GEOTECHNICAL ENGINEERING FORMULAS A handy reference for use in geotechnical analysis and design

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1. SOIL CLASSIFICATION

1.1 USCS: Unified Soil Classification System

Coarse Grained soils have less than 50% passing the # 200 sieve:

Symbol	Passing the #200	$\frac{D}{D}_{30}$ Cu=	$Cc = \frac{\left(D_{30}\right)^2}{D_{10} \times D_{60}}$	Soil Description
GW	< 5%	4 or higher	1 to 3	Well graded gravel
GP	< 5%	Less than 4	1 to 3	Poorly graded gravel
GW-GM	5 to12%	4 or higher	1 to 3 but with <15% sand	Well graded gravel with silt
GW-GM	5 to12%	4 or higher	1 to 3 but with ≥15% sand	Well graded gravel with silt and sand
GW-GC	5 to12%	4 or higher	1 to 3 but with <15% sand	Well graded gravel with clay or silty clay
GW-GC	5 to12%	4 or higher	1 to 3 but with ≥15% sand	Well graded gravel with clay and sand
GC	>12%	N/A	N/A,<15%sand	Clayey Gravel
GC	> 12%	N/A	N/A,>15%sand	Clayey Gravel with sand
GM-GC	>12%	N/A	N/A,<15%sand	Clayey Silt with gravel
GM-GC	>12%	N/A	N/A,≥15%sand	Clayey Silt with sand
SW	< 5%	6 or higher	1 to 3	Well graded sand
SP	< 5%	Less than 6	1 to 3	Poorly graded sand
SM	>12%	N/A	N/A	Silty Sand or Sandy Silt
SC	>12%	N/A	N/A	Clayey Sand or Sandy Clay
SC-SM	>12%	N/A	N/A	Silty Clay with Sand

Where:

Cu = Uniformity Coefficient; gives the range of grain sizes in a given sample. Higher Cu means well graded.

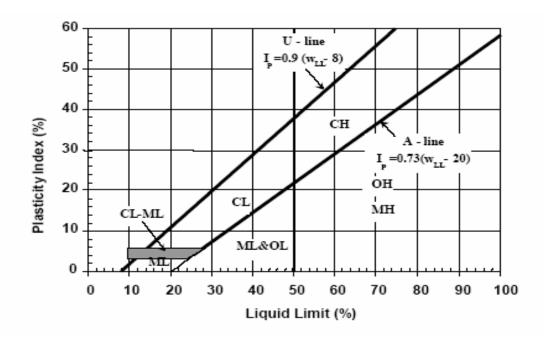
Cz = Coefficient of Curvature is a measure of the smoothness of the gradation curve. Usually less than 3.

D10, D3, & D60 are the grain size diameter corresponding to 10%, 30% and 60% passing screen.

1.1.1 Relative Density of Cohesionless Soils:

SPT or N value	Relative Density	% Relative Density
0 – 3	Very loose	0 – 15
4 – 10	Loose	15 – 35
11 – 30	Medium dense	35 – 65
31 – 50	Dense	65 -85
> 50	Very dense	85 - 100

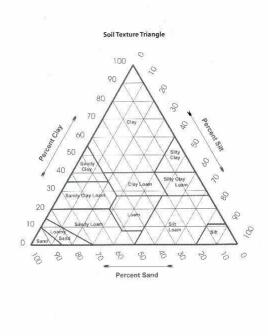
1.1.2 Fine Grained(Cohesive) Soil Charts using the USCS System:



1.1.3 Consistency of Fine Grained Soils:

SPT or N value	Cohesion, C or Su	Consistency
< 2	< 500 psf	Very soft
2 – 4	500 – 1000 psf	Soft
5 – 8	1000 – 2000 psf	Firm
9 – 15	2000 – 4000 psf	Stiff
16-30	4000 – 8000 psf	Very stiff
>30	> 8000 psf	Hard

1.2 USDA Soil Classification System

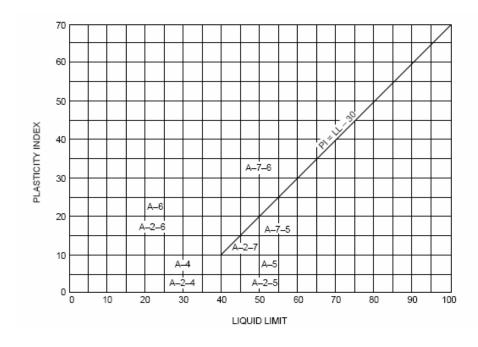


The percent SAND, SILT, and CLAY lines are drawn and their intersection gives the soil classification.

1.3 AASHTO Soil Classification System:

CLASSIFICATION OF HIGHWAY SUBGRADE MATERIALS (With suggested subgroups)											
General Classification		Granular Materials (35% or less passing No. 200) Silt-Clay Materials (More than 35% passing #200)									
Crown Classification	А	-1	A-3		А	-2		A-4	A-5	A-6	A-7
Group Classification	A-1-a	A-1-b	A-3	A-2-4	A-2-5	A-2-6	A-2-7	A-4	A-5	А-ь	A-7-5 A-7-6
Sieve Analysis, Percent Passing:											
No. 10 No. 40 No. 200	0-50 0-30 0-15	0-50 0-25	51-100 0-10	0-35	0-35	0-35	0-35	36-100	36-100	36-100	36-100
Characteristics of fraction passing #40:											
Liquid Limit Plasticity Index	0	-6	N.P.	0-40 0-10	41+ 0-10	0-40 11+	41+ 11+	0-40 0-10	41+ 0-10	0-40 11+	41+ 11+
Group Index		0	0		0		-4	0-8	0-12	0-16	0-20
Usual Types of Significicant Constituent Materials	Stone Fr Gravel a	agments, and Sand	Fine Sand Silty or Clayey C		Gravel and	Sand		Soils		y Soils	
General Rating as Subgrade	Excellent to Good					Fair to	o Poor				

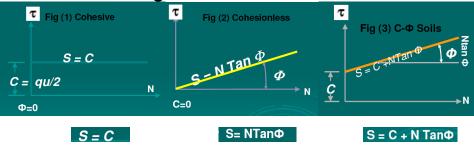
Cohesive soils classification in AASHTO System:



2. PHASE RELATIONSHIP EQUATIONS:

		•
Dry Unit	Bulk or Wet or Total	Saturated Unit
Weight, γd	Unit Weight, γm or	Weight, γs or γsat
	γw or γt or γ	
ν – γ	$(1+w)G_s\gamma_w$	$\gamma_{\text{sat}} = \frac{(G_{\text{s}} + e)\gamma_{\text{w}}}{1 + e}$
$\frac{\gamma_d}{1+w}$	1 + e	$\frac{1}{1+e}$
$\gamma_d = \frac{G_s \gamma_w}{1 + e}$	$\frac{(G_s + Se)\gamma_w}{1 + e}$	$\gamma_{\text{sat}} = \left(\frac{e}{w}\right) \left(\frac{1+w}{1+e}\right) \gamma_{w}$
$\gamma_d = \frac{eS\gamma_w}{(1+e)w}$	$\frac{(1+w)G_{s}\gamma_{w}}{1+\frac{wG_{s}}{S}}$	$\gamma_{\rm sat} = [(1-n)G_{\rm s} + n]\gamma_{\rm w}$
$\gamma_d = G_s \gamma_w (1 - n)$	$G_s \gamma_w (1-n)(1+w)$	$\gamma_{\text{sat}} = \gamma_d + \left(\frac{e}{1+e}\right) \gamma_w$

2.1 Shear Strength of Soils



2.2 Bearing Capacity of Soils

Hansen B.C. Factors:

Ø	Ne	Nq	Νγ
0	5.10	1.00	0.00
4	6.19	1.43	0.05
8	7.53	2.06	0.22
12	9.28	2.97	0.63
16	11.63	4.34	1.43
20	14.83	6.40	2.95
24	19.32	9.60	5.75
26			7.94
28	25.80	14.72	10.94
30	30.14	18.40	15.07
32	35.49	23.18	20.79
34	42.16	29.44	28.77
36	50.59	37.75	40.05
38	61.35	48.93	56.18
40	75.32	64.20	79.54

φ′	N^{d}	N_c	N_{γ}
0	1.00	5.70	0.0
2	1.22	6.30	0.2
4	1.49	6.97	0.4
6	1.81	7.73	0.6
8	2.21	8.60	0.9
10	2.69	9.60	1.2
12	3.29	10.76	1.7
14	4.02	12.11	2.3
16	4.92	13.68	3.0
18	6.04	15.52	3.9
20	7.44	17.69	4.9
22	9.19	20.27	5.8
24	11.40	23.36	7.8
26	14.21	27.09	11.7
28	17.81	31.61	15.7
30	22.46	37.16	19.7
32	28.52	44.04	27.9
34	36.50	52.64	36.0
35	41.44	57.75	42.4
36	47.16	63.53	52.0
38	61.55	77.50	80.0
40	81.27	95.66	100.4
42	108.75	119.67	180.0
44	147.74	151.95	257.0
45	173.29	172.29	297.5

Allowable Gross Bearing Capacity- 9 Equations

In Cohesionless(granular) Soils

qu = $\gamma D(Nq) + 0.6\gamma R(N \gamma)$ --for circular footings

qu = γ D(Nq) + 0.4 γ B(N γ)-- for square or rectangular footings

qu = γ D(Nq) + 0.5 γ B(N γ)—for continuous footings

In Cohesive (clayey) Soils

qu = 1.3 C(Nc) + γ D--for circular footings

qu = CNc (1+ (0.3B/L)) + γ D-- for square or rectangular footings

qu = $CNc + \gamma D$ —for continuous footings

In Mixed soils (C-Φ)

qu = 1.3C(Nc) + γ D(Nq) + 0.6 γ R(N γ)--for circular footings

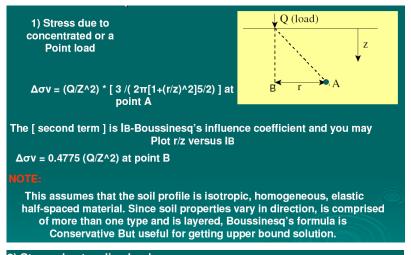
qu = CNc (1+ (0.3B/L)) + γ D(Nq) + 0.4 γ B(N γ) - for sq/rect. footings

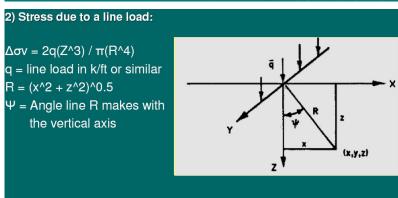
qu = $CNc + \gamma DNq + 0.5 \gamma B(N \gamma)$ -—for continuous footings

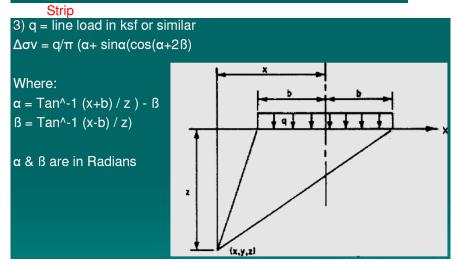
Cohesion Surcharge Friction 1st term 2nd term 3rd term Note:If Df/B > 1, terzaghi's B.C. factors do not apply. Use Hansen's B.C. factors. For example, if depth of footing (Df) is 3 ft but footing width (B) is 2.75 ft.

3. STRESSES IN SOILS

3.1 Various Loading Conditions:







4. SHALLOW FOUNDATIONS

Df

Q

Layer 1

Layer 2

Layer 3

g all

4.1 Conventional Footings

4.11Geotechnical Analysis

gall = Q / Bx1 for Continuous Footings

qall = Q / BxL for Rectangular Footings

gall = Q / BxB for Square Footings

qall < qu / 3 from Bearing Capacity Calculations

e < B/6, where e=eccentricity

Df > 1.0 ft minimum

Df > frost depth

Df > setback distance for footings on slope

Df > scour depth

Df > high moisture variations depth(expansive soils)



Given: A Continuous footing with $\gamma m = 100$ pcf, Df = 5 ft, qall = 4,000 psf, D.L=22 k/ft, L.L.=12 k/ft, f'c=3 ksi, fy= 60 ksi. Design the footings using the ACI code:

Weight of footing =
$$(19)(150)/12 = 237.5 \text{ lb./ft.}^2$$

Weight of soil = $(5-19/12)(100) = 341.7 \text{ lb/ft.}^2$
 $q_e = 4000-238-342 = 3420 \text{ lb./ft.}^2 = 3.42 \text{ k/ft.}^2$

, use 10 ft. footing.

3.) Net upward pressure =
$$P_u$$
 / Area

$$P_u = 1.2D + 1.6L = (1.2)(22) + (1.6)(12) = 45.6 \text{ k/ft.}$$

 $q_u = 45.6 / (10 \cdot 1) = 4.56 \text{ k/ft.}$

4.) Check one-way shear:
$$d = 19 - 3.5 = 15.5$$
 in.

$$V_u = q_u \left(\frac{L}{2} - \frac{a}{2} - d\right) = 4.56 \left(\frac{10}{2} - \frac{12}{2 \cdot 12} - \frac{15.5}{12}\right) = 14.63 \text{ kip}$$

$$d = \frac{V_u}{\phi 2\sqrt{f_c'b}} = \frac{14.63 \cdot 1000}{0.75 \cdot 2\sqrt{3000} \cdot 12} = 14.8 \text{ in.} < 15.5 \text{ in.}$$

use actual d = 15.5 in.

5.) Calculate B.M. and A_s :

$$M_u = \frac{q_u}{2} \left(\frac{L}{2} - \frac{a}{2}\right)^2 = \frac{4.56}{2} \left(\frac{10}{2} - \frac{12}{2 \cdot 12}\right)^2 = 46.17 \text{ k.ft.}$$

, Assume a = 1.5 in.

$$A_{s} = \frac{M_{u}}{\phi f_{y} \left(d - \frac{a}{2} \right)} = \frac{46.17 \cdot 12}{0.9 \cdot 60 \left(15.5 - \frac{1.5}{2} \right)} = 0.7 \text{ in.}^{2}$$

$$a = \frac{A_s f_y}{0.85 f_c b} = \frac{0.7 \cdot 60}{0.85 \cdot 3 \cdot 12} = 1.364 \text{ in., o.k.}$$

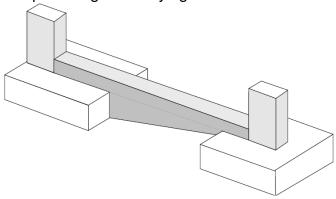
Chec

$$A_{s(min)} = 0.0018bh = (0.0018)(12)(19) = 0.41 \text{ in.}^2 < 0.70 \text{ in.}^2$$

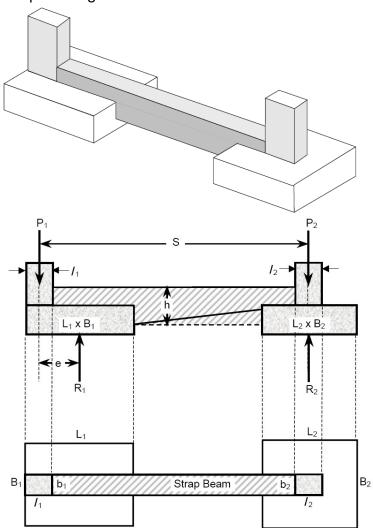
Use #7 bars @ 9 in. $(A_s = 0.80 \text{ in.}^2)$

- 6.) l_{dav} = available development length = $(10 \cdot 12/2) (12/2) 3 = 51$ in. Required $l_d = 48$ in. < 51 in. $(l_d \text{ from Table 7.2, chapter 7})$
- 7.) Longitudinal reinforcement = $A_{s(min)}$ = 0.41 in.², use #5 bars @ 9 in. (A_s = 0.41 in.²)

4.2 Strap or Cantilever Footings: Strap Footing with varying beam thickness



Strap Footings with constant beam thickness



DIMENSION FOOTINGS (Determine L₁, B₁, L₂ and B²)

Allowable load $P = P_1 + P_2$

Ultimate load $P_u = [1.4DL_1 + 1.7LL_1] + [1.4DL_2 + 1.7LL_2]$

Ultimate ratio $r_u = \frac{P_u}{P}$, Ultimate applied pressure $q_u = q_a \times r_u$

 $\Sigma M_{\text{col. 2}} = 0$

$$R_1 (S - e) - P_{u1} S = 0$$
(1)

$$\Sigma M_{R1} = 0$$

$$P_{u2}(S-e) - R_2(S-e) - P_{u1}e = 0$$
 (2)

 $\Sigma F = 0$

$$P_{u1} + P_{u2} - R_1 - R_2 = 0$$
 (3)

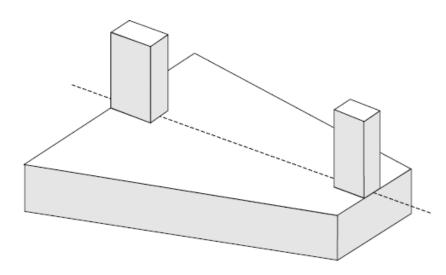
 $q_u = r_u q_a$

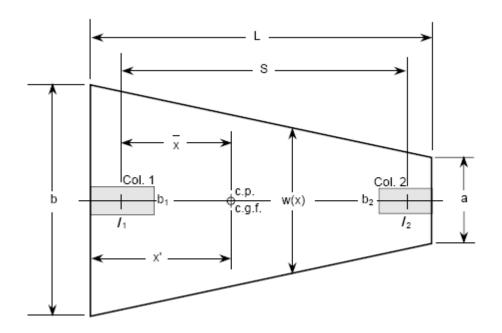
Footing 1:
$$L_1 = 2 \times \left(e^{-t} + \frac{I_1}{2}\right)$$
 and $B_1 = \frac{R_1}{q_{ij} L_1}$

Footing 2: let $k_2 = \frac{L_2}{B_2}$ (1.0 means footing 2 is square)

$$B_2 = \sqrt{\frac{R_2}{k_2 q_u}}$$
 and $L_2 = k_2 B_2$

4.3 Trapezoidal Footings:





Allowable load $P = P_1 + P_2$

Ultimate load $P_u = [1.4DL_1 + 1.7LL_1] + [1.4DL_2 + 1.7LL_2]$

 $\mbox{Ultimate ratio} \qquad \mbox{r_u} = \frac{\mbox{P_u}}{\mbox{P}} \ \ \, , \ \, \mbox{Ultimate applied pressure q_u} = \mbox{q_a} \ \, x \ \, r_u$

$$\Sigma M_{co1.1} = 0$$
 $\overline{x} = \frac{P_{u2} \times s}{P_u}$ and $x' = \overline{x} + \frac{I_1}{2}$

For a trapezoidal solution, $\frac{L}{3} < x' < \frac{L}{2}$

For a trapezoidal solution, $\frac{L}{3}$ < x' < $\frac{L}{2}$

From trapezoidal geometry,

$$A = \frac{a+b}{2} \quad L \quad \text{where A = Area} = \frac{P_u}{q_u}$$

$$d \quad x' = \frac{L}{3} \left(\frac{2a+b}{a+b} \right)$$

From these equations, we solve for a and b

SOIL CONSOLIDATION EQUATIONS 5.

Instant Settlement of footings:

In Continuous footings: $\Delta H = 8q(B^2) / Kv(B+1)^2$ In Square footings: $\Delta H = 4q(B^2) / Kv(B+1)^2$

Where:

Kv = modulus of subgrade reaction in Tons per cubic foot(Ton/ft^3)

B = footing width in feet, B is less than 20'

q = applied stress at base of footings in Tons per square foot

Kv = 50 - 80 loose cohesionless soils

Kv = 80 − 150 in medium dense soils—most common value in design

Kv= 150 - 230 in Dense soils &

Kv = 230 - 300 in very dense soils

 S_i = immediate settlement of a point on the surface

 C_s = shape and rigidity factor

q = equivalent uniform stress on the footing (total load/footing area)

B =characteristic dimension of the footing

 $S_i = C_s q B \left(\frac{1 - v^2}{E_u} \right)$ v = Poisson's ratio

 E_u = undrained elastic modulus (Young's modulus)

Primary Consolidation: 5.2

 $S=(Cc/1+eo)H \times Log(\sigma o + \Delta q)/\sigma o$

$$s_c = \frac{C_c H}{1 + e_o} \log \left(\frac{\sigma_o^2 + \Delta q}{\sigma_o^2} \right)$$

5.3 Overconsolidated Soils

Settlement of Overconsolidated soils Case 1:($\sigma f < Pc$)

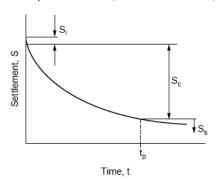
ΔH = (Cr/1+eo)H x Log Pc/σο

$$s_c = \frac{C_r H}{1 + e_o} \log \left(\frac{\sigma_o^2 + \Delta q}{\sigma_o^2} \right)$$

Settlement of Overconsolidated soils Case 2:(of > Pc) $\Delta H = (Cr/1+eo)H \times Log(Pc/\sigma o) + (Cc/1+eo)H \times Log(\sigma f/\sigma o)$

$$s_c = \left(\frac{H}{1 + e_o}\right) \left(C_r \log \left(\frac{P_c}{\sigma_o^2}\right) + C_c \log \left(\frac{\sigma_o^2 + \Delta q}{P_c}\right) \right)$$

5.4 Time rate of settlement (i=immediate, c=consolidation, & s=secondary)



5.41 Coefficient of consolidation, Cv:

k = hydraulic conductivity

$$c_v = \frac{k(1+e_o)}{\gamma_w a_v} \qquad \begin{array}{l} \gamma_w = \text{unit weight of water} \\ e_o = \text{initial void ratio} \\ a_v = - \frac{de}{d\sigma_v'} = \text{coefficient of compressibility} \end{array}$$

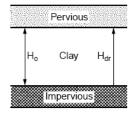
$$Z = \frac{z}{H_{dr}}$$

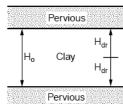
$$T = \frac{c_v t}{H_d^2}$$

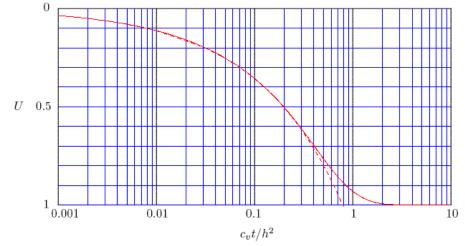
z= depth below top of the compressible stratum $H_{dr}=$ length of the longest pore water drainage path

Single-Drained Layer

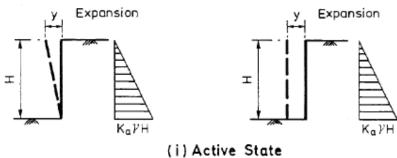
Double-Drained Layer

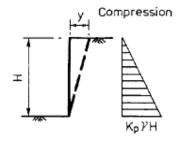


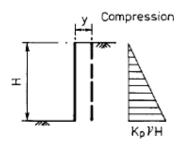




RETAINING STRUCTURES: 6.

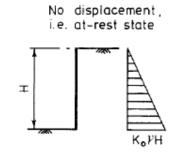


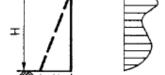




(ii) Passive State

(a) Rigid Retaining Wall Free to Translate or Rotate about Its Base



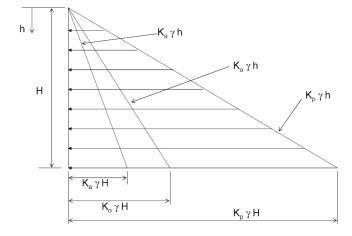


Expansion with soil arching

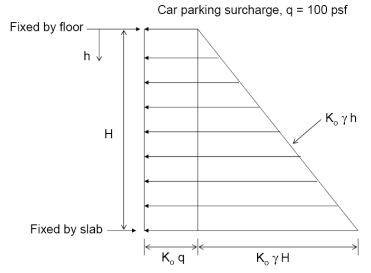
(b) Restrained Rigid Wall

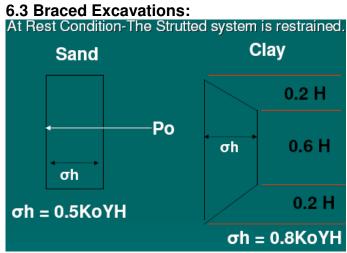
(c) Top of Wall Restrained

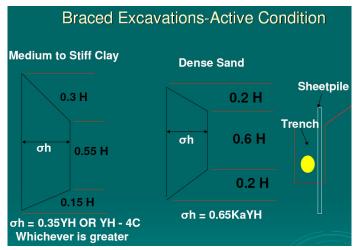
6.1 Horizontal Stresses: Active, At Rest and Passive



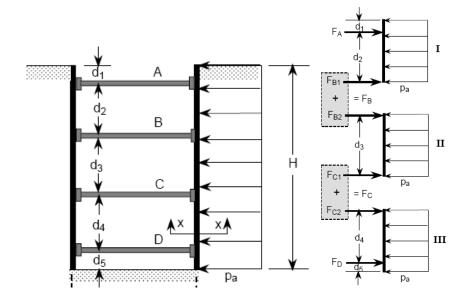
6.2 Basement Wall with surcharge:







6.4 Forces on Struts:



Note that the first strut A must be placed at a depth $d_1 < z_c$ (depth of tension crack) where $z_c = \frac{2~c}{\gamma}$.

Forces on Struts and Selection of Section

(Designed as column, pined at both ends)

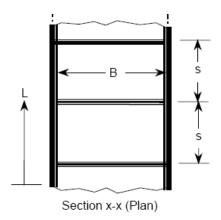
- Draw the pressure diagram p_a
- 2) Assume that the sheet pile is hinged at all levels of struts
- 3) Calculate F_A , F_{B1} , F_{B2} , F_{C1} , F_{C2} , and F_D which are the reaction in the load distributions I, II and III.
- 4) The loads in the struts are calculated as:

$$P_A = (F_A) \times S$$

 $P_B = (F_{B1} + F_{B2}) \times S$
 $P_C = (F_{C1} + F_{C2}) \times S$
 $P_D = (F_D) \times S$

Maximu Moment on Sheet Pile and selection of Section

- For each of the load distributions I, II and III find M_{max} i.e. where the shear is equal to zero.
- 2) The design moment for the sheet pile is the maximum of step (1)
- 3) Calculate the section modulus $S = \frac{M_{max}}{\sigma_{all}}$ where σ_{all} = allowable stress for sheet pile
- 4) Select the sheet pile section based on S in Step 3



Maximu Moment on Wales and selection of Section

(Designed as beams pined at the struts)

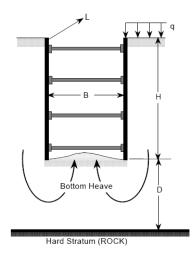
At level A:
$$M_{A, max} = \frac{F_A \left(s^2\right)}{8}$$

At level C:
$$M_{C, max} = \frac{(F_{C1} + F_{C2})(s^2)}{8}$$

At level B:
$$M_{B, max} = \frac{(F_{B1} + F_{B2}) (s^2)}{8}$$

At level D:
$$M_{D, max} = \frac{F_D(s^2)}{8}$$

Bottom Heave Calculations:



Safety Factor against bottom heave $SF_H \ge 1.5$

$$SF_{H} = \frac{1}{H} \left(\frac{5.7 \text{ c}}{\gamma - \frac{\text{c}}{0.7B}} \right)$$

If D \leq 0.7 B

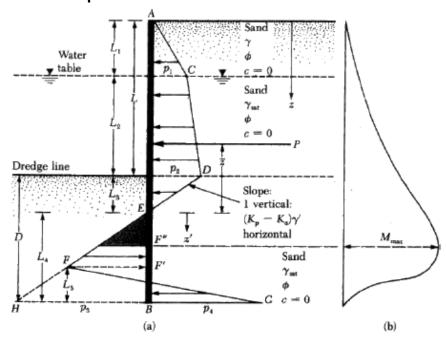
$$SF_{H} = \frac{1}{H} \left(\frac{5.7 \text{ c}}{\gamma - \frac{\text{c}}{0.7B}} \right)$$

OR

$$SF_{H} = \frac{c N_{c}}{\gamma H + q}$$

Whichever is larger

6.5 Cantilever Sheetpiles in Sand



(a) Sheet Pile in Sand with dimensions and pressure distribution(b) Moment distribution over the sheet pile.

1. Calculate:
$$k_a = tan^2 \left(45 - \frac{\phi}{2} \right)$$
 and

$$k_p = \frac{1}{k_a}$$
 Note: Some designers use $k_{p(Design)} = \frac{k_p}{SF}$ where SF = 1.5 -2.0

2. Calculate:
$$p_1 = \gamma L_1 k_a$$
 and

$$p_2 = (\gamma L_1 + \gamma L_2) \times k_a$$
, $\gamma = \gamma_{SAT} - \gamma_w$

3. Calculate:
$$L_3 = \frac{p_2}{\left[\gamma \left(k_p - k_a\right)\right]}$$

4. Calculate: P = Area ACDE =
$$\frac{1}{2}p_1L_1 + p_1L_2 + \frac{1}{2}(p_2 - p_1)L_2 + \frac{1}{2}p_2L_3$$

5. Calculate: Z by taking moment about E of area ACDE = PZ

6. Calculate:
$$p_5 = (\gamma L_1 + \gamma L_2) \times k_p + \gamma L_3 (k_p - k_a)$$

7. Calculate: L4 by trial and error from the equation:

$$L_4^4 + A_1L_4^3 - A_2L_4^2 - A_3L_4 - A_4 = 0$$
,

where.

$$A_{1} = \frac{p_{5}}{\left[\gamma\left(k_{p} - k_{a}\right)\right]}$$

$$A_{2} = \frac{8P}{\left[\gamma\left(k_{p} - k_{a}\right)\right]}$$

$$A_{3} = 6P\left[\frac{\left[2\overline{z}\gamma\left(k_{p} - k_{a}\right) + p_{5}\right]}{\left[\gamma\left(k_{p} - k_{a}\right)\right]^{2}}\right]$$

$$A_{4} = P\left[\frac{\left(6\overline{z}p_{5} + 4P\right)}{\left[\gamma\left(k_{p} - k_{a}\right)\right]^{2}}\right]$$

8. Calculate:
$$p_4=p_5+\gamma \dot{L}_4\!\left(\!k_p-k_a\right)$$
 , $p_3=\gamma \dot{L}_4\!\left(\!k_p-k_a\right)$ and

$$L_{5} = \frac{\left(p_{3}L_{4} - 2P\right)}{\left(p_{3} + p_{4}\right)}$$

Draw the sheet pile (similar to page 1) with the estimated values in steps 1-8

9. Calculate: D = 1.3 to
$$1.6(L_3 - L_4)$$

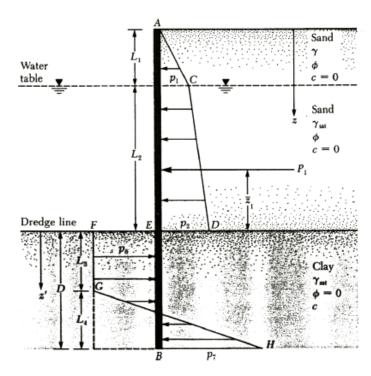
10. Calculate:
$$z' = \sqrt{\frac{2P}{\gamma (k_p - k_a)}}$$

11. Calculate:
$$M_{max} = P(\overline{z} + z') - \left[\frac{1}{2} \gamma' z'^2 (k_p - k_a) \right] \left(\frac{1}{3} \right) z'$$

12. Calculate:
$$S = \frac{M_{max}}{\sigma_{all}}$$

where S = Minimum section modulus of sheet pile σ_{all} = allowable stress for sheet pile

6.6 Cantilever Sheetpiles in Clay



Sheet Pile in Clay with dimensions and pressure distribution

Design Steps (refer to figure above for terms)

1. Calculate:
$$k_a = tan^2 \left(45 - \frac{\phi}{2} \right)$$

2. Calculate:
$$p_1 = \gamma L_1 k_a$$
 and

$$p_2 = (\gamma L_1 + \gamma L_2) \times k_a$$
 where $\gamma = \gamma_{SAT} - \gamma_w$

$$3. \quad \text{Calculate: } \ \frac{1}{2} p_1 L_1 + p_1 L_2 + \frac{1}{2} \big(p_2 - p_1 \big) \ L_2$$

4. Calculate:
$$\overline{Z}_1$$
 by taking moment about E of area ACDE = $\overline{P}_1\overline{Z}_1$

$$D^{2} \Big[4c - \Big(\gamma L_{1} + \gamma \dot{L}_{2} \Big) \Big] - 2DP_{1} - \frac{P_{1} \Big(P_{1} + 12c \overline{Z_{1}} \Big)}{\Big(\gamma L_{1} + \gamma \dot{L}_{2} \Big) + 2c}$$

7. Calculate:
$$p_6 = 4c - (\gamma L_1 + \gamma L_2)$$
 and $p_7 = 4c + (\gamma L_1 + \gamma L_2)$

8. Calculate:
$$L_4 = \frac{(Dp_6 - P_1)}{4c}$$
 and

Draw the sheet pile (similar to page 1) with the estimated values in steps 1-8

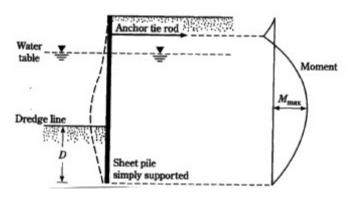
9. Calculate:
$$z' = \frac{P_1}{p_6}$$

10. Calculate:
$$M_{max} = P_1(\overline{z_1} + z') - \frac{p_8 z'^2}{2}$$

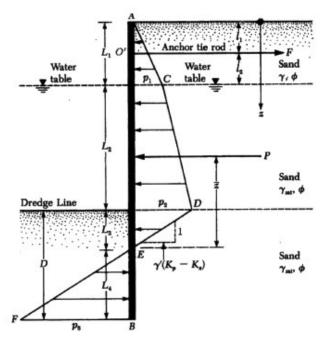
11. Calculate:
$$S = \frac{M_{max}}{\sigma_{all}}$$

where S = Minimum section modulus of sheet pile σ_{all} = allowable stress for sheet pile

6.6 Anchored Sheetpiles in Sand (Also called Bulkheads)



Deformation and moment distribution over the sheet pile.



Sheet Pile in Sand with dimensions and pressure distribution Design Steps (refer to figure above for terms)

1. Calculate:
$$k_a = tan^2 \left(45 - \frac{\phi}{2}\right)$$
 and
$$k_p = \frac{1}{k_a} \quad \text{Note: Some designers use } k_{p(Design)} = \frac{k_p}{SF} \text{ where SF} = 1.5 - 2.0$$

2. Calculate:
$$p_1 = \gamma L_1 k_a$$
 and $p_2 = \left(\gamma L_1 + \gamma L_2\right) \times k_a$, where $\gamma = \gamma_{SAT} - \gamma_w$

3. Calculate:
$$L_3 = \frac{p_2}{\gamma (k_p - k_a)}$$

4. Calculate:
$$P = Area \ ACDE = \frac{1}{2}p_1L_1 + p_1L_2 + \frac{1}{2}(p_2 - p_1)L_2 + \frac{1}{2}p_2L_3$$

6. Calculate: L4 by trial and error from the equation:

$$L_4^3 + 1.5L_4^2(I_2 + L_2 + L_3) - \frac{3P(L_1 + L_2 + L_3) - (\bar{z} + I_1)}{[\gamma(k_p - k_a)]} = 0$$

8. Calculate: Force in anchor rod
$$F = P - \frac{1}{2} \left[\gamma' (k_p - k_a) \right] \times L_4^2$$

9. Calculate:
$$p_8 = \gamma (k_p - k_a) \times L_4$$
 and

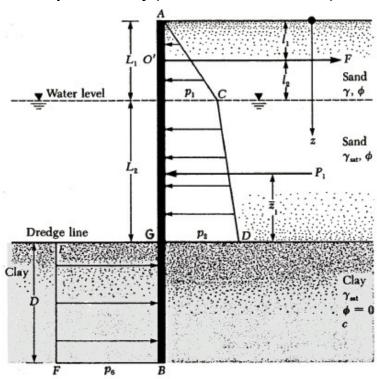
Draw the sheet pile (similar to page 1) with the estimated values in steps 1-8

$$\frac{1}{2}p_1L_1 - F + p_1(z_m - L_1) + \frac{1}{2}k_a\gamma(z_m - L_1)^2 = 0$$

$$L_1 < Z_m < (L_1 + L_2)$$

12. Calculate:
$$S = \frac{M_{max}}{\sigma_{all}}$$

6.7 Anchored Sheetpiles in Clay (Also called Bulkheads)



Anchored Sheet Pile in Clay with dimensions and pressure distribution

Design Steps (refer to figure above for terms)

1. Calculate:
$$k_a = tan^2 \left(45 - \frac{\phi}{2} \right)$$

2. Calculate:
$$p_1 = \gamma L_1 k_a$$
 and

$$\textbf{p_2} = \left(\gamma \textbf{L_1} + \gamma \dot{\textbf{L}_2}\right) \! \times \textbf{k_a}$$
 , where $\gamma^{'} = \gamma_{\text{SAT}} - \gamma_{w}$

3. Calculate: $P_1 = \text{Area ACDG} = \frac{1}{2}p_1L_1 + p_1L_2 + \frac{1}{2}(p_2 - p_1)L_2$

4. Calculate: \overline{z}_1 by taking moment about G of area ACDG = $P_1\overline{z}_1$

5. Calculate: $p_6 = 4c - (\gamma L_1 + \gamma L_2)$

6. Calculate: D from the following equation:

$$p_6D^2 + 2p_6D(L_1 + L_2 - I_1) - 2P_1(L_1 + L_2 - I_1 - \overline{z}_1) = 0$$

7. Calculate: Force in anchor rod F= P₁ - p₈ D

Draw the sheet pile (similar to page 1) with the estimated values in steps 1-6

8. Calculate: $D_{actual} = 1.3 \text{ to } 1.6 D$,

9. Calculate: z_m for zero shear, hence, M_{max} by :

$$\frac{1}{2}p_1L_1 - F + p_1(z_m - L_1) + \frac{1}{2}k_a\gamma'(z_m - L_1)^2 = 0$$

$$L_1 < Z_m < (L_1 + L_2)$$

10. Calculate: M_{max} by summing moments at a point z_m from surface.

11. Calculate: $S = \frac{M_{max}}{\sigma_{all}}$

where S = Minimum section modulus of sheet pile $\sigma_{all} = allowable$ stress for sheet pile

7. PILE FOUNDATIONS

7.1 Single Piles Equations:

Q_U = Ultimate Pile Capacity

Qp = Load-capacity of pile point

Q_s = Skin friction resistance

L = Pile length

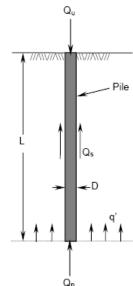
q' = overburden pressure at pile tip

D = Pile dimension



Pile cross-section

Ap = Area of the pile cross-section



Load-Capacity of Pile Point

$$Q_p = A_p q' N_q^*$$

or

$$Q_p = A_p q_I$$

whichever is smaller

where q_I(kPa) = 50 N_q* tanφ

and N_q* = Bearing capacity factor obtained from Fig. 1

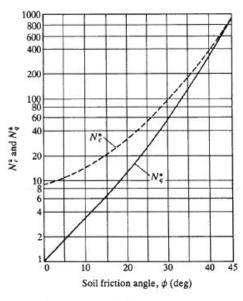


Fig. 1 Variation of N_c^* and N_q^* with ϕ

$$Q_s(1) = p L' f_{av}$$

 $Q_s(2) = p (L - L') f$
 $Q_s = Q_s(1) + Q_s(2)$

where

 $f = K \sigma_v' tan\delta$ (skin friction)

 σ_V ' = overburden pressure

K = lateral earth pressure = 0.5 + 0.008 D_r (D_r = relative density in percent)

 δ = friction angle between soil and pile (usually 0.6 ϕ)

p = perimeter of the pile

 f_{av} = average skin friction from 0 to L = f/2

Skin Friction Resistance

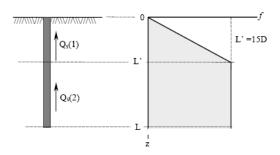


Fig. 2. Variations of skin friction with depth

7.2 Group capacity of piles:

Efficiency of Pile Group

Efficiency,
$$\eta = \frac{2 (n_1 + n_2 - 2) d + 4D}{p n_1 n_2}$$

where the parameters are as defined in Fig. 3

Ultimate load capacity of pile group:

$$Q_u(g) = \Sigma Q_u * \eta$$

if $\eta \geq 1.0$ use η = 1.0

Allowable load capacity of pile group:

$$Q_a(g) = Q_u(g)/SF$$

where $2.5 \le SF \le 4.5$

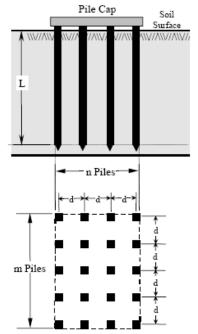


Fig. 3. Pile group

Example:

Estimate the total load that the pile group shown below can carry. Note that the every single pile is identical to that in Examples 1 and 2.

SOLUTION:

Total load that the a single pile can carry Q = 844.73 kN (Examples 1 and 2)

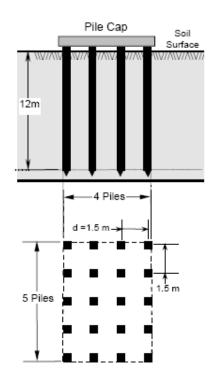
The efficiency of the group:

$$\eta = \frac{2 \left(m + n - 2\right) d + 4D}{p m n}$$

$$\eta = \frac{2(5 + 4 - 2)(1.5) + 4x(0.305)}{(4x0.305) \times 5 \times 4}$$
= 0.91 (or 91%)

The total load that can be carried by The pile group is, therefore

$$Q_{TG} = 16894.6 \times 0.91 = 15374.09 \text{ kN}$$



7.3 Settlement of Group Piles:

The settlement for group piles in coarse-grained soils from SPT and cone penetration tests can be estimated from:

SPT:
$$\rho_{es} = \frac{360\Delta\sigma_z I\sqrt{B_g}}{N_{cor}}$$
 (mm) (8.44)

Cone
$$\rho_{es} = \frac{\Delta \sigma_z I B_g}{2q_c}$$
 (8.45)

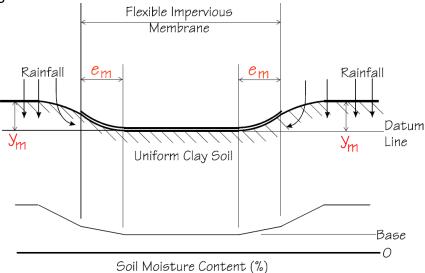
where: I=1-0.08 $\frac{L}{B_g} \ge 0.5$, $\Delta \sigma_z$ is the stress increase at a depth of $\frac{2L}{3}$, i.e.,

$$\left(\Delta\sigma_{z} \approx \frac{Q_{a}}{\left(B_{g} + \frac{2}{3}L\right)\left(L_{g} + \frac{2}{3}L\right)}\right)$$

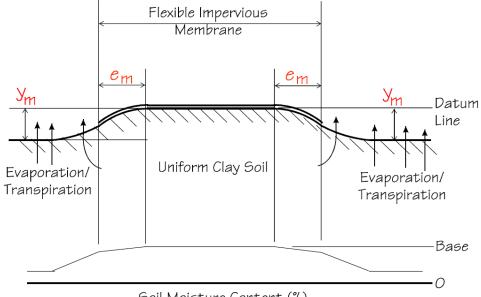
 B_g and L_g are the width and length of the pile group, L is the embedded length of the pile, and q_c is the arithmetic average of the cone resistance over two pile diameters below the cone tip.

8. Post Tensioned Slabs:

Edge Lift:



Center Lift:



Soil Moisture Content (%)

Edge Lift

□ Soils are wetter at slab edge than at any point inside slab edge.

Center Lift

☐ Soils are drier at slab edge than at any point inside slab edge.

Edge Moisture Variation Distance $\mathbf{e}_{\mathbf{m}}$

- ☐ Thornthwaite Moisture Index (climate)
- □ Soil Permeability
- Vegetation

Unrestrained Differential Swell $\mathbf{y}_{\mathbf{m}}$

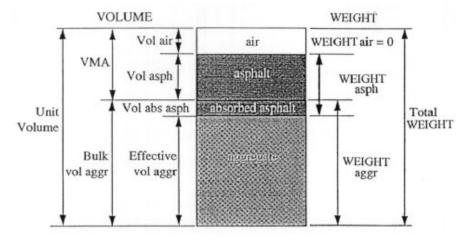
- □ Properties (activity) of clay
- □ Depth of clay (active zone)
- ☐ Soil suction

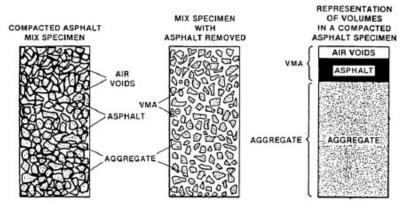
One set of e_m & y_m values established for each swell mode (edge and center lift)
Design cannot be done without these parameters

	Edge Lift	Center Lift		
e _m	2.0 ft	5.0 ft		
y _m	0.75 in	3.0 in		

The Structural Engineer also needs Kv (given in immediate settlement section), effective PI(pp 138 of Geotechnical DVD book) and other climatic constants that are from building codes(given).

9. Asphalt Mix Design:





LOW STABILITY

Causes	Effects
Excess binder in HMA	Washboarding, rutting, and flushing or bleeding
Excess medium size sand in HMA	Tenderness during rolling and for a period after construction, and difficulty in compacting
Rounded aggregate, little or no crushed surfaces	Rutting and channeling

POOR DURABILITY

Causes	Effects
Low binder content	Dryness or ravelling
High void content through design or lack of compaction	Early hardening of binder followed by cracking or disintegration
Water susceptible (hydrophillic) aggregate in HMA	Films of binder strip from aggregate leaving an abraded, ravelled, or
	mushy pavement

MIX TOO PERMEABLE

Causes	Effects
Low binder content	Thin binder films will cause early aging and ravelling
High void content in design HMA	Water and air can easily enter pavement causing oxidation and disintegration
Inadequate compaction	Will result in high voids in pavement leading to water infiltration and low strength

POOR FATIGUE RESISTANCE

Causes	Effects
Low asphalt binder content	Fatigue cracking
High design voids	Early aging of binder followed by fatigue cracking
Lack of compaction	Early aging of binder followed by fatigue cracking
Inadequate pavement thickness	Excessive bending followed by fatigue cracking

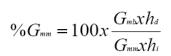
AC Mix design Formulas:

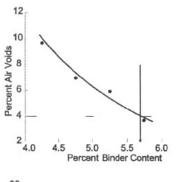
 G_{mb} = bulk specific gravity at N_{des}

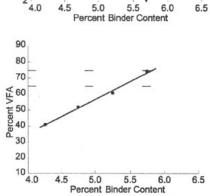
 G_{mm} = theoretical maximum specific gravity at N_{des}

= height of specimen at N_{des}

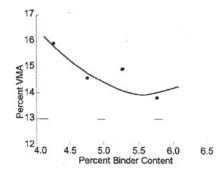
= height of specimen at N_{ini} h_i

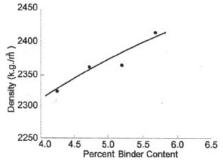






6.0





When weighing in Water:

Maximum Specific Gravity
$$(Gmm) = \frac{A}{A - (C - B)}$$

where

A = weight of oven dry sample in air, g

B = weight of container in water, g

C = weight of container and sample in water, g

When weighing in Air:

Maximum Specific Gravity (*Gmm*) =
$$\frac{A}{A+D-E}$$

where:

A = weight of oven dry sample in air, g

D = weight of container filled with water at 77°F, g

E = weight of container filled with sample and water at 77°F, g

Bulk Specific Gravity (
$$Gmb$$
) = $\frac{A}{B-C}$

where:

A = weight of specimen in air, g

B = weight of surface-dry specimen in air, g

C = weight of specimen in water, g

Percent Water Absorbed by Volume =
$$\left(\frac{B-A}{B-C}\right)$$
 x 100

Open Graded Mixtures:

$$Gmb = \frac{A}{B - E - \left(\frac{B - A}{F_{\tau}}\right)}$$

where:

A = weight of dry specimen in air, g

B = weight of dry, sealed specimen, g

E = weight of sealed specimen in water, g

(weight of absorbed water is subtracted)

Ft =apparent specific gravity of plastic sealing material at 77°F

Water Absorption, percent =
$$\left(\frac{A_1 - A}{A}\right) \times 100$$

where

Dust Proportion =
$$\frac{P_{200}}{Pbe}$$
 $\frac{P_{200}}{Pbe}$ = aggregate content passing the No. 200 sieve, percent by weight of aggregate $Pbe = Pbe = Pbe$

Air Voids (
$$Va$$
) = 100 x $\left(\frac{Gmm - Gmb}{Gmm}\right)$

Voids in the Mineral Aggregate (VMA) = 100 - $\left(\frac{Gmb \times Ps}{Gsb}\right)$

where:

where:

Gmm = Maximum Specific Gravity of HMA Gmb = Bulk Specific Gravity of HMA Gmb = Bulk Specific Gravity of HMA
Gsb = Bulk Specific Gravity of aggregate
(Obtained from design mix formula)
Ps = Aggregate, percent by total weight of HMA

The percent of aggregate by total weight of HMA (Ps) is determined by subtracting the actual binder content by total weight of HMA (Pb) supplied on the design mix formula from 100.

$$Ps = 100 - Pb$$

Voids Filled with Asphalt (
$$VFA$$
) = $\left(\frac{VMA - Va}{VMA}\right)$ x 100

10. Concrete Mix Design:

Free Moisture (%) =
$$\frac{WW - Ws}{Wd}$$
 x100

Where: Ws = Saturated surface-dry weight.

Absorbed Moisture (%) =
$$\frac{WW - Wd}{Wd} \times 100$$

Fineness modulus:

The standard size sieves are 6 inch, 3 inch, 1 1/2 inch, 3/4 inch, 3/8 inch, No. 4, No. 8, No. 16, No. 30, No. 50, and No. 100. In this series, the size of each opening, beginning with the 100-mesh sieve, is one-half that of the next larger size used. The percent of material passing the 100-mesh sieve is not used in calculating the fineness modulus. For example, the fineness modulus of a fine aggregate, such as would be used in concrete, may be as follows:

FINE AGGREGATE - SAND, GRADING A

Sieve Size	Percent Retained	Cumulative Percent Retained
3/8" (9.5 mm)	0.0	0.0
No .4 (4.75 mm)	0.0	0.0
No. 8 (2.36 mm)	12.0	12.0
No. 16 (1.18 mm)	15.0	27.0
No. 30 (0.600 mm)	32.0	59.0
No. 50 (0.300 mm)	18.0	77.0
No. 100 (0.150 mm)	13.0	90.0
No. 200 (0.075 mm)	10.0	100.0 (Not Include.)
		Total = 265.0

$$\frac{2.65}{100} = 2.65 = F.M.(Ans.)$$

Yield:

The yield of concrete produced per batch shall be calculated as follows:

$$Y = \frac{(N \times 94) + Wf + Wc + Ww}{W}$$

Where: Y = Yield of concrete produced per batch, in cu. ft.,

N = Number of bags of cement in the batch,

94 = Net weight of a bag of cement, in lbs.,

Wf = Total weight of fine aggregate in batch in condition used, in lbs.,

Wc = Total weight of coarse aggregate in batch in condition used, in lbs.,

Ww = Total weight of mixing water added to batch, in lbs., and

W = Weight of concrete, in lbs. per cu. ft.

Relative Yield:

$$Ry = \frac{Y}{Yd}$$

Y = Yield of concrete produced per batch, in yd3,

Yd = Theoretical Yield (yd3)

NOTE: A value for Ry greater than 1.00 indicates an excess of concrete being produced, while a value less than 1.00 indicates the batch to be "short" of its designed volume.

Where: Ry = Relative Yield of concrete

(a) Unit Weight

The net weight of the concrete shall be calculated by subtracting the weight of the measure used in the test from the gross weight. The unit weight shall be calculated by multiplying the net weight by the factor for the measure used. The method of determining this factor is given in AASHTO T121.

Modulus of Rupture:

=
$$(7.5\sqrt{f'c})$$
 or

Modulus of Rupture, in psi =
$$\frac{3WL}{2bd^2}$$

Where: W = Maximum indicated load, in lbs.,

L = Distance between supports, in in., and

b & d = Breadth and depth of beam, in in.

With a 6" x 6" x 40" beam, this formula resolves to: $\frac{72W}{432} = \frac{W}{6}$

Therefore, 1/6 of the gage reading equals the modulus of rupture of the tested specimen, in lbs. per sq. in.