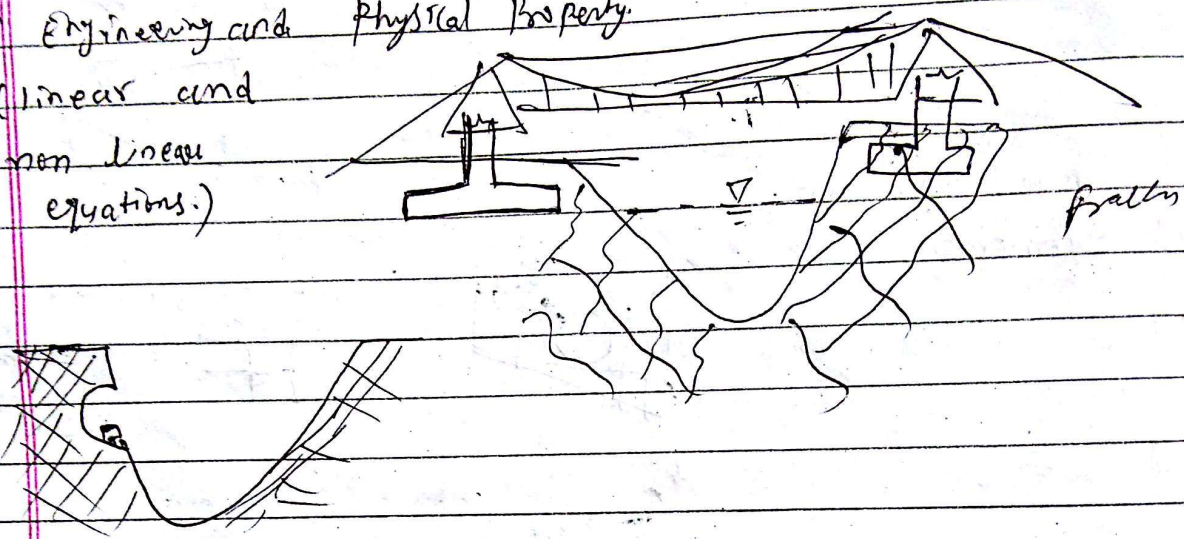


Rock Engineering

- Introduction:-
- Stereographic projection
- Engineering and Physical Property (linear and non linear equations.)



UCS - uniaxial compressive strength
 Indirect Methods of getting UCS

- determination of Tensile Strength
- Tri-axial Test

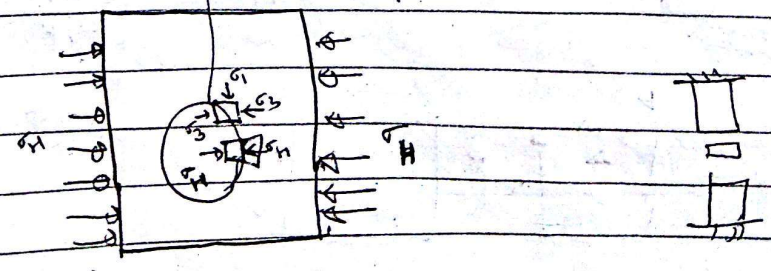
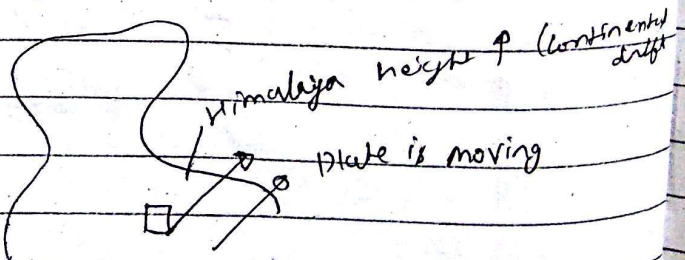


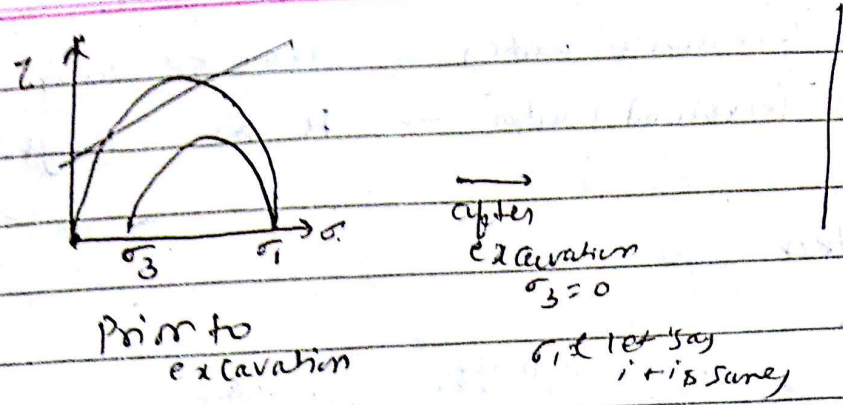
$$\sigma_c = f(\sigma_3)$$

$$\tau_f = f(\sigma_n)$$

→ Strength Criteria

Redistribution of stresses





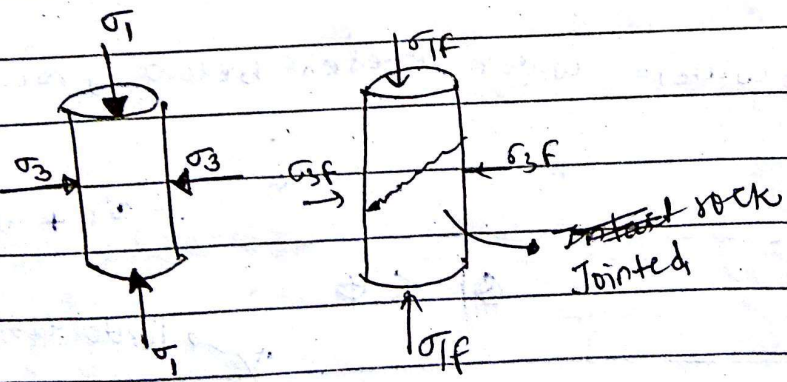
Engineering Behaviour of Rocks Mass :-

Soil Mass :-
 We tested soil mass specimens

Rock Mass :-
 Not possible to Rock mass specimens.

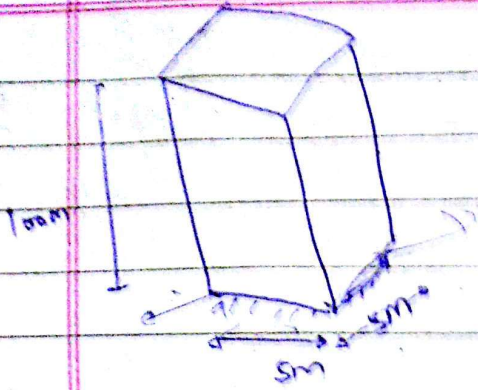
- ⇒ Start with intact rock
- ⇒ then scale it down.
- ⇒ Incorporate effect of fracture

Rock Mass = f (intact rock + fracture)



$\gamma = 26 \text{ kN/m}^3$

Page 20/21

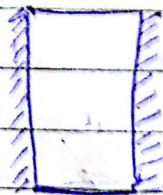


$E = 20000 \text{ Pa}$

$\nu = 0.2$

- ① Plot Variation of vertical stress due to self weight with depth
- ② Plot Variation of vertical strain and lateral strain with depth (consider base frictionless)
 - * Comment what will happen if base is not frictionless.

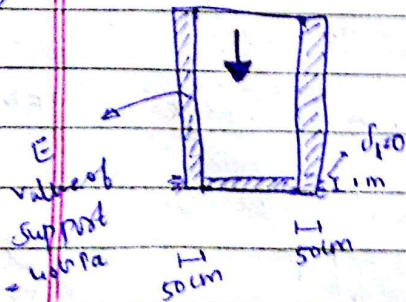
③



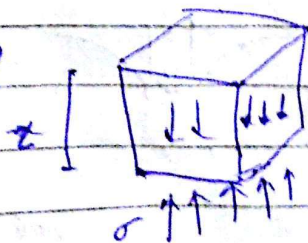
Rigid support in one direction. (frictionless)

* will there be any stress in the support (-Horizontal stress)

④

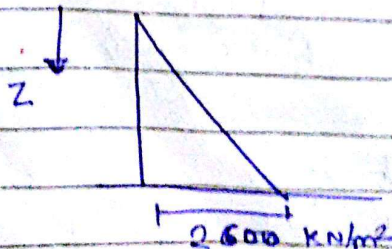


Answer :- ①



$\sigma_z (5 \times 5) = 26 (z) (5 \times 5)$

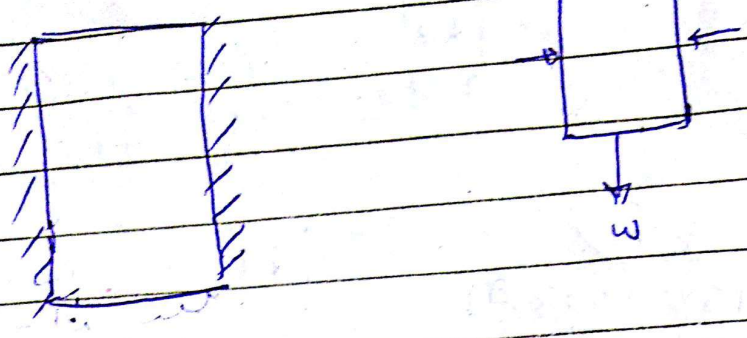
$\sigma_z = 26z$



Vertical stress due to self load.

depth z from Top.

Figure 3



Horizontal stress = $-v\sigma_z \Rightarrow -(0.2)(2.6\sigma_z) \Rightarrow -5.2\sigma_z$

Next Tut class?

A-4 Paper = 4-5

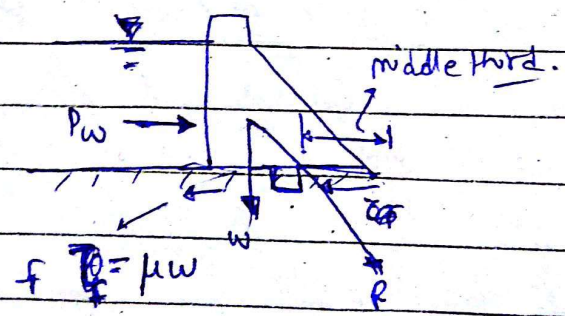
board pin, eraser, sharpener, pencil, pen

Field Applications:-

1960s \Rightarrow "Two failure"

Malpasset Dam }
Vajont Reservoir } \Rightarrow Rock Mech

① Large Dam



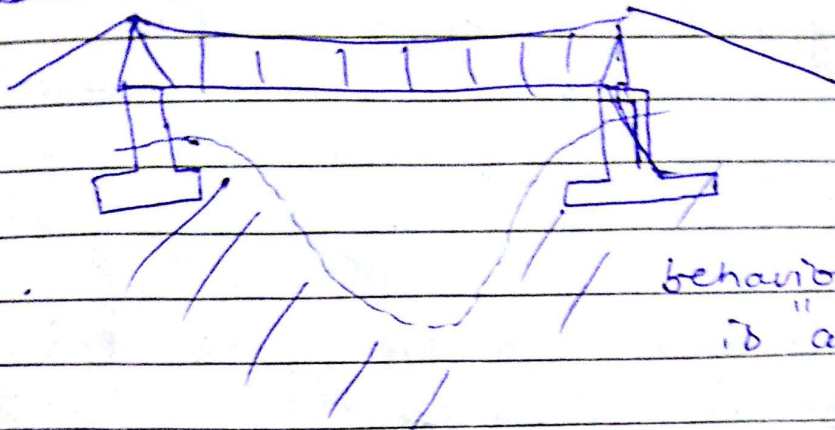
failure at:
Toppling
sliding

Discontinuity \Rightarrow Joints / fracture

Continuous joint is dangerous.

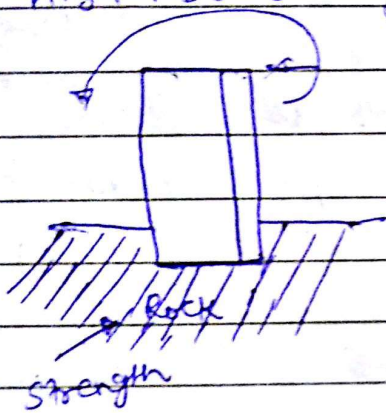
* Systematic and irregular joints.

② Large Bridges:-



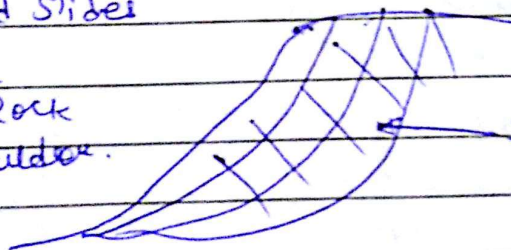
behaviour of mass is "anisotropic".

③ High Rise Buildings



④ Land Slides

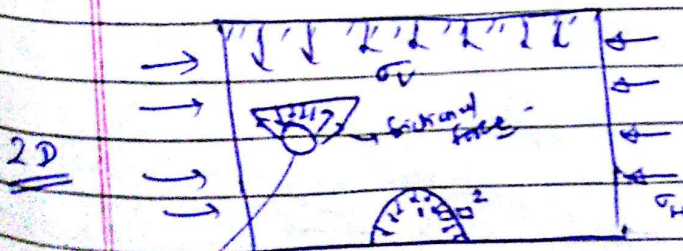
* Complex
= Soil + Rock
+ Boulder.



strength along the failure surface.

⑤ Tunnels

+ stress induced instability



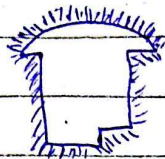
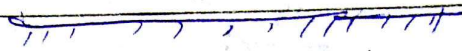
1+2 (equilibrium before excavation)

after excavation, 2 will be unstable.

Tunnel [frictional force, cave in due to ...]

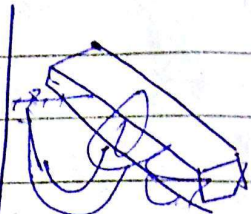
⑥ Underground Power Houses

[extremely difficult structure, difficult to construct]



* Structural Instability

analyze in 3D
→ size
→ not analysed in plane strain case



Tenzaghi theory does not valid at ends

* Tunnel is analysed through plane strain case.

⑦ Defence

① Underground utilities.

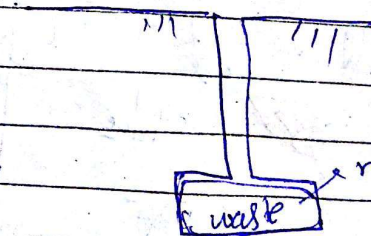
② Nuclear Power Plants

- very sensitive

- "waste"

- Nuclear waste disposal

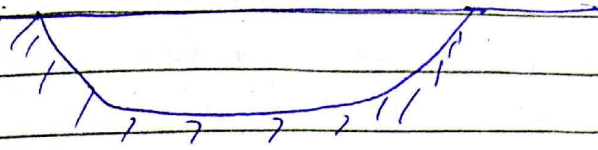
["dispose underground"]



⑧ Open Cast Mining

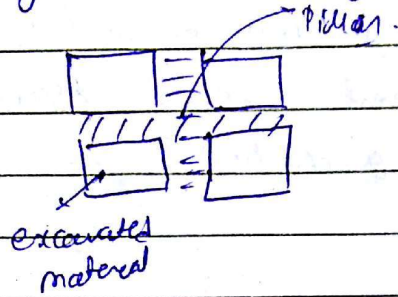
- Huge excavation (big pit)

* Stability of Slopes is concerned



* ^{CE} Slope Stability

(12) underground Mining :-



* mining slope (FOS = 1)

* failure with own weight

Definition

Rock Mechanics:- It is the theoretical and applied science of the mechanical behaviour of the rock. It is that branch of mechanics which is concerned with response of the rock to the ~~rock~~ ^{force} of its physical environment.

Rock:- It is a competent materials. It is a aggregate of minerals.

$$\sigma_{ci} > 1 \text{ MPa}$$

if $\sigma_{ci} > 1 \text{ MPa}$ then it is rock.

* High strength and low strain than soil

17/1/17

Rock Engineering

Date: ___/___/___

Definitions / Terminology

① Rock Mechanics

② Rock

③ Micro fissure - Very fine crack in the rock.

Always exist

in rock at

boundary of Crystals.

width $< 10^{-6}$ m
width $< 1 \mu$ (0.001 mm)
length ≈ 1 to 2 Crystals

④ Micro fracture

May exist

in intact rock

Little greater than Micro fissure

width ≈ 0.01 mm

Not visible to naked eye

⑤ Micro fracture

wider than ≈ 0.1 mm

Very large length (tens of meters)

Exist in Rock Mass

not in Intact Rock

⑥ Intact Rock :-

Individual Rock specimen free from any flaw (discontinuity)

⑦ Fault :- It is a fracture in the rock along which, displacements have already taken place. Displacements ~~for~~ may be very large, there is a zone where large displacements might occur in the

Past. It is very difficult to ~~post~~ measure the ~~measure~~ width of the fault zone. Because of large displacements, material is crushed in walls, and sinks inside.

Sticker Side

Gouge Material is formed. Shear strength along this plane will be very low depending on the amount of displacements, and type of gouge material formed.

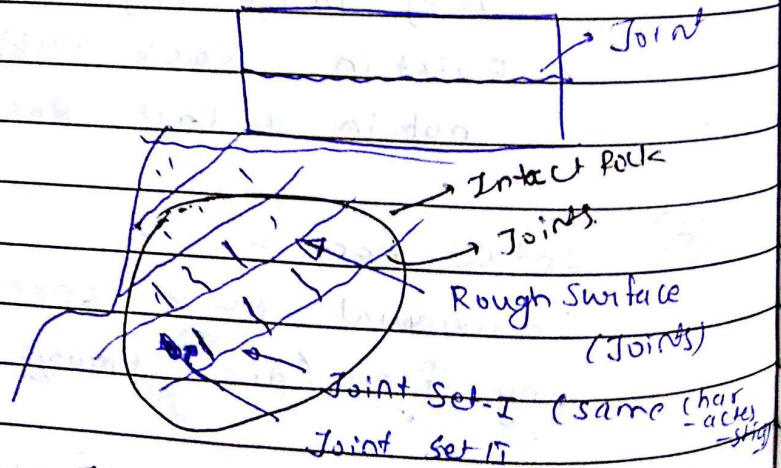
powdered material of crushed rock.

- * large displacement → gouge material fine
- * small displacement → gouge material coarse

⑧ Joint :-

It is a fracture along which there has been no displacements or almost nil displacement.

* Generally joints will be occurring in sets. A number of sets will form joint system.



⑨ Rock Mass Joint System = 2 Joint Sets + Random Joints.

Large mass where blocks of intact rock are separated by discontinuity.

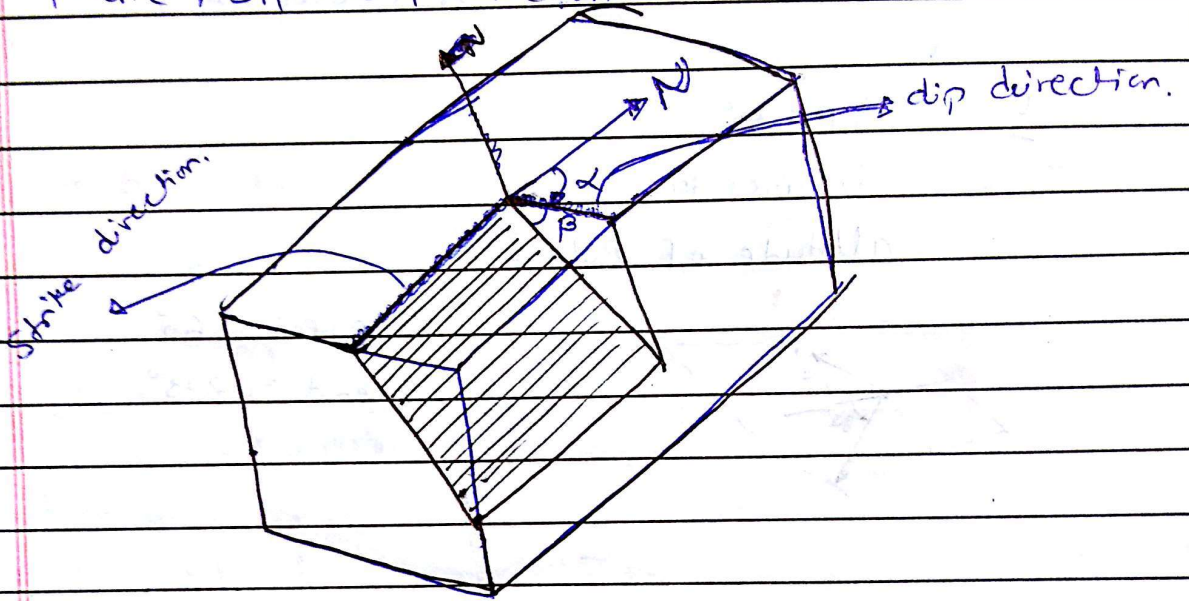
Large mass (intact rock + joints)

Stereographic Projections:-

Joints/"plane" (only angles will be considered)

How to present - Planer features graphically.

Plane; Dip; Dip direction \rightarrow Altitude of a plane.



Angle in horizontal plane \Rightarrow w.r.t N (WCB)
(Whole Circle Bearing)

$\beta = \underline{\text{Dip}} \Rightarrow$ Angle the plane makes with Horizontal Plane.
 \times Slope of line of steepest gradient.

\times Dip direction:- Angle to which Horizontal Plane is making w.r.t North
 \times line \perp to strike direction.

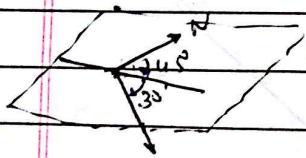
Strike direction - Common line b/w Horizontal plane and fracture plane.

Pole is "Normal to the plane"

Plunge - Angle of Pole with respect to Horizontal

Trend - w.r.t North
 * Plan of plunge line w.r.t North direction

Q. dip = 30°
 dip direction = 45°
 altitude of Pole =



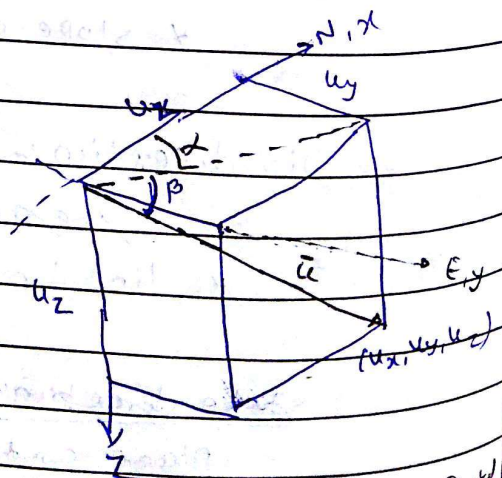
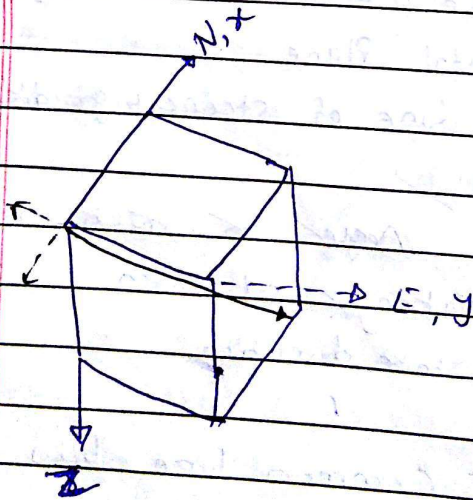
Plunge = 60°

Trend = 225°

Mathematics Representation:-

Planes/Lines \Rightarrow Angles

3D :- Plane :- Altitude $\rightarrow xy^\circ / ABC^\circ$
 $= 05 / 010^\circ$



clockwise direction

Trend is α

Plunge is β

\rightarrow measured in horizontal plane
 \rightarrow measured in vertical plane.



Trend $\Rightarrow \tan \alpha = \frac{u_y}{u_x}$, $\alpha = \tan^{-1} \left(\frac{u_y}{u_x} \right) + \text{Q}$

If $\alpha = 0-90^\circ$; $u_x \rightarrow +ve$, $u_y \rightarrow +ve$, $\text{Q} = 0$

If $\alpha = 90-180^\circ$, $u_x = -ve$, $u_y = +ve$, $\text{Q} = +180^\circ$

If $\alpha = 180-270^\circ$, $u_x = -ve$, $u_y = -ve$, $\text{Q} = 180^\circ$

If $\alpha = 270-360^\circ$, $u_x = +ve$, $u_y = -ve$, $\text{Q} = 360^\circ$

plunge $\Rightarrow \tan \beta = \frac{u_z}{\sqrt{u_x^2 + u_y^2}} \Rightarrow \beta = \tan^{-1} \left(\frac{u_z}{\sqrt{u_x^2 + u_y^2}} \right)$

$u_x = |\vec{u}| \cos \beta \cos \alpha$, $u_y = |\vec{u}| \cos \beta \sin \alpha$

$u_z = |\vec{u}| \sin \beta$

direction cosine: $l = \frac{u_x}{|\vec{u}|} \Rightarrow \cos \beta \cos \alpha$

$m = \frac{u_y}{|\vec{u}|} \Rightarrow \cos \beta \sin \alpha$

$n = \frac{u_z}{|\vec{u}|} \Rightarrow \sin \beta$

\downarrow β is always +ve

Consider always, n to be +ve i.e. β is always +ve

direction Ratio $\Rightarrow l : m : n$

If it is given $1 : -2 : -3$

then consider it as $-1 : 2 : 3$ ✓

Example

Plane altitude $\Rightarrow 30^\circ / 110^\circ$

Normal $\Rightarrow 60^\circ / 290^\circ$

Example Normal to the plane is having direction ratios $1, -2, 3$.

\Rightarrow $1, 2, 3$

$$\sin \beta = \frac{3}{\sqrt{1^2+2^2+3^2}} = \frac{3}{\sqrt{14}}, \quad \beta = \sin^{-1}\left(\frac{3}{\sqrt{14}}\right)$$

si

Stereographic Projection

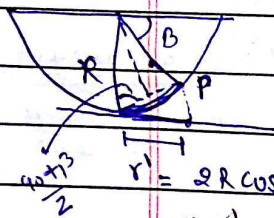
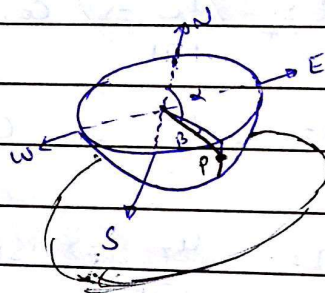
Graphical presentation of angles (No distances)

equal area projection

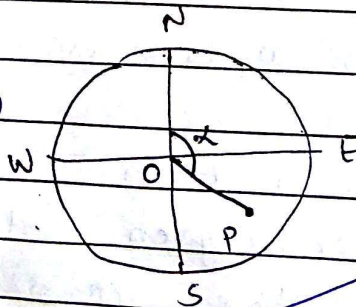
Line lower Hemisphere

assumption

- every line is passing through center



$r' = 2R \cos\left(\frac{90+\beta}{2}\right)$
= chord distance



'OP' is measure of 'beta'

$OP = r' = 2R \cos\left(\frac{90+\beta}{2}\right)$

$r'' =$ Just to make simple r' is divided by $\sqrt{2}$

for $\beta = 0$

$r' = \sqrt{2} R$

$r'' = \sqrt{2} R \cos\left(\frac{90+\beta}{2}\right)$

So, they divided by $\sqrt{2}$.

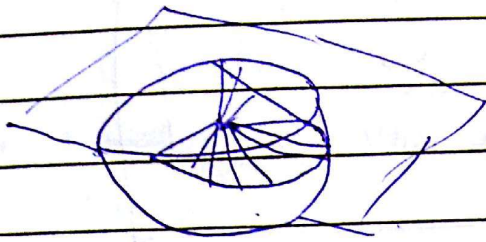
finally

$r'' = \sqrt{2} R \cos\left(\frac{90+\beta}{2}\right)$

After equal area projection

Plane

assumption:- every plane is passing through center



Intersecting Plane will be a circle.

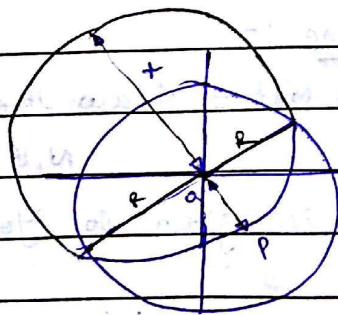
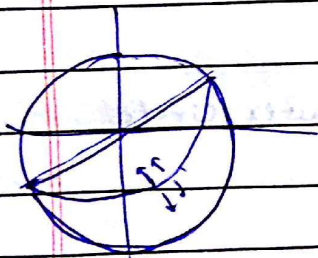
Intersecting plane will have many number of line passing through center.

If plane attitude

is having $\alpha = \text{dip direction}$
 $\beta = \text{dip}$

Trend of line of steepest slope = $\alpha = \text{dip direction}$,
Plunge of line of steepest slope = $\beta = \text{dip}$

A stereographic projection of intersecting circle will be also be a part of circle.



$$2 \left[R \frac{\sqrt{2} \cos(90 + \beta)}{2} \right] = R(R)$$

$$\Rightarrow x = \frac{R}{\sqrt{2} \cos(90 + \beta)}$$

Radius of bigger circle \Rightarrow $\frac{R + R \sqrt{2} \cos(90 + \beta)}{2}$

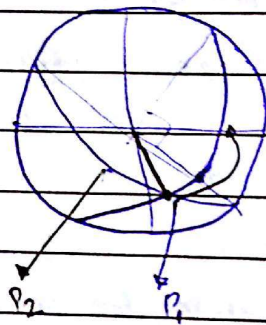
$$\text{radius} = \frac{2R + 2R \cos^2\left(\frac{90+\beta}{2}\right)}{2\sqrt{2} \cos\left(\frac{90+\beta}{2}\right)}$$

$$R_0 = R \left[\frac{1 + 2 \cos^2\left(\frac{90+\beta}{2}\right)}{2\sqrt{2} \cos\left(\frac{90+\beta}{2}\right)} \right]$$

Example P_1 - plane altitude 30/120

$$R = 3 \text{ cm}$$

P_2 - plane altitude = 45/240



To draw Plane :-

(1) Stereo net \Rightarrow draw the outer circle
N, E, S, W

(2) β/α is given to you

(3) Mark point 'P' at angle α to the north direction.

(4) Rotate the circle and bring the point 'P' either in East or West direction.

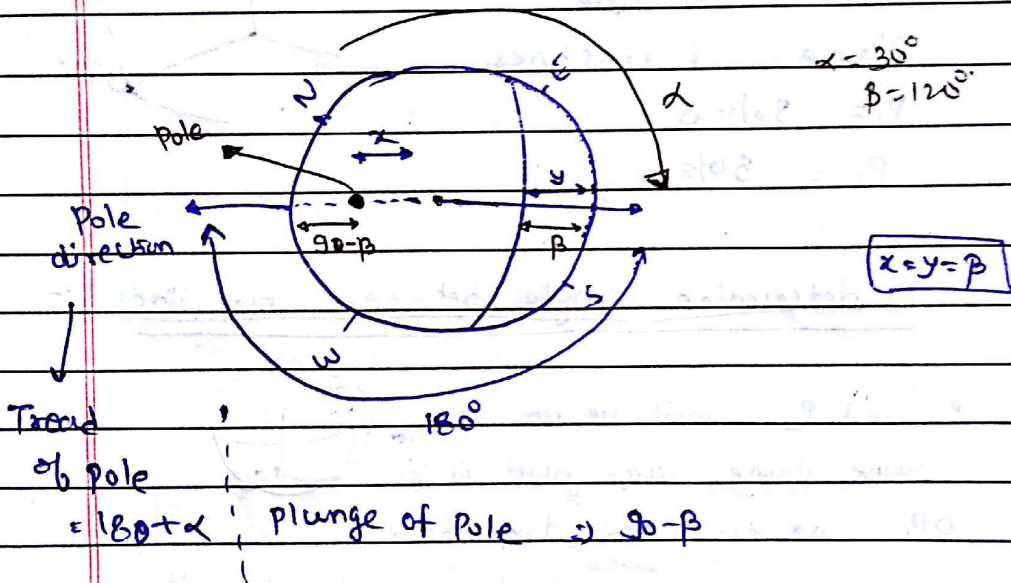
Because we can measure dip only in East and West direction. from periphery.

α Pole is the normal to the Plane.

To draw Pole

suppose attitude of Plane $30/120$
 attitude of Pole $\rightarrow 60/300$

- (1) Mark the Point 'P' at 300° .
- (2) Rotate the point 'P' and bring the point 'P' either on 'East or west direction', mark the point corresponding to dip.
- (3) Measure dip from periphery.



To get line of Intersection of Plane :-

Two Planes are given

$30/120$
 $50/300$

plot two planes on stereonet and marked the intersection between these two planes. you will get a point that will be a line.

- * Mark the pole and rotate the P_1 and P_2 in such a way that both lies on great circle.
- * So pole of that great circle will be the line of intersecting.

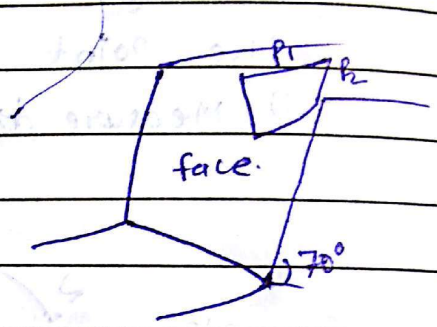
Problem :-

Slope face = $70^\circ/135^\circ$

P_1 and P_2 are plane of fractures.

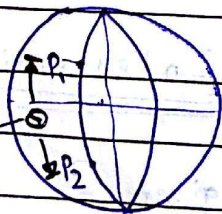
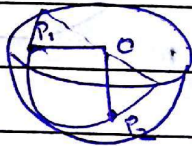
$P_1 = 30^\circ/120^\circ$

$P_2 = 50^\circ/200^\circ$



To determine angle between two lines :-

P_1 and P_2 will lie on same plane (same great circle)
 OP_1 and OP_2 are two lines

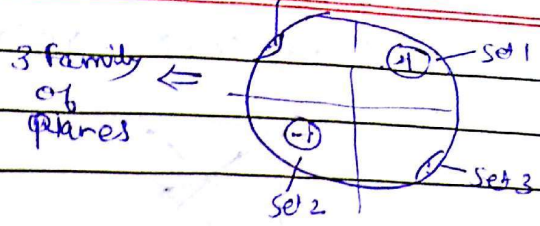


angular distance 'O' - this will be the angle b/w lines

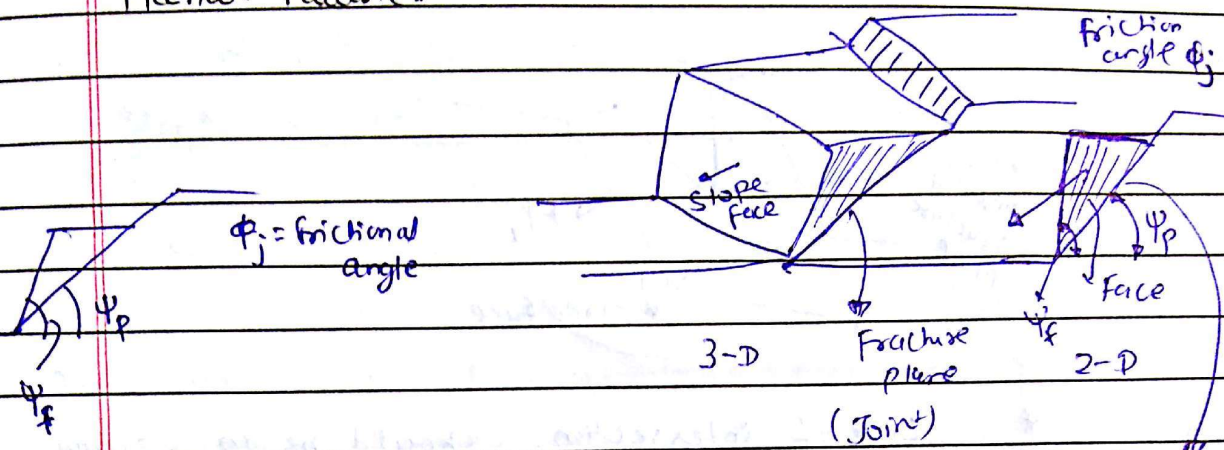
How to get Potential Failure mode in field :-

Field :- assume field attitude $30^\circ/120^\circ$

Pole - ...



Planar Failure:-



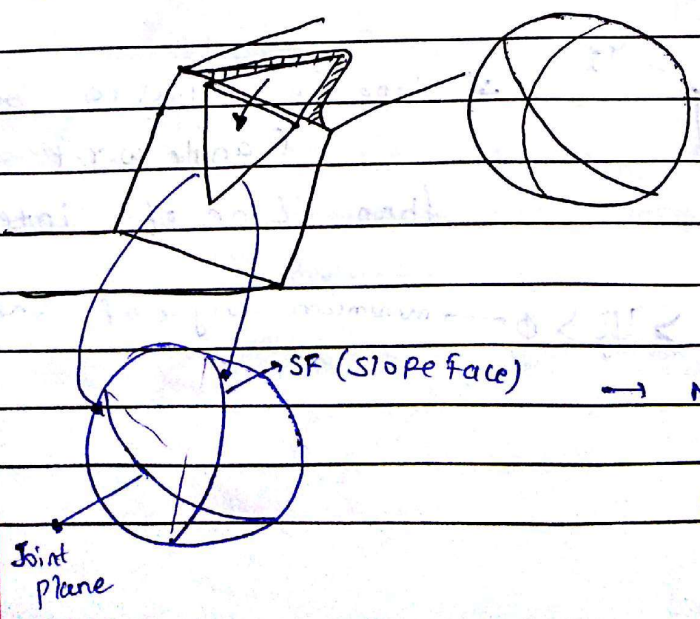
IF $\psi_f > \psi_p > \phi_j$ [for Planar Failure] (Joint plane)

* (slope face should be in same direction as Joint plane)

kinematic analysis of Rock Slope:-

(Potential Failure Mode)

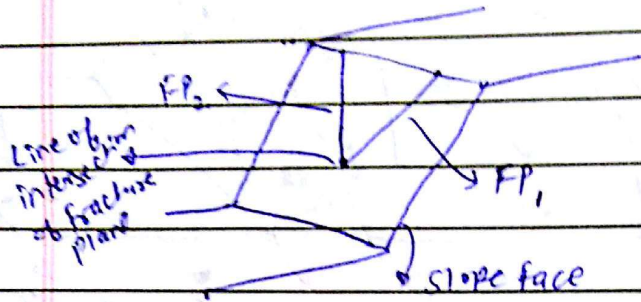
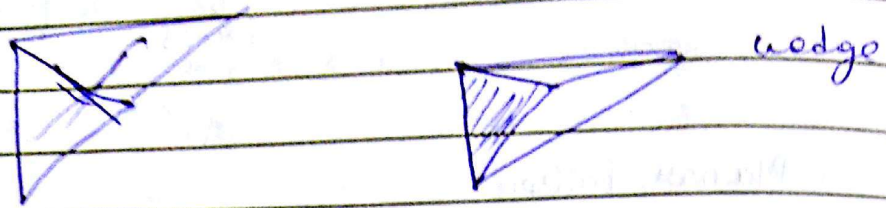
- a) Plane Failure
- b) Wedge Failure
- c) Toppling failure
- d) circular failure



→ No Planar failure can possible (not in same direction) (not in same dipping)

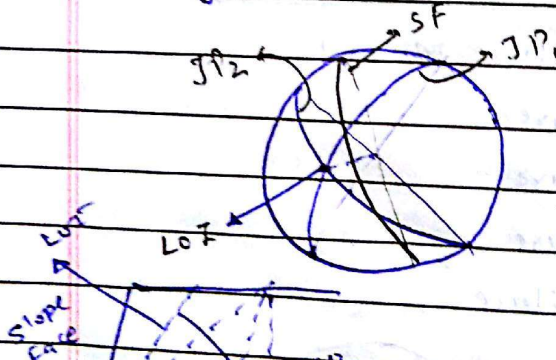
b) wedge failure

Above case represent wedge failure

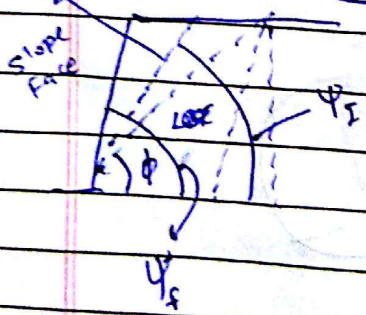


* Line of intersection should be in same direction as of slope face. (Trend should be same)

* Line of intersection should be steeper than ϕ_j



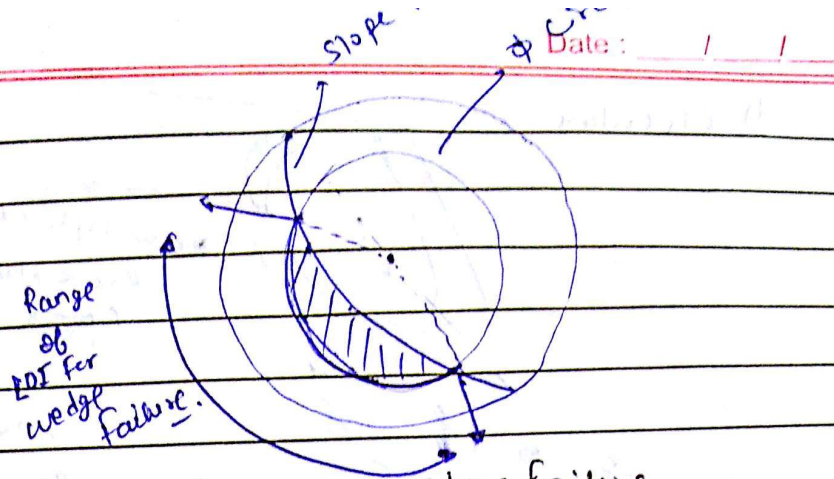
Line of intersection and slope face is having same trend. (Most Ideal case)



* Slope face has to be steeper (angle w.r.t Horizontal) than line of intersection.

$$\psi_f > \psi_i > \phi \rightarrow \text{minimum angle of line of intersection}$$

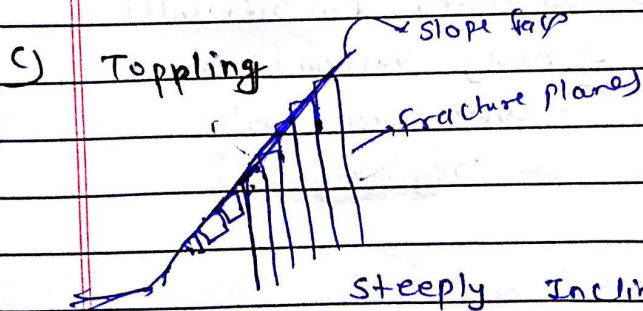
Example



Steps for determining wedge failure

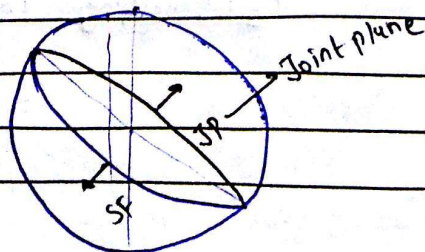
- 1) Draw slope face
- 2) Measure distance ϕ (Plunge) and draw a circle passing through that point.
- 3) Get the ~~line of intersection~~ Shaded area. (area b/w ϕ circle and slope face)
- 4) Get the line of intersection and if this line lies in the shaded area then there will be possibility of wedge failure. [line of intersection of two joint faces]

$\psi_F > \psi_J > \phi$ for wedge failure



steeply inclined joint planes towards the slope face causes toppling.

Case of Toppling

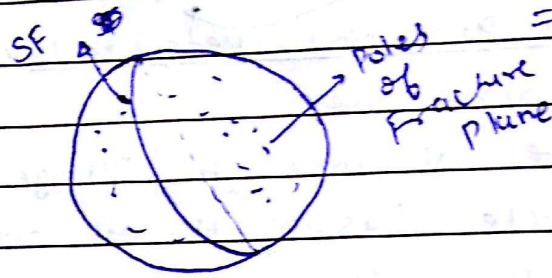


(Joint plane trend will be somewhat opposite to slope face but having more dip angle than slope face.)

d) Circular



heavily fractured
Rock mass
(~~also~~) | fracture planes
in every direction
(no family of fractures)
=> Random fracture.



Intact Rocks :-

1. Physical and engineering Properties :-

Index Properties :-

Soils \rightarrow Porosity / Void Ratio

Void index = $\frac{\text{wt. of water absorbed}}{\text{wt. of dry rock}}$

- * SP is saturated (v. difficult)

Wipe
Surface wt

- Apply vacuum / boil

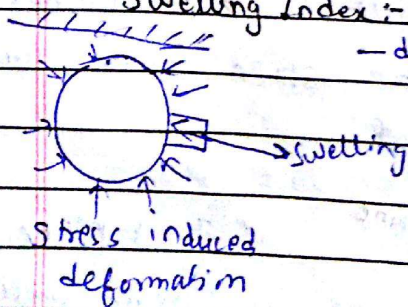
- over 12hr $- 105^{\circ}\text{C}$

- Dry weight

Swelling Index :-

- due to presence of clay mineral

tunnel



E_s (Swelling index) = $\frac{\Delta l}{l}$

= $\frac{\text{change in length}}{\text{original dimension}}$

Hydraulic Permeability:-

depends on
Fractures,
Interconnected
fractures

$$q = k i A$$

$$q = \frac{k}{\mu} \frac{dh}{dx} \cdot A$$

Hydraulic Permeability (conductivity)
(Properties of medium and fluid)

=> Pressure Development phenomenon

$$q = \frac{k}{\mu} \frac{dh}{dx} \cdot A$$

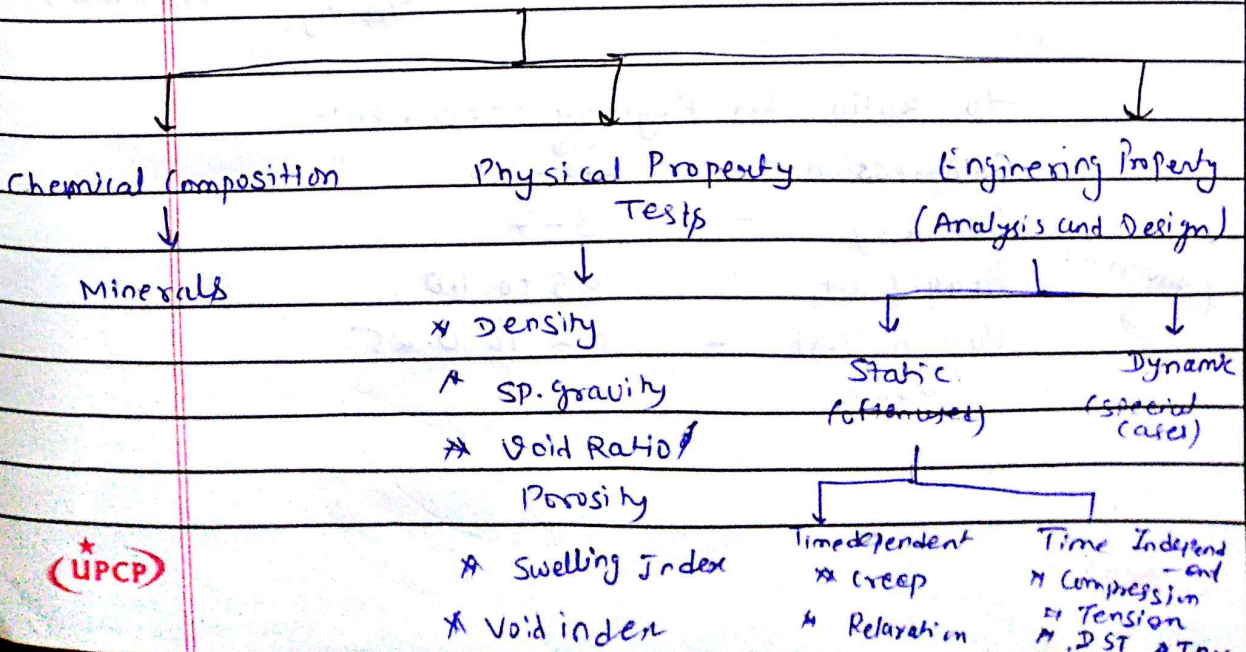
$$q = \frac{k}{\mu} \frac{dh}{dx} \cdot A$$

where k = Hydraulic Permeability (m^2)
 ↓
 Viscosity (Permeant) ($N \cdot s / m^2$)
 (Fluid Property) ↓
 (Properties of medium)

$$1 \text{ Darcy} = 9.86 \times 10^{-9} \text{ cm}^2$$

for Hydraulic Permeability

Lab Tests

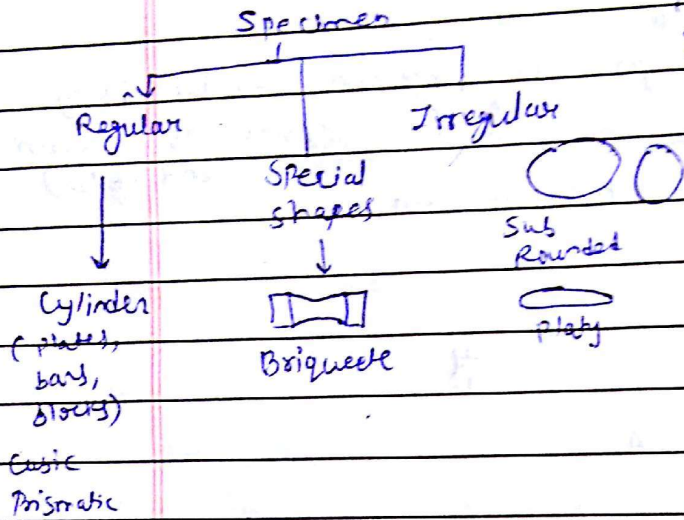


Date: 02 02 2017

Specimen:-

- * Definite shape
- * It may be of Irregular

Shape



Sample:-

- * Statistical quantity
- Rocks :- wide scatter in results.

Preparation

= STRINGENT

For getting result we need to use Probabilistic approach

(Rocks is uncertain material) (2)

Drilling Sizes:-

EX - 7/8" ≈ 23mm

AX - 1 1/8" ≈ 28mm

BX - 1 5/8" ≈ 41mm

NX - 2 1/8" ≈ 50mm → for projects

(minimum dia of specimen)

while Testing ←

L/D Ratio for Regular Specimen:-

Compression - ^{L/D} 2.0 - 3.0

Bending - 3 - 7

(circular) (2) → Bragiliari - 0.5 to 1.0

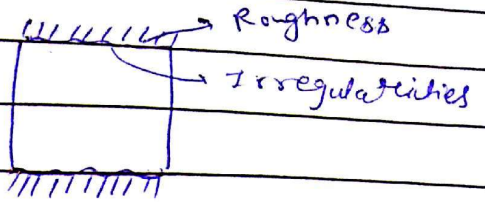
Punch Test (disc) - 0.2 to 0.25

Tolerance limits :-

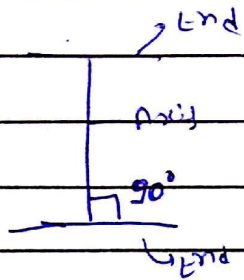
Compression Test :-

(a) ^{first} Tolerance limit ends should be

Smooth to 0.09 mm or $0.002 D$
whichever is minimum

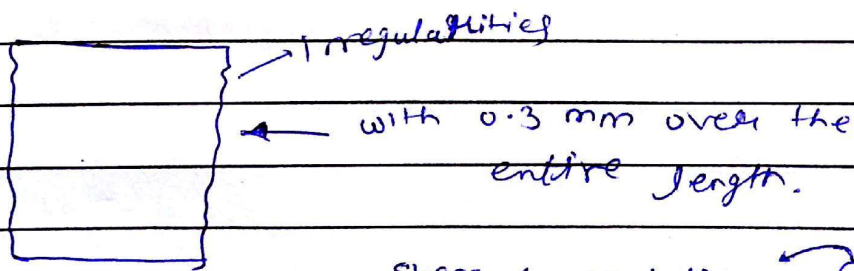


(b)



deviation in angle should be less than 0.001 radian
It should be perfectly 90° if not then load will ~~not~~ not passing through axis. Some eccentricity will comes in the column so moment and bending will occur.

(c)



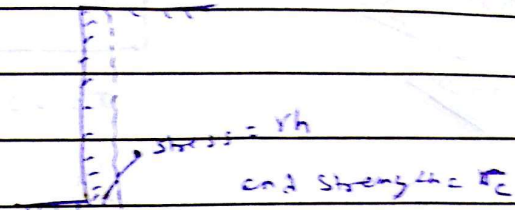
Stress concentration will occur due to irregularities resulting premature failure.

Intact Rock Properties:-

UCS:-

Probabilistic analysis:-

when $\gamma_h < \sigma_c$ (safe)



assume γ_h value is 100% correct

σ_c is having some scattering. Inst (or safe)

* So, what is the Probability that $\sigma_c < \gamma_h$?
(Probabilistic analysis)

Tests:-

Same lithological unit

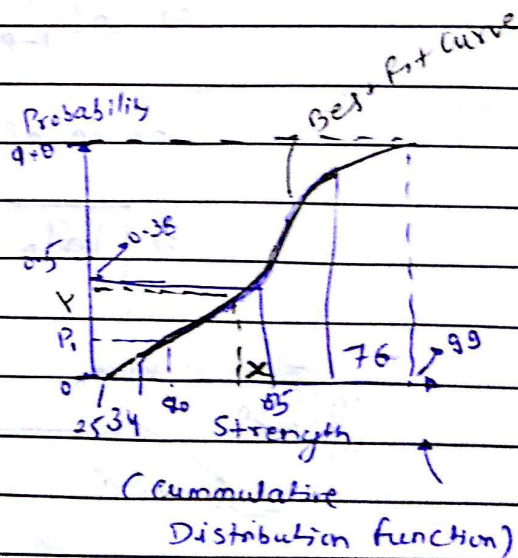
Values:-

50, 52, 34, 76, 74, 65, 68, 56, 63 (in MPa)

Probability, that strength is less than or equal to γ_h . let $\gamma_h = 55$ MPa

(m) Order	Probability
1 → 34	0.1
2 → 50	0.2
3 → 52	0.3
4 → 56	0.4
5 → 63	0.5
6 → 65	0.6
7 → 68	0.7
8 → 74	0.8
9 → 76	0.9

Graphical Method



Probability = $\frac{m}{n+1}$ = $\frac{\text{order}}{\text{total values} + 1}$

Y : Probability of strength to be equal to or less than X_1

from this curve

* $P(X \leq 20) = 0$
 $P(X \leq 40) = P_1$

for failure \rightarrow Probability of failure
 $P(X \leq 53) = 0.38$

$P(X \leq x) = Y$

Best equation for strength data:-

Weibull distribution

$$P = 1 - e^{-\left(\frac{\sigma - \sigma_u}{\sigma_0}\right)^m}$$

Remember
this

exam

σ = variable

σ_u = minimum value of variable
below which probability of occurrence
of the event is zero,

σ_0 = Scale Parameter = It determines
the magnification of the plot

m = Steepness of the plot

To fit into data:-

$$1 - P = e^{-\left(\frac{\sigma - \sigma_u}{\sigma_0}\right)^m}$$

$$\Rightarrow \frac{1}{1 - P} = e^{\left(\frac{\sigma - \sigma_u}{\sigma_0}\right)^m}$$

$$\Rightarrow \ln\left(\frac{1}{1 - P}\right) = \left(\frac{\sigma - \sigma_u}{\sigma_0}\right)^m$$

$$\Rightarrow \ln \ln\left(\frac{1}{1 - P}\right) = m \ln\left(\frac{\sigma - \sigma_u}{\sigma_0}\right)$$

$$\Rightarrow \ln \ln\left(\frac{1}{1 - P}\right) = m \ln(\sigma - \sigma_u) - m \ln \sigma_0$$

$$Y = AX + B$$

three
unknown

two unknown

* Least Square Method

for minimizing the error

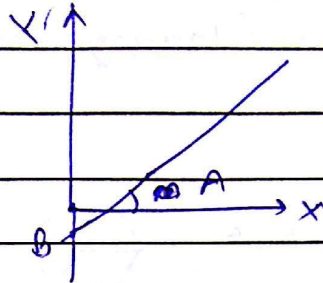
* Best Fitting method

Steps

① assume any value of σ_u

② calculate X and Y.

m	Strength	Y	X



* Always
(B will
be -ve)

$\sigma_0, \quad A = m, \quad B = -A \ln \sigma_0$

$$m = A$$

$$\ln \sigma_0 = -\frac{B}{A}$$

$\sigma_0: \checkmark$

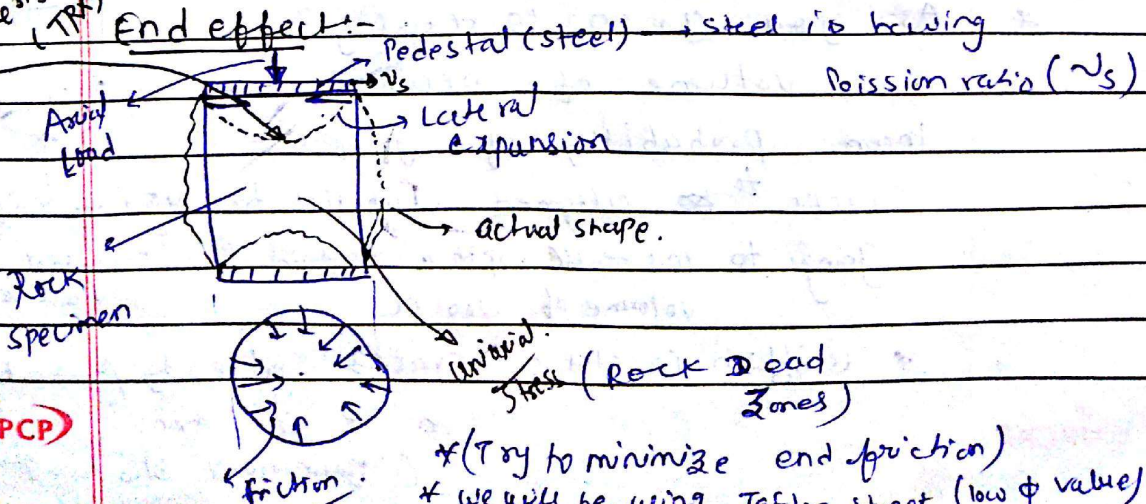
Factors Affecting UCS:-

UCS is a index property. It is not a constant quantity. It depends on so many factors. These factors are:-

- 1) End effect
- 2) Specimen geometry
- 3) Rate of Loading
- 4) Environmental Factor
- 5) In situ stress
- 6) Mineralogy / Grain size / Porosity

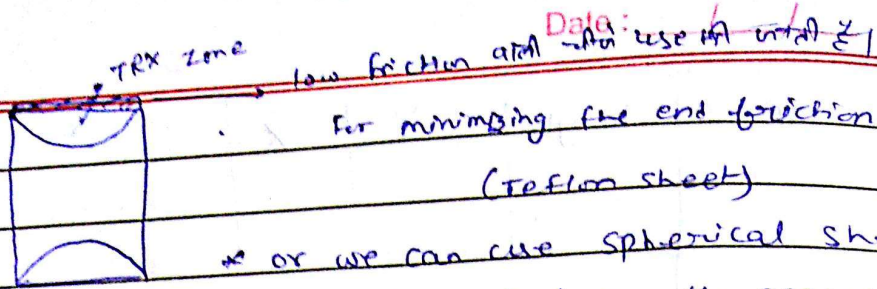
friction.
This zone is stress (TRX)

End effect:-



UPCP

* (Try to minimize end friction)
* We will be using Teflon sheet (low ϕ value)



For minimizing the end friction.

(Teflon sheet)

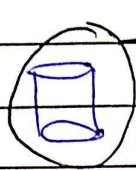
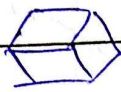
or we can use spherical sheet

(For Reducing the eccentricity)

(Loading will be in axial direction)

2) specimen geometry

→ depends on shape



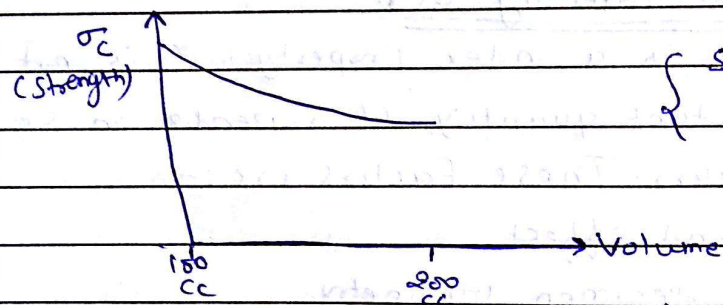
(Sharp corners)

(Concentration of stress, premature failure)

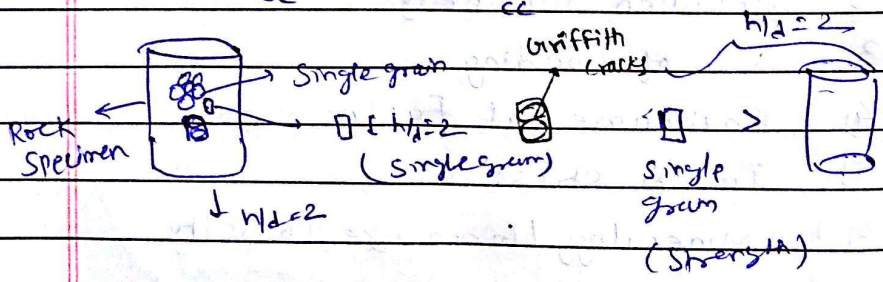
→ depends on size

→ $\frac{h}{d}$ ratio

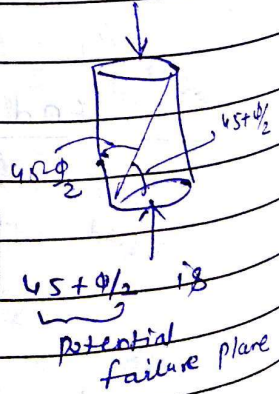
$\frac{h}{d} = ?$



Strength decrease
increase volume

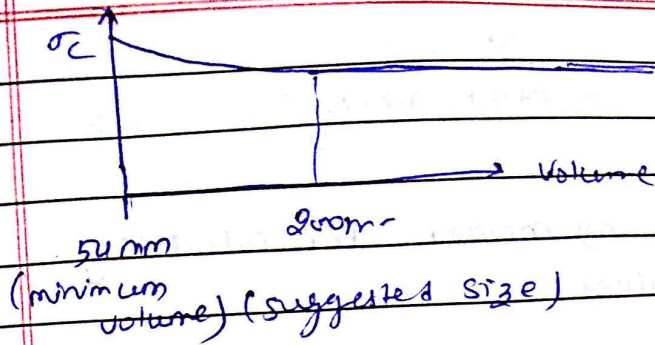


* As you go on increasing the volume of specimen than probability of Griffith cracks to be aligned parallel to $45^\circ \pm \phi/2$ is going to increase upto a certain volume of 200 cc.



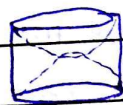
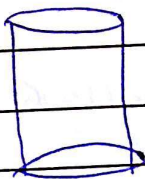
* Griffith cracks :- cracks already present in the specimen. (Boundary b/w two grains)

Remember column



$\frac{h}{d}$ ratio

$\frac{h}{d} = \text{large}$



$\frac{h}{d} = \text{small}$

$\frac{h}{d} \approx 2$ → is used
 $\frac{h}{d} \approx 2 \text{ to } 2.5$

if $\frac{h}{d} \neq 2$, then correct the values

Remember exam

Correction for $\frac{h}{d}$ values if it is not equals 2.

① ASTM ⇒

$$\sigma_{c1} = \frac{\sigma_c}{0.778 + \frac{0.222}{(h/d)}}$$

$$\sigma_{c1} \neq \frac{\sigma_c}{h/d} = 1$$

$$\sigma_c \neq \frac{\sigma_{c1}}{h/d} \neq 1$$

② Protodyakonov

$$\sigma_{c2} = \frac{8\sigma_c}{7 + \frac{2}{(h/d)}} \quad \frac{h}{d} = 2$$

$$\sigma_{c2} \neq \frac{\sigma_c}{h/d} = 2$$

* example

If $h = 90\text{mm}$, $d = 50\text{mm}$

and $\sigma_c = 50\text{MPa}$

$$\frac{h}{d} = \frac{90}{50} \Rightarrow \frac{9}{5}$$

(UPCP)

So, corrected UCS values for $\frac{h}{d} = 2$

$$\sigma_{c2} = \frac{8 \times 50}{7 + \frac{2 \times 5}{9}}$$

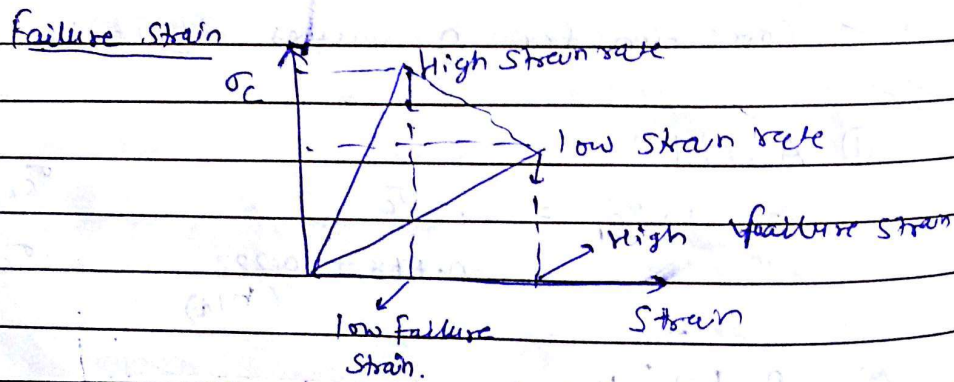
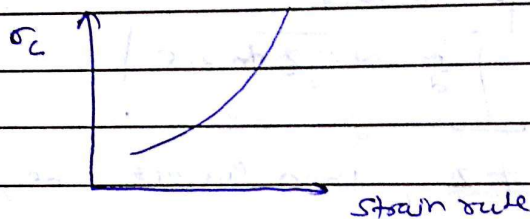
$$\Rightarrow \frac{200 \times 9}{73} \approx 25\text{MPa}$$

$$\sigma_c = \frac{30}{0.778 + \frac{0.222 \times 5}{9}}$$

We can use any method for calculating corrected values.

3) Rate of Loading

- * If you apply higher loading rate, strength will be more
- * loading rate is small than strength will be less.



- * Strain Rate Specimen should be such that failure occurs within about 5 to 10 minutes

4) Environmental Factors:-

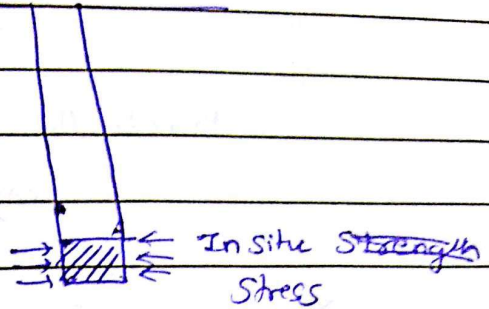
- Moisture Content
- Type of Liquid
- Temperature

- * Moisture Content may result in low strength
- * Type of Liquid - chemicals.
- * at large depth - Temperature increases.

- On High temp. low strength

5) In-situ Stress:-

* Removal of stress occurs when you take out from the ground



* Induce cracks and result in lower strength.

6) Mineralogy:-

Porosity, grain size

well graded material \Rightarrow Higher strength

* strength will depend on minerals, grain size distribution, Porosity etc.

UCS:-

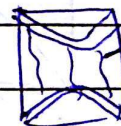
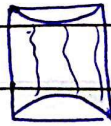
Factors affecting UCS:-

* Mode of failure :-

(brittle/ductile) \rightarrow Type of material

(i) Cone Formation

(ii) constrained shear due to end friction.



vertical fracture

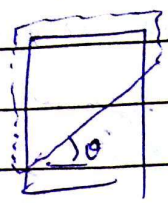
(ii) If there is no end-friction and material is brittle



vertical cracks

It is called splitting / slabbing / vertical splintering

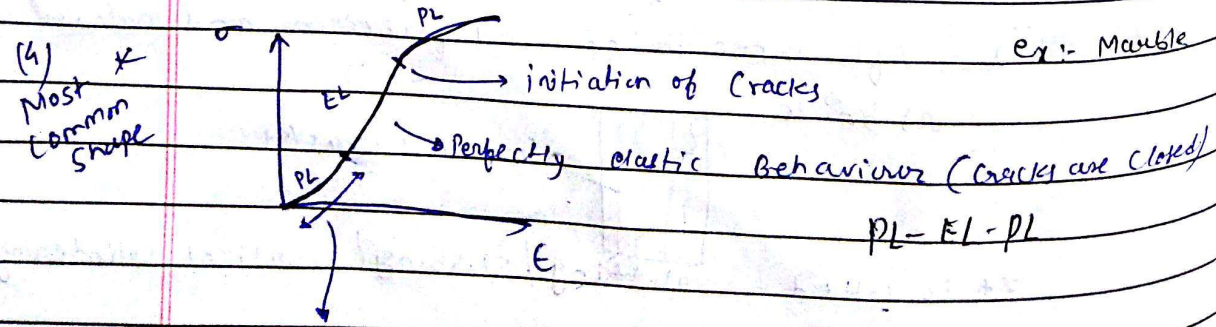
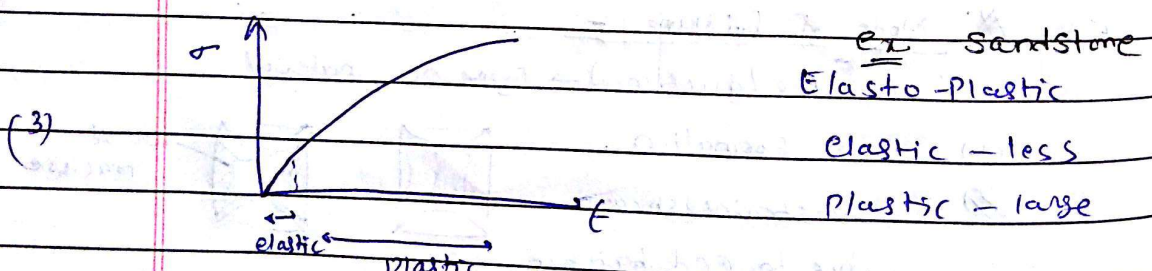
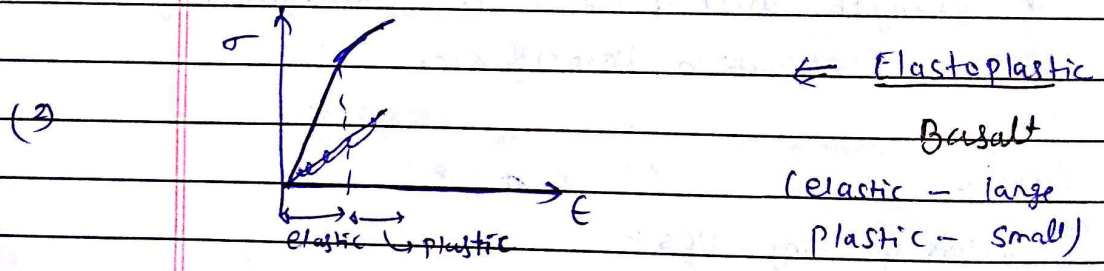
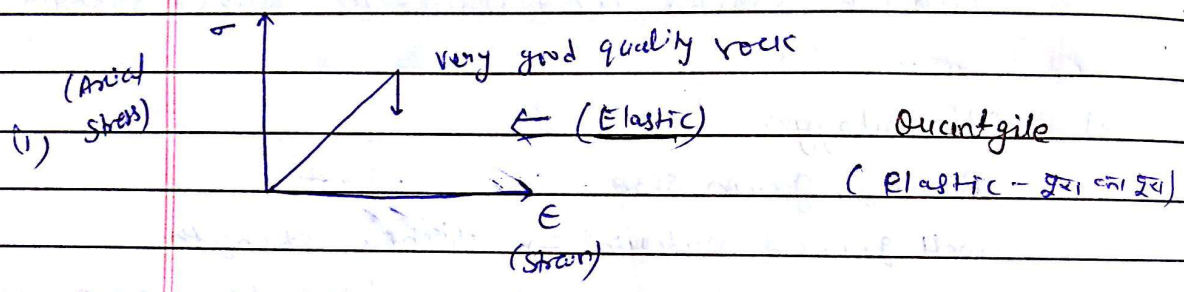
(iii) Shear Failure



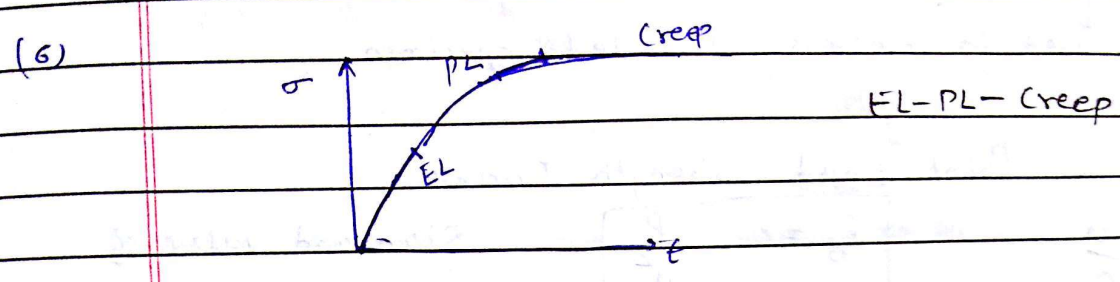
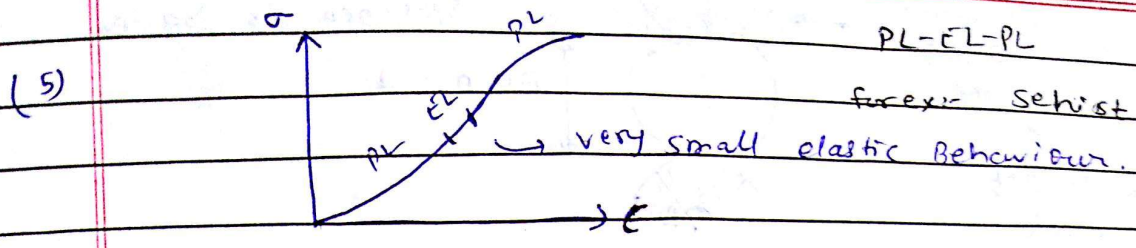
$\theta = 45^\circ + \phi/2$

Practically \Rightarrow one mode is dominating, others may also be present

Shape of stress strain Curve



- * Initial seating (adjustment)
- * Closure of microcracks



Indirect Methods of UCS:-

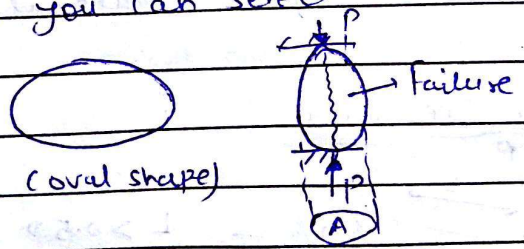
In field symmetrical specimens are not always available.

Basic Principle is energy required to split the Particle that will give you idea about the strength.

For that purpose there is a test:

(A) ISRM (International Standards for Rock Mechanics)

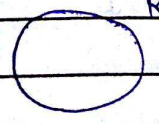
So, you can select

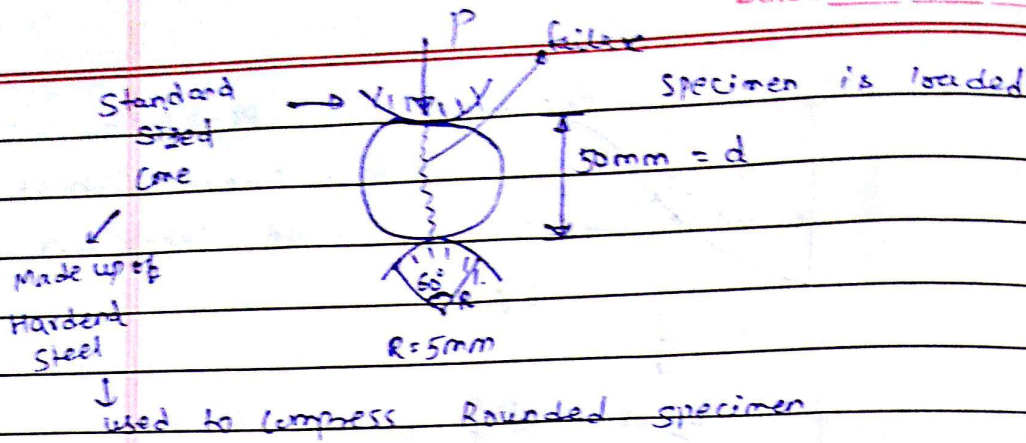


Remember
$$\sigma_c \approx \frac{1}{0.19} \frac{P}{A}$$
 $A = \text{Area Projected on Horizontal plane}$

(B) Point Load Test (Common)

Rounded Regular Specimens





Point Load strength Index

Remember Exam

$$I_s = \frac{P}{d^2}$$

Standard value of $d = 50\text{mm}$

$$I_{S50} = \frac{P}{(50)^2}$$

So, E_{50} I_{S50}
 $d = \text{distance b/w cones}$

$$\sigma_c \approx 24 I_{S50}$$

what if $d \neq 50$, then you have to apply the correction.

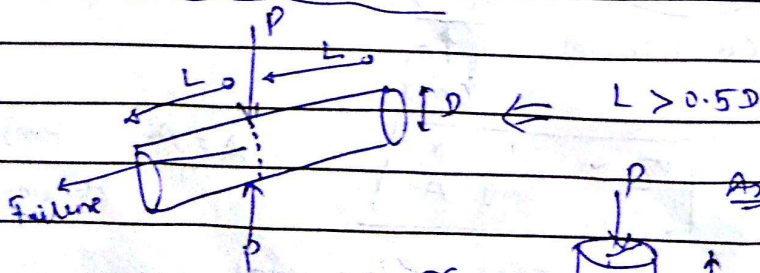
Correction Factor (F) = $\left[\frac{d}{50} \right]^2$

0.45 = Based on experience (general value)

So, $I_{S50} = F \left[\frac{P}{d^2} \right]$

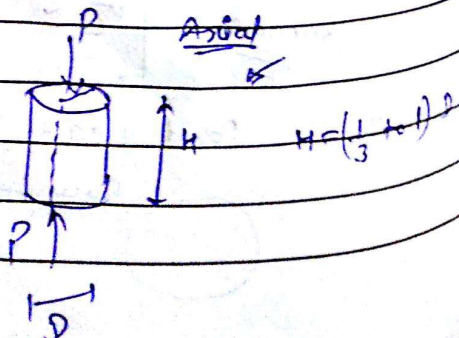
Applicable for Isotropic Material

For cylindrical specimens



Dia Metral \uparrow

or,



From of tests Com

Above Tests are not used in for projects. These are just index test. For Project we need to do actual UCS test.

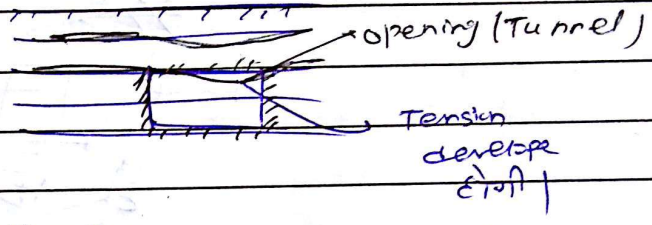
Tensile strength:-

Field-soil →

Tensile strength of soil ≈ 0

but Rocks do have some tensile strength, but not always.

For example:-

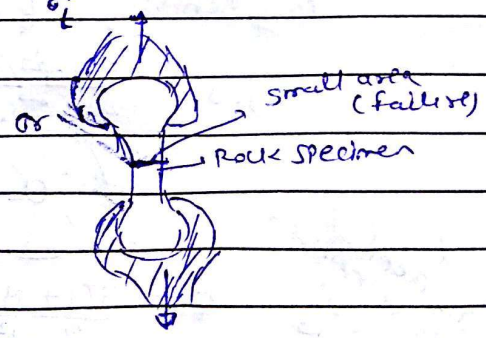
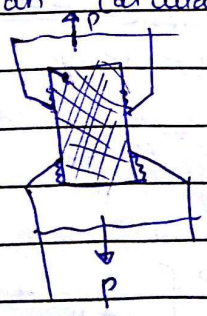


So, in some situation Tension strength important (σ_t)

test ← direct
 indirect

direct test:- Tensile stress is induced and specimen is going to fail due to tensile stress.

So, we can calculate σ_t
Problem is



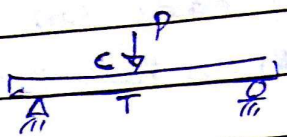
From practice point of view, these tests are not common.

indirect tests:- (Theory of elasticity)

- (1) Bending test
- (2) Hydraulic Extension
- (3) Diametral Compression

(4) Miscellaneous

→ bending test

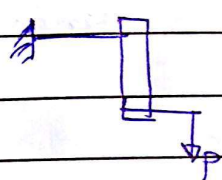
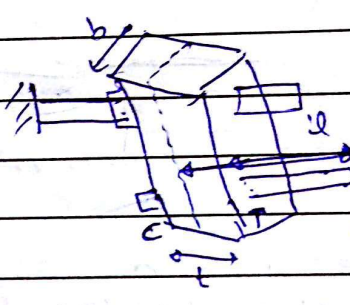


max. Bending Moment $\equiv \frac{PL}{4} = M$

$$\sigma = \frac{M y}{I} = \frac{M b}{bd^2}$$

→ Hydraulic extension

→ Protodyakonov



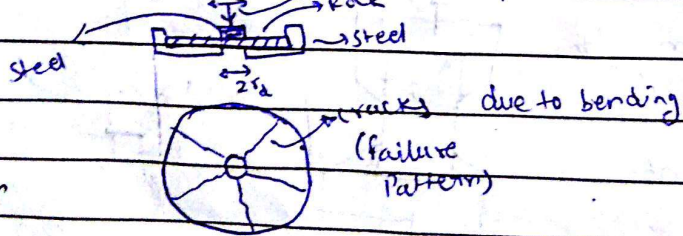
moment will be generated

$M = Pl$ = moment

$$\sigma = \frac{My}{I} + \frac{P}{A}$$

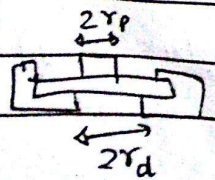
$$\sigma = \frac{My}{I} + \frac{P}{A}$$

Bending of disc :- $2r_p$

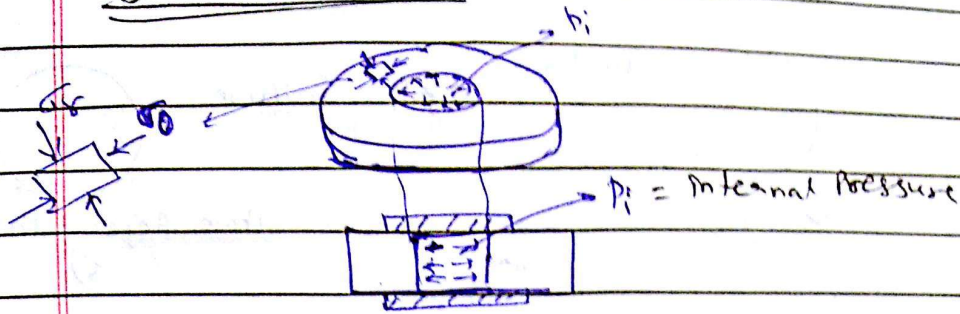


* No need to remember Sir will give this in exam

$$\sigma = \frac{3(1+\mu)P}{2\pi t^2} \left[\frac{1}{1+\mu} + \log_e \frac{r_d}{r_p} \right] \rightarrow - \frac{(1-\mu) r_p^2}{(1+\mu) 4r_d^2}$$



Hydraulic expansion



assumption - Failure is due to tension

Remember

$$\sigma_\theta = -p_i \left(\frac{r_i^2}{r_0^2 - r_i^2} \right) \left[\frac{r_0^2}{r^2} + 1 \right] \quad \text{-ve means Tensile}$$

r_i = internal Radius

r_0 = outer Radius

r = distance from center

$$\sigma_r = p_i \left(\frac{r_i^2}{r_0^2 - r_i^2} \right) \left(\frac{r_0^2}{r^2} - 1 \right)$$

at $r = r_i$ (failure will occur) (Maximum tensile stress here)

$$\sigma_\theta \text{ at } r=r_i = -p_i \left(\frac{r_i^2}{r_0^2 - r_i^2} \right) \left[\frac{r_0^2 + r_i^2}{r_i^2} \right]$$

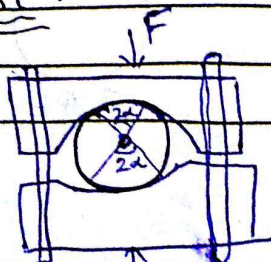
$\sigma_\theta = -p_i \frac{(r_0^2 + r_i^2)}{(r_0^2 - r_i^2)}$	or σ_T	tangential tensile stress
--	---------------	---------------------------

Above test is practical difficult

So, most common test in practical purpose is

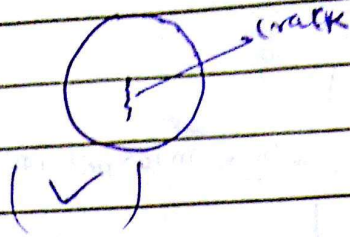
Most Popular

Brazilian Test :-

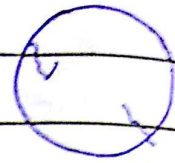


$\alpha = 11^\circ$
 $2\alpha = 22^\circ$

Failure should initiate from center



if failure



than reject this test (X)

Remember

$$\sigma = \frac{F}{\pi d t}$$

(sum)

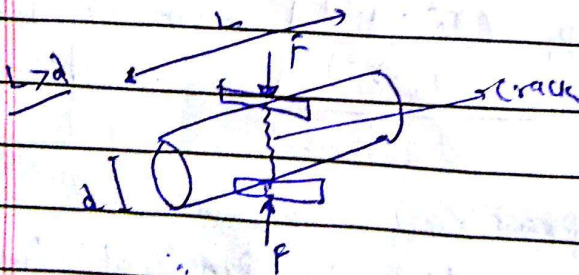
* Loaded minimum NX size of specimen is preferred.

* Thickness 't' is kept between 0.5 to 1 times Diameter (D).

If 't' is small than buckling will takes place.

Miscellaneous Test :-

(a) Diametral Compression of Cylinder



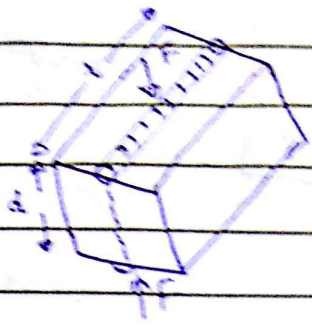
$$\sigma_t = \frac{0.96 F}{d^2}$$

(b) Sphere diametral Compression

$$\sigma_t = \frac{0.90 F}{d^2}$$

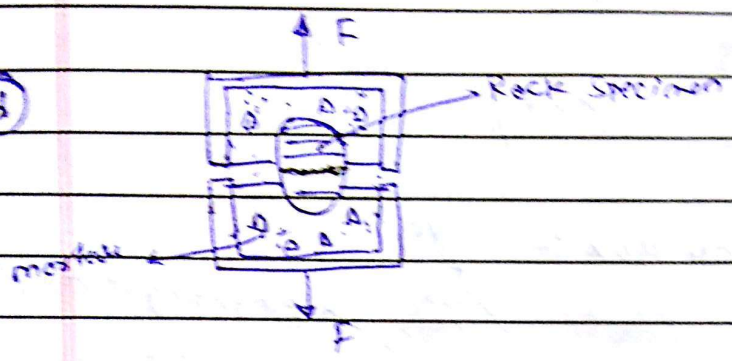


(c)



$$\tau = \frac{F}{A}$$

(d)



only when Rock is curved

(e)

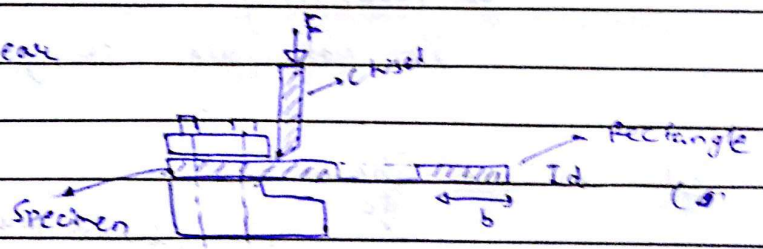
Point load Test

$$\sigma_c \approx 1.25 I_{550}$$

Shear Test:-

①

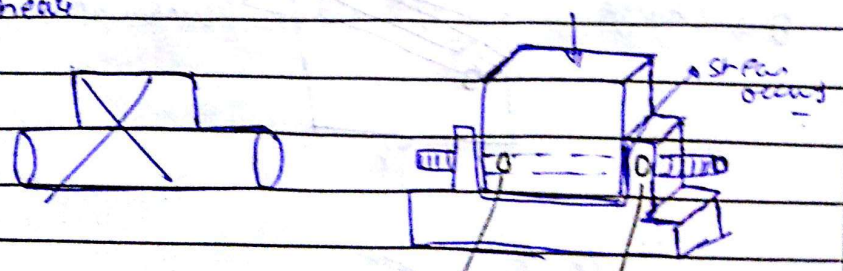
Single Shear



Shear area = bd

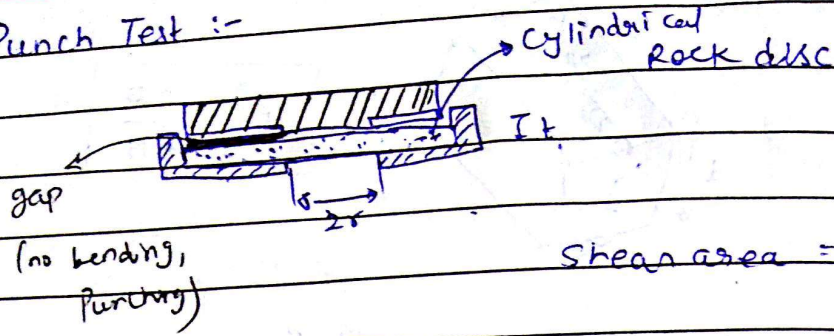
②

Double shear



two circular shear failure

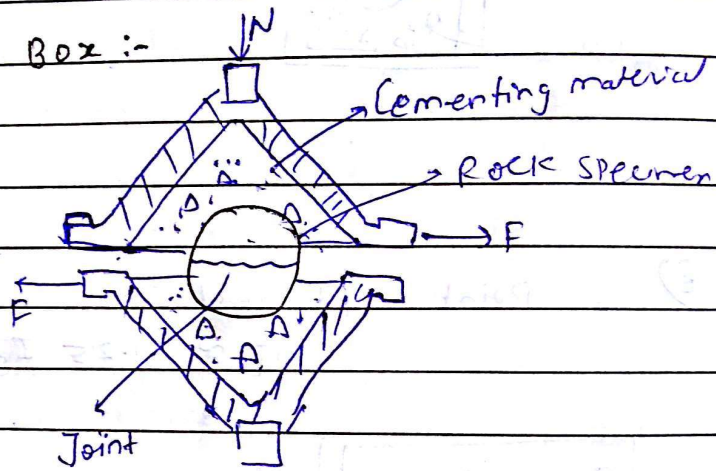
③ Punch Test :-



$$\sigma_c = \frac{F}{2\pi r t}$$

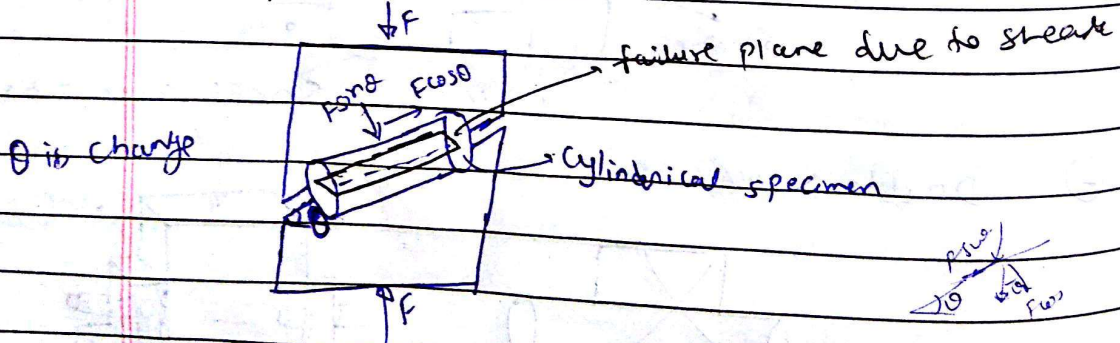
④ Field shear Box :-

For very very weak rock

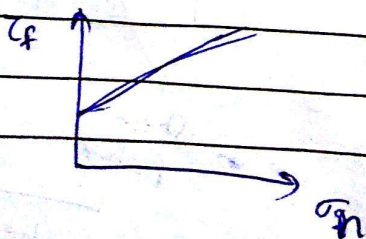


* Generally used for jointed rock. To calculate strength of joint.

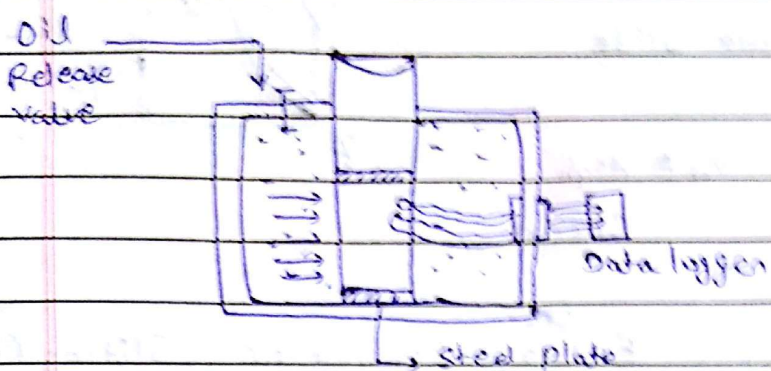
⑤ Oblique Shear Test



Shear area = Dl



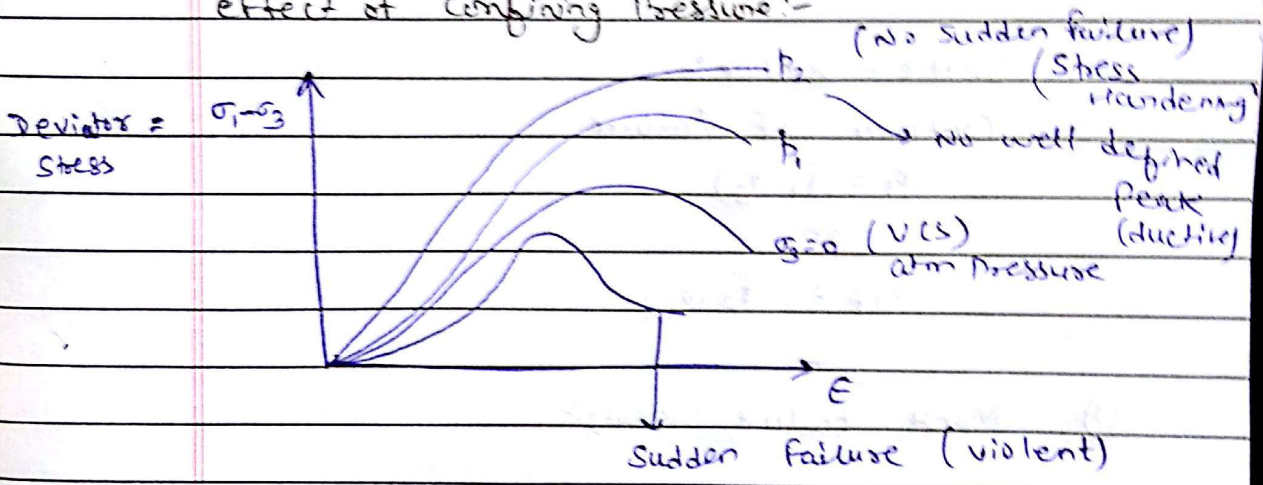
Tri-axial Behaviour of Rocks → Intact Rocks:-



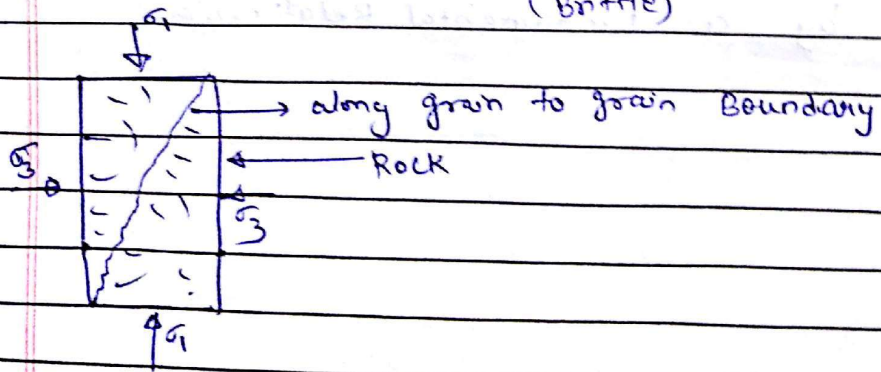
$\sigma_{1f} = f(\sigma_3)$

σ_{1f}	σ_3
50	1
70	2
85	3

effect of confining pressure:-

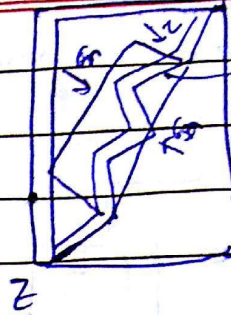


Sudden failure of energy (Brittle)



Date: σ_1 / σ_2 / σ_3

(a) If $\sigma_n = \text{low}$
Side Slide



with Shear failure or slide over (work done adjust σ_n)

(b) If $\sigma_n = \text{High}$
Shear

Low σ_n :- Energy Required for Sliding (dilatancy) is low brittle failure.

High σ_n :- E (Shear failure) = Low
: Shear failure
Ductile failure.

Brittle - Ductile :-

Criteria of Failure

$$\sigma_1 = f_1(\sigma_3)$$

or

$$\sigma_{1f} = f_2(\sigma_3)$$

① Mohr failure theory :-

The Failure of a material may be represented by a fundamental Relationship.

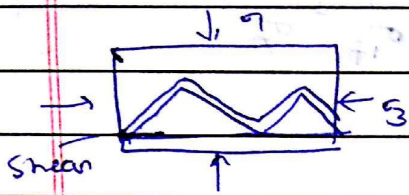
Tri-axial Strength:-

Failure Criteria:-

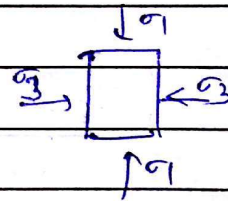
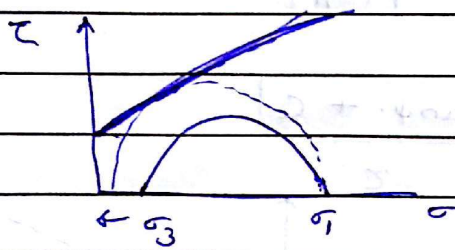
→ strength increases on increasing confining pressure

→ If σ_3 is high, at failure, change in volume for dense sand will decrease.

It is not going to dilate.

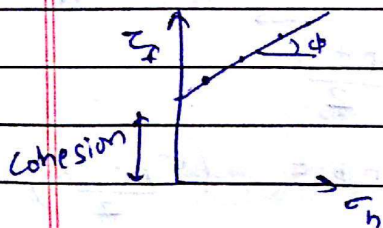


Mohr Theory:-



Coulomb Theory:-

"Direct Shear Test"



τ_f is stress dependent & stress independent } two components

$$\tau_f = \sigma_n \tan \phi + C$$

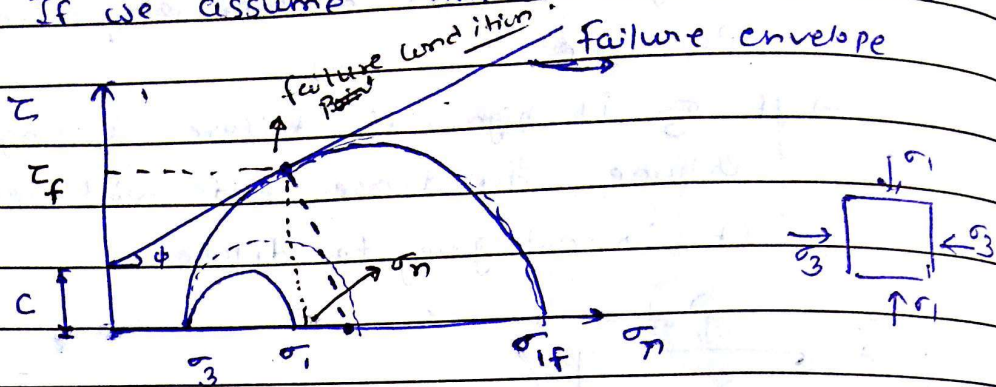
Stress dependent

Cohesion

Stress independent

Mohr - Coulomb Theory:-

If we assume linear failure criteria failure envelope

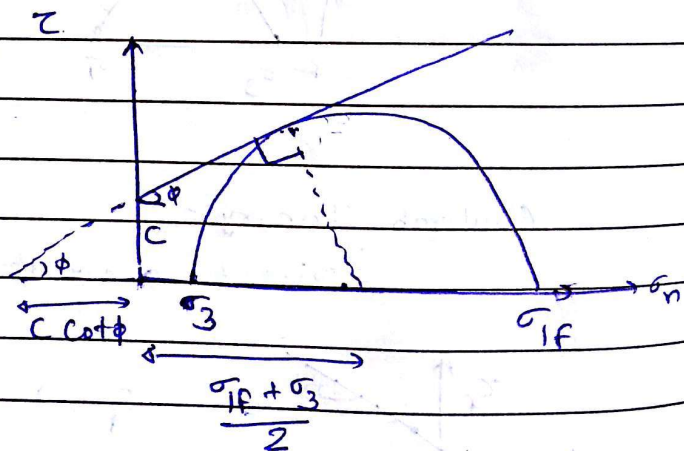


failure criteria

$$\sigma_{1f} = f(\sigma_3)$$

$$\tau_f = f(\sigma_n)$$

$$\tau_f = \sigma_n \tan \phi + c$$



$$\left(c \cot \phi + \frac{\sigma_{1f} + \sigma_3}{2} \right) \sin \phi = \left(\frac{\sigma_{1f} - \sigma_3}{2} \right)$$

$$\Rightarrow c \cos \phi + \left(\frac{\sigma_{1f} + \sigma_3}{2} \right) \sin \phi = \left(\frac{\sigma_{1f} - \sigma_3}{2} \right)$$

$$\Rightarrow 2c \cos \phi + \sigma_{1f} [\sin \phi - 1] + \sigma_3 [\sin \phi + 1] = 0$$

$$\sigma_{1f} = \frac{\sigma_3 (1 + \sin \phi) + 2c \cos \phi}{1 - \sin \phi}$$

$\sigma_{1f} = f(\sigma_3)$

$$\sigma_{1f} = \sigma_3 \tan^2(45 + \frac{\phi}{2}) + \frac{2c \cos \phi}{1 - \sin \phi}$$

Mohr
Coulomb
Criterion

$$\sigma_{1f} = \sigma_3 \left(\frac{1 + \sin \phi}{1 - \sin \phi} \right) + 2c \left(\frac{\cos \phi}{1 - \sin \phi} \right)$$

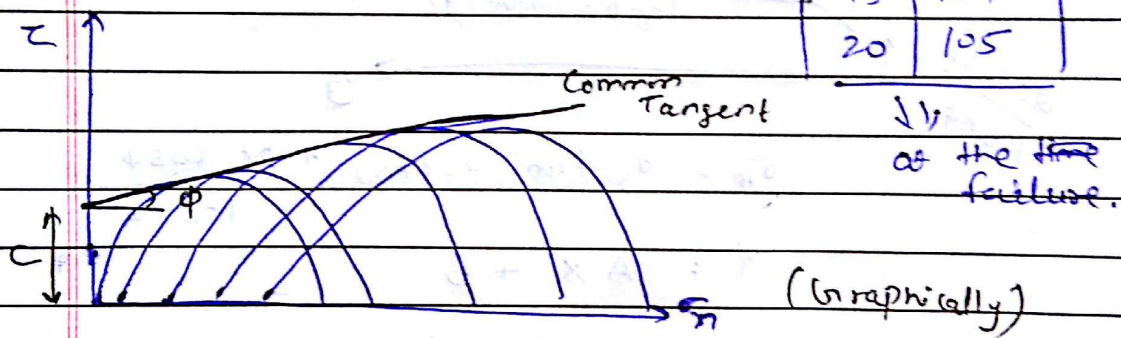
Exam Practice

Tri-axial Test Result:-

Lab $\Rightarrow c, \phi = ?$

σ_3	σ_1
0	70
5	85
10	97
15	107
20	105

(i) Draw Mohr Circles



It is very difficult to get best fitting common tangent

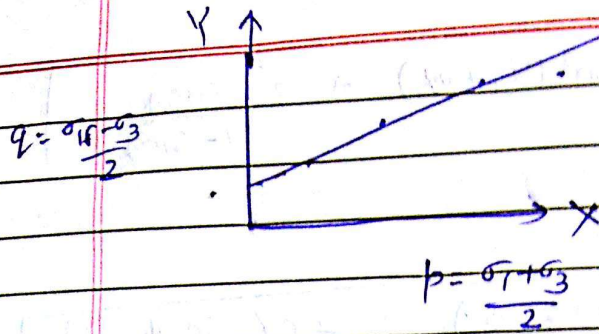
(ii) So, we plot p-q plot.

p-q plot

$$\frac{\sigma_{1f} - \sigma_3}{2} = c \cos \phi + \left(\frac{\sigma_{1f} + \sigma_3}{2} \right) \sin \phi$$

$$Y = B + AX \quad \text{Eqn of Straight line.}$$

Data: fit / line /



$$A = \sin \phi$$

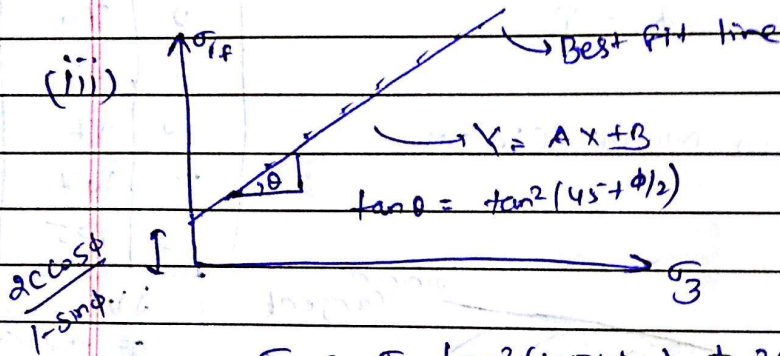
$$B = c \cos \phi$$

$$B = c \sqrt{1-A^2}$$

$$\phi = \sin^{-1} A$$

$$\sin \phi = \frac{A}{\sqrt{1-A^2}}$$

$$\Rightarrow c = \frac{B}{\sqrt{1-A^2}}$$

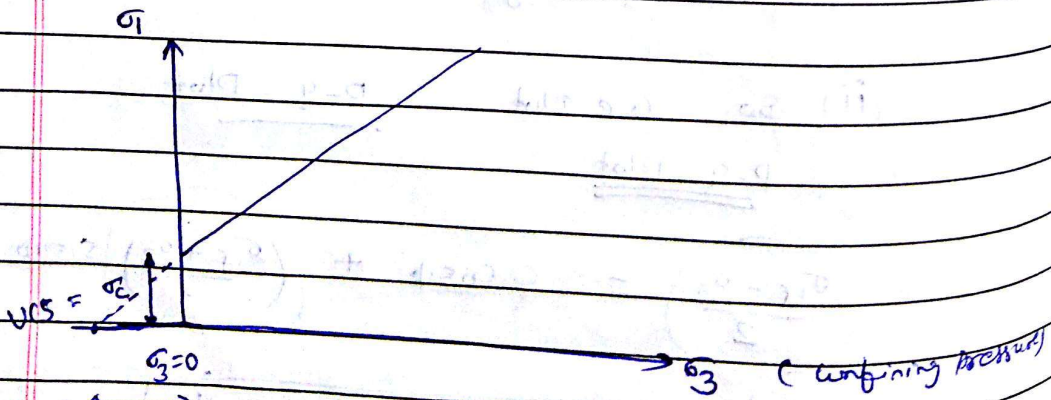


$$\sigma_{1F} = \sigma_3 \tan^2(45 + \phi/2) + \frac{2c \cos \phi}{1 - \sin \phi}$$

$$Y = AX + B$$

$$\tan^2(45 + \phi/2) = A$$

$$\frac{2c \cos \phi}{1 - \sin \phi} = B$$



$\sigma_1 = \text{tensile strength}$ (failure at $\sigma_1 = 0$)

(failure due to tension)

$$\boxed{\frac{\sigma_c}{\sigma_t} = \tan^2(45 + \phi/2)}$$

$$\sigma_f = \sigma_3 \frac{1 + \sin\phi}{1 - \sin\phi} + \frac{2c \cos\phi}{1 - \sin\phi}$$

$$\Rightarrow \sigma_c = \sigma_3 \left(\frac{1 + \sin\phi}{1 - \sin\phi} \right) + \frac{2c \cos\phi}{1 - \sin\phi}$$

$$\boxed{\sigma_c = \frac{2c \cos\phi}{1 - \sin\phi}}$$

$$0 = (-\sigma_t) \left(\frac{1 + \sin\phi}{1 - \sin\phi} \right) + \frac{2c \cos\phi}{1 - \sin\phi}$$

$$\boxed{\sigma_t = \frac{2c \cos\phi}{1 + \sin\phi}}$$

$$\boxed{\frac{\sigma_c}{\sigma_t} = \frac{1 + \sin\phi}{1 - \sin\phi} = \tan^2(45 + \phi/2)}$$

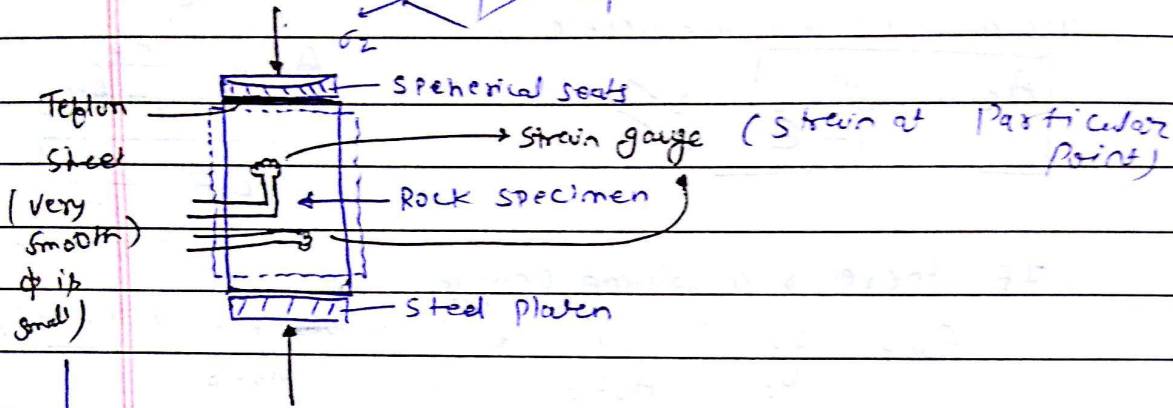
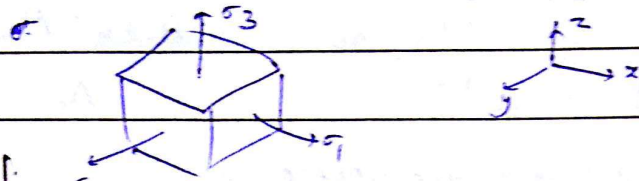
$\sigma_c, \sigma_t, c, \phi$ → four things (rest other things we can find)

Lab Tests:

1) UCS (Uniaxial Compression Test)

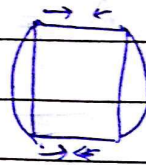
(Uniaxial compressive strength Test)

In Rocks all three principle stress are acting.



(To minimize the end friction)

If end friction is not minimize then deformed shape will be like this. Bending will be there.



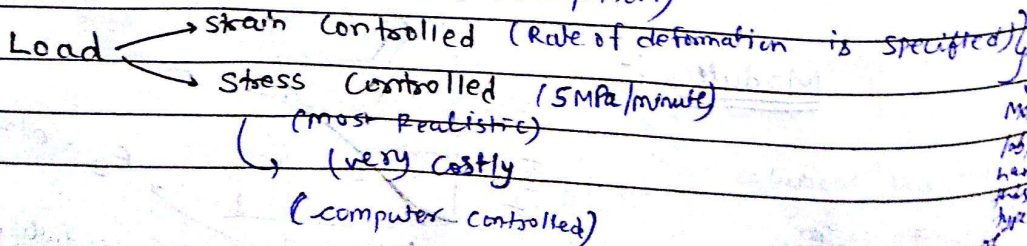
So, vertical stress will not be more a principal stress. because due to friction shear will develop.

* Strain gauge is used to measure the strain at Particular point. [Costly]

$$\epsilon = \frac{\Delta L}{L}$$

(or we can use dial gauge)

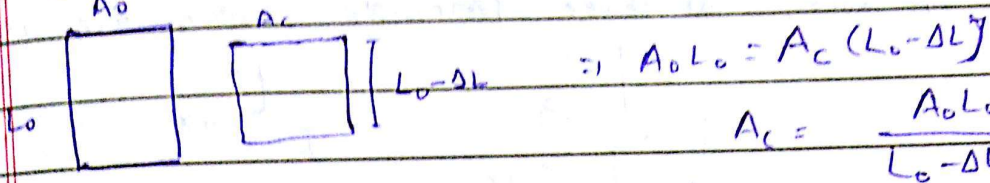
→ strain is uniform throughout the length (assumption)



Load

Deformation

Stress (Area correction of 4.3116%)
Strain
 A_0



$$A_c = \frac{A_0 L_0}{L_0 - \Delta L}$$

assumption: No volume change

$$A_c = \frac{A_0}{1 - \epsilon}$$

$$A_c = \frac{A_0}{1 - \Delta L / L_0}$$

$$A_c = \frac{A_0}{1 - \epsilon}$$

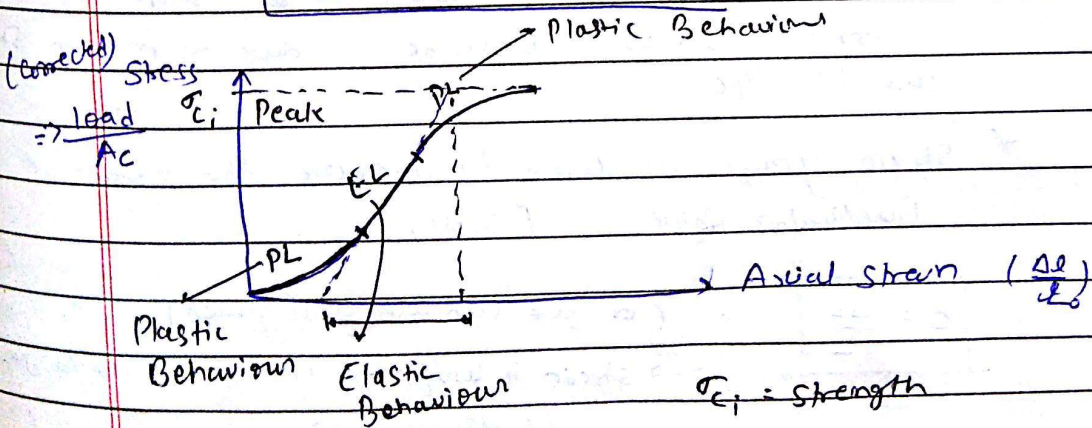
If there is a volume change

$$\epsilon_v = \frac{\Delta V}{V_0} = \frac{\Delta V}{A_0 L_0} \times (\text{final volume} - \text{initial vol.})$$

$$A_0 L_0 + \epsilon_v A_0 L_0 = A_c (L_0 - \Delta L)$$

$$A_c = \frac{A_0 + \epsilon_v A_0}{1 - \epsilon} \Rightarrow \frac{A_0 (1 + \epsilon_v)}{1 - \epsilon}$$

$$A_c = \frac{A_0 (1 + \epsilon_v)}{1 - \epsilon}$$

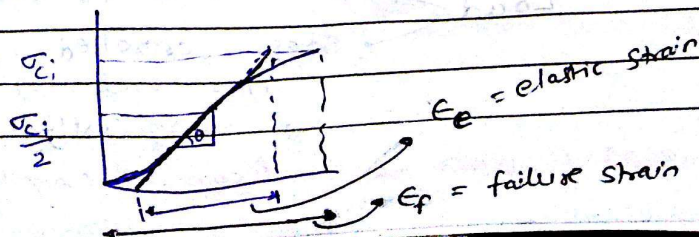


Modulus :-

tangent Modulus



σ_c
slope of curve at any point



$$\tau = \sigma_n \tan \phi + c$$

$\phi =$ Increase in shear strength per unit increase in normal stress

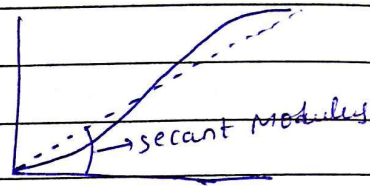
Date: ___/___/___

tangent Modulus = slope of curve at any value

$$E_{E50} = \frac{\sigma_{ci}}{\epsilon_e} \quad (\text{at } 50\% \text{ peak stress})$$

initial Tangent Modulus:- E_{E0}

Secant Modulus = E_{sec}



Get :- $\sigma_{ci} =$ Strength
 $E_{E50} =$ Tangent Modulus

$$\text{Modulus Ratio} = \frac{E_{E50}}{\sigma_{ci}} \Rightarrow \frac{1}{\epsilon_e}$$

It gives the idea about elastic strain at failure.

* It gives the idea about elastic strain at time of failure and gives some idea about failure strain (ϵ_f) depending on the shape of curve. It is lower limit of failure strain.

$$\boxed{\epsilon_f \geq \epsilon_e}$$

(corrected)

$$\Rightarrow \frac{1}{\epsilon_e}$$

Classify the Rocks:-

Based on "Strength" & "Modulus Ratio".

(This is called Deere-Miller classification)
 (Two lettered)

Remember
 This Table
 (UPCP)

Strength		Modulus Ratio	
A - Very High	> 200 MPa	H - High	> 500
B - High	100-200 MPa	M - Medium	200-500
C - Medium	50-100 MPa	L - Low	< 200
D - Low	25-50 MPa		
E - Very low	< 25 MPa		

tangent M

(UPCP)

Sto

Strength \rightarrow XY

E₂ = Rock which is classified as DL

low strength and Low modulus Ratio

CM \Rightarrow Medium, Medium

(50 to 100)

(200-500)

$\rightarrow E_c = \left(\frac{1}{500} \text{ to } \frac{1}{200} \right)$

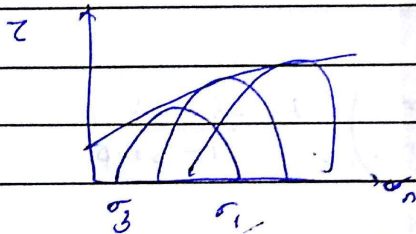
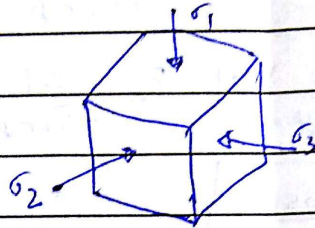
28/2/17

Date: ___/___/___

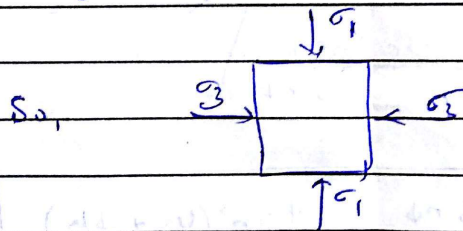
After Mid Term

Failure Criteria for Intact Rocks:-

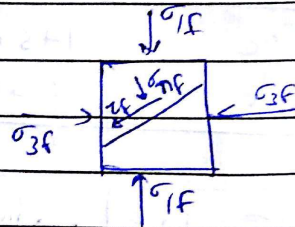
Mohr Theory - already discussed



no effect of σ_2

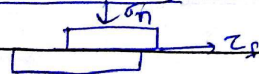


at time of failure



Coulomb Theory

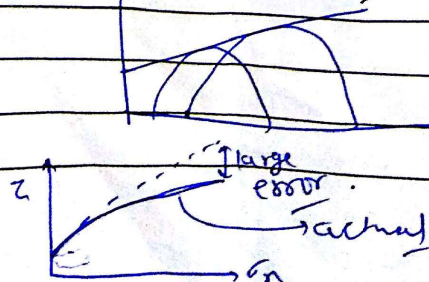
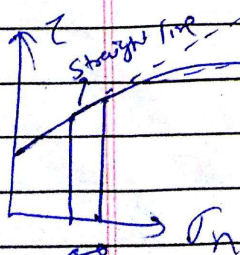
(at failure)



$$\tau_f = \sigma_n \tan \phi + c$$

(Coulomb's eqn)

Combined (Mohr Coulomb Theory) Criteria



this straight line is valid only for small stress conditions

Small stress conditions

(UPCP) change in sigma_n

(UPCP)



Shear strength τ_f = $\sigma_n \tan \phi + c$

Stress depends \rightarrow $\sigma_n \tan \phi$

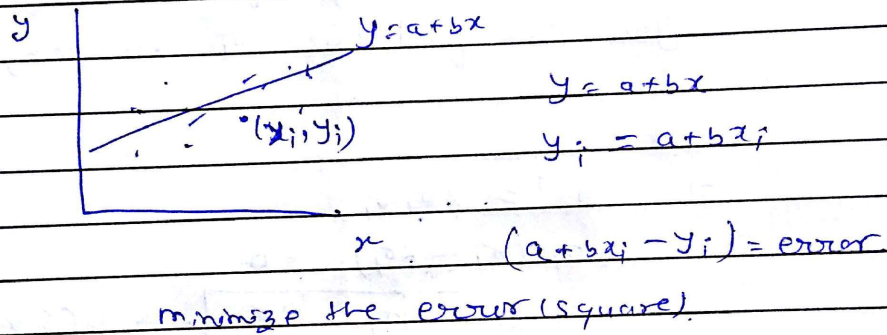
Stress independent \rightarrow c

Date: _____

$$\tau_f = \left(\frac{2c \cos \phi}{1 - \sin \phi} \right) + \sigma_f \left(\frac{1 + \sin \phi}{1 - \sin \phi} \right)$$

Parameters: $c, \phi \rightarrow$ Shear strength (stress dependent)
 shear strength (stress independent)

Least square method for optimal Result.
 (Straight line fitting)

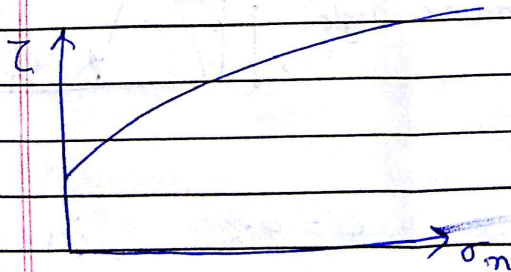


$$y = a + bx$$

$$\sum y = \sum a + b \sum x \rightarrow \textcircled{1}$$

$$\sum xy = a \sum x + b \sum x^2 \rightarrow \textcircled{2}$$

$$b \sum x = \sum xy - a \sum x$$

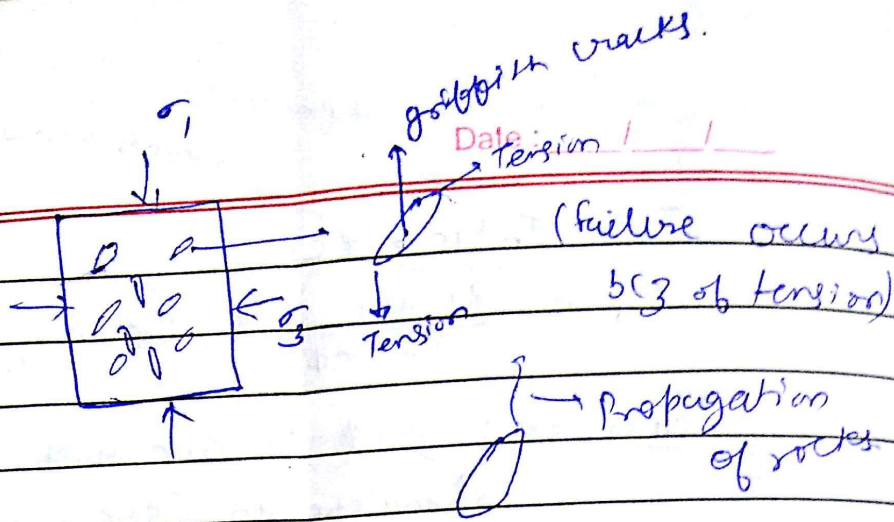


Non linear failure Criteria:-

① Griffith criteria

(UPCP)

for Brittle Material - "glass"



Date: Tension / /

exam

$$(\sigma_1 - \sigma_3)^2 = 8\sigma_t(\sigma_1 + \sigma_3) \quad \text{for } \sigma_1 + 3\sigma_3 > 0$$

Parameter is only ' σ_t '
 you can find out σ_t for any σ_3

For $\sigma_3 = 0$
 $\Rightarrow \sigma_1^2 = 8\sigma_t\sigma_1$
 $\sigma_1(\sigma_1 - 8\sigma_t) = 0$

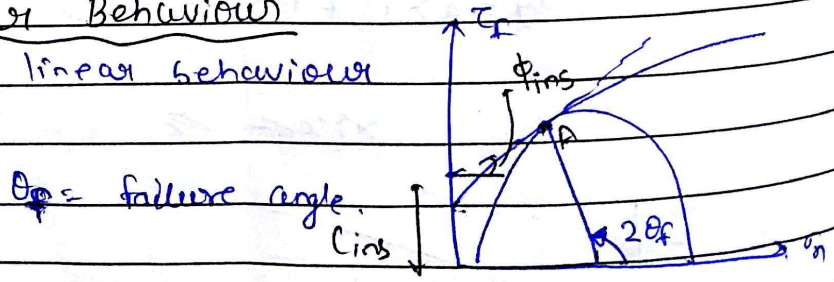
$$\sigma_c = \sigma_1 = 8\sigma_t \rightarrow \text{UCS value taken } \sigma_3 = 0$$

$$\frac{\sigma_c}{\sigma_t} = 8$$

Non linear Behaviour

If non linear behaviour

exam



for Plane A,

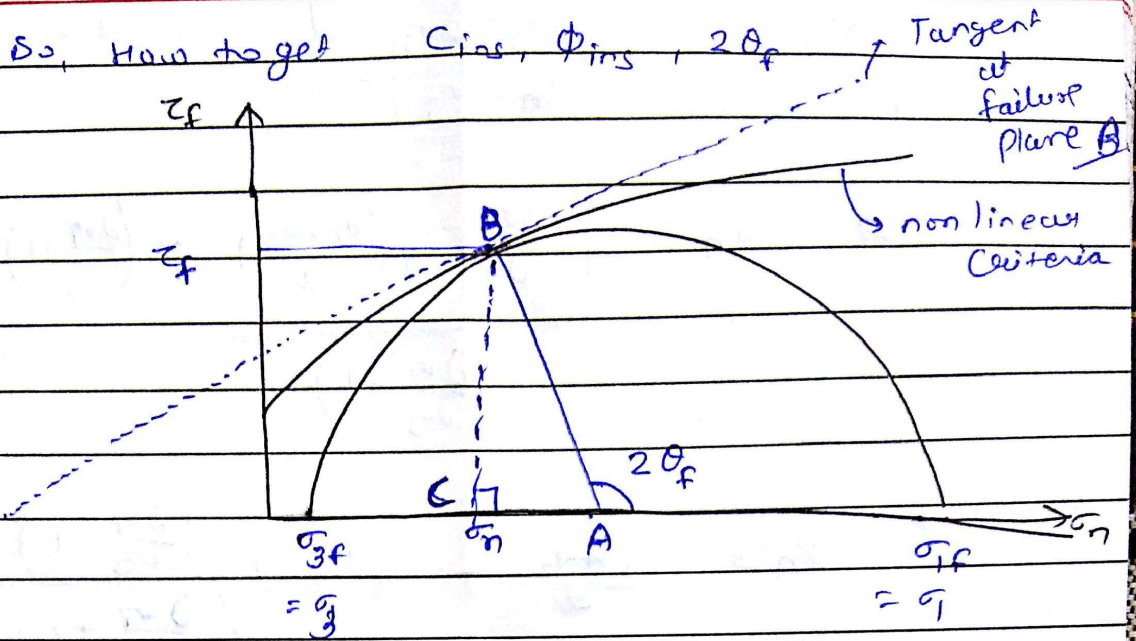
$$C = C \sin \delta, \quad \phi = \phi_{ins} \quad \text{ins = instantaneous}$$

corresponding to $2\theta_f$

actually in practice C and ϕ is not constant, because of non linear behaviour

exam

(at time of failure)



exam

Balmer expressions:

$\theta_f = \text{angle of failure plane w.r.t horizontal}$

$$\sigma_{1f} = f(\sigma_{3f}) \rightarrow (1)$$

From ΔABC

$$\Rightarrow \left(\frac{\sigma_{1f} - \sigma_{3f}}{2}\right)^2 = (\tau_f)^2 + \left(\frac{\sigma_{1f} + \sigma_{3f}}{2} - \sigma_n\right)^2 \rightarrow (2)$$

Partially differentiate above eqn w.r.t σ_{3f}

$$\Rightarrow 2 \left[\frac{\sigma_{1f} - \sigma_{3f}}{2} \right] \left[\frac{d\sigma_{1f}}{d\sigma_{3f}} - 1 \right] = 0 + 2 \left[\frac{\sigma_{1f} + \sigma_{3f}}{2} - \sigma_n \right] \left[\frac{d\sigma_{1f}}{d\sigma_{3f}} + 1 \right]$$

$$\Rightarrow 2 \left[\frac{\sigma_1 - \sigma_3}{2} \right] \left[\frac{\partial \sigma_1}{\partial \sigma_3} - 1 \right] = 0 + 2 \left[\frac{\sigma_1 + \sigma_3}{2} - \sigma_n \right] \left[\frac{\partial \sigma_1}{\partial \sigma_3} + 1 \right]$$

$$\Rightarrow \left(\frac{\sigma_1 - \sigma_3}{2}\right) \left(\frac{\partial \sigma_1}{\partial \sigma_3} - 1\right) = 0 + \left[\frac{\sigma_1 + \sigma_3}{2} - \sigma_n\right] \left[\frac{\partial \sigma_1}{\partial \sigma_3} + 1\right]$$

$$\frac{\frac{\partial \sigma_1}{\partial \sigma_3} - 1}{\frac{\partial \sigma_1}{\partial \sigma_3} + 1} = \frac{\sigma_1 + \sigma_3 - 2\sigma_n}{\sigma_1 - \sigma_3}$$

$$\frac{\partial \sigma_1}{\partial \sigma_3} = \frac{\sigma_1 + \sigma_3 - 2\sigma_n}{\sigma_1 - \sigma_3 - \sigma_1 - \sigma_3 + 2\sigma_n}$$

$$\Rightarrow \frac{\partial \sigma_1}{\partial \sigma_3} = \frac{\sigma_1 + \sigma_3 - 2\sigma_n}{2\sigma_n - 2\sigma_3}$$

$$\frac{\partial \sigma}{\partial \sigma_3} = \frac{\sigma_1 + \sigma_3 - 2\sigma_n}{\sigma_n - \sigma_3}$$

$$\Rightarrow \sigma_n = \frac{\left(\frac{\partial \sigma}{\partial \sigma_3} - 1 \right) \left(\frac{\sigma_1 - \sigma_3}{2} \right) - \left(\frac{\partial \sigma}{\partial \sigma_3} + 1 \right) \left(\frac{\sigma_1 + \sigma_3}{2} \right)}{\left(\frac{\partial \sigma}{\partial \sigma_3} + 1 \right)}$$

$$\sigma_n = \frac{\sigma_1 + \sigma_3}{2} - \frac{\left(\frac{\sigma_1 - \sigma_3}{2} \right) \left(\frac{\partial \sigma}{\partial \sigma_3} - 1 \right)}{\left(\frac{\partial \sigma}{\partial \sigma_3} + 1 \right)}$$

Final expression

$$\sigma_n = \left\{ \frac{\sigma_1 + \sigma_3 \frac{\partial \sigma}{\partial \sigma_3}}{1 + \frac{\partial \sigma}{\partial \sigma_3}} \right\}$$

$$\sigma_n = \frac{\sigma_1 + \sigma_3 \left[\frac{\partial \sigma}{\partial \sigma_3} + 1 - 1 \right]}{1 + \frac{\partial \sigma}{\partial \sigma_3}}$$

Remember

$$\sigma_n = \frac{\sigma_3 + \sigma_1 - \sigma_3}{1 + \frac{\partial \sigma}{\partial \sigma_3}} \quad \rightarrow (2)$$

Put in eqn (1)

$$\tau_f^2 = \left(\frac{\sigma_1 - \sigma_3}{2} \right)^2 - \left[\frac{\sigma_1 + \sigma_3}{2} - \frac{\sigma_1 + \sigma_3 \frac{\partial \sigma}{\partial \sigma_3}}{1 + \frac{\partial \sigma}{\partial \sigma_3}} \right]^2$$

$$\tau_f^2 = \left(\frac{\sigma_1 - \sigma_3}{2} \right)^2 - \left[\frac{\sigma_1 + \sigma_3 \frac{\partial \sigma_1}{\partial \sigma_3} + \sigma_3 + \sigma_3 \frac{\partial \sigma_3}{\partial \sigma_3} - 2\sigma_1}{2 \left(1 + \frac{\partial \sigma_1}{\partial \sigma_3} \right)} \right]$$

$$\tau_f^2 = \left(\frac{\sigma_1 - \sigma_3}{2} \right)^2 - \left[\frac{\sigma_1 \frac{\partial \sigma_1}{\partial \sigma_3} - \sigma_3 \frac{\partial \sigma_3}{\partial \sigma_3} + \sigma_3 - \sigma_1}{2 \left(1 + \frac{\partial \sigma_1}{\partial \sigma_3} \right)} \right]$$

$$\frac{\tau_f^2}{4} = \left(\frac{\sigma_1 - \sigma_3}{2} \right)^2 - \frac{(\sigma_3 - \sigma_1) \left[\frac{\partial \sigma_1}{\partial \sigma_3} - 1 \right]}{2 \left(1 + \frac{\partial \sigma_1}{\partial \sigma_3} \right)}$$

$$\tau_f = \frac{\sigma_1 - \sigma_3}{1 + \frac{\partial \sigma_1}{\partial \sigma_3}} \sqrt{\frac{\partial \sigma_1}{\partial \sigma_3}}$$

$$\tan(\alpha - 2\theta_f) = \frac{\tau_f}{CA} = \frac{\tau_f}{\frac{\sigma_1 + \sigma_3 - \sigma_n}{2}}$$

$$\Rightarrow \tan(\alpha - 2\theta_f) = \frac{2(\sigma_1 - \sigma_3)}{\left[1 + \frac{\partial \sigma_1}{\partial \sigma_3} \right] \left[\sigma_1 + \sigma_3 - 2\sigma_n \right]} \sqrt{\frac{\partial \sigma_1}{\partial \sigma_3}}$$

$$\tan(\alpha - 2\theta_f) = \left(\frac{\partial \sigma_1}{\partial \sigma_3} \right) \left(\frac{\sigma_1 - \sigma_3}{\sigma_1 + \sigma_3 - 2\sigma_n} \right)$$

$$\tan 2\theta_f = \frac{2 \sqrt{\frac{\partial \sigma_1}{\partial \sigma_3}}}{1 - \frac{\partial \sigma_1}{\partial \sigma_3}}$$

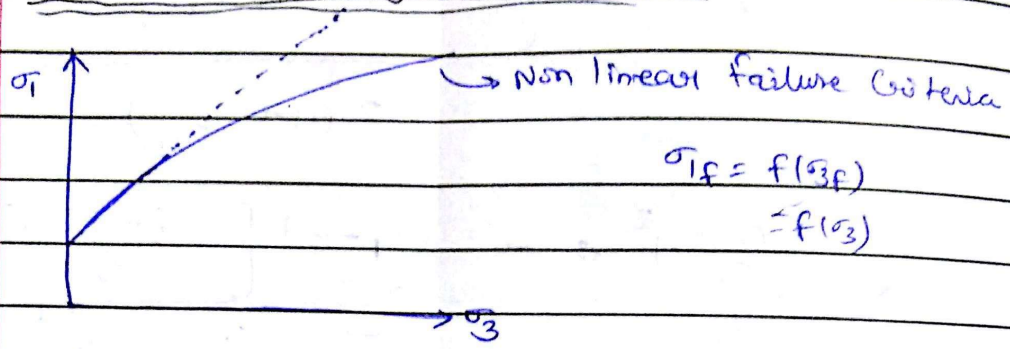
$$\tan \left(\frac{\alpha}{2} + \theta_f \right) = \frac{\sqrt{\frac{\partial \sigma_1}{\partial \sigma_3}} + \sqrt{\frac{\partial \sigma_1}{\partial \sigma_3}}}{1 - \sqrt{\frac{\partial \sigma_1}{\partial \sigma_3}} \sqrt{\frac{\partial \sigma_1}{\partial \sigma_3}}}$$

Remember all expression and try to derive it yourself.

$$\tan \theta_f = \sqrt{\frac{\partial \sigma_1}{\partial \sigma_3}}$$

1/1/17

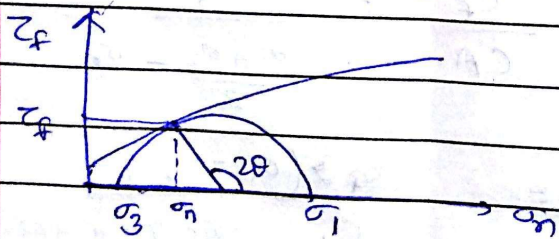
Non linear Strength Criteria:-



at failure

$$\sigma_n = \sigma_3 + \frac{\sigma_1 - \sigma_3}{1 + \frac{\partial \sigma_1}{\partial \sigma_3}}$$

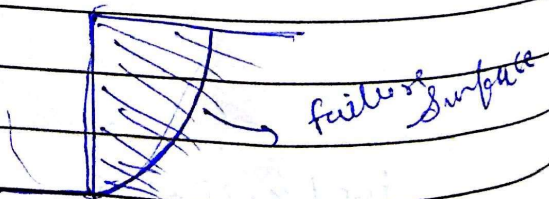
$$\tau_f = \frac{\sigma_1 - \sigma_3}{1 + \frac{\partial \sigma_1}{\partial \sigma_3}} \sqrt{\frac{\partial \sigma_1}{\partial \sigma_3}}$$



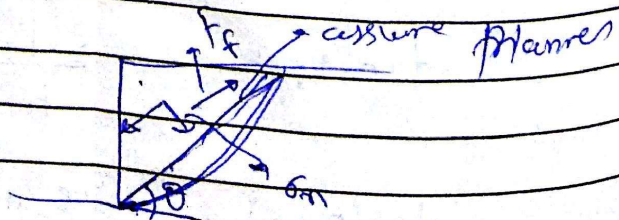
$$\tan \theta = \sqrt{\frac{\partial \sigma_1}{\partial \sigma_3}}$$

Example

Heavily fractured Rock Mass



assumption:- failure surface is circular



$$F_f = \text{max } \tau_f$$

$$\sigma_f = A\sigma_3^2 + B\sigma_3 + C$$

$$\frac{\partial \sigma_f}{\partial \sigma_3} = 2A\sigma_3 + B$$

$$\tau_f = \left(\frac{\sigma_1 - \sigma_3}{1 + 2A\sigma_3 + B} \right) \sqrt{(2A\sigma_3 + B)}$$

$$\sigma_n = \sigma_3 + \frac{\sigma_1 - \sigma_3}{1 + 2A\sigma_3 + B}$$

first calculate σ_3 then τ_f

σ_n is known to me

$$\tau_f = f(\sigma_3) \quad \text{and} \quad \sigma_n = f(\sigma_3)$$

If $\sigma_n = 5 \text{ MPa}$ what is $\tau_f = ?$

$$\text{if } A = -0.02, B = 4, C = 6$$

$$\sigma_n = 5 = \sigma_3 + \frac{-0.02\sigma_3^2 + 4\sigma_3 + 6 - \sigma_3}{1 - 0.04\sigma_3 + 4}$$

$$25 - 0.2\sigma_3 = 5\sigma_3 - 0.04\sigma_3^2 - 0.02\sigma_3^2 + 3\sigma_3 + 6$$

$$\Rightarrow 0.06\sigma_3^2 - 8.2\sigma_3 + 19 = 0$$

$$\sigma_3 = 134.30 \quad \text{and} \quad \sigma_3 = 2.35$$

$$\sigma_1 = 182.47 \quad \tau_f = 15.289$$

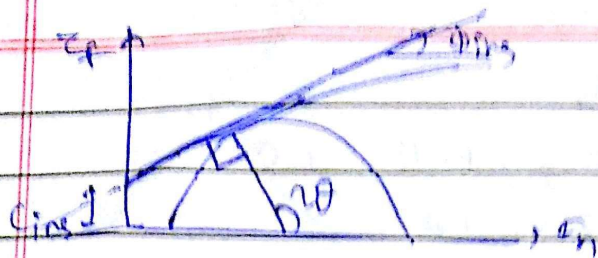
So discard

$$\tau_f = \frac{182.47 - 134.30}{1 - 0.04 \times 134.30 + 4} \sqrt{-0.04 \times 134.30 + 4}$$

$$\tau_f = \frac{15.289 - 2.35}{1 - 0.04 \times 2.35 + 4} \sqrt{-0.04 \times 2.35 + 4}$$

$$\tau_f = 5.21 \text{ MPa} \quad \checkmark$$

$$\tan \theta = \sqrt{-0.02 \times 2.35 + 4} \Rightarrow 1.976$$



$$\tau_f = \sigma_n \tan(\phi_{intst}) + C_{intst}$$

$$90 = 90 + \phi_{intst}$$

$$\phi_{intst} = 20 - 90, \quad \phi_{intst} = 36.32^\circ$$

for, $\tan(\phi_{intst}) = \tau_f$

$$C_{intst} = \tau_f - \sigma_n \tan(\phi_{intst})$$

$$C_{intst} = 5.21 - 5 \tan(36.32^\circ)$$

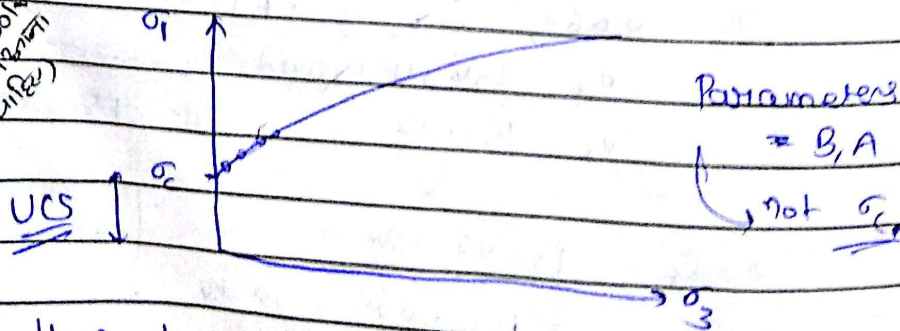
$$C_{intst} = 1.534$$

Empirical Criteria:-

a) Murrell (1965)

$$\sigma_1 = \sigma_c + B(\sigma_3)^A$$

(no need to remember, But solve in lab)



Parameters = B, A, not σ_c

How to get B, A

TRX test - in-lab

σ_3	σ_1
2	70
5	100
8	120
10	130

find out B, A = ?

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Fit in Murrell's model.

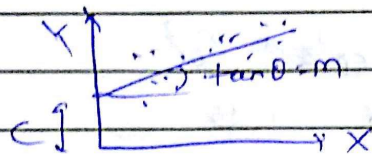
$$\sigma_1 - \sigma_c = B \sigma_3^A$$

$$\ln(\sigma_1 - \sigma_c) = \ln B + A \ln \sigma_3$$

$\sigma_c = \text{UCS}$

$$Y = \text{---} + AX$$

$$Y = C + mX \quad \text{linear}$$



$$C = \ln B \Rightarrow B = e^C$$

$$m = A \Rightarrow A = m$$

(b) Bleniaswski

(no need to remember)

$$\frac{\sigma_1}{\sigma_c} = 1 + B \left(\frac{\sigma_3}{\sigma_c} \right)^\alpha$$

$$\ln \left(\frac{\sigma_1 - \sigma_c}{\sigma_c} \right) = \ln B + \alpha \ln \left(\frac{\sigma_3}{\sigma_c} \right)$$

$$Y = C + mX$$

$$B = e^C, \quad \alpha = m$$

$\alpha \approx 0.74 \Rightarrow B$ will be different (more or less constant) for different plots

B depends on type of rock

	B
Middlestone / Silt Stone	3
Sand Stone	4
Quartzite	4.5
Granite	5

(c) Hoek-Brown (1980)

Remembers exam

$$\sigma_1 = \sigma_3 + \sqrt{m \sigma_c \sigma_3 + S \sigma_c^2}$$

most widely used (UPCP)

It is parameter here.

σ_{ci} = intact rock UCS

Parameters = m, σ_{ci}, S

For intact Rocks

$S=1$

for intact Rocks ←

$$\sigma_1 = \sigma_3 + \sqrt{m \sigma_{ci} \sigma_3 + \sigma_{ci}^2}$$

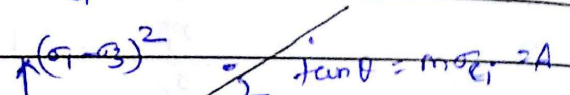
$$(\sigma_1 - \sigma_3)^2 = m \sigma_{ci} \sigma_3 + \sigma_{ci}^2$$

$$Y = AX + B$$

$$A = m \sigma_{ci}$$

$$B = \sigma_{ci}^2$$

$$Y = (\sigma_1 - \sigma_3)^2, X = \sigma_3$$



σ_3	σ_1
2	70
5	100
8	120
10	130
0	40
0	50

Total 6 data points

→ for Muirell $\Rightarrow \sigma_c = \frac{40+50}{2} = 45$ vs (it is not Parameter) (only 4 data point required) for best fit

→ but for Hoek brown, it is Parameter

→ Take all data point (6 data points)

σ_{ci} is Parameter now. (don't take as)

σ_3	σ_1	$Y = (\sigma_1 - \sigma_3)^2$	$X = \sigma_3$
0	40	1600	0
0	50	2500	0
2	70	4624	2
5	100	9025	5
8	120	12544	8
10	130	14400	10

$B = 2171.90$

$A = 1266.46$

$$Y = 1266.46X + 2171.90$$

$$\sigma_{C_i} = 46.60$$

$$m = 27.17$$

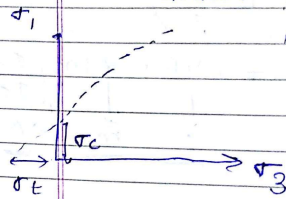
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Rough value of m .

Greas - 24.5

Granite - 29.2



Physical significance of m .

$$\text{For } \sigma_3 = -\sigma_t$$

$$\sigma_1 = 0$$

$$0 = -\sigma_t + \sqrt{m \cdot \sigma_c (-\sigma_t) + \sigma_c^2}$$

$$t^2 = c^2 - m \sigma_c \sigma_t$$

$$m = \frac{\sigma_c}{\sigma_t} - \frac{\sigma_t}{\sigma_c}$$

Remember

$$m \leq \frac{\sigma_c}{\sigma_t}$$

$$\sigma_1 - \sigma_3 = \frac{2c \cos \phi_0}{1 - \sin \phi_0} + \frac{2 \sin \phi_0}{1 - \sin \phi_0} \sigma_3$$

Non-linear criterion (MMC)

$$(\sigma_1 - \sigma_3) = \frac{2c \cos \phi_0}{1 - \sin \phi_0} + \frac{2 \sin \phi_0}{1 - \sin \phi_0} \sigma_3 - A' \sigma_3^2$$

$$\frac{\partial (\sigma_1 - \sigma_3)}{\partial \sigma_3} = \frac{2 \sin \phi_0}{1 - \sin \phi_0} - 2A' \sigma_3 = 0$$

$$\frac{\sin \phi_0}{1 - \sin \phi_0} = A' \sigma_3$$

Experimentally

$$\sigma_{crit} = \frac{1}{A'} \left[\frac{\sin \phi_0}{1 - \sin \phi_0} \right]$$

$$\sigma_{crit} \approx \sigma_{ci}$$

$$A' = \frac{\sin \phi_0}{\sigma_{crit} (1 - \sin \phi_0)}$$

$$A' = \frac{\sin \phi_0}{\sigma_{ci} (1 - \sin \phi_0)}$$

~~Eq~~

$$\sigma_1 - \sigma_3 = \frac{2c \cos \phi_0}{1 - \sin \phi_0} + \frac{2 \sin \phi_0}{1 - \sin \phi_0} \sigma_3 - \frac{1}{\sigma_{ci}} \frac{\sin \phi_0}{1 - \sin \phi_0} \sigma_3^2$$

$$(\sigma_1 - \sigma_3) = \sigma_{ci} + B \sigma_3 + A \sigma_3^2 \quad \text{OR} \quad 0 \leq \sigma_3 \leq \sigma_{ci} \quad \left(B = -2 \sigma_{ci} A \right)$$

where $B = \frac{2 \sin \phi_0}{1 - \sin \phi_0}$, $A = \frac{1}{\sigma_{ci}} \frac{\sin \phi_0}{1 - \sin \phi_0}$

for $\sigma_3 > \sigma_{ci}$

$$\sigma_1 - \sigma_3 = \sigma_{ci} + \frac{2 \sin \phi_0}{1 - \sin \phi_0} (\sigma_{ci}) - \frac{1}{\sigma_{ci}} \frac{\sin \phi_0}{1 - \sin \phi_0} \sigma_{ci}^2$$

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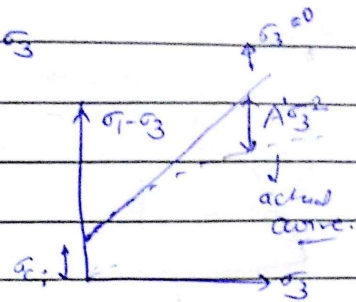
Strength Criteria:-

MMC (Modified Mohr Coulomb Criteria)

$$\sigma_1 = \frac{2C_0 \cos \phi_0}{1 - \sin \phi_0} + \frac{1 + \sin \phi_0}{1 - \sin \phi_0} \sigma_3$$

Subtracting σ_3 (LHS and RHS)

$$\sigma_1 - \sigma_3 = \frac{2C_0 \cos \phi_0}{1 - \sin \phi_0} + \frac{2 \sin \phi_0}{1 - \sin \phi_0} \sigma_3$$



Proposed Equation

$$\sigma_1 - \sigma_3 = \sigma_{c1} + B \sigma_3 - A \sigma_3^2$$

$$\sigma_{c1} = \frac{2C_0 \cos \phi_0}{1 - \sin \phi_0}$$

$$B = \frac{2 \sin \phi_0}{1 - \sin \phi_0}$$

0 → indicates $\sigma_3 \rightarrow 0$

Another way of representation

Exam

$$(\sigma_1 - \sigma_3) = \sigma_{c1} + B \sigma_3 + A \sigma_3^2$$

Boundary Condition

from experience $\sigma_{3 \text{ out}} = \sigma_{c1}$

$$\frac{\partial(\sigma_1 - \sigma_3)}{\partial \sigma_3} = 0 + B + 2A \sigma_3$$

$$\text{at } \sigma_3 \rightarrow \sigma_{c1} = \sigma_{c1}, \quad \frac{\partial(\sigma_1 - \sigma_3)}{\partial \sigma_3} \rightarrow 0$$

$$\text{So, } B + 2A \sigma_{c1} = 0$$

$$B + 2A \sigma_{c1} = 0$$

$$B = -2A \sigma_{c1}$$

So, for the range $\sigma_3 \leq \sigma_{c1}$

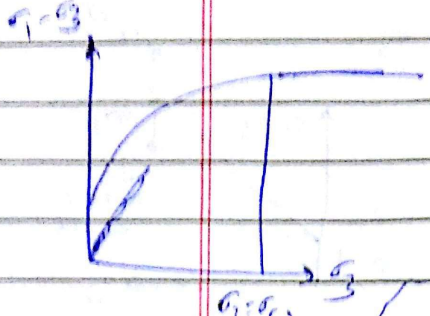
$$(\sigma_1 - \sigma_3) = \sigma_{c1} - 2A \sigma_{c1} \sigma_3 + A \sigma_3^2$$

A = empirical coefficient

UPCP

For $\sigma_3 > \sigma_{ci}$

for $\sigma_3 = \sigma_{ci}$ $(\sigma_1 - \sigma_3) = \sigma_{ci} - 2A\sigma_{ci}\sigma_{ci} + A\sigma_{ci}^2$
 $(\sigma_1 - \sigma_3) = \sigma_{ci} - A\sigma_{ci}^2$



$$A = \frac{-1 \sin \phi_0}{\sigma_{ci} (1 - \sin \phi_0)}$$

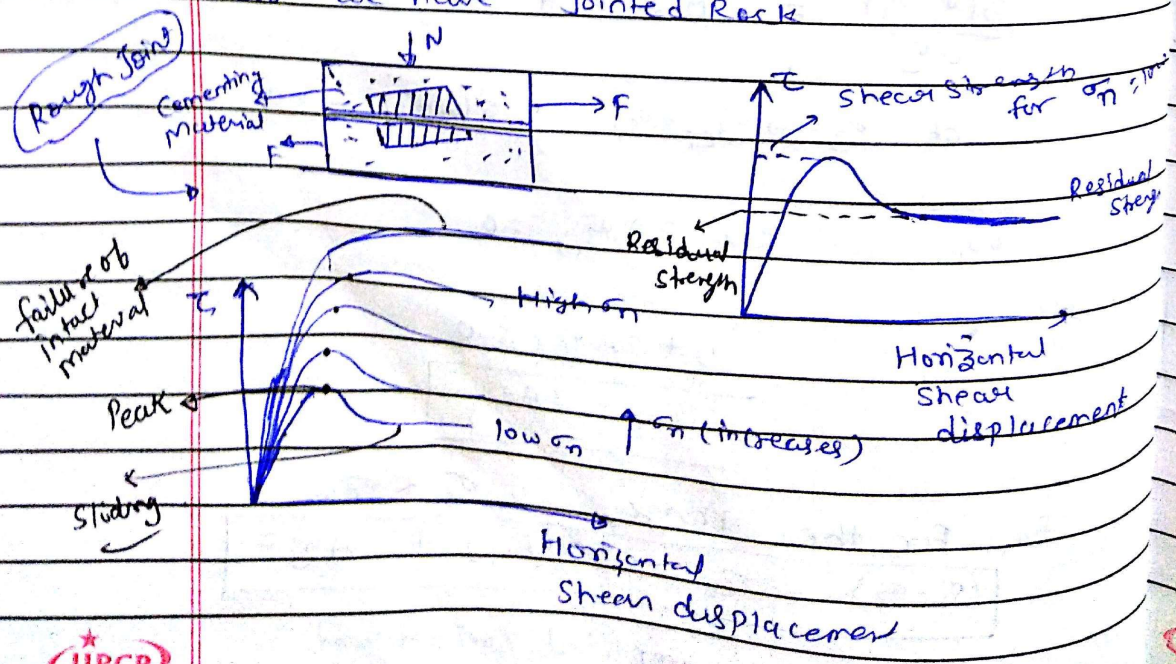
$(\sigma_1 - \sigma_3 - \sigma_{ci}) = A(\sigma_{ci}^2 - 2\sigma_{ci}\sigma_3)$

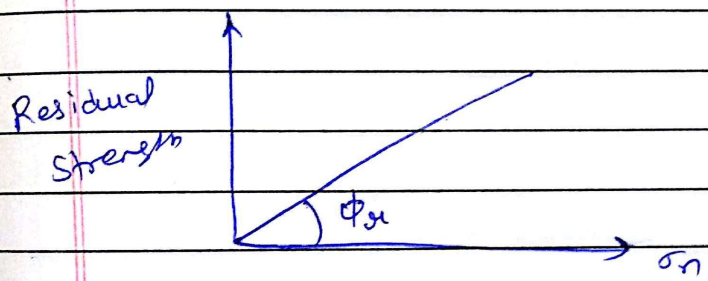
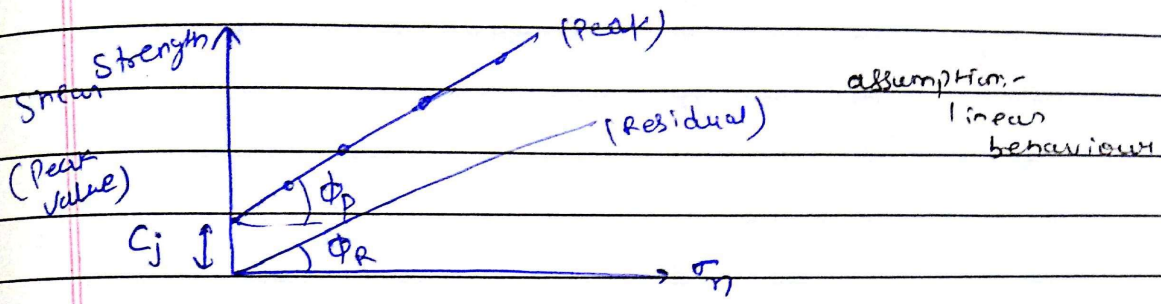
$\sigma_3 < \sigma_{ci} \Rightarrow$ $\underbrace{\hspace{10em}}_Y \quad \underbrace{\hspace{10em}}_X \quad \boxed{Y \geq AX}$

$$A = \frac{\sum (\sigma_1 - \sigma_3 - \sigma_{ci})}{\sum (\sigma_{ci}^2 - 2\sigma_{ci}\sigma_3)}$$

Shear Strength Behaviour of Rock Joints:-

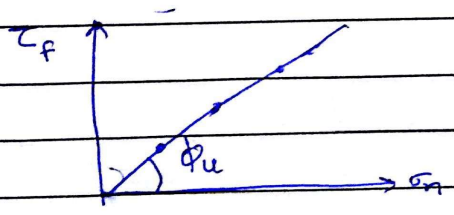
lets us say
lets we have jointed rock





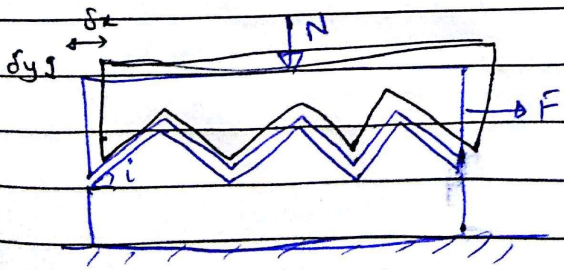
Shear strength Models :-

Saw Cut Joint (Smooth Joint)



$\phi_u = \frac{\text{friction coefficient}}{\mu_s \tan \phi_u}$

intercept = 0 (for smooth surface)
intercept ≠ 0 (rough surface)



$N = \text{Low} \Rightarrow$ Pure sliding (but there is shear strength $\tau_f(\sigma_n)$)

$\delta x, \delta y$ displacement

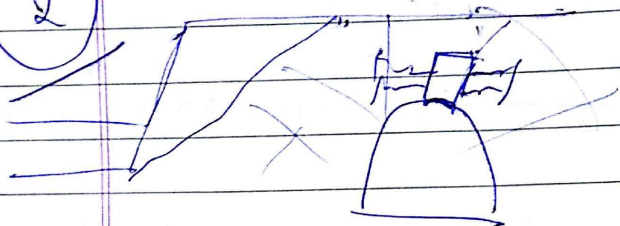


10/03/17 Shear strength of Rock Joints \rightarrow Normal load constant during shear

CNL - Const. Normal Load Condition

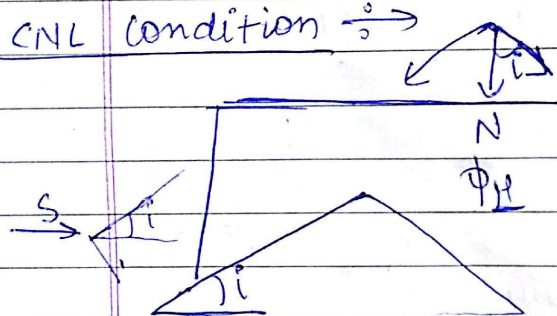
CNS - Const. Normal stiffness condition. \rightarrow k_s will change with displacement (enhanced strength)

2



Patton's Model \rightarrow

CNL condition \rightarrow



$N = \text{low} = \text{only sliding}$

Component causing sliding
 $= S \cos i - N \sin i$

Normal to sliding surface
 $= S \sin i + N \cos i$

3

At the verge of sliding

$$\tau = \sigma_n \tan \phi_u$$

Basic friction angle.

$$S \cos i - N \sin i = S \sin i + N \cos i \times \tan \phi_u$$

$$S (\cos i - \tan \phi_u \sin i) = N (\sin i + \cos i \tan \phi_u)$$

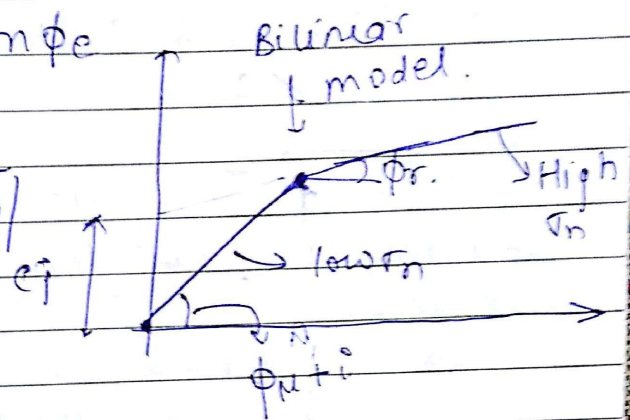
$$S = N \cdot \tan(\phi_u + i)$$

$$\tau_f = \sigma_n \cdot \tan(\phi_u + i) \quad \rightarrow \text{For low normal load}$$

$$\tau_f = \sigma_n \tan \phi_e$$

For high $\sigma_n \rightarrow$

$$\tau_f = c_j + \tan(\phi_j^s) \cdot \sigma_n$$

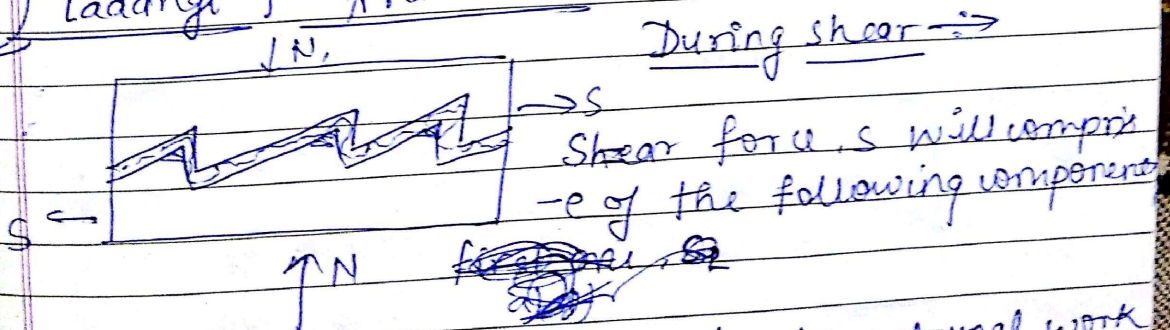


1) Coulomb's Equation \rightarrow

$$\tau_f = c_j + \sigma_n \tan \phi_j^s$$

Angle of approach is limiting value of ϕ when σ_n tends to zero.

3) Ladanyi & Archambault \rightarrow



- 1) S_1 - This is the component due to external work in dilating against the external force N
- 2) S_2 = Additional internal work done in friction, due to dilation only.
- 3) S_3 = Component due to work done in internal friction if the specimen does not change in volume.

$S_4 =$ work done in shearing the asperities.

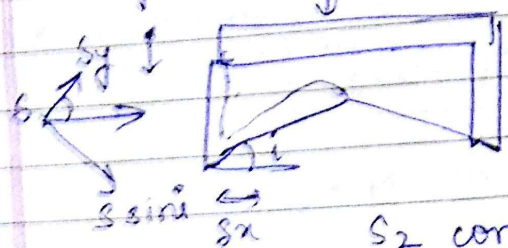
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Adami, Archambault-based, on energy principles, shear strength model for rock joint -



$$S_1 = \frac{N \cdot S_y}{dx} = N \tan i \quad \text{--- ①}$$

S_2 component
 $(S \sin i) \times \tan \phi_u = \text{Frictional force} = S_2 \cos i$
 $S_2 = S \tan i \tan \phi_u \quad \text{--- ②}$

First 2 components \therefore of geometry

If dilation was not there, first two components will be zero.

$$S_3 = N \tan \phi_u$$

No shearing \rightarrow Only sliding $\rightarrow S_4 = 0$

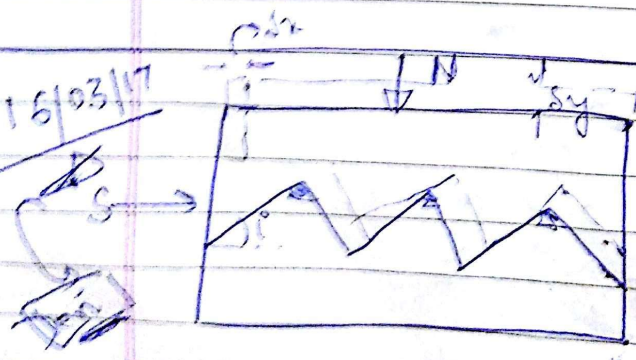
$$S = S_1 + S_2 + S_3 + S_4$$

$$S = N \tan i + S \tan i \tan \phi_u + N \tan \phi_u$$

$$S = \frac{N (\tan i + \tan \phi_u)}{1 - \tan i \tan \phi_u}$$

$$S = N \tan(\phi_u + i)$$

16/03/17



$$S = S_1 + S_2 + S_3 + S_4$$

$$S_1 dx = N dy$$

$$S_1 = N \tan i$$

$$S_2 = S \tan i \tan \phi_u$$

$$S_3 = N \tan \phi_u$$

$S_4 =$ shear = Mohr Coulomb Equation

$$\tau = c_i + \sigma \tan \phi_i \quad \text{for intact material failure}$$

$$S_4 = A (c_i + \sigma \tan \phi_i) \quad (c_i, \phi_i)$$

$$S = (S_1 + S_2 + S_3) (1 - a_s) + S_4 \cdot a_s$$

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$a_s = \text{Shear Area Ratio} = \frac{A_s}{A}$

$$S = (N \tan i + S \cdot \tan i \cdot \tan \phi_u + N \tan \phi_u) (1 - a_s) + A (c_i + \sigma_n \tan \phi_i) a_s$$

$$S - S \tan i \cdot \tan \phi_u (1 - a_s) = (N \tan i + N \tan \phi_u) (1 - a_s) + A (c_i + \sigma_n \tan \phi_i) a_s$$

$$S = \frac{N (\tan i + \tan \phi_u) (1 - a_s) + A (c_i + \sigma_n \tan \phi_i) a_s}{[1 - \tan i \tan \phi_u (1 - a_s)]}$$

$$\tau = \frac{S}{A} = \frac{[\sigma_n (1 - a_s) (\tan i + \tan \phi_u) + (c_i + \sigma_n \tan \phi_i) a_s]}{[1 - \tan i \tan \phi_u (1 - a_s)]}$$

If σ_n is very very low $\rightarrow a_s \approx 0$



$$\tau = \frac{\sigma_n (\tan i + \tan \phi_u)}{(1 - \tan i \tan \phi_u)} \rightarrow \tau = \sigma_n \tan (\phi_u + i)$$

If σ_n is very very large

$$\tau = c_i + \sigma_n \tan \phi_i$$

$$S \cdot S_x = \left[N S_y + S \cdot S_m \cdot \tan \phi_u \cdot \frac{S_x}{\cos i} + N \cdot \tan \phi_u \cdot S_x \right] (1 - a_s) + [A (c_i + \sigma_n \tan \phi_i)] a_s \cdot S_x$$

$$a_s = 1 - \left(\frac{1 - \sigma_n}{\sigma_{TRN}} \right)^{k_1} \rightarrow 1.5$$

$$\tan i = \left(\frac{1 - \sigma_n}{\sigma_{TRN}} \right)^{k_2} \cdot \tan i_0$$

Transition stress (Brittle to ductile)

a_s is a function of Normal stress

Barton's Model \rightarrow (Most widely used) \rightarrow JRC

- Tests on specimens with different roughness and normal stresses.

JRC \rightarrow Joint Roughness Coefficient.

$$\tau = \sigma_n \tan \phi_u + \frac{JRC \log_{10} \frac{JCS}{\sigma_n}}{L \text{ (degrees)}}$$

Standard profiles are available.



(R)

$$\tau_{1f} = \tau_3 + \frac{2(c_j + \tau_3 \tan \phi_j)}{\sin 2\theta \left(1 - \frac{\tan \phi_j}{\tan \theta}\right)}$$

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→ strength of jointed rock would be smaller of (I) & (II)

$$\frac{d(\tau_{1f})}{d\theta} = 0 \Rightarrow 2 \cos 2\theta \left(1 - \frac{\tan \phi_j}{\tan \theta}\right) + \sin 2\theta \left(0 + \frac{\tan \phi_j}{\tan^2 \theta}\right) = 0$$

$$\Rightarrow 2 \cos 2\theta - \left(\frac{2 \cos 2\theta}{\tan \theta}\right) \tan \phi_j + \frac{2 \sin \theta \cos \theta}{\tan \theta \sin^2 \theta} \tan \phi_j = 0$$

$$2(\cos 2\theta)(\tan \theta) = \tan \phi_j (-2 + 2 \cos 2\theta)$$

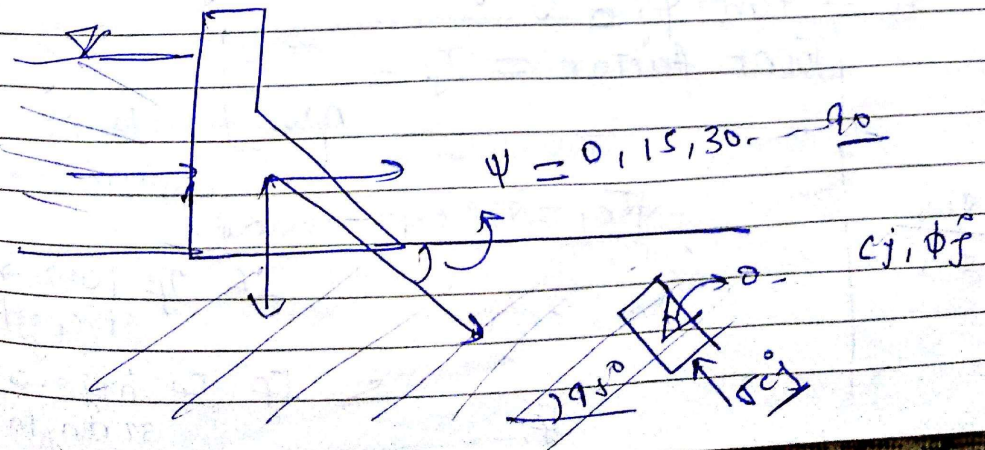
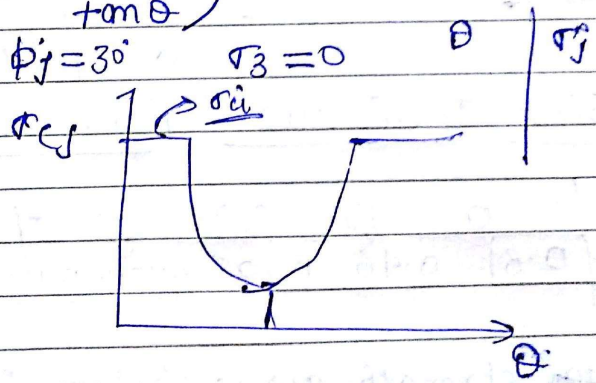
$$\tan \phi_j (\cos 2\theta - 1) = (\cos 2\theta) \tan \theta$$

$$\tan \phi_j \cdot \tan 2\theta = -1$$

$$2\theta = 90 + \phi_j \rightarrow \theta = 45 + \frac{\phi_j}{2}$$

$$\tau_{c_j} = \frac{2c_j}{\sin 2\theta \left(1 - \frac{\tan \phi_j}{\tan \theta}\right)}$$

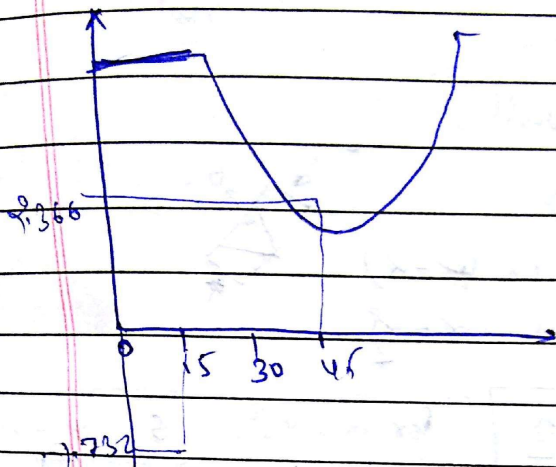
ex) $c_j = 0.5 \text{ MPa}$, $\phi_j = 30^\circ$, $\tau_3 = 0$



Example

$C_j = 0.5 \text{ MPa}$

$\phi_j = 30^\circ$



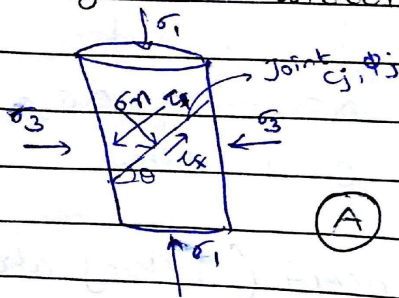
$\sigma_j \Rightarrow (\text{tension}) (\sigma_3 = 0)$
UCS

$\sigma_j = \frac{2 C_j}{\sin \theta (1 + \frac{\tan \theta_j}{\tan \theta})}$

θ	σ_j	
15	-1.732	(Tension)
45	2.366	(Compression)
60	1.732	
75	2.366	

21/3/17

Strength Behaviour of Jointed Rocks



For failure $\tau \geq \tau_f$

(A) Single plane of weakness

Theory :-

- ① σ_1 - failure along the joint plane
- ② σ_1 - failure of intact material

UPCP

Strength of rock

Physical minimum

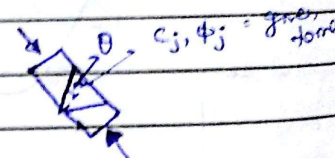
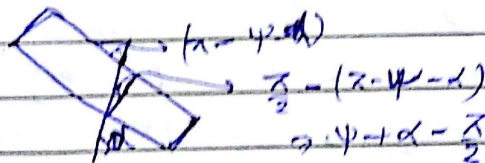
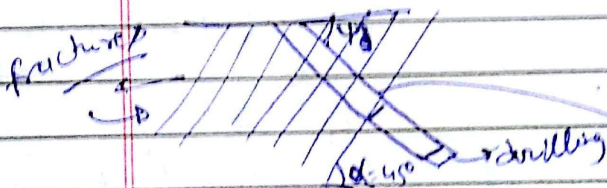
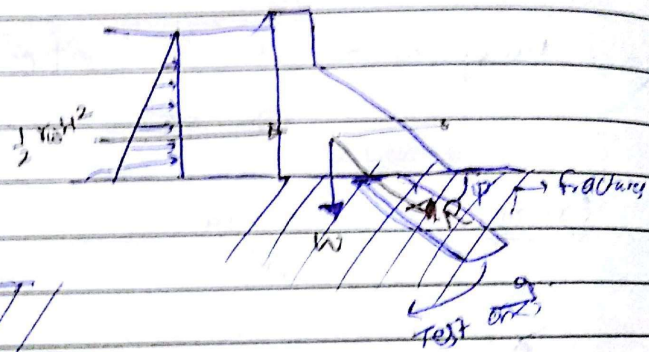
$$\sigma_1 = \sigma_3 + \frac{2(c_j + \sigma_3 \tan \phi_j)}{\sin 2\theta (1 - \frac{\tan \phi_j}{\tan \theta})}$$

When $\sigma_3 = 0$ ($\sigma_1 = \text{UCS} = \sigma_{c_j}$)

$$\sigma_{c_j} = \frac{2c_j}{\sin 2\theta (1 - \frac{\tan \phi_j}{\tan \theta})}$$

no. of joints per meter = J_n

for example



β	η
0	0.81
10	0.46
20	0.105
30	0.04
40	0.01
50	0.30
60	0.4
70	0.634
80	0.81
90	1.0

$$\theta = \psi + \alpha - \frac{\pi}{2} \quad \text{for } \alpha = 45^\circ, \quad \theta = \psi - 45^\circ$$

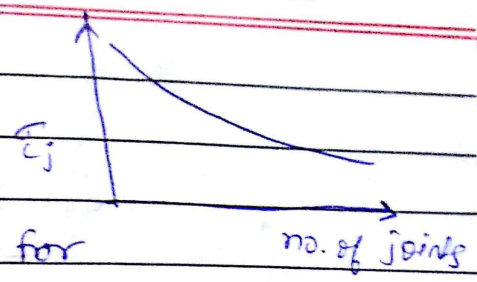
ψ	σ_1	σ_2	σ_1	σ_2	σ_{c_j}	Select minimum

assumption: no interference of planes with each other.

(B) Joint factor concept (Talking about UCS only)
 1) frequency of joint is important

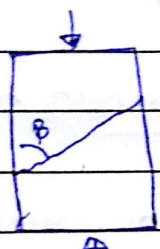
no. of joints per meter depth = J_n

So, we should account for frequency of joints.

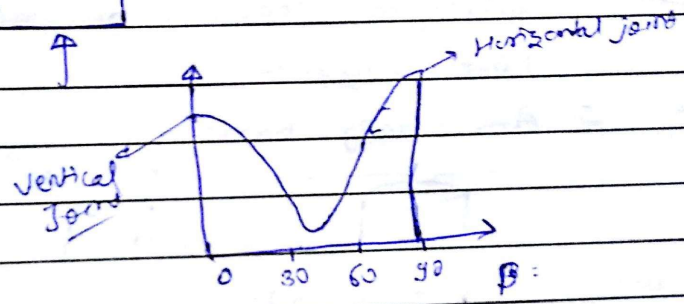


2) orientation is important

β	η
0	0.81
10	0.46
20	0.105
30	0.046
40	0.071
50	0.306
60	0.465
70	0.634
80	0.814
90	1.0



at what β , you will get minimum UCS
 $\beta = 45 - \phi/2$



Joint inclination parameter

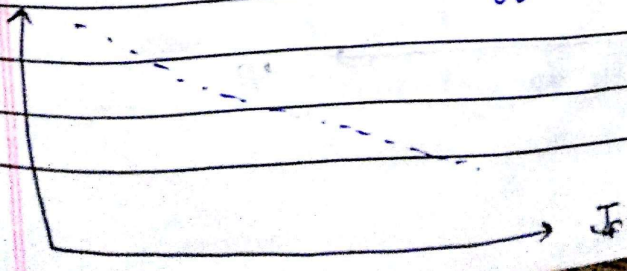
for β , strength = n times of UCS

3) shear strength along joint = ϕ_j (roughness)
 $\mu = \tan \phi_j$

Joint factor (J_f) = $\frac{J_n}{n \cdot \mu}$

Labels: J_n is 'No. of joints per meter depth', $n \cdot \mu$ is ' $\tan \phi_j$ '.

for different β , there is a 'n'



(UPCP)

if $J_f \Rightarrow$ High \Rightarrow Strength = low
 $J_f =$ low \Rightarrow Strength = High
 $J_f = 0 \Rightarrow$ strength of intact Rock

(no need to remember)

$$\sigma_{eq} = \sigma_c \cdot e^{-(0.008 \cdot J_f)}$$

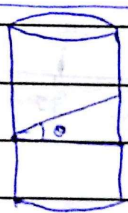
Upper limit of joints strength = UCS of intact Rock

due to defects strength has decreased.

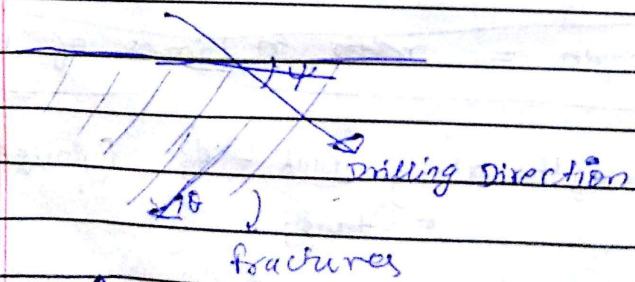
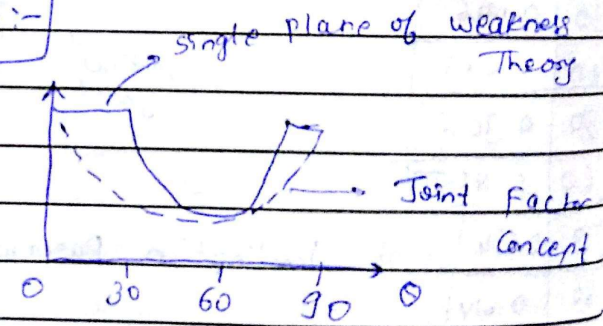
23/3/17

Strength Behaviour of Jointed Rocks and Rock Masses :-

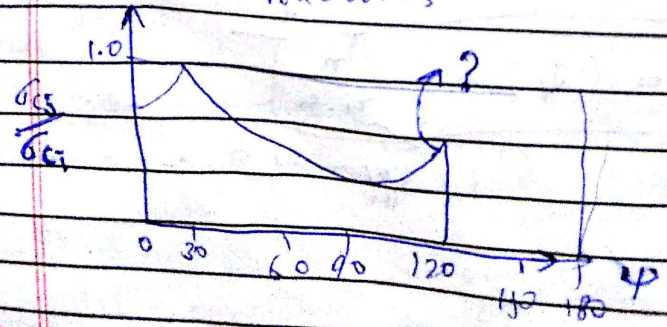
$$\frac{\sigma_{max}}{\sigma_{min}} = \text{Anisotropy Ratio :-}$$



$$\frac{\sigma_{cj}}{\sigma_c}$$



$$J_s = \frac{J_n}{n-1}$$



ϕ of
 NOTE
 ϕ_{jo}
 Tangent
 at
 30°

Ramamurthy :- (for jointed rocks)

$$\left(\frac{\sigma_1 - \sigma_3}{\sigma_3}\right) = \beta_j \left(\frac{\sigma_{ci}}{\sigma_3}\right)^{\alpha_j}$$

no need to remember

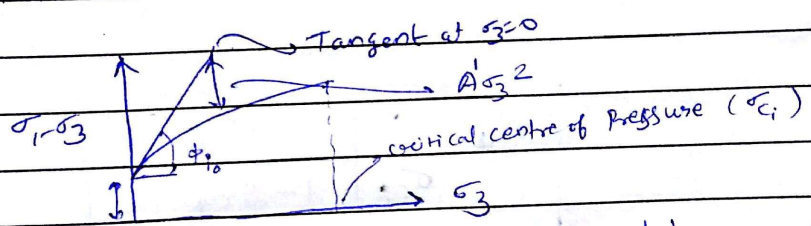
$$\alpha_j = \left(\frac{\sigma_{ci}}{\sigma_{ci}}\right)^{0.5} \times \alpha_i \rightarrow \text{TRX test intact}$$

$$\beta_j = \beta_i \cdot 0.13 e^{\{2.03 \alpha_j / \alpha_i\}}$$

IITR

Singh + Singh (2005) RMRE
 Singh et al - 2011
 - 2012

MMC :- Intact rock =



$$\sigma_1 = \frac{2 \cos \phi_0 \sigma_3}{1 - \sin \phi_0} + \sigma_3 \frac{(1 + \sin \phi_0)}{(1 - \sin \phi_0)}$$

$$(\sigma_1 - \sigma_3) = \frac{2 \cos \phi_0}{1 - \sin \phi_0} \sigma_3 + \frac{2 \sin \phi_0}{1 - \sin \phi_0} \sigma_3$$

$$(\sigma_1 - \sigma_3) = A \sigma_3^2 + B \sigma_3 + \sigma_{ci}$$

$$(\sigma_1 - \sigma_3 - \sigma_{ci}) = A (\sigma_3^2 - 2 \sigma_{ci} \sigma_3)$$

phi_0
 note phi_0
 Tangent at sigma_3 = 0

(UPCP)

$$A = \frac{\sum (\sigma_1 - \sigma_3 - \sigma_{ci})}{\sum (\sigma_3^2 - 2 \sigma_{ci} \sigma_3)}$$

$$B = -2A\sigma_c$$

$$B = \frac{2\sin\phi_{j_0}}{1 - \sin\phi_{j_0}}$$

$$\Rightarrow \boxed{\sin\phi_{j_0} = \frac{B}{2+B}}$$

ϕ_{j_0} is a parameter

Example

For jointed rocks

$$(\sigma_1 - \sigma_3) = \sigma_{c_j} + \frac{2\sin\phi_{j_0}}{1 - \sin\phi_{j_0}} \sigma_3 - A'_j \sigma_3^2$$

$$\frac{\partial(\sigma_1 - \sigma_3)}{\partial \sigma_3} = 0$$

$$\Rightarrow \frac{2\sin\phi_{j_0}}{1 - \sin\phi_{j_0}} = 2A'_j \sigma_3 \quad \left| \text{for } \sigma_3 = \sigma_{c_j} \right.$$

$$\Rightarrow \frac{2\sin\phi_{j_0}}{1 - \sin\phi_{j_0}} = 2A'_j \sigma_{c_j}$$

$$A'_j = \frac{1}{\sigma_{c_j}} \left[\begin{array}{c} \sin\phi_{j_0} \\ 1 - \sin\phi_{j_0} \end{array} \right]$$

$$(\sigma_1 - \sigma_3) = \sigma_{c_j} + \frac{2\sin\phi_{j_0}}{1 - \sin\phi_{j_0}} \sigma_3 - \frac{1}{\sigma_{c_j}} \left[\begin{array}{c} \sin\phi_{j_0} \\ 1 - \sin\phi_{j_0} \end{array} \right] \sigma_3^2$$

To apply theory σ_{c_j} and ϕ_{j_0} is unknown
 To get ϕ_{j_0}

For jointed rocks: $(\sigma_1 - \sigma_3)_{max}$ Put $\sigma_3 = \sigma_{c_j}$

$$(\sigma_1 - \sigma_3)_{max} = \sigma_{c_j} + \frac{2\sin\phi_{j_0}}{1 - \sin\phi_{j_0}} \sigma_{c_j} - \left[\frac{\sin\phi_{j_0}}{1 - \sin\phi_{j_0}} \right] \sigma_{c_j}^2$$

$$= \sigma_{c_j} + \frac{\sin\phi_{j_0}}{1 - \sin\phi_{j_0}} \sigma_{c_j}$$

no need to remember but write

28/3/11

ex: Basalt

Granite

for intact rocks

$$(\sigma_1 - \sigma_3)_{max} = \sigma_c + \frac{\sin \phi_0}{1 - \sin \phi_0} \sigma_c \rightarrow (2)$$

eqn ① and ② are equal

• no need to remember but write it

$$\sin \phi_0 \Rightarrow \left(\frac{1 - \frac{\sigma_c}{\sigma_c}}{\frac{\sigma_c}{\sigma_c}} \right) + \frac{\sin \phi_0}{1 - \sin \phi_0}$$

$$\left(\frac{2 - \frac{\sigma_c}{\sigma_c}}{\frac{\sigma_c}{\sigma_c}} \right) + \frac{\sin \phi_0}{1 - \sin \phi_0}$$

$\sigma_3 = \sigma_c$

28/3/17 Classification of Rocks and Rock Masses

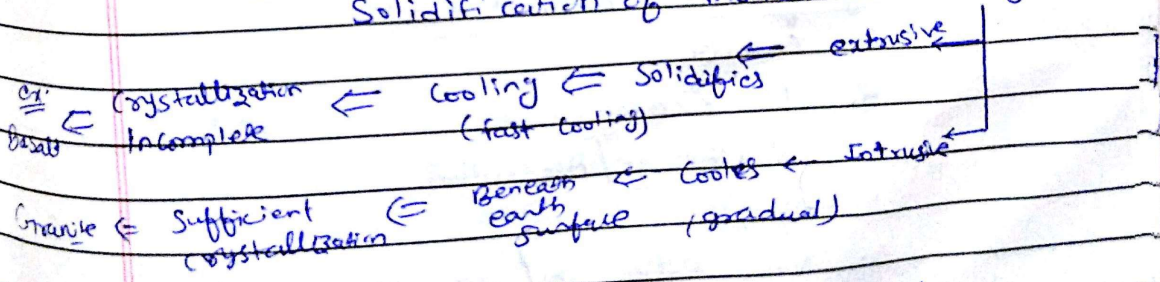
Intact rock

Geological classification:-

Tells about "origin"

Igneous → formed due to cooling and solidification of molten mass (magma)

$\frac{\phi_0}{\sin \phi_0}$ σ_3



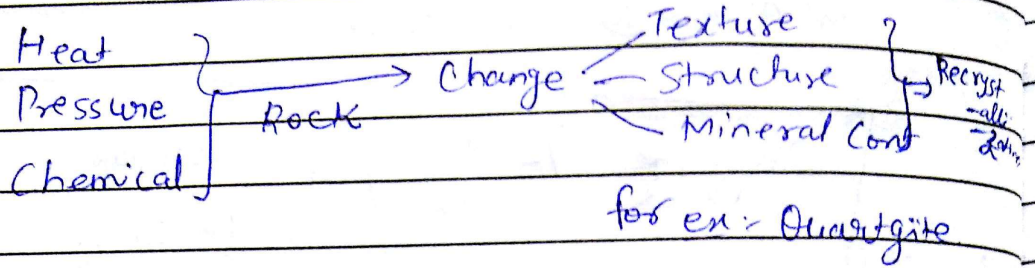
Sedimentary sediments consolidated

- ① weathering or erosion
- ② Transportation
- ③ Deposition
- ④ Consolidation and cementation

ex: Sandstone, Siltstone

(UPCP)

Metamorphic



Engineering Classification (Rock Engineering)

Purpose

- 1) Identify "Most Significant Parameters"
- 2) Divide into classes
- 3) Experience at one site \Rightarrow Site conditions at other site.
- 4) Derive "quantitative" data and guidelines for eyes design - numbers / strength / E = ?

Factors Affecting

Significant Parameters:-

1. Colour
2. Mineralogy
3. Weathering / Alteration
4. Density
5. Void Ratio \Rightarrow / Void Index
6. Wave Velocity \Rightarrow Competency
- * 7. Compressive strength
- * 8. Modulus value
- * 9. Joint spacing
- * 10. Joint Inclination
- * 11. Joint surface characterization

Most Important



Density \rightarrow

Class	Dry Density	Description
1	< 1.8	v. low
2	1.8 - 2.2	low
3	2.2 - 2.55	Moderate
4	2.55 - 2.75	High
5	> 2.75	v. High

Some Wave Velocity \rightarrow

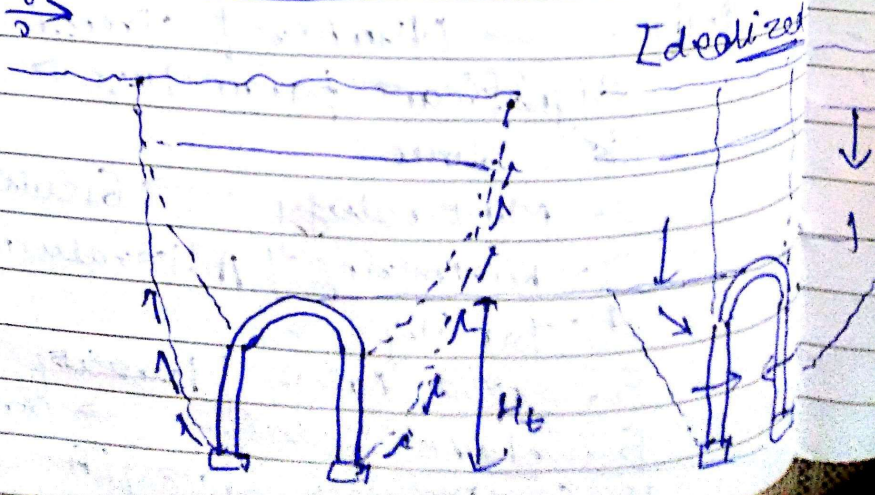
Class	Velocity (m/s)	Description
1	< 1500	v. low \rightarrow Incompetent
2		
3		
4		
5	> 5000	more competent

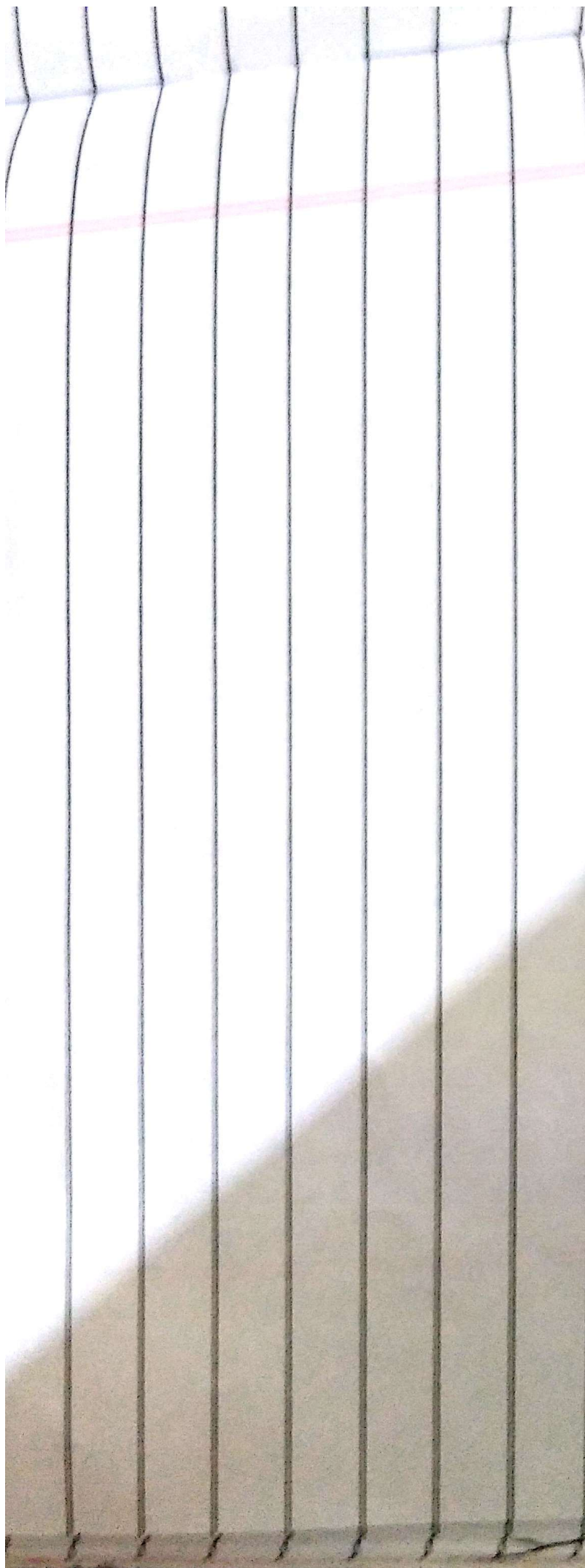
Deere - Miller

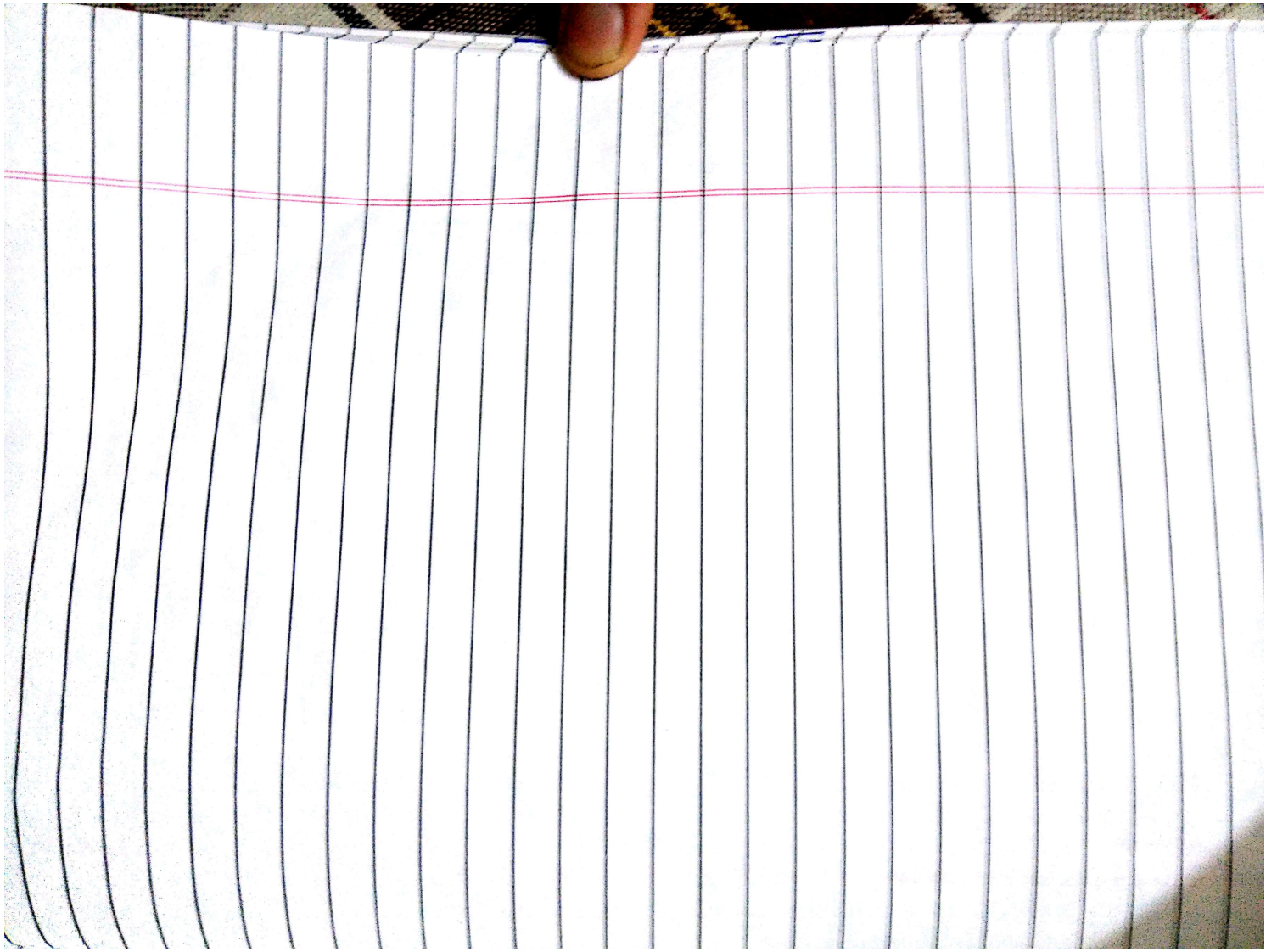
v. high strength $\rightarrow > 200 \text{ MPa} \rightarrow A$	mod. ratio
High $\rightarrow 100 - 200 \text{ " } \rightarrow B$	$M_r > 500 \Rightarrow \text{High (H)}$
medium $\rightarrow 50 - 100 \text{ " } \rightarrow C$	$200 - 500 \Rightarrow \text{Medium (M)}$
low $\rightarrow 25 - 50 \text{ " } \rightarrow D$	$< 200, \Rightarrow \text{Low (L)}$
v. low $\rightarrow < 25 \text{ " } \rightarrow E$	

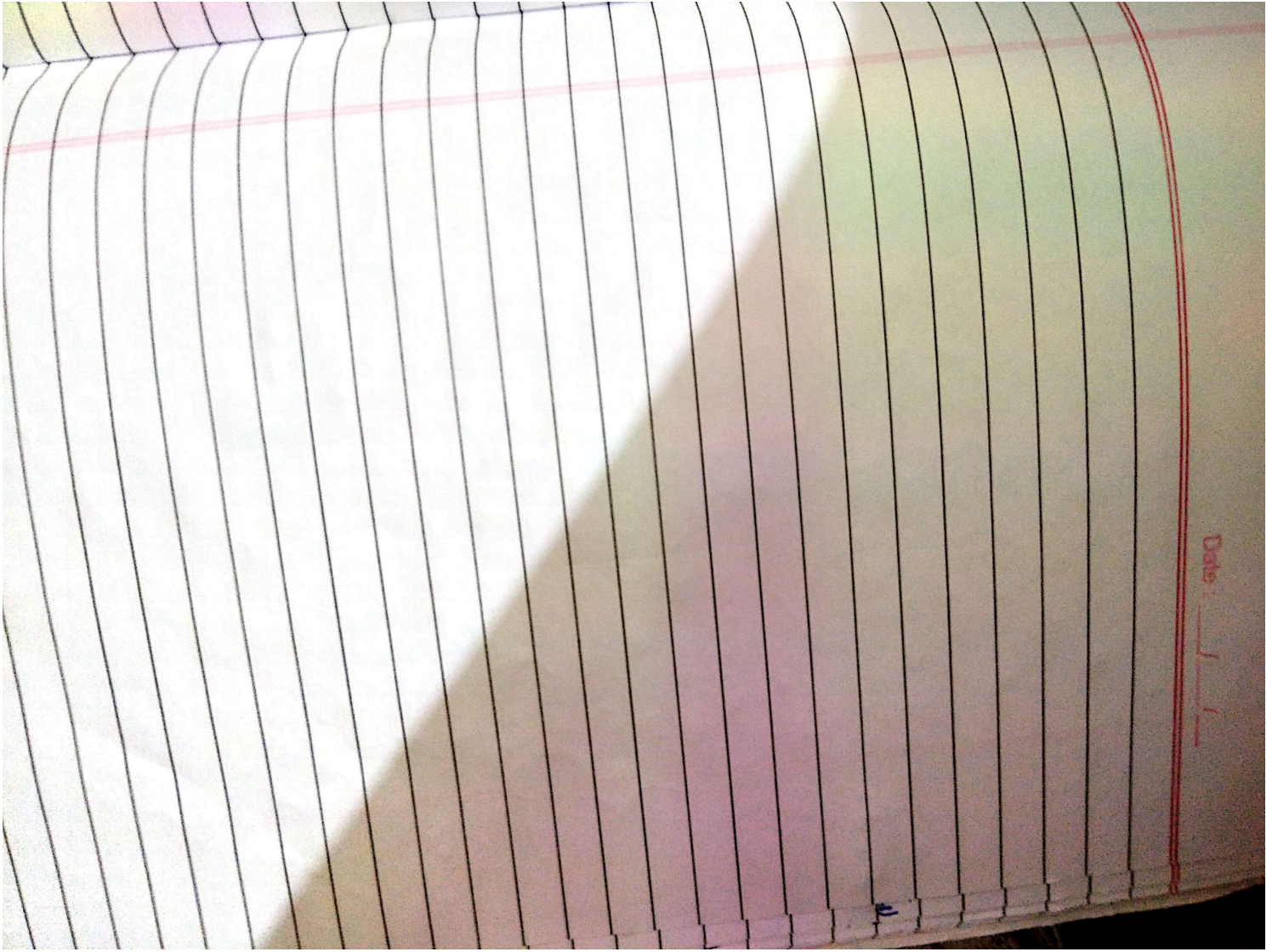
Rock Mass Classification \rightarrow

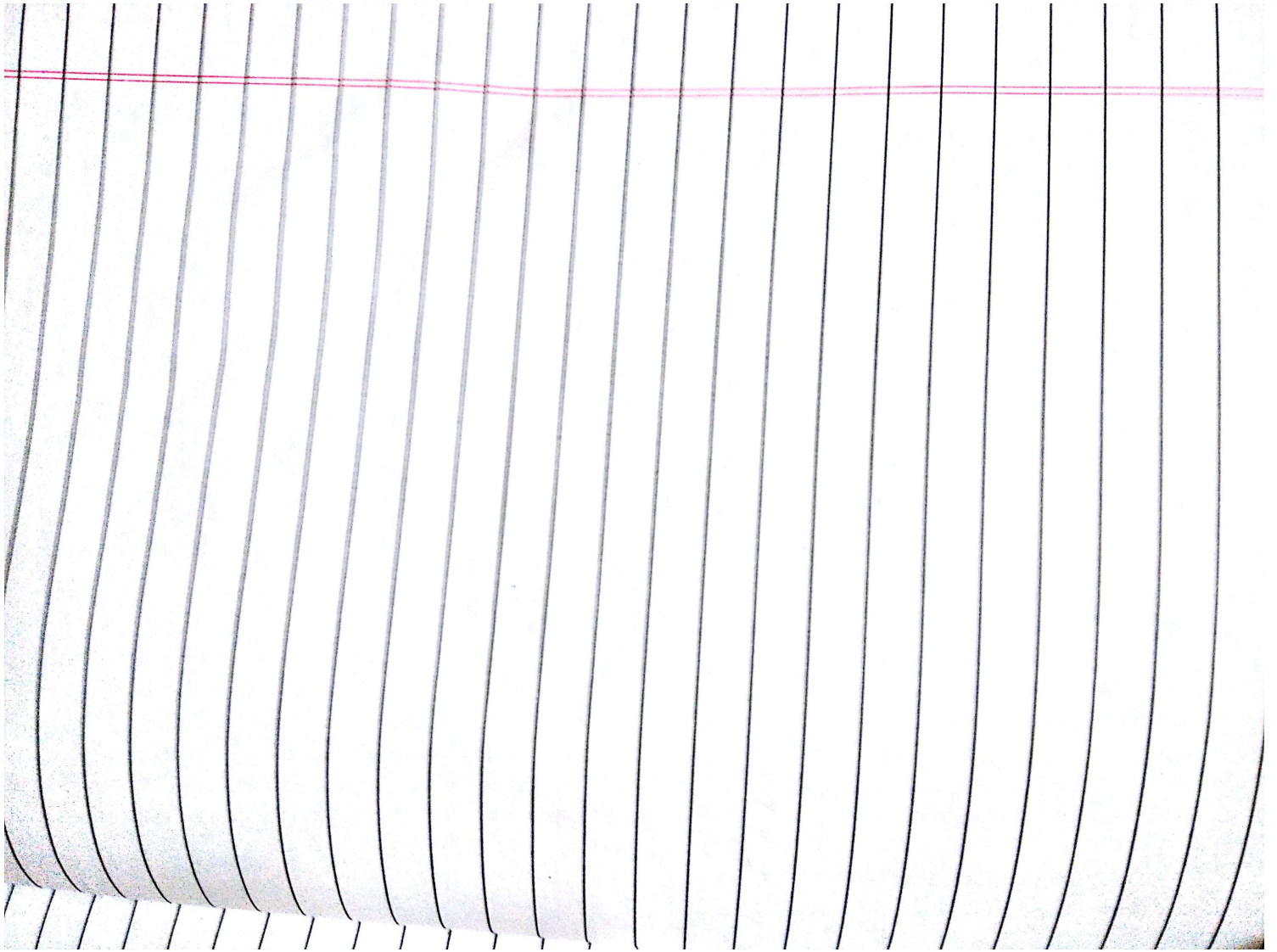
1) Terzaghi (1946) \rightarrow
"Tunnels"











31-3-17

Rock Mass Classification Systems

Date: ___/___/___

- (i) Terzaghi's Road Load Classification System
- (ii) RQD
- (iii) Rock Structure Rating (RSR)

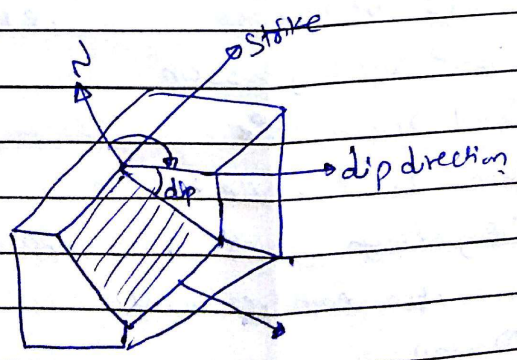
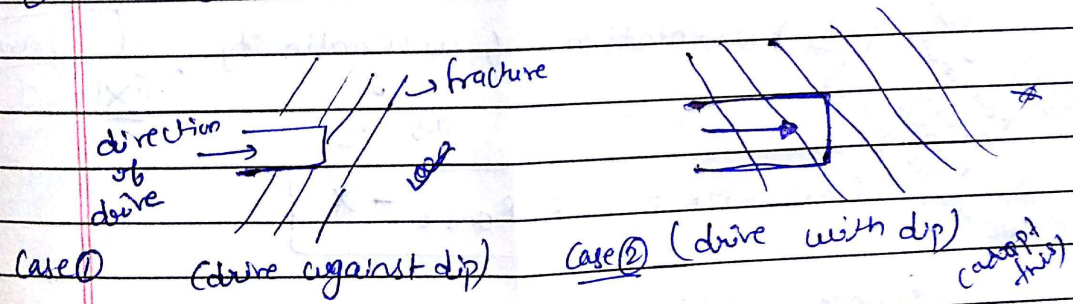
- Parameters :-

Geological Parameters	Construction Constant Parameters for
a) Rock Type	a) direction of drive (tunnels)
b) Joint Pattern	b) Size of tunnel
c) Joint Orientation	c) Method of excavation
d) Faults/folds	d) (Drill-Blast/TBM)
e) Rock Material Property	
f) Weathering	

(Use Table and chart)

Parameter A :- General Area Geology

Parameter B :- Joint Pattern, direction of drive



(with the dip better case)

example dip = 40°, joint spacing = 15cm
2-6 inch.

Driving with Dip, Strike \perp to Axis of tunnel

Parameter B weightage (Rating) from Tables \Rightarrow 16

Parameter C > Ground water, joint condition

ex $A = 24, B = 36$
 $C = A + B = 60$

Joint condition = good

500 gpm / 100 ft of tunnel = Anticipated water flow

6/4/17 So, Rating = 21 (weightage)

(iv) Rock Mass Rating (RMR) (Table 4.4)

Parameters*

- a) UCS of intact rock → 15% max. weightage
- b) RQD → 20% max. weightage
- c) spacing of discontinuity → 20% "
- d) Condition of discontinuity → 30% "
- e) ground water condition → 15% "
- f) Orientation of discontinuity → 100%
 adjustment

RMR basic

Exam Remember These names and Parameters

$\text{Final RMR} = \text{RMR}_{\text{basic}} - X$

Exam Remember

Example UCS = 55 MPa, RQD = 70%

Joint spacing = 33 cm

Joint dip = 35°

Tunnel axis parallel to strike

Very rough joints, no separation not continuous, blow weathered wall

Damp.

a = 7	d = 30	f = -5	(fair condition)
b = 13	e = 10		for tunnels at max
c = 10			

(UPCP)

Final RMR = 7 + 13 + 10 + 30 + 10 - 5
 = 70 - 5 = 65 R_c

(UPCP)

Rock Mass Classes

=> Good Rock [Rating = (61, 80)]

class number = II (Good Rock)

So, average stand up time = 1 year for 10m^{span} ^{openly}

(one year dis support (70/73))

otherwise tunnel ~~is~~

Cohesion of

(c) Rock Mass = 300 - 400 kPa

(φ) = 35 - 45

Inclined case

Penalty for foundations = (φ)

(V) Rock Mass Quality Index, Q (1973)

(Q system) (most popular system)

exam
Remember

$$Q = \left[\frac{RQD}{J_n} \right] \left[\frac{J_r}{J_a} \right] \left[\frac{J_w}{SRF} \right]$$

Table = 4.6

$$Q = \frac{RQD}{J_n} \times \frac{J_r}{J_a} \times \frac{J_w}{SRF}$$

Parameters

i) RQD = Rock Quality Designation ^(spacing)

ii) J_n = Joint Set numbers (no. of joint set)

if J_n increase than RQD will decrease

$\frac{RQD}{J_n}$ = Block Size (gives you idea about)

iii) J_r = Joint Roughness number

iv) J_a = Joint Alteration number

$\frac{J_{oi}}{J_{oa}}$ → give you idea about ϕ (friction)

v) J_{wi} = Joint water Reduction

vi) SRF = Stress Reduction Factor.

Range of Q = 0.001 to 1×10^3

7/4/13

Q system:-

Six Parameters

Table 4.6

(Remember)

ESR = Excavation Support Ratio

→ depending upon the importance of structure

Excavation
category

ESR

(vi)

Temporary

A. Temporary Mine Opening 3-5

B. Vertical shaft

- Circular 2.5

- Rectangle 2.0

C. Permanent Mine opening → 1-6

D. Storage Rooms,

Water Treatment Plant → 1-3

Minor Highway

E. Power Station → 1.0

Major Highway

F. Nuclear Power Station → 0-8

Support Design using Q System

See Figure 4.3

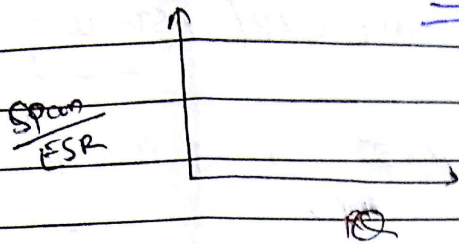


Figure 4.3

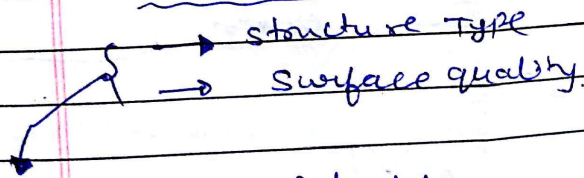
$$\text{Corrected span} = \frac{\text{Span or Height in m}}{\text{ESR}}$$

* We make the Rock self supporting by providing Bolts and Reinforcement

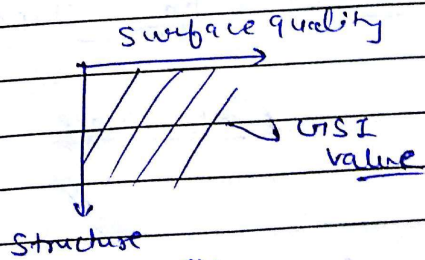
* These Empirical design is oversafe $\left\{ \begin{array}{l} \text{RMR} \\ \text{RSR} \\ \text{Q} \end{array} \right.$

(vi) Geological Strength Index (GSI) (Latest classification system) for blocky jointed Rock mass (1996) (i)

2 important Parameters (very simple)



from graph we calculate GSI value.



This GSI value is used for Hoek Brown Criterion (Jointed Rock mass)

$$\sigma_1' = \sigma_3' + \sigma_c \left[m_j \left(\frac{\sigma_3}{\sigma_{c_j}} \right) + S_j \right]^a$$

stress are effective stress

for intact rock $\rightarrow a = 0.5, S_j = S_i = 1$

$$\sigma_1' = \sigma_3' = \sigma_c$$

(UPCP)

For jointed Rock mass

$$m_j = m_i \exp\left(\frac{GSI-100}{28}\right)$$

$$GSI > 25, S_j = \exp\left(\frac{GSI-100}{9}\right); \alpha = 0.5$$

$$GSI < 25, S_j = 0, \alpha = 0.65 - \frac{GSI}{200}$$

example

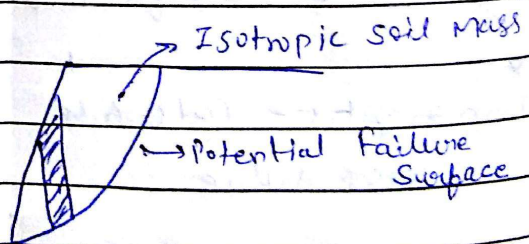
- fair surface quality
- structure looks ~~linear~~ blocky

	<u>Structure</u>	<u>surface</u>	<u>GSI</u>	<u>Approx GSI</u>
1.	Blocky	fair	45-65 ✓	~50
2.	Disintegrated	Poor	15-25	~40
3.	Sheared	Very poor	0-15	~20
4.	Very Blocky	Good	45-65	~50
5.	Intact & massive	Good	70-80	~35
6.	Blocky/disturbed	Poor	25-35	~30

11-4-17

Rock slopes :-

Soil slope



Rock slope → Joints → Anisotropic

- Planar failure
- Wedge failure
- Toppling failure

(heavily fractured rock mass)

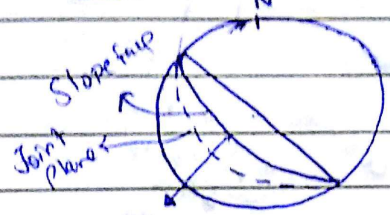
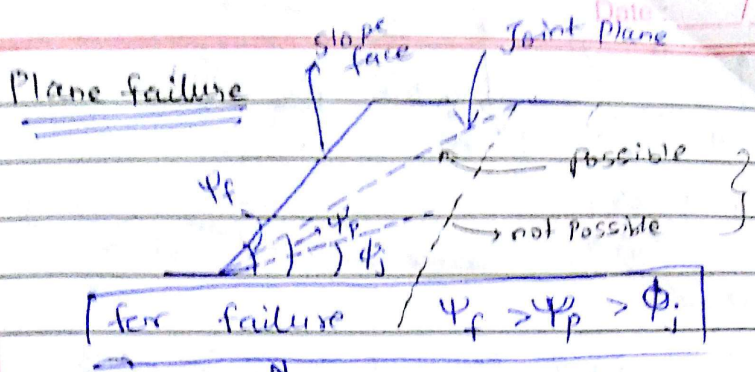
- Circular failure

(Behave as soil)

UPCP

- * If there is 2-3 sets of joints set then rock mass will behave as isotropic
- * But in case of circular failure (large no. of joint set) - anisotropic

Plane failure



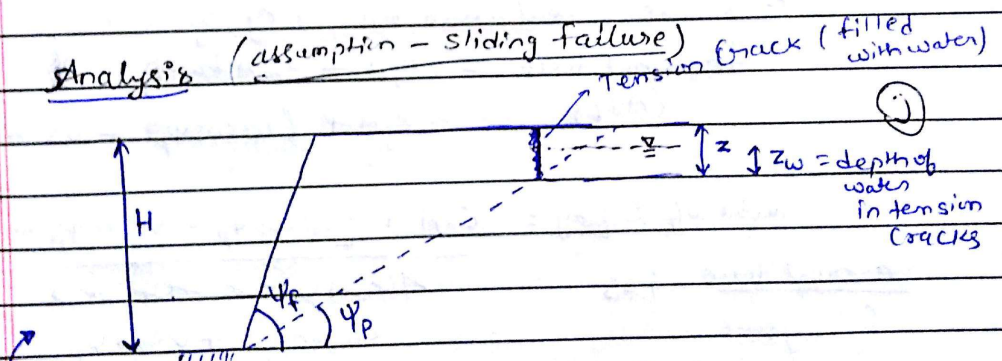
Conditions

- 1) Dip direction of Joint Plane } failure is possible
 dip direction of Slope face } should be in same direction
 difference = $(\pm 20^\circ)$

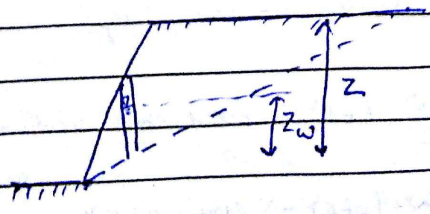
- 2) failure plane "daylights" in slope face ($\psi_f > \psi_p$)
- 3) $\psi_p > \phi_j$

Analysis

(assumption - sliding failure)



- Case 1) Tension cracks in upper surface
- Case 2) Tension cracks in slope face



(UPCP)

$a = 0.5$

st

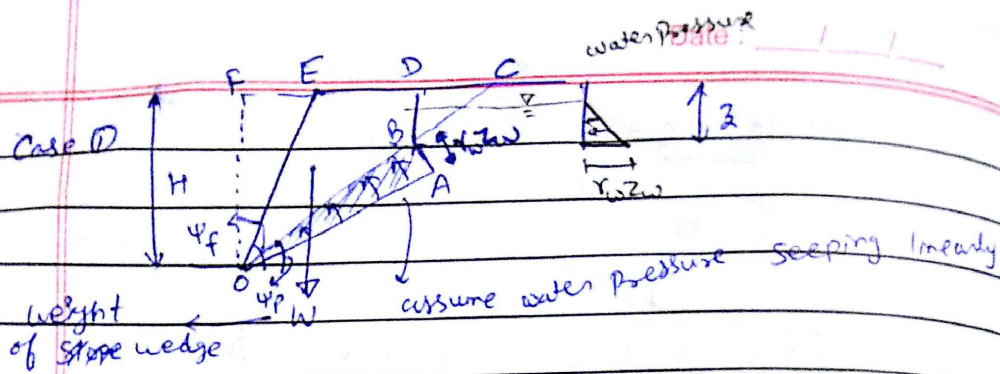
5

mass

acc

e as soil

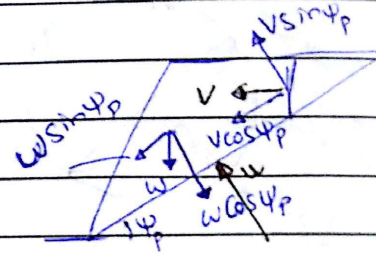
isotropic joint set anisotropic



act(OAB) = force per unit length on OB.

Sliding force

$$\Rightarrow W \sin \psi_p + V \cos \psi_p$$



normal force on potential failure surface

$$\Rightarrow W \cos \psi_p - V \sin \psi_p - u$$

Shear strength of Joint Plane :- $\left\{ \begin{array}{l} \text{Coulomb's} \\ \text{Rankin} \\ \text{Baker} \\ \text{LA} \end{array} \right.$

lets use Coulomb's eqn $\Rightarrow c_j, \phi_j$

$$\text{frictional force} = c_j A + \sigma_n A \tan \phi_j$$

$$(A \tau_f) = c_j A + [W \cos \psi_p - V \sin \psi_p - u] \tan \phi_j$$

$$\text{factor of safety} = \frac{c_j A + [W \cos \psi_p - V \sin \psi_p - u] \tan \phi_j}{W \cos \psi_p - V \sin \psi_p - u}$$

$$\frac{\text{frictional force}}{\text{Sliding force}} = \text{FOS}$$

Sliding force

$$W \cos \psi_p - V \sin \psi_p - u$$

$$W \sin \psi_p + V \cos \psi_p$$

$$\text{FOS} = \frac{c_j A + [W \cos \psi_p - V \sin \psi_p - u] \tan \phi_j}{W \sin \psi_p + V \cos \psi_p}$$

against sliding

for case (D) let's find the values

$$\text{act(OEC)} = \text{act(OFC)} - \text{act(OFE)}$$

$$= \frac{1}{2} \cdot H \cdot \frac{H}{\tan \psi_p} - \frac{1}{2} H \cdot \frac{H}{\tan \psi_p}$$

$$A_1(OEC) = \frac{H^2}{2} [\cot \psi_p - \cot \psi_f]$$

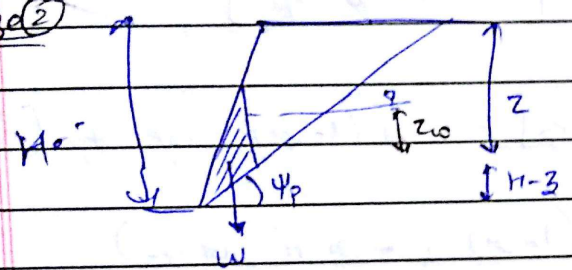
$$A_2(BDC) = \frac{z^2}{2} \cot \psi_p$$

So, Area of wedge [OEDA]

$$= \frac{H^2}{2} [\cot \psi_p - \cot \psi_f] - \frac{z^2}{2} \cot \psi_p$$

$$\text{So, weight of wedge (W)} = \frac{1}{2} \gamma H^2 \left(1 - \frac{z^2}{h^2}\right) (\cot \psi_p - \cot \psi_f)$$

For Case (2)



we can calculate weight of shaded part

Tension crack:- (depth of tension cracks)

- i) either from field
- ii) Theoretical

Critical Tension Crack (for dry condition)

for dry condition FOS ($u=0, v=0$)

$$FOS = \frac{c_j A + (W \cos \psi_p) \tan \psi_j}{W \sin \psi_p}$$

$$F = \frac{c_j A}{W \sin \psi_p} + \cot \psi_p \tan \psi_j \quad \text{(for dry condition)}$$

for critical tension crack :- say $\frac{z}{H} = \alpha$

$$F = f(A, W)$$

for minimum tension crack FOS.

$$\frac{\partial F}{\partial W} = 0 \Rightarrow \frac{c_j \frac{\partial A}{\partial W} - \sin \psi_p \frac{\partial W}{\partial W}}{(W \sin \psi_p)^2} = 0$$

(UPCP)

$$\frac{\partial F}{\partial W} = \frac{c_j}{\sin \psi_p} \left[W \frac{\partial A}{\partial W} - A \frac{\partial W}{\partial W} \right] = 0$$

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$$W \frac{\partial A}{\partial x} = A \frac{\partial W}{\partial x}$$

Homework

$$W = \frac{1}{2} \gamma H^2 [(1 - \alpha^2) \cot \psi_p - \cot \psi_p]$$

[Derive
and get
 $\frac{\partial W}{\partial x}$]

$$A = (H - z) \operatorname{cosec} \psi_p$$

$$A = H [1 - \alpha^2] \operatorname{cosec} \psi_p$$

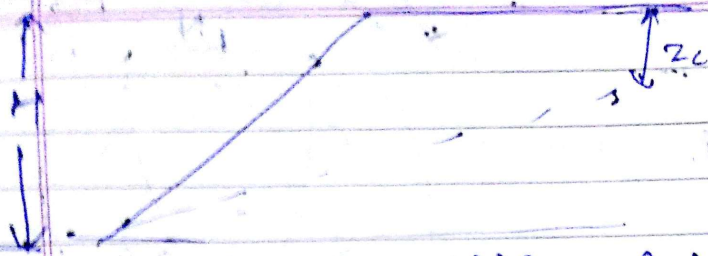
$$\frac{\partial A}{\partial x} = H (-\operatorname{cosec} \psi_p)$$

$$\frac{\partial W}{\partial x} = \frac{1}{2} \gamma H^2 [-2 \alpha \cot \psi_p]$$

$$\Rightarrow W H [-\operatorname{cosec} \psi_p] = H (1 - \alpha^2) \operatorname{cosec} \psi_p \left[\frac{1}{2} \gamma H^2 (-2 \alpha) \cot \psi_p \right]$$

$$= -W = (1 - \alpha^2) (-\alpha^2 H^2 \cot \psi_p)$$

13/07/17



$$W \frac{\partial A}{\partial \alpha} = A \frac{\partial W}{\partial \alpha}$$

$$\alpha = \frac{z_c}{H}$$

$$W = \frac{1}{2} \gamma H^2 \left[(1 - \alpha^2) \cot \psi_p - \cot \psi_f \right]$$

$$A = H(1 - \alpha) \cos \psi_p, \quad \frac{\partial A}{\partial \alpha} = H(-1) \cos \psi_p$$

$$\frac{\partial W}{\partial \alpha} = \frac{1}{2} \gamma H^2 \{-2\alpha\} \cot \psi_p$$

$$\alpha^2 - 2\alpha + 1 - \cot \psi_f \cdot \tan \psi_p = 0$$

$$\alpha = \frac{2 \pm \sqrt{4 - 4 \times 1 \times (1 - \cot \psi_f \tan \psi_p)}}{2}$$

$$\alpha = 1 - \sqrt{\cot \psi_f \tan \psi_p}$$

$$\frac{z_c}{H} = 1 - \sqrt{\cot \psi_f \tan \psi_p}$$

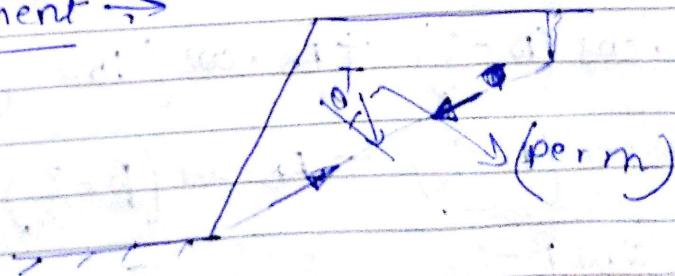
Position of tension crack \rightarrow

$$b = H \cdot \cot \psi_p - H \cot \psi_f - z_c \cdot \cot \psi_p$$

$$\frac{b}{H} = \left(1 - \frac{z_c}{H}\right) \cot \psi_p - \cot \psi_f$$

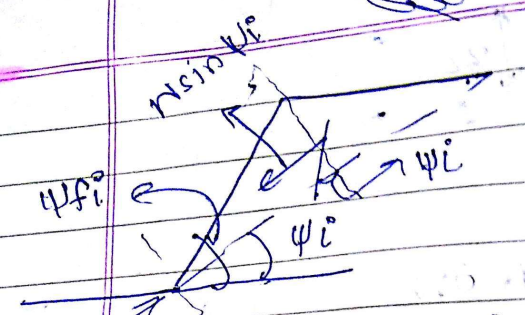
$$\frac{b_c}{H} = \sqrt{\cot \psi_f \tan \psi_p} - \cot \psi_f$$

Reinforcement \rightarrow

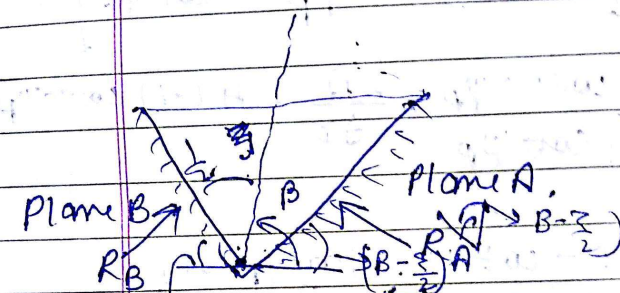
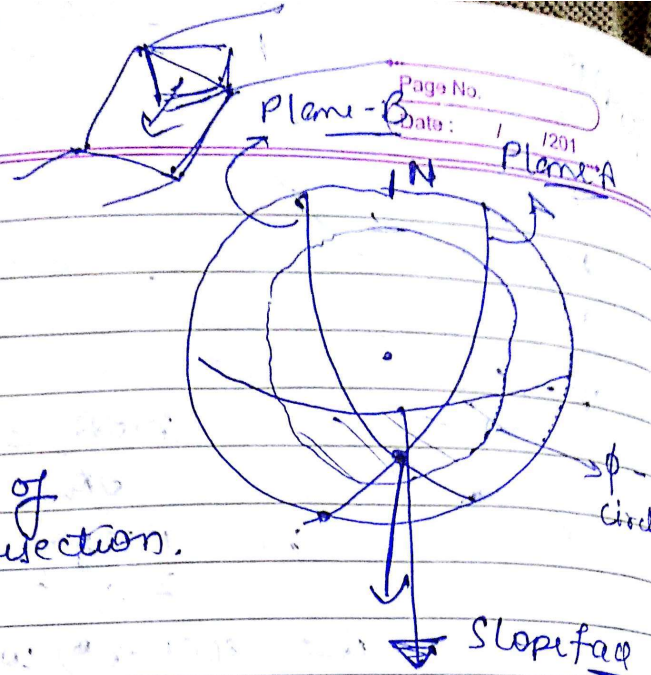


$$FOS = \frac{c_f \Delta + (W \cos \psi_p - U \sin \psi_p + T \cos \alpha \tan \phi_f)}{W \sin \psi_p + U \cos \psi_p - T \sin \alpha}$$

Nedge failure →



$\psi_i =$ Plunge of line of intersection.



Plane A → flatter
Plane B → steeper

$$FOS = \frac{(C_A \cdot A_A + R_A \tan \phi_A) + (C_B \cdot A_B + R_B \tan \phi_B)}{N \sin \psi_i}$$

$180 - (\beta + \frac{\epsilon}{2})$

$C_A = 0, C_B = 0, \phi_A = \phi_B = \phi$

$$FOS = \frac{(R_A + R_B) \tan \phi}{N \cdot \sin \psi_i}$$

To find R_A, R_B →

$$R_A \cdot \sin(\beta - \frac{\epsilon}{2}) = R_B \cdot \sin\{180 - (\beta + \frac{\epsilon}{2})\}$$

$$R_A \cdot \sin(\beta - \frac{\epsilon}{2}) = R_B \cdot \sin(\beta + \frac{\epsilon}{2}) \quad \text{--- (1)}$$

$$R_A \cdot \cos(\beta - \frac{\epsilon}{2}) + R_B \cdot \cos\{180 - (\beta + \frac{\epsilon}{2})\} = W \cos \psi_i$$

$$R_A \cdot \cos(\beta - \frac{\epsilon}{2}) - R_B \cdot \cos(\beta + \frac{\epsilon}{2}) = W \cos \psi_i$$

$$R_B [\sin(\beta + \frac{\epsilon}{2}) \cot(\beta - \frac{\epsilon}{2}) - \cos(\beta + \frac{\epsilon}{2})] = W \cdot \cos \psi_i$$

SHOW

$$R_B = \frac{W \cos \psi_i \cdot \sin(\beta - \frac{\xi}{2})}{\sin \xi}$$

$$FOS = \frac{\sin \beta \cdot \tan \phi}{\sin \frac{\xi}{2} \tan \psi_i}$$

20-02

1/1