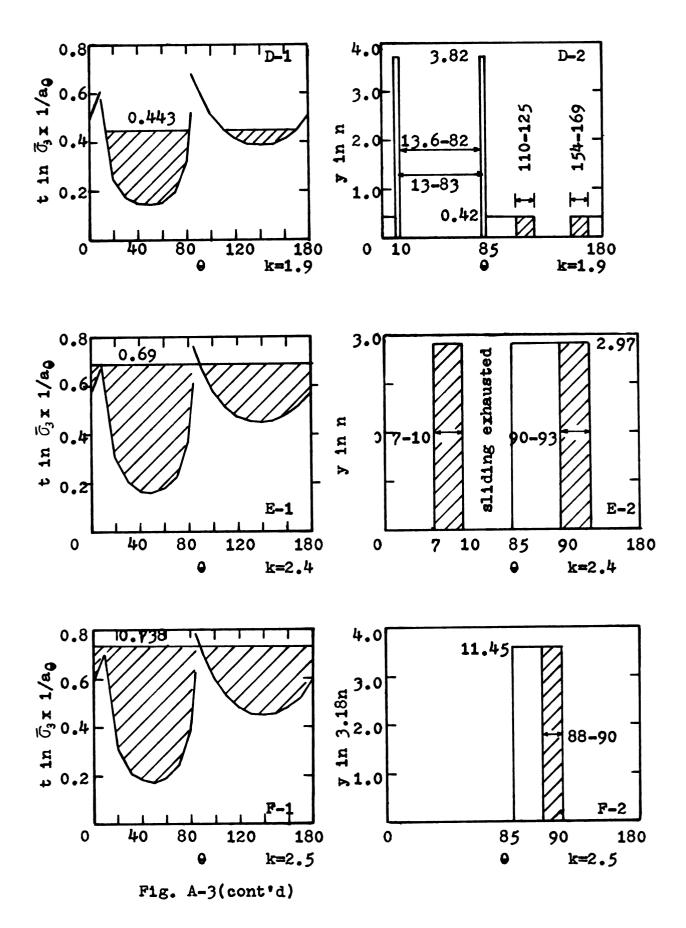
BEHAVIOR OF THE PARTICLE MODEL

C. Illustrations









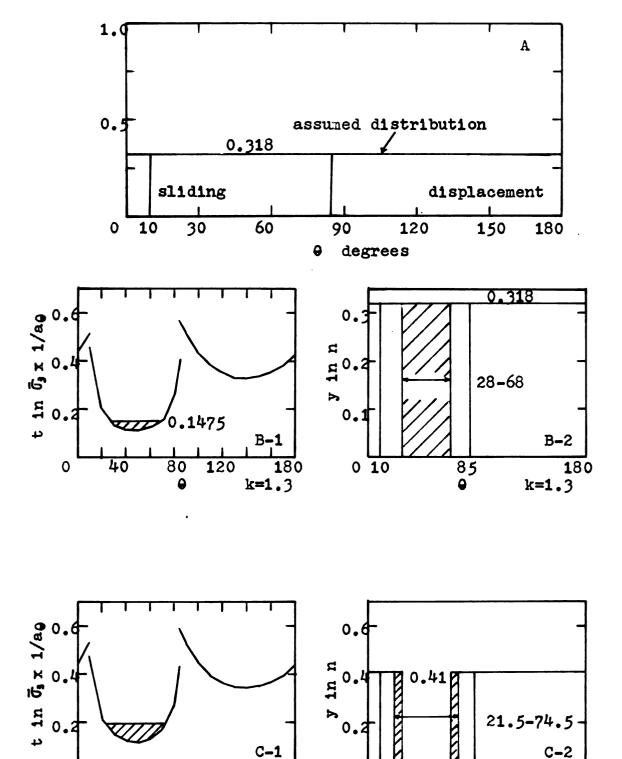
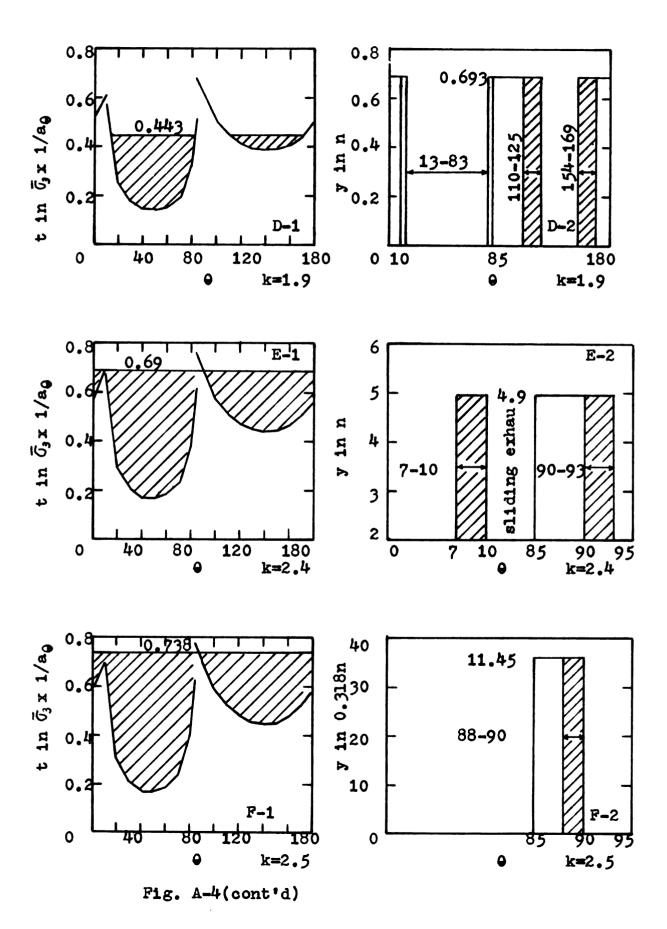


Fig. A-4 (d) Complete Re-orientation

k=1.4

5

k=1.4



APPENDIX II

BEHAVIOR OF THE PARTICLE MODEL

D. Results of Computations

Table A-8(a) Behavior of Particle Model

(a) Partial Remolding $\rho = 0.3$

k	Ta	Sliding t ₁₀		Displaces t ₂₀	Displacement t20		Stress-
	⊡ ₃	M deg.	n _m	M deg.	n _m	n	factor
1.1	0.0493	-	0	_	0	0	1.00
1.2	0.0985	-	0	-	0	0	1.00
1.3	0.1475	28-68	.235	-	0	0.235	1.31
1.4	0.2580	16.75-79	.360	-	0	0.360	1.56
1.5	0.384	12.75- 83	.405	116-164	.264	0.669	3.02
1.6	0.890	all	.430	all	-570	1.000	
				,			

Table A-8(b) Behavior of Particle Model

(b) Complete Remolding $\rho = 0.3$

	Ta	Sliding t ₁₉		Displace t ₂₀		Σn _m	Stress-
k	, v	М	n _m	М	n_{m}		factor
	G _a	deg.	n	deg.	n	n	
1.1	0.0493	-	0	-	0	0	1.00
1.2	0.0985	-	0	-	0	o	1.00
1.3	0.1475 0.1930 0.2155 0.2200 0.2230	28-68 21-75 19-76.8 18.5-77 18.25-77	.235 .315 .336 .339 .334	-	0 0 0 0	0.235 0.315 0.336 0.339 0.344	1.31 1.46 1.49 1.51 1.52
1.4	0.3000 0.3190	14.6-81 14.0-81	•384 •389	-	0	0.384 0.389	1.62 1.64
1.5	0.4025 1.4600	12.4-82.8 all	.405 .430	111-170 all	•326 •570	0.731 1.000	5.92
	·						

Table A-8(c) Behavior of Particle Model

(c) Partial Re-orientation $\rho = 0.3$

Sliding Displacement $\sum n_m$ t10 t20 Tal Stressk factor M M $n_{\rm m}$ n_{m} σ_{a} deg. deg. n n 1.2 0.0985 0 0 1.00 0 1.3 0.1475 28 - 68 0 0 1.00 .235 .160 1.4 0.1970 1.00 21.5-28 0 0.395 68 - 74.5 18.3-21.5 .121 1.5 0.2460 0 0.516 1.00 74.5-77.5 14.8-18.3 .165 0 0.681 1.00 1.6 0.2950 77.5-80 14.5-14.8 1.7 0.3450 0 0.769 1.00 .088 80 - 81.7 1.109 13.6-14.5 .149 1.00 1.8 0.3940 125-154 .191 81.8-82 1.9 0.4430 13 -13.6 .225 .104 110-125 1.438 1.00 82 - 83 154-169 2.0 0.4925 1.704 1.00 12.3-13 .129 105-110 .137 83-83.4 169-175 11.5-12.3 .161 2.0435 1.00 2.1 0.5410 100-105 .179 83.8-84.**5** 175-180 0-1.0 1.0-4.0 2.2 0.5900 10.8-11.5 .258 .166 .4677 1.00 84.5-85 96-100 2.9647 1.00 2.3 0.5900 10.3**-1**0.8 .296 4.0-7.0 .201 93- 96 2.4 0.6900 .430 7.0-10.0 all .3110 3.7057 1.00 90-93 2.5 0.7380 NONE 88-90 400 4.105 1.00 2.6 0.7870 NONE 86.5-88 .500 4.605 1.00

٠,

Table A-8(d) Behavior of Particle Model

(d) Complete Re-orientation $\rho = 0.3$

	TX	Sliding t ₁₀		Displace t ₂ 0		$\sum n_{m}$	Stress-
k	\overline{G}_3	M deg.	n _m n	M deg.	n _m n	n	factor
1.3	0.1475	28 - 68	.222	•	-	0.222	1.00
1.4	0.1970	21.5-28 68-74.5	.093	-	-	0.315	1.00
1.5	0.2460	18.3-21.5 74.5-77.5	.0488	-	-	0.3638	1.00
1.6	0.2950	14.75-18 77.5-80	.0501	-	-	0.414	1.00
1.7	0.3450	14.5-14.7 80-81.75	.0174	-	-	0.432	1.00
1.8	0.3940	13.6-14.5 81.75-82	.0102	125-154	0.257	0.699	1.00
1.9	0.4430	1313.6 82-82	.0194	110 -1 25 1 <i>5</i> 4 -1 69	0.363	1.08	1.00
2.0	0.4925	12.3-13.0 83-83.75	.0294	105 -1 10 169 -1 75	0.216	1.339	1.00
2.1	0.5410	11.5-12.3 83.8-84.5	.0390	0-1 100-105 175-180	0.286	1.662	1.00
2.2	0.5900	10.8 -11.5 84.5 - 85	.0460	1-4 96-100	0.192	1.900	1.00
2.3	0.6400	10.3-10.8	.0278	4 - 7 93 - 96	0.303	2.221	1.00
2.4	0.6900	all	.0480	7 -1 0 90 - 93	0.520	2.799	1.00
2.5	0.7380		NONE	88-90	0.400	3.199	1.00
2.6	0.7870		NONE	86.5-88	0.500	3.699	1.00
2.7	0.836		NONE	all	1.000	4.699	1.00

- 3

Table A-9(a) Behavior of Particle Model

(a) Partial Remolding

P = 0.5

	Ta	Sliding	t ₁₉	Displace t20	ement	\sum_{m}	Stress-
k	$\bar{\mathcal{C}}_3$	M deg.	n _m n	M deg.	n _m n	ı n	factor
	V 3			.			
1.3	0.1475	28 - 68	.235	-	0	.235	1.31
1.4	0.25 80	16.8-79	.360	-	0	.360	1.56
1.5	0.384	12.8-83	.405	-	0	.405	1.58
1.6	0.495	11 - 85	.425	-	0	.425	1.74
1.7	0.600	10 - 85	.430	-	0	.430	1.76
1.8	0.691	all	.430	117-163	.254	.684	3.16
1.9	1.400	all	.430	all	-570	1.000	

Table A-9(b) Behavior of Particle Model

(b) Complete Remolding

P = 0.5

	T _d Sliding t10			Displacement t20			Stress-
k		M	$n_{\mathbf{m}}$	M	n_{m}		factor
	Ī,	deg.	n	deg.	n	n	
1.3	0.1475 0.1930 0.2155 0.2200 0.2230	28 - 68 21 - 75 19-76.8 18.5-77 18.3-77.5	.235 .315 .336 .339 .344	- - - -	0 0 0 0	.235 .315 .336 .339 .344	1.31 1.46 1.49 1.51 1.52
1.4	0.3000 0.3190	14.6-81 14-81.5	.384 .389	- -	0	.384 .389	1.62 1.64
1.5	0.4120 0.413 0.4185	12.4 -8 2.8 12 - 83.8 11.9 - 84	,405 .411 .415	- - -	0 0 0	.425 .411 .415	1.68 1.70 1.71
1.6	0.5050 0.5160	10.3-85 10 -85	.428 .430	<u>-</u> -	0	.428 .430	1.75 1.76
1.7	0.6050	all	.430	-	0	.430	1.76
1.8	0.6910 1.2450	all all	.430 .430	117-163 all	.254 .570	.684 1.00	3.16

Table A-9(c) Behavior of Particle Model

(c) Partial Re-orientation

P = 0.5

	Td	Sliding t ₁₀		Displace t ₂₀		∑n _m	Stress-
k		M	n _m	M	$n_{\mathbf{m}}$		factor
	~	deg.	n	deg.	n	n	
1.3	0.1475	28 - 68	.235	-	0	.235	1.00
1.4	0.1970	2 1. 5 - 28 68 - 74 . 5	.160	-	0	•395	1.00
1.5	0.2460	18.3-21.5 74.5-75.5	.121	-	0	.516	1.00
1.6	0.2950	14.75-18 77.5-80	.165	-	0	.681	n
1.7	0.3450	14.5-14.7 80-81.8	.088	-	0	.7693	11
1.8	0.3940	13.6-14.5 81.8-82	.149	-	0	.9178	n
1.9	0.4430	13-13.6 82 - 83	.1042	-	0	1.022	a
2.0	0.4925	12.3-13 83-83.8	.129	-	0	1.151	0)
2.1	0.5410	11.5-12.2 83.8-84.5	.161	-	0	1.312	n
2.2	0.5900	10.8 -11. 5 84 .5- 85	.258	-	0	1.570	n
2.3	0.6400	10.3 -10. 8 85	.296	-	0	.886	n
2.4	0.6900	all	430	-	0	2.296	**
2.5	0.7380	NONE	NONE	-	0	2296	
2.7	0.7870	***		120-128 152-160	.198	2.723	10
2.8	0.8850	11	"	115-120	.154	2.877	"

Table A-9(c) -- cont'd

	$ au_{lpha}$	Sliding t ₁₉		Displace		$\sum n_{m}$	Stress-
k	ত ₃	M deg.	n _m n	M deg.	n _m n	n	factor
2.9	0.935	NONE	NONE	111-115	.170	3.047	1.00
3.0	0.985	**	n	108-169 169 - 172	.128	3.175	n
3.1	1.035	99	11	106-108 172-174	•0975	3.272	
3.2	1.083	99	n	104-106 174-176.5	.121	3.394	n
3.3	1.132	91	n	101-104 177-179	.154	3.548	89
3.4	1.180	**	11	100-101.5 179-180	.100	3.648	10
3.5	1.231	91	**	0 -1. 5 98 .5-1 00	.120	2.768	99
3.6	1.280	91	n	1.5-2.5 97.5-98.5	.091	3.859	**
3.7	1.330	**	**	2.5-4.0 97 - 97.5	.100	3.959	n
3.8	1.380	97	**	4.0-5.0 95.5 - 97	.139	4.098	**
3.9	1.430	20	"	5.0-6.0 95-95.5	•097	4.195	**
4.0	1.480	91	n	6.0-7.0 94-94	.143	4.338	n

Table A-9(d) Behavior of Particle Model (d) Complete Re-orientation $\rho = 0.5$

k	ا _م ان	Sliding t ₁₉ M deg.	n _m	Displacer t ₂₉ M deg.	nent n _m	∑n _m	Stress- factor
345678901234567890 111111222222222333333333333334	0.1470 0.14760 0.14950 0.24950 0.344910 0.55490 0.55490 0.55490 0.559490 0.559490 0.559490 0.59855 1.12380 1.38380 1.4811.4911.4811.4811.4811.4811.4811.481		.093 .049 .050 .017 .010 .029 .046 NONE	•	.000 .000 .000 .000 .000 .000 .000 .00	0.3277 0.3277 0.44554 0.55486 0.61061 0.55816 0.61061	1.0

Calculation of Average Stress-Displacement Curves Table A-10

		Curve B	2.600 2.600 2.600 2.600 2.600 2.600	(c)+(d) ues of N
. 0.5		Curve A	44460000000000000000000000000000000000	= (a)+(b)+(c)+(d) 4 at all values of
Β <i>d</i>	Ä	(q)	40000000444 700000000000000000000000000	전 • 차 # g
		(c)	44777777777777777777777777777777777777	Ourve
	ı	(৭)	44440 ••••••============================	O
		(a)	44444 	(a) (b)
		Curve B	11111111111111 22222 22222 22222 22222 22222 22222 2222	(a)+(b)+(c)+(d) (a)+(b)+(c)+(d)
•3		Curve A	22222222222222222222222222222222222222	(a) +(
0 = d	×	(q)	40000000000000000000000000000000000000	. .
,		(c)	44444444444444444444444444444444444444	ф ц
		(p)	44440 	n V
		(a)	4444 	z z
Contact		n e	00044999999999999999999999999999999999	Curve A

APPENDIX III

TEST DATA

A. Figures

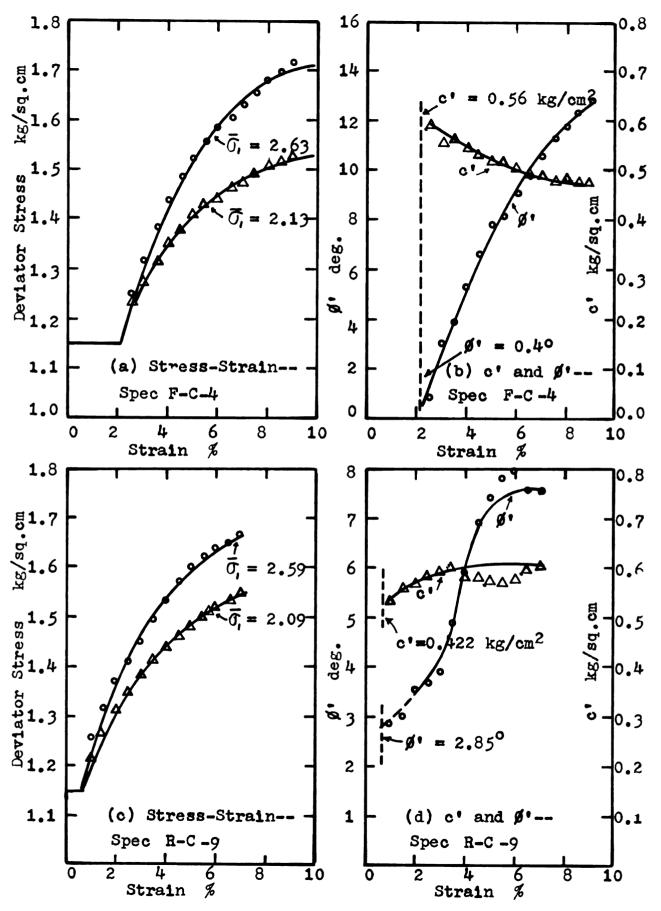


Fig. A-5 Typical Creep-CFS test data for Sault Clay

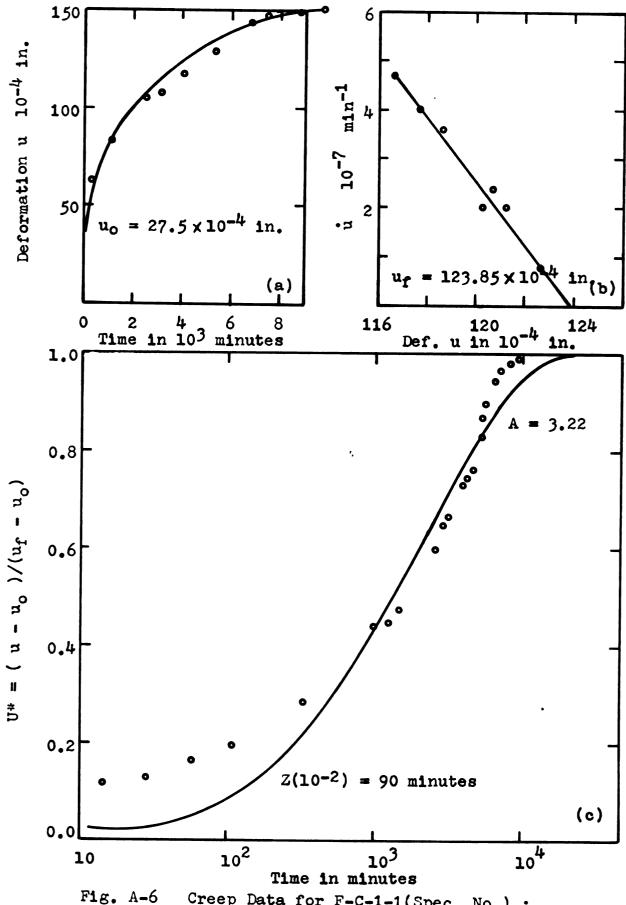


Fig. A-6 Creep Data for F-C-1-1(Spec. No.): $D = 0.0 - 0.5 \text{ kg/cm}^2$

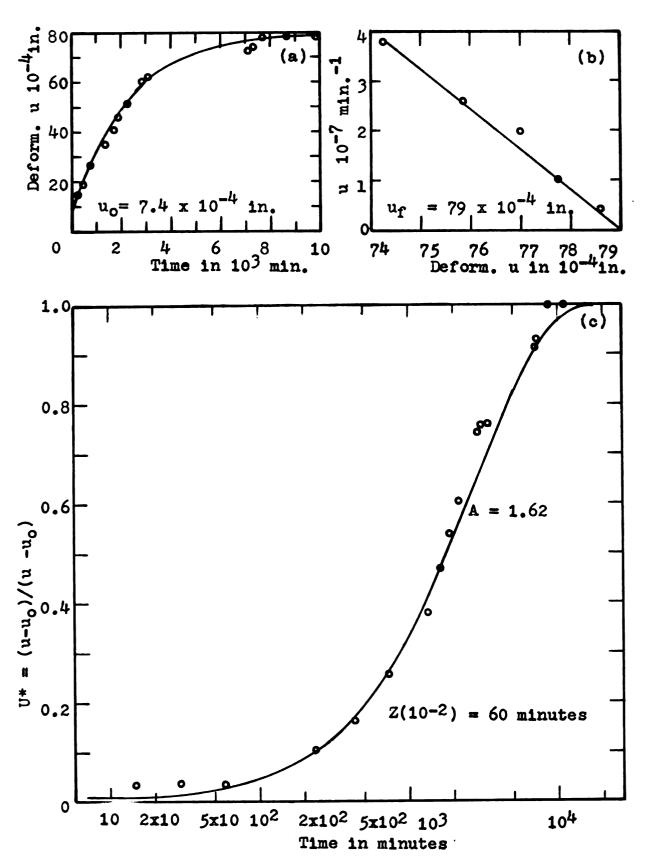
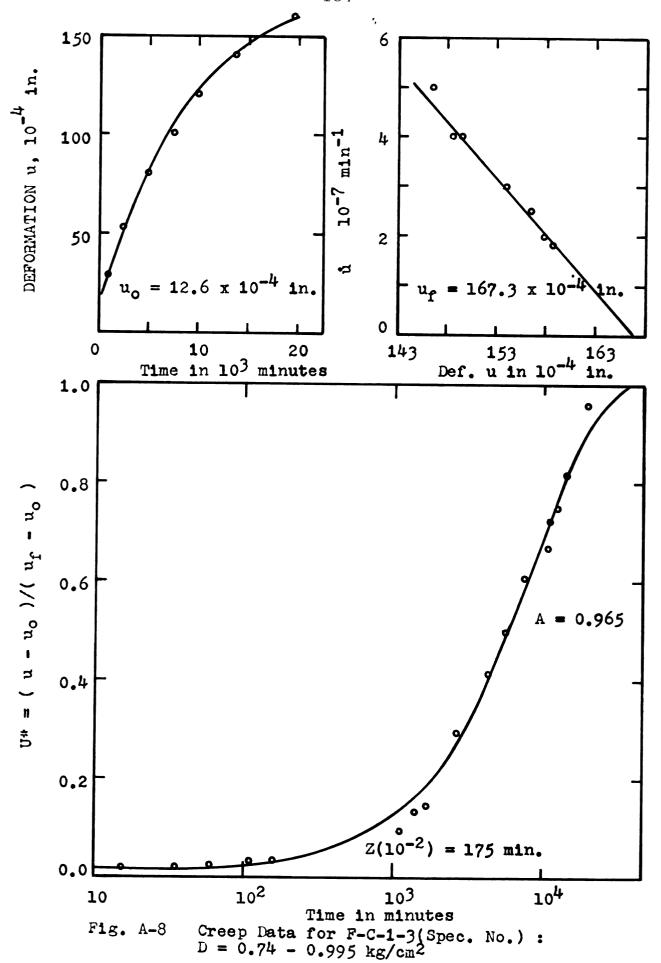
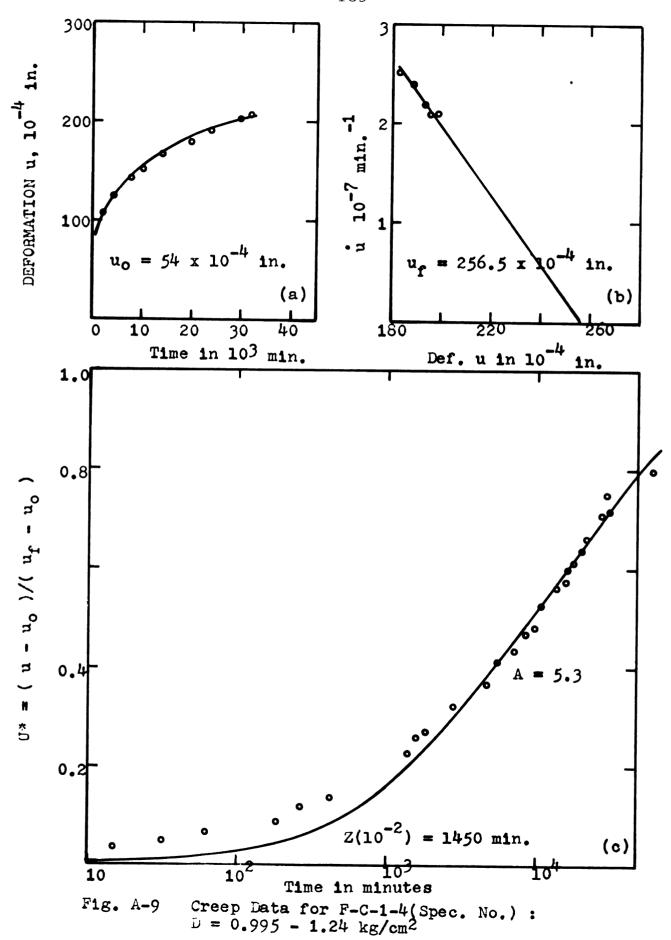


Fig. A-7 Creep Data for F-C-1-2(Spec. No.): $D = 0.5 - 0.74 \text{ kg/cm}^2$





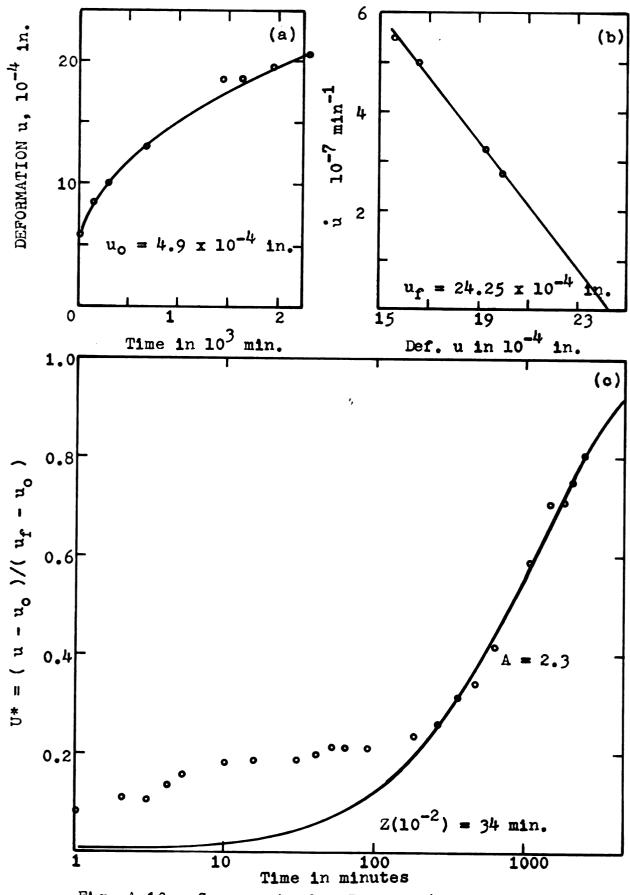


Fig. A-10 Creep data for F-OC-1-2(Spec. No.): $D = 0.495 - 0.75 \text{ kg/cm}^2$

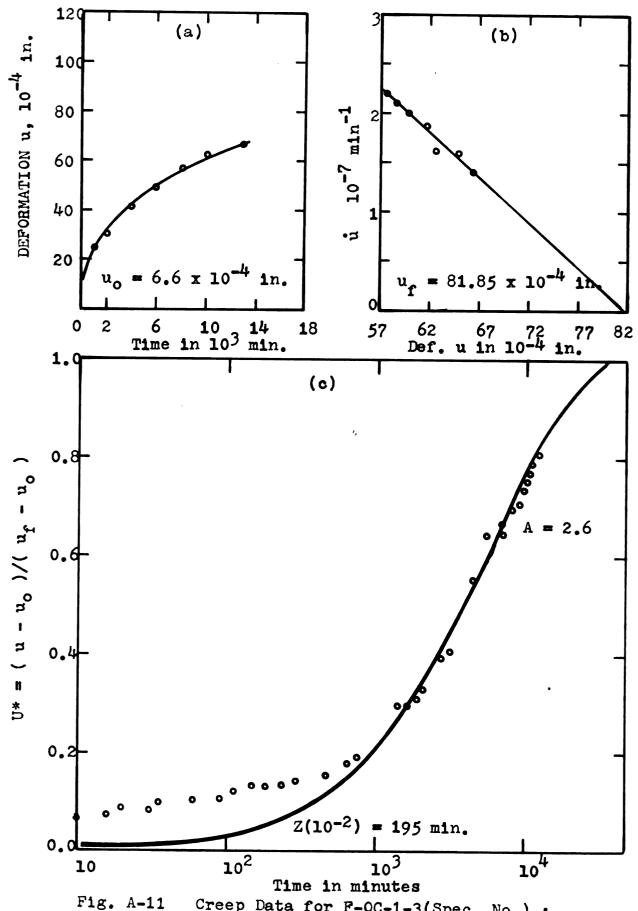


Fig. A-11 Creep Data for F-OC-1-3(Spec. No.): $D = 0.75 - 1.06 \text{ kg/cm}^2$

1

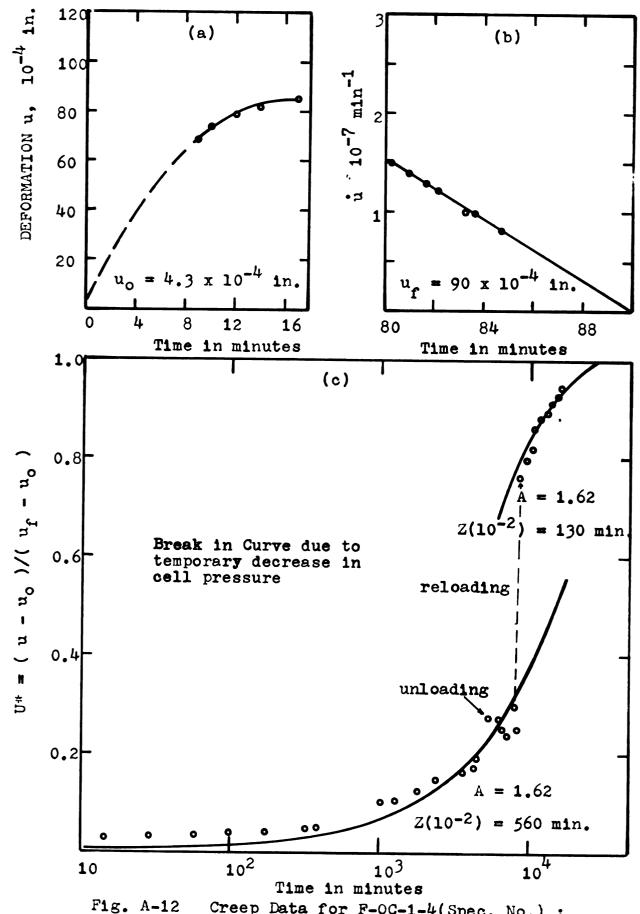
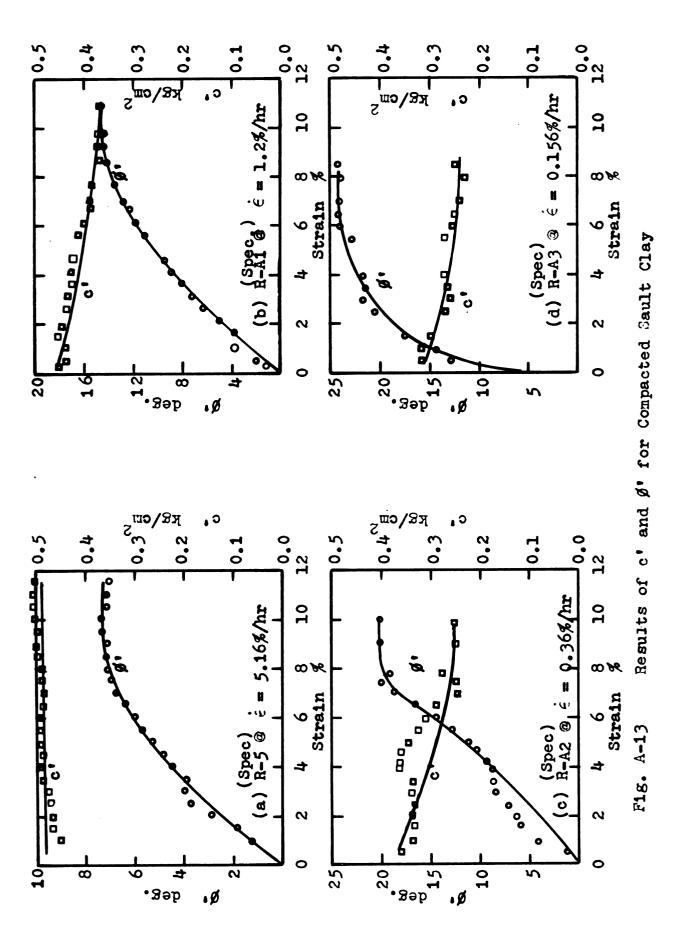
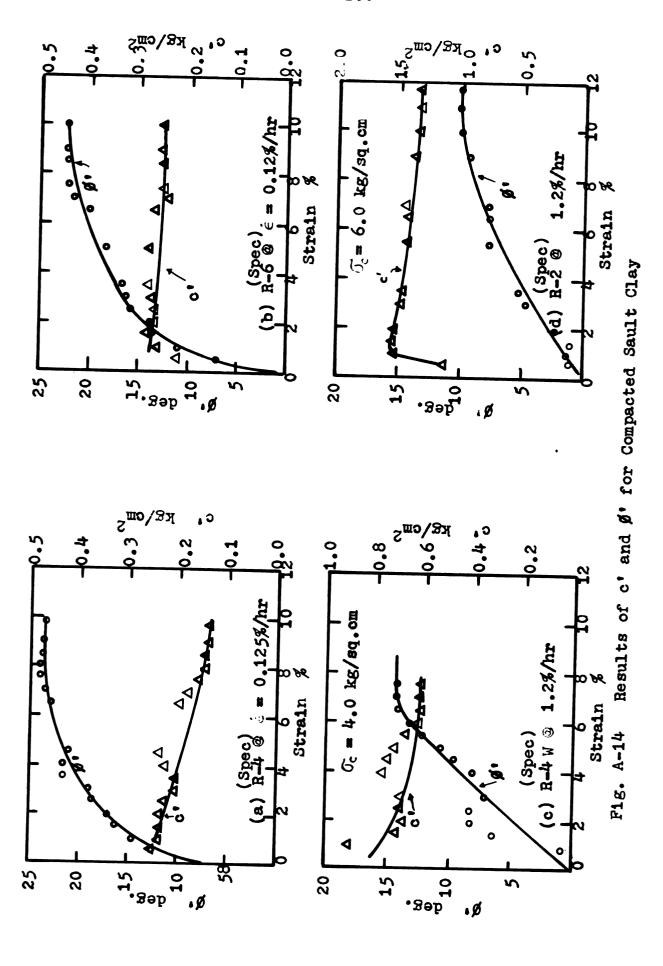
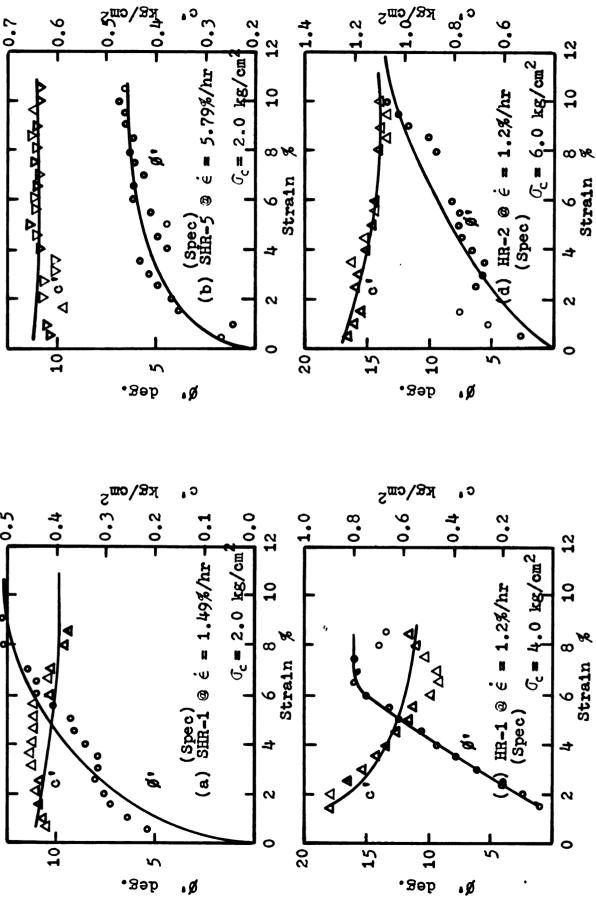


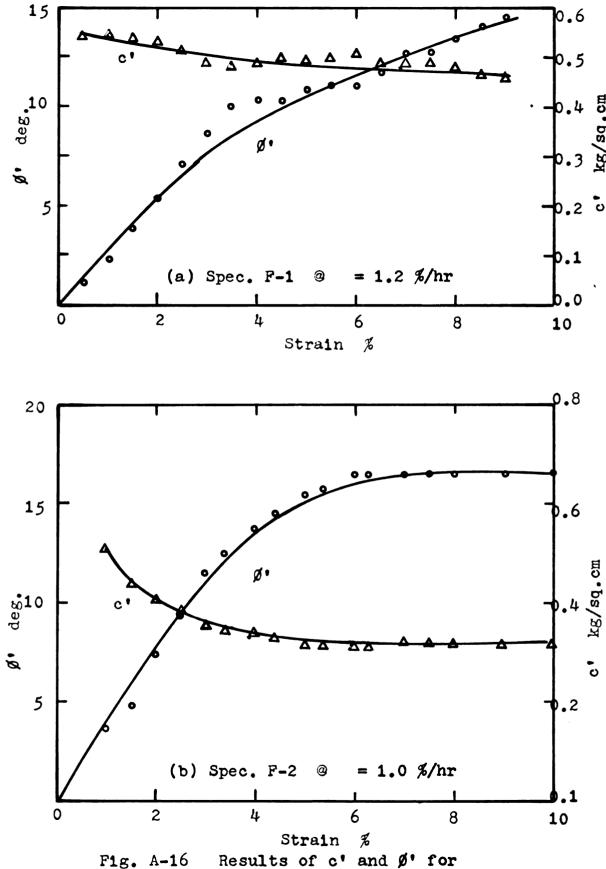
Fig. A-12 Creep Data for F-OC-1-4(Spec. No.): $D = 1.06 - 1.255 \text{ kg/cm}^2$







Results of c' and $ot \emptyset$ ' for Remolded Sault Clay F18. A-15



Results of c' and Ø' for Consolidated Sault Clay

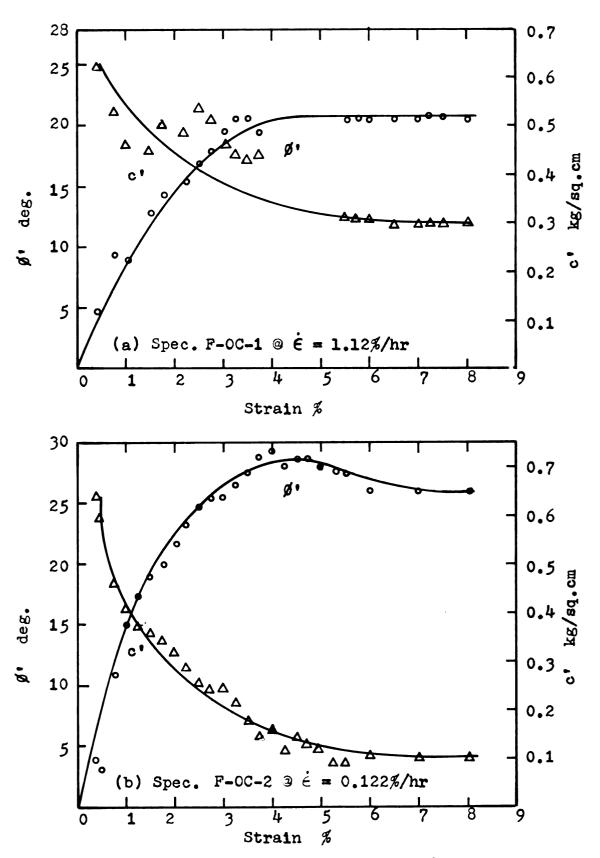


Fig. A-17 Results of c' and Ø' for Overconsolidated Sault Clay

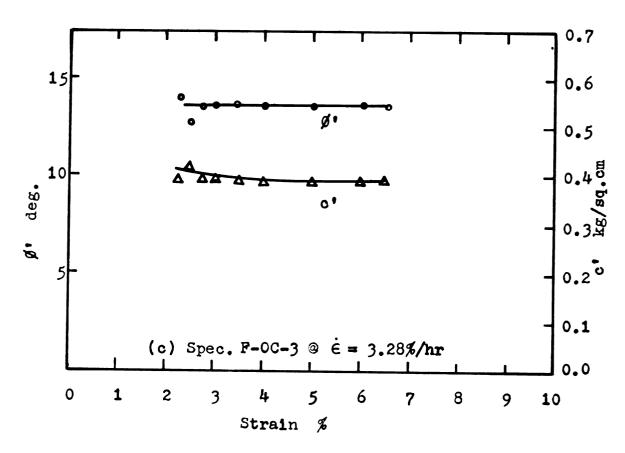
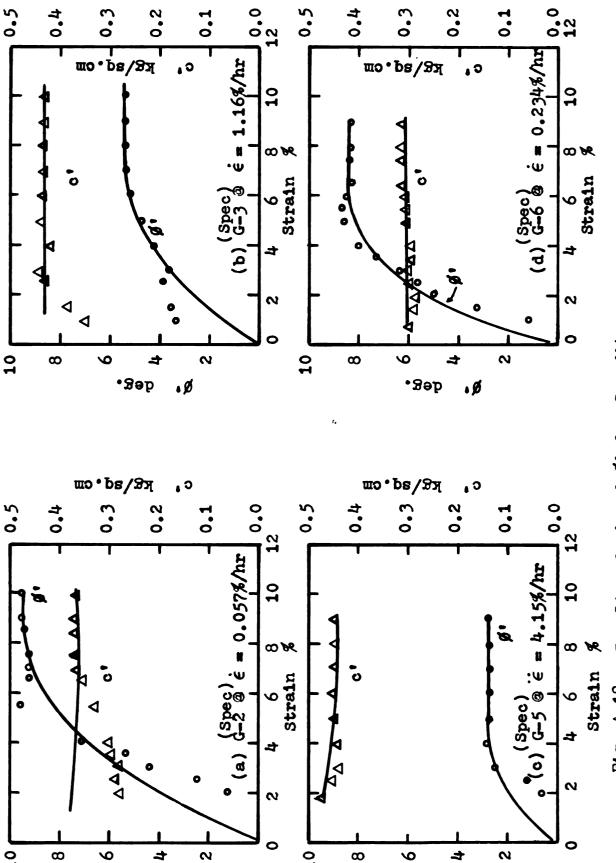


Fig. A-17(cont'd)



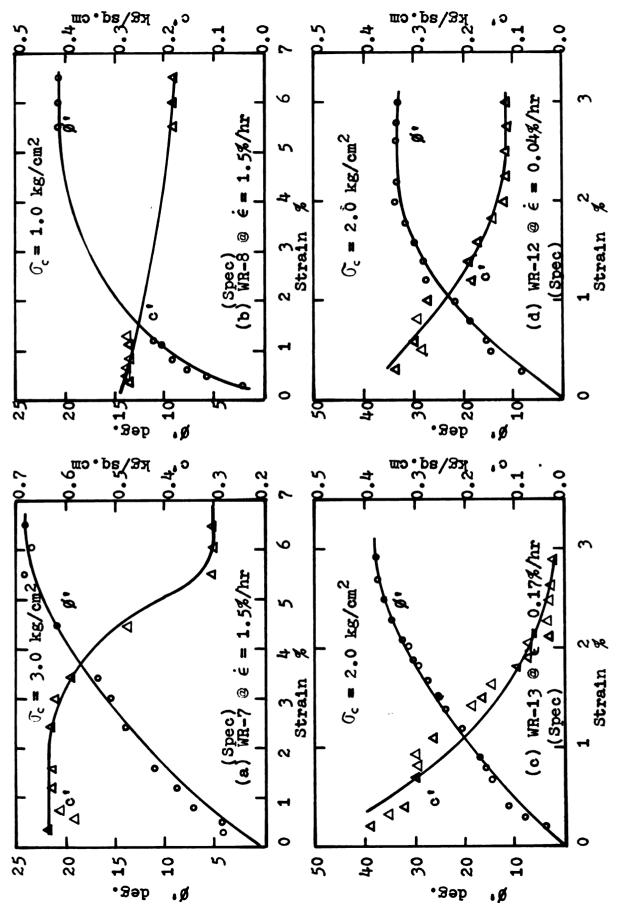
deg.

ø

deg.

ø

Fig. A-18 Results of c' and β ' for Grundite



Results of c' and \$' for Undisturbed Willow Run Clay F18. A-19

		,
		<i> </i>

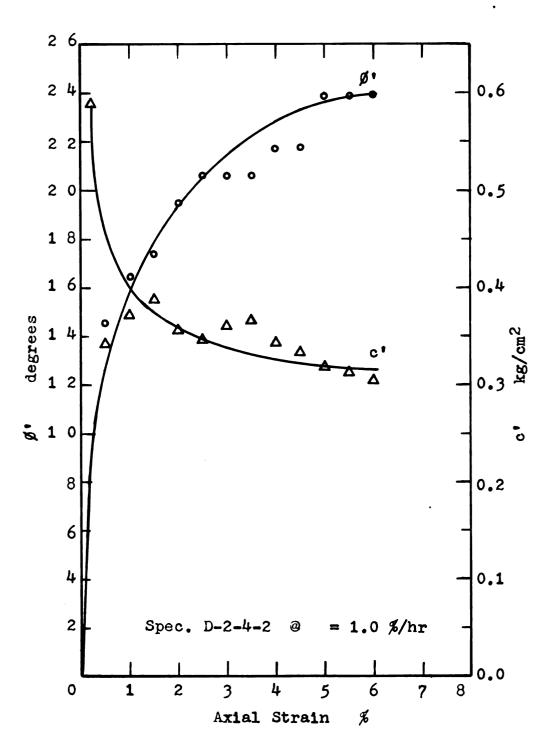


Fig. A-20 Results of c' and Ø' for Undisturbed Marine City Clay

APPENDIX III

TEST DATA

B. Tables

'
ı
· · · · · · · · · · · · · · · · · · ·
i
, t
i
i de la companya de
1
'
· · · · · · · · · · · · · · · · · · ·
!
l
ļ
i
1
ľ
<u>'</u>
•
1
1
,
1
İ
i

Table A-11 Results of CF3 Tests on Compacted Sault Clay

€		R-A3 (Spec.No)	R-A2 (Spec.No)	R-Al (Spec.No)	R-5 (Spec.No)
1.0	<pre></pre>	0.156 261.000 0.310 15.400	0.360 116.000 0.335 4.100	1.200 81.600 0.447 2.200	5.600 22.600 0.465 1.200
2.0	<pre>%/hr t min. c' kg/cm² ø' deg.</pre>	0.156 638.000 0.290 19.000	0.360 266.400 0.335 6.300	1.200 148.400 0.440 3.700	5.600 32.200 0.465 2.800
3.0	ė %/hr t min. c' kg/cm ² g' deg.	0.156 1020.000 0.275 21.600	0.360 453.400 0.340 8.250	1.200 201.900 0.430 6.300	5.600 43.900 0.470 4.000
4.0	é %/hr t min. c' kg/cm ² ø' deg.	0.15 6 1402.000 0.265 21.600	0.360 620.400 0.330 9.000	1.200 255.400 0.425 8.000	5.600 55.600 0.485 4.600
5.0	é %/hr t min. c' kg/cm ² ø deg.	0.156 1784.000 0.260 21.600	0.360 787.400 0.315 11.200	1.200 308.900 0.420 9.400	5.600 67.300 0.490 5.300
6.0	é %/hr t min. c' kg/cm ² ø' deg.	0.156 2174.000 0.245 24.000	0.360 954.400 0.300 14.500	1.200 358.900 0.410 11.000	5.600 78.900 0.490 6.000
7.0	<pre> %/hr t min. c' kg/cm² ø' deg.</pre>	0.156 2410.000 0.230 25.400	0.360 1089.400 0.290 18.400	1.200 408.900 0.390 12.600	5.600 90.500 0.485 6.500
8.0	<pre>%/hr t min. c' kg/cm² ø' deg.</pre>	0.156 2646.000 0.225 25.400	0.360 1189.400 0.280 19.000	1.200 458.900 0.375 13.600	5.600 101.900 0.490 7.000

1

Table A-12 Results of CFS Tests on Over-consolidated Sault Clay

ϵ		Ι	[
1		F-0C-1	F-0C-2	F-0C-3	
%		(Spec.No)	(Spec.No)	(Spec.No)	
1.0	<pre>%/hr t min. c' kg/cm² ø' deg.</pre>	1.120 53.500 0.540 8.800	0.122 491.500 0.420 15.200	3.280 7.600	
2.0	<pre></pre>	1.120 107.000 0. 445 14.800	0.122 983.000 0.285 22.400	3.280 0.420 11.400	
3.0	<pre> %/hr t min. c' kg/cm² ø' deg.</pre>	1.120 160.070 0.385 18.600	0.122 1475.000 0.206 26.500	3.280 0.394 13.580	
4.0	<pre></pre>	1.120 214.000 0.345 20.600	0.122 1965.00 0.160 28.500	3.280 0.394 13.580	
5.0	<pre>% %/hr t min. c' kg/cm² ø deg.</pre>	1.120 268.000 0.320 20.800	0.122 2460.000 0.125 28.500	3.280 0.394 13.58	·
6.0	<pre> %/hr t min. c' kg/cm² ø' deg.</pre>	1.120 321.500 0,310 20.80	0.122 2950.000 0,110 26.90	3.280 0.394 13.580	
7.0	<pre>% %/hr t min. c' kg/cm² ø' deg.</pre>	1.120 375.000 0.300 20.800	0.122 3440.000 0.100 26.000	3.280 0.394 13.580	
·					

Table A-13 Results of CFS Tests on Grundite

ϵ		0.2	C 2	C F	C 6
1 %		G-2	G-3	G-5	G=6
		(Spec.No)	(Spec.No)	(Spec.No)	(Spec.No)
2.0	€ %/hr t min.	0.057	1.160	4.150	0.234
j	t min.	2100.000	103.500 0.430	28.900 0.470	512.000 0.300
	ø' deg.	4.000	2.700	2.000	5.000
1					
3.0	<i>•</i> %/hr	0.057	1,160	4.150	0.234
	t min.	3160.000	155.000	43.400	769.000
	c kg/cm	0,3700		0.460	0.302
	Ø' deg.	5.600	3.700	2,500	6.370
4.0	ė %/hr	0.057	1.160	4.150	0.234
1.0	t min.	4220.000	207.000	57.800	1024.000
	c' kg/cm ²	0.370	0.430	0.455	0.299
	Ø' deg.	6.900	4.400	2.650	8.000
50	ė %/hr	0.057	1.160	4,150	0 004
5.0	t min.	0.057 5260.000	258.500	72,500	0.234 1280.000
	c' kg/cm ²	0.370	0.430	72.500 0. 445	0.304
	Ø' deg.	7.900	5.000	2.700	8.600
6.0	ė %/hr	0.057	1.160	4.150	0.234
	t min.	6320.000	31 0.000	86.800	1538.000
	c' kg/cm ²	0.370	0.430	0.445	0.310
	Ø' deg.	8.600	5.300	2.700	8.590
7.0	ė %/hr	0.057	1.160	4.150	0.234
	t min.	7380.000	362.000	101.200	1792.000
	c' kg/cm ²	0.370	0.430	0.445	0.317
	Ø' deg.	9.100	5.500	2.700	8.350
9.0	€ %/hr	0.057	1.160	4.150	0.234
	t min.	9480.000	465.000	130.000	2305.000
	c' kg/cm ²	0.370	0.430	0.445	0.317
	Ø' deg.	9.400	5.500	2.700	8.350
		1	1		l

Table A-14 Results of CFS Tests on Undisturbed Willow Run Clay

€ %		WR-7 (Spec.No)	WR-8 (Spec.No)	WR-13 (Spec.No)	WR-12 (Spec.No)
1.0	<pre>% %/hr t min. c' kg/cm² ø' deg.</pre>	1.500 40.000 0.635 7.200	1.500 40.000 0.265 9.750	0.170 353.000 0.210 18.500	0.040 1500.000 0.235 22.500
2.0	<pre>% %/hr t min. c' kg/cm² ø' deg.</pre>	1.500 80.000 0.630 12.25	1.500 80.000 0.240 15.000	0.170 706.000 0.065 32.000	0.040 3000.000 0.130 32.800
3.0	<pre> %/hr t min. c' kg/cm² ø' deg.</pre>	1.500 120.000 0.610 16.000	1.500 120.000 0.225 18.000	0.170 1059.000 0.020 38.000	0.040 4500.000 0.112 33.500
4.0	é %/hr t min. c' kg/cm ² ø' deg.	1.500 160.000 0.540 19.000	1.500 160.000 0.205 19.800		
5.0	<pre> %/hr t min. c' kg/cm² deg.</pre>	1.500 200.000 0.410 22.25	1.500 200.000 0.190 20.500		
6.0	<pre></pre>	1.500 240.000 0.305 24.000	1.500 240.000 0.180 20.750		

APPENDIX IV

SAMPLE CALCULATION--DETERMINATION OF PARAMETERS

IN RHEOLOGIC MODEL

Sample Calculation-Determination of Model Parameters

$$(\frac{1}{1} - \frac{1}{3})_{i} = 0.495 \text{ kg/cm}^{2}$$

 $(\frac{1}{1} - \frac{1}{3})_{f} = 0.7515 \text{ kg/cm}^{2}$
 $(\frac{1}{1} - \frac{1}{3})_{f} = 0.2565 \text{ kg/cm}^{2}$

From Fig. A-10 (a) & (b) in Appendix III:

$$u_0 = 4.9 \times 10^{-4} \text{ in.}$$

$$u_f = 24.25 \times 10^{-4} \text{ in.}$$

$$(k_1 + k_2) = \frac{24.25 \times 10^{-4} \text{ in.}}{3.00} \times 1_0$$

$$= \frac{0.2565}{3(4.9 \times 10^{-4})} \times 2.96 = 516 \text{ kg/cm}^{2}$$

$$k_{2} = \frac{2(1-3)}{3 \times u_{f}} \times 1_{o} = \frac{0.2565}{3 \times (24.25 \times 10^{-4})} \times 2.96$$

$$= 104.3 \text{ kg/cm}^{2}$$

$$k_1 = (\kappa_1 + \kappa_2) - \kappa_2 = 516 - 104.3 = 411.7 \text{ kg/cm}^2$$

$$\frac{k_1}{k_1 + k_2} = \frac{411.7}{515} = 0.798$$

From Fig A-10(c):

for A = 2.30, Z(t) =
$$10^{-2}$$
 t = 34.0 minutes

$$= \frac{3}{\sqrt{2}} \times \frac{A}{2(1-3)} \times \frac{k_1 + k_2}{k_1}$$

$$= \frac{3}{\sqrt{2}} \times \frac{2.30}{0.2555} \times \frac{1}{0.798} = 23.8 \text{ cm}^2/\text{kg}$$

$$= \frac{2Z(t) (k_1 + k_2)}{k_1 k_2} = \frac{2 \times 0.01 \times 516}{411.7 \times 104.3 \times 23.8 \times 34.0}$$

$$= 2.97 \times 10^{-7} \text{ min}^{-1}$$

APPENDIX V

ANALYSIS OF ${}^{\dagger}_{C}$ and ${}^{\dagger}_{C}$

			· İ
			!
			:
			1
			1
			<u> </u>
			1
			•
			!
			Ì

Friotion-Displacement Rate Characteristics at Failure-CFS Tests on Sault Clay Table A-15

CLAY	Test No.	ψ.	επ	tα	•n	/ = 0.3	P = 0.5 N= 6 ne
		%/hr	₹ર	hr	gep	ne/hr	ne/hr
Compacted	R-6	0.120	0.6	75.00	22.30	0.800	0.800
	H T	0.125	8.5	00°89	23.60	0.0883	0.0883
	B-A3	0.156	0.6	57.70	24.30	0,1040	0,1040
	R-A2	0,360	10.0	32.20	20.00	0.1862	0.1862
	R-A1	1.200	10.0	8.34	14.50	0.7200	0.7200
	R-5	2.600	11.0	1.97	7.20	3.050	3.050
Remolded	SHR-1	1.490	9.5	6.37	12.70	246.0	0.942
	SHR-5	5.790	11.0	1.90	6.50	3.160	3.160
Consol1da	F-1	1,200	0.6	7.50	14.60	0.800	008.0
	F-2	1.000	10.0	10.00	16.40	009.0	009*0
			•				

,
1
i
1
!
1

Table A-16 Predicted \$ 1n Creep-CFS Tests on Sault Clay

	Spec	৸	ů ¢	Observed	64		P = 0.3			,	P = 0.5	
ey.	o	@ end		- U	N C	N _o	. 64. 8 N = Nc	Predic #8	Nc	N _C	5 5 7 7 7 7 7 7 7 7 7 7	Predio
ເວ			hr	deg	ne	ne/hr	deg	gep	ne	ne/hr	ne/hr	deg
	FC5	1.96	6.0	00°0	2.05	4.100	6.20	3.04	06.0	1.800	10.0	06.4
	FC4	2.17	1.0	04.0	3.30	3.300	2.00	3.43	1.25	1.25	11.6	5.70
•	FC2	2.25	0.4	6.20	00.4	1.000	12.50	6.13	1.40	0.35	17.5	8.58
4 Tn e	PC6	2.14	4.0	3.75	3.15	0.789	13.80	6.75	1.20	0.30	18.1	8.87
95 P	FC9	2.10	6.5	5.10	2.75	0.423	16.60	8.15	1.13	0.1585	21.0	10.30
əte	FC8	1.96	8.0	2.40	2.05	0.256	18.90	9.25	06.0	0.1125	22.6	11.10
977 9	FC7	2.09	12.0	6.20	2.75	0.229	19.40	9.50	1.10	0.0916	23.5	11.50
suo	FC3	2.85	24.0	00.6	00.9	0.250	19.00	9.32	2.61	0.1087	22.8	11.20
၁	FC1	3.26	120.0	12.20	00.9	0.050	25.50	12.50	3.80	0.0316	25.5	12.50
	RC8	1.85	2.0	1.00	1.60	0.800	13.70	4.03	0.72	96.0	17.4	5.11
pəq	вс9	1.92	8.0	2.85	1.90	0.238	19.30	5.67	0.85	0.106	22.9	47.9
Compac	RC10 RC11 RC7	2.02 2.43 2.56	16.0 24.0 72.0	5.00 0.003 7.40	2.35 6.00 6.00	0.147 0.250 0.084	21.40 19.0 24.0	6.30 7.60 7.05	1.75	0.0625 0.073 0.032	24,9 14,3 25.5	7.32 7.15 7.50

Validity of Equation (25')

The exponents of the hyperbolic tangent in equation (25) for the following clays are evaluated as follows.

Compacted Sault Clay--For the range of constant strain rates in the CFS tests, from Table 5(a), it can be shown that $20 \text{ cm}^2/\text{kg} \quad '= \frac{1}{\text{oct}}/\frac{1}{\text{oct}} =

The hyperbolic tangent therefore approaches unity.

Over-consolidated Sault Clay--From Table 5(b),

For
$$\sqrt{2} \times 0.655 \% = 0.925 \% = 0.925 \times 10^{-2}$$
,

$$B = \frac{\sqrt{1 + 3.2}}{2} \times \frac{k_1}{2} = 2.25$$

$$A + B = 2.945$$

Therefore $\tanh 2.946 \stackrel{.}{=} 1$

Similarly, it can be shown that for the consolidated Sault clay, Grundite and undisturbed Marine City clay:

$$\tanh^{-1} \frac{1 - \frac{1}{\sqrt{1 + \frac{1}{\sqrt{2}}}}}{2} \stackrel{\cdot}{=} 3$$

Thus we have shown that Equation (25') can be used to compute the predicted values of cohesion for the series of CFS tests at the different constant strain rates on the compacted Sault, consolidated Sault, overconsolidated Sault, Grundite and undisturbed Marine City clays.

Table A-17 Predicted Values of Cohesion(f₁)-Compacted Sault Clay

					r		
			Spec. R-4	Spec. R-A3	Spec. R-A2	Spec. R-A2	Spec. R-5
Strain	Ė	%/hr	0.125	0.15	0.360	1.20	5.16
Rate	Yoot 10	-5 _{min} -1	2.94	3.78	8.50	28.30	122.00
$D = 0$ $\epsilon = 0$ $\beta = 1.8$	- 0.1081 95 x 10-7	7 min-1					·
	β'=	Yoot/B	151.000	194.50	448.000	1490.00	6420.0
d =21.	$3 \text{ cm}^2/\text{kg}$	f1=	0.268				0.44
€ = 0.	5 - 0.769 1081-0.29 03 x 10-7	7 min-1			_	_	
d =28.	$8 \text{ cm}^2/\text{kg}$	Yoct/B	196.00 0.209	245.00 0.22	565.00 0.246	1890.0 0.228	8160.0 0.339
€ =0.2	65-0.999 44-0.658 5 x 10-7	%		05 50	221 0	725 0	21/15 0
ø = 21.	3 om ² /kg	f ₁₌		95.50 0.249		0.345	3145.0 0.414
€=0. 65	9-1.24 8-1.43 4 x 10-7	min-1					
a =20.8	cm^2/kg	roct/B f ₁ =		27.80 0.196	64.00 0.236		
D=1.24 ε=1.43 β=19.4	-1.49 -2.55 8 x 10-7	kg/cm ²					
	$\beta' = 8 \text{ cm}^2/\text{kg}$	foct/B	15.10 0.250	18.70 0.266	43.50 0.327	145.0 0.41	625.0 5 0.520
D=1.49 €=2.55 β=15.9	-1.708 1 -3.85 5 x 10-7	kg/cm ² min-1	18.50	23.0	53.5	178.0	765.0
	•	roct/β f ₁ =		0.173		0.263	0.329
D=1.70 €=3.85	8-1.89 1	min ⁻¹	0.10)	0.17)	0,210	0,203	0, 129
	β'=	roct/B	7.07	8.85	20.4	68.2	292.0
d=50.9	cm ² /kg	f1=	0.053	0.058	0.074	0.0975	0.1268

Table A-18 Predicted Values of Cohesion(f₁) -- Over-consolidated Sault Clay

			Spec. F-0C-2	Spec. F-OC-1	Spec. F-0C-3
Strain	Ė	%/hr	0.122	1.120	3.280
Rate	Yout 10	-5 min-1	2.880	26.400	77.300
€ = 0.	.495 - 0.7515 .0 - 0.0822 .97 x 10-7	% min-1			
	_	roct/β	102.000	892.000	2600.000
Ø = 19	0.45 cm ² /kg	f ₁ =	0.226	0.317	0.362
	7515 - 1.06				
	.082 - 0.358 .545 x 10 ⁻⁷	% min ⁻¹			
	β ['] =	root/B	196.000	1710.00	5000.000
or = 19	.45 cm ² /kg	f ₁ =	0.310	0.421	0.477
$\epsilon = 0$.			7.380	64.500	188.000
o = 17	%7 cm ² /kg	f ₁ =	0.155	0.278	0.338

Table A-19 Predicted Values of Cohesion(f₁)--Grundite

			Spec.	Spec.	Spec.	Spec.
			G-2	G-6	G - 3	G-5
Strain	Ė	%/hr	0.057	0.234	1.160	4.150
Rate	Foct 10	-5min-	1.325	3.50	27.400	97.200
l	0 - 0.5		2			
β = 32	2 x 10-7 β'=			17.30	85.50	306.00
d = 12	2.2 cm ² /kg	g f ₁ =	0.326	0.352	0.483	0.587
D = 0	5 - 0.8	kg/cm ²				
$\epsilon = 2$.	.01 -2.93	%				
B = 2	$\beta' = 34 \times 10^{-7}$	roct/3	8.42	34.50	171.00	610.00
$\alpha = 12$	2.2 cm ² /kg	g f ₁ =	0.236	0.352	0.483	0.587
D = 0.	0 - 0.7	kg/cm ²				
€ = 0. B = 7.	$0 - 2.52$ 4×10^{-1}	%-1				
	β'=	Yoct/B	18.20	74.50	377.0	132.00
d= 20	0.7 cm ² /1	sg f ₁ =	0.299	0.245	0.322	0.384

Table A-20 Predicted Values of Cohesion(f₁)-Consolidated Sault Clat

			
 			Spec.
			F-2
Strain	ė	%/hr	1.000
Rate	roct 10	-5 min-1	23.500
D = 0.0	- 0.5	kg/cm ²	
€ = 0.0	0.57	%	
	4 x 10-7	min-1	
		τoct/β	435.000
d = 14.	0 cm ² /kg	f ₁ =	0.483
D = 0.5	0.75	lan /a-2	
	5 - 0.75		
	67 - 1.065	% min-1	
β = 8.7	'3 x 10-7		
	β ['] =	roct/B	270.000
$\alpha = 15.$	5 cm ² /kg	f ₁ =	0.405
D = 0.7	75 - 1.00	kg/cm ²	
	65 - 1.693		
B = 5.6	58 x 10-7	min-1	
	β'=	roct/B	415.000
め= 1.4	6 cm ² /kg	f ₁ =	0.460
D = 1.0	- 1.25	kg/cm ²	
€ = 1.6	93 - 2.66	%	
$\beta = 0.2$	214 x 10-7	min-1	
	β' =	_{roct} /β	11000.000
$\alpha = 67.$	35 cm ² /kg	f ₁ =	0.149

Table A-21 Predicted Values of Cohesion(f₁)-Undisturbed Marine City Clay

			Spec. D-2-4-2
Strain	Ė	%/hr	1.000
Rate	roct 10	-5 min-1	24.000
D=0.27 ϵ = 0.0 β = 0.9	- 0.703	kg/cm ² % min-1	
	β'=	Ϋ́oct/β	2660.000
α= 65 c	m ² /kg	f ₁ =	0.133
$\epsilon = 0.7$	x 10-7	% min-1	
o = 12		$\dot{\gamma}_{\text{oct}/\beta}$ $f_1 =$	510.000 0.580
$\epsilon = 2.2$	07 - 0.87 03 - 3.068 0 x 10-7	%	
d = 30	_	Ϋ́oct/β f1 =	1840.000 0.275
€ = 3.0	7 - 1.07 68 - 4.693 x 10-7	kg/cm ² % min-1	
d= 26		$\dot{r}_{\text{oct}}/\beta$ $f_1 =$	1270.000 0.303

Predicted Cohesion (4) on Sault Clay--Creep-CFS Tests Table A-22

$f^* = f_V f_0$	0.6540 0.6100 0.4460 0.3970 0.3760 0.3360 0.2665	0.5220 0.3040 0.1945 0.1290 0.0224
$p\left(-\frac{\kappa k_1^{-1}}{k_1^{-1}}k_2\right)\rfloor \cdot kg/cm^2$	0.287 0.268 0.196 0.1143 0.11475 0.1170	0.2320 0.1350 0.0865 0.0573 0.00996
$f_1 = \frac{1}{\sigma} \log \tanh \left[\frac{1}{2} \cos \frac{k_1 \cdot k_2}{k_1 + k_2} t + \tanh^{-1} \exp \left(-\frac{\omega k_1^{-1}}{k_1^{-1} + k_2} \right) \right] \cdot kg/cm^2 f^* = f/f_0$	F-C-9 (Spec. No.)	C-C-7 (Spec. No.) (2) (2) $(2 - c_3) = 0.987 \text{ kg/cm}^2$ $0 = 13.96 \text{ cm}^2/\text{kg} - 1$ (assumed) $\frac{k_1 \text{ k}}{8} = 0 \times 10^{-7} \text{ min} - 1$ (assumed) $\frac{k_1 \text{ k}}{1} = \frac{k_2}{2} = 3.11 \times 10^{-4} \text{ min}^{-1}$ $\tau = \sqrt{2}/3 (c_1 - c_3) = 0.465 \text{ kg/cm}^2$ $\exp(-\frac{v \text{ k}_1}{1} + \frac{v}{k_2}) = 0.002$ $f_0 = 0.445 \text{ kg/cm}^2$
Creep Time	30 60 240 390 480 720 1440	120 480 480 960 1140 4320
CIAY	Consolidated	Compacted

on co -- Creep-CFS Tests Table A-23 Influence of $\dot{\epsilon}$

Observed	kg/cm ²	694°0	0.422	0.372	0.376	0.356
ಕ್ಕ	kg/om ²	0.180-0.26	0.15-0.22\$	0.125-0.2	0.125-0.2	0.125-0.2
$\dot{\epsilon}_c$ at end	%/hr	0.1270	0.0535	0.0050	t/0€00°0	0.0010
Creep Time	te min	120	480	960	1440	4320
Sample No.		R-C-8	R-C-9	R-C-10	B-C-11	B-C-7
, ; ; ; ; ; ; ; ; ; ; ; ; ; ; ; ; ; ; ;	CLAI		41	lus2 i	pacte	Com

Values of c* in Creep-CFS Tests on Sault Clay Table A-24

$c^* = \frac{c^{1-c} - c^{1/c}}{c^{1-c} - c^{1/c}}$	1.000 0.950 0.296 0.560 0.0228 0.2180 0.1730	0.7860 0.4620 0.1170 0.1450 0.0690
c'o-c' kg/cm ²	0.2200	0.1450
c' - c' _z kg/cm ²	0.2200 0.2090 0.0650 0.1230 0.0770 0.0050 0.0480 0.0380	0.1140 0.0670 0.0170 0.0210 0.0010
c' kg/cm	0.3550	0.3550
с' kg/сm	0.5750 0.5640 0.4200 0.4320 0.3500 0.3930 0.3930	0.4690 0.4220 0.3720 0.3760 0.3560
Creep Time min.	30.00 60.00 240.00 240.00 390.00 480.00 720.00 1440.00	120.00 480.00 960.00 1440.00 4320.00
Clay	Consolidatee Sault	Compacted Sault

