

ENGINEERING TRIPOS PART IB - PAPER 8

SOLUTIONS - 1997

1. a) A $\rho = 7.9/4 = 1.975 \text{ kg/l} = 1975 \text{ kg/m}^3$
 $\rho_d = 6.63/4 = 1.658 \text{ kg/l} = 1658 \text{ kg/m}^3$

Data sheet: $\rho = \rho_w \frac{G_s + S_r e}{(1+e)}$; $\rho_d = \rho_w \frac{G_s}{(1+e)}$

$\rho_w = 1000 \text{ kg/m}^3$, $G_s = 2.65$, $\therefore e = 0.598$
 $\therefore 3157 = 1000 (2.65 + 0.598 S_r)$
 $\therefore S_r = 0.847$

B $\rho = 1820 \text{ kg/m}^3$, $\rho_d = 1325 \text{ kg/m}^3$
 $\therefore e = 1.00$, $\therefore S_r = 0.99$

b) A $\sigma_v = 2 \times 1.975 \times 9.81 = 59 \text{ kPa}$
B $\sigma_v = 4 \times 1.975 \times 9.81 + 2 \times 1.820 \times 9.81$
 $= 113 \text{ kPa}$

c) A is 2m above the water table. If it had been saturated, its pore pressure would have been

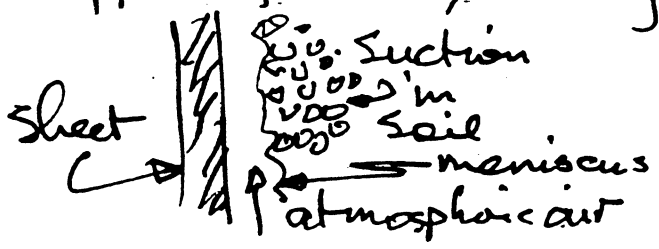
$u = -2 \times 9.81 \approx -20 \text{ kPa (gauge)}$

If it had been dry ($S_r = 0$), its pore pressure would have been zero (atmospheric). The answer lies somewhere in between, probably closer to -20 kPa

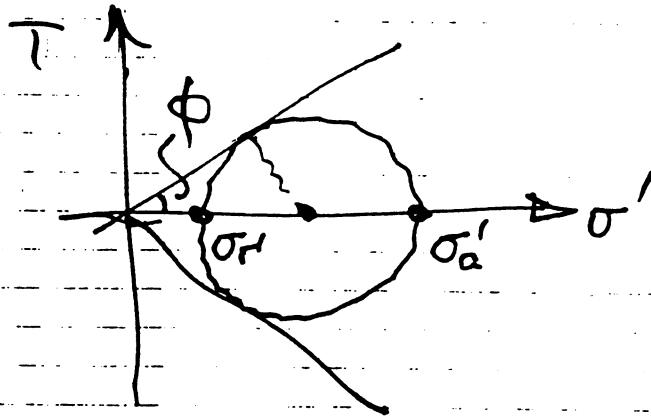
B is 2m below the water table.

$u = 2 \times 9.81 \approx 20 \text{ kPa}$

Negative pressures can be eliminated by rain, or leaking pipes etc. They need not act as suction on supporting sheets, in any event.



d) (i) $\sigma_r' = 50 \text{ kPa}$; $\sigma_a' = 290 \text{ kPa}$, $\sigma_a' = 190 \text{ kPa}$
max ult

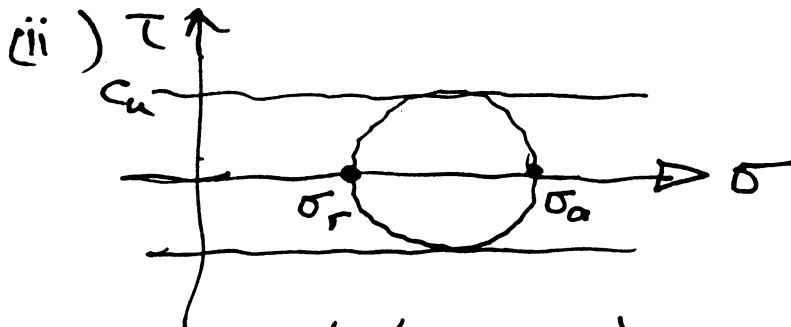


$$\sin \phi = \frac{\text{radius}}{\text{distance to centre}}$$

$$= \frac{(\sigma_a' - \sigma_r')/2}{(\sigma_a' + \sigma_r')/2}$$

$$\sin \phi_{\text{max}} = 240/340 ; \phi_{\text{max}} = 44.9^\circ$$

$$\sin \phi_{\text{ult}} = 140/240 ; \phi_{\text{ult}} = 35.7^\circ$$



$$c_u = \frac{1}{2} (\sigma_a - \sigma_r) = 60 \text{ kPa}$$

e) For A: Take $\sigma_v' = 39 \text{ kPa}$

Use $\frac{\sigma_h'}{\sigma_v'} = \frac{50}{190}$ corresponding to the

ultimate shear ratio in the triaxial test.

Then $\sigma_h = \sigma_h' = 10 \text{ kPa}$

For B: Take $\sigma_v = 113 \text{ kPa}$

Use $\sigma_h = \sigma_v - 2c_u = 113 - 120 = -7 \text{ kPa}$

e) cont.

The 102 kPa at A mobilises the ultimate strength, which is less than the peak strength. It is consistent with massive sliding failure. If supports prevent rupture, this is safe. But rainfall or leaks would tend to pond above the clay and the water level might rise. Either punch drain holes, or allow water table to rise, for safety.

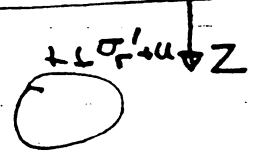
The -72 kPa at B invokes undrained strength, with negative pore pressures mobilised temporarily. Water could certainly drain into a crack against the sheets to give $u = +20 \text{ kPa}$, $\sigma_h = 20 \text{ kPa}$. Even higher pressures would occur if the clay drained; a ϕ calculation would be needed.

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- 2 a) * Water is mobile
- if restricted it causes pressure
 - if permitted it leaks into cavities

- * Consider a sealed tunnel etc.
- full water pressure $u = \gamma_w Z$
 - reduced effective earth pressure $\sigma'_r \ll (\gamma - \gamma_w) Z$

e.g. water table at surface



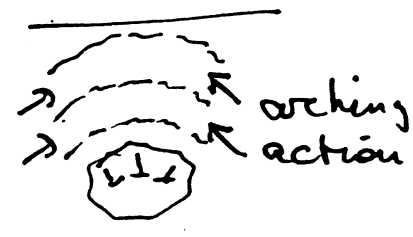
- Terzaghi's effective stress σ' applied like Dalton's Law of Partial Pressures

$$\sigma = \sigma' + u$$

total stress effective stress pore pressure
 For level uniform ground:
 σ obtained by vertical equilibrium $\sigma = \gamma Z$
 u obtained by hydrostatics $u = \gamma_w Z$
 $\therefore \sigma' = (\gamma - \gamma_w) Z$

So if u is high σ' is low.
 But stiffness & strength proportional to σ' .
 So high water table is bad.

For collapsing tunnel etc. $\sigma < \gamma Z$



But if $u = \gamma_w Z$ still the water component of lining stress is not reduced, only the effective component. The proportional effect of water at collapse is

therefore even greater.

- * Consider a leaky tunnel etc.
- seepage into the face as it is constructed can bring soil with it, leading to collapse & void migration
 - seepage into lining during service can cause corrosion, and needs pumping away.

- * What can be done?
- during construction; pump to lower water table, grout or freeze to restrict inflow, use compressed air (or pressurised bentonite) in headings, shafts etc.
 - in service; design lining etc to resist pressure or prevent leakage.

2b) * NATM: excavate freely eg with back hoe in soil or with roadheader in rock; spray primary concrete lining as soon as final surface is exposed; monitor movement and follow up with secondary lining system sufficient to reduce movement towards zero. Advantages - quick, cheap, permits ground to mobilise strength, therefore leads to minimal tunnel lining. Disadvantages - hard to control failure if collapse begins, vulnerable to brittle soils, requires skill & real-time intelligence, may lead to excessive surface settlements in clay.

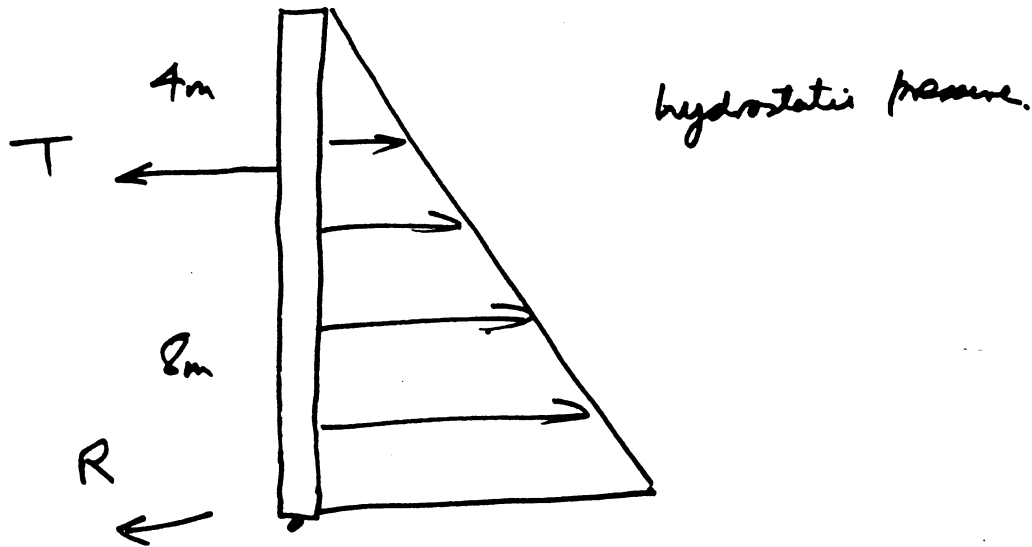
* TBM - FFB: tunnel boring machine with picks housed on a head in a chamber pressurised by bentonite mud, or air etc. Muck has to pass either through a sealed port into the (unpressurised) tunnel, or through a circulation system under pressure which can segregate the muck from the bentonite mud, before the mud is recirculated.

Advantages: elimination of surface subsidence, safe working conditions. Disadvantages: high cost, problems with blocking of muck route.

* Ideal for NATM - tunnel in weak rock, subsidence not critical. Ideal for TBM - tunnel in homogeneous soft clay with high water table, with a city above which demands small subsidence.

* NATM - fault zone leads to unheralded collapse; TBM - erratic boulders jam the picks; one example of unforeseen problems.

Qn 3 (a)



Density of water = 10 kN/m^3

Water pressure at depth $x = 10 \cdot x \text{ kN/m}^2$

\therefore On a metre wide strip, total horizontal force

$$= \frac{120}{2} \cdot 12 \cdot 1 = 720 \text{ kN}$$

The resultant will act $\frac{2}{3}$ of total height from the top i.e. 8 m down

$\therefore R$ & T will both be 360 kN

\therefore The bending moment will be

$$M = \frac{10x}{2} \cdot x \cdot \frac{x}{3} - 360 \{x - 4\} \quad \text{Macaulay brackets.}$$

$$\text{Shear Force } F = \frac{10x}{2} \cdot x - 360 \{x - 4\}^0$$

which is zero when $x = 9.48 \text{ m}$

when $M = -596 \text{ kNm/m width}$

-89-

Possible bar size $32 \text{ mm bar} \approx 804 \text{ mm}^2/\text{bar}$
 $= 7.9 \text{ bars/m}$
 $\approx 127 \text{ mm spacing.}$

40 mm bar $= 1256 \text{ mm}^2/\text{bar}$
 $= 5.03 \text{ bars/m}$
 $\approx 198 \text{ mm spacing.}$

Thus, $32 \phi @ 125 \text{ mm}$ } both acceptable.
 or $40 \phi @ 175 \text{ mm}$

- Practical considerations. Bigger bar requires more thickness to give both required depth and cover. Bigger spacing gives more possibility of cracking.

$\therefore \underline{32 \phi @ 125}$ more likely

In bending the other way, at the tie position Neutral axis will be at a much smaller depth, since moment much lower. ($M = 106.7 \cdot 10^6 \text{ Nmm}$)

Initially, guess $\alpha = 0.2$ (value not critical)

$$0.87 \cdot 460 \cdot A_s \cdot 316 \left(1 - \frac{0.2}{2}\right) = 106.7 \cdot 10^6$$

$$\Rightarrow A_s = 937 \text{ mm}^2/\text{m.}$$

Recalculate.

-90-

$$\alpha = 2.175 \cdot \frac{460}{40} \cdot \frac{937}{1000 \cdot 316}$$
$$= 0.074$$

Recalculate A_s

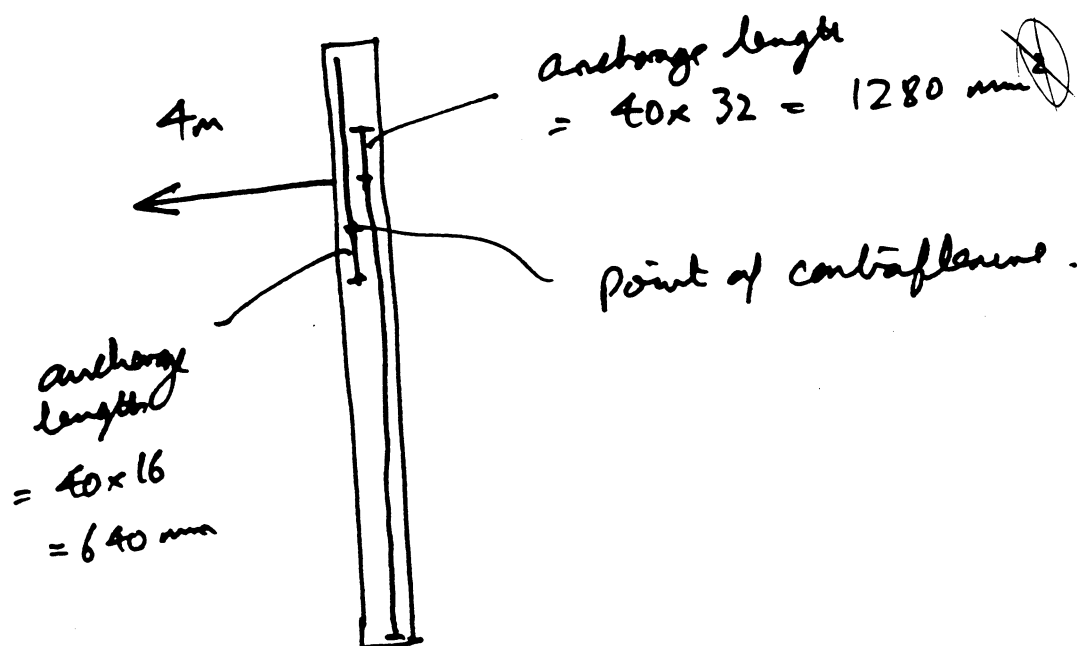
$$0.87 \cdot 460 \cdot A_s \cdot 316 \left(1 - \frac{0.074}{2}\right) = 106.7 \cdot 10^6$$

$$\Rightarrow \underline{A_s = 876 \text{ mm}^2/\text{m}}$$

(could iterate further, but not really necessary as integral No. of bars is needed).

$$16 \text{ mm bar} = 201 \text{ mm}^2 \equiv 4.35 \text{ bar/m}$$
$$= 229 \text{ mm spacing.}$$

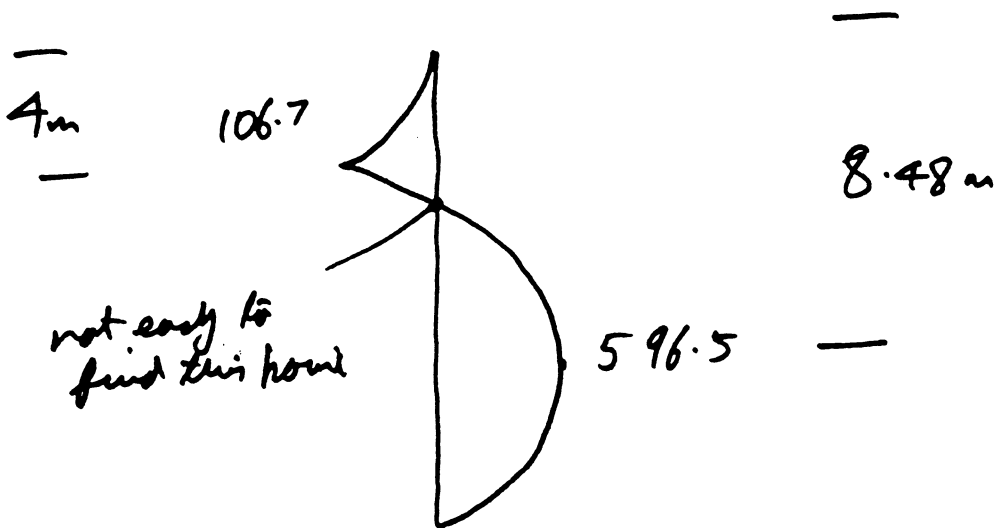
or Seems reasonable 16 ϕ @ 225 mm



Moment distribution is

(in kNm/m)

-91-



This is the loading with no safety factors applied to the load. However, it is impossible to overload a water tank, since it would overflow. \therefore No increase needed.

[note to matter - this argument should be explicitly stated. They could argue that the factor includes an allowance for ignorance and general uncertainty in modelling, in which case a value up to about 1.2 should be accepted].

$$(b) \quad M = 0.15 f_{cu} b d^2 \quad -92-$$

$$\text{Say } M = 600 \text{ kNm/m} \quad (\text{rounded})$$

$$f_{cu} = 40 \text{ N/mm}^2$$

$$b = 1000 \text{ mm}$$

$$\therefore 600 \cdot 10^6 = 0.15 \cdot 40 \cdot 1000 \cdot d^2$$

$$\Rightarrow \underline{d = 316 \text{ mm}}$$

This is effective depth.

Allow: 50 mm cover (since water retaining)

+ 15 mm for bar radius

$$\Rightarrow \text{Total thickness} \approx \underline{380 \text{ mm}}$$

can be confirmed after steel has been designed.

(c) If we take the effective depth as 316 mm,

then the neutral axis will be at half depth

(i.e. $x_d = 158 \text{ mm}$)

$$\therefore 0.87 \cdot f_y \cdot A_s d \left(1 - \frac{x}{2}\right) = 600 \cdot 10^6$$

$$0.87 \cdot 460 \cdot A_s \cdot 316 \cdot \frac{3}{4} = 600 \cdot 10^6$$

$$\Rightarrow 6326 \text{ mm}^2 \text{ of steel /m width}$$

Q 3(d)

- (i) A SLS criterion limiting the stresses in the concrete would be a better design philosophy than ULS in a water retaining structure such as this since this would result in smaller crack widths + thus reduce the likelihood of corrosion of the reinforcement & the possibility of leakage through the wall. ULS design only considers strength + does not examine cracking at all.
- (ii) This would require a thicker section resulting in reduced curvatures. The amount of reinforcement required would depend on calculations however you would expect more steel in the section. This would result in lower steel stresses + narrower crack widths. In reinforced concrete cracking is inevitable hence the aim of the designer is to limit the size of these cracks.

Q 3 (d)

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1997
P.I.A Paper 8 Section B Answer

- 4. b) $\sim 0.065 \text{ kgm}$
- d) $\sim 12 \text{ Hz}$ for a 1% speed fluctuation

5. ~~a) vibration from pig motion, ocean currents or noise from electronic circuits~~

- b) low-pass $\sim 72 \text{ Hz}$
high-pass $\sim 3.4 \text{ Hz}$

c) range $\sim 0.5 - 2.5 \text{ kNm}^{-1}$ depending on choice of c and 3dB frequency

d) ~~mechanical low-pass only~~
~~electronic easier to design~~

6. -

Answers for (4)+(5)

+ Cries 4, 5, 6.

Please add + give to
Ann Dawling's See -
see you next Friday Ru

Q4 (a) The eccentric moment for the eccentric mass may be derived by considering only the semi-cylindrical part of the mass, since the cylindrical part has a centre of mass on the axis of rotation and therefore zero eccentric mass.

$$m.r = (m.r)_{\text{semi-cylinder}} + \cancel{(m.r)_{\text{cylinder}}}$$

$$\text{For a semi-cylinder } m = \rho \cdot \frac{\pi}{2} \cdot \left(\frac{D}{4}\right)^2 \cdot \frac{D}{2} = \frac{\rho \pi D^3}{64}$$

$$r = \frac{2}{3} \cdot \frac{D}{4} \cdot \frac{\sin \frac{\pi}{2}}{\pi/2} = \frac{D}{3\pi} \quad \left(\begin{array}{l} 5.4.2 \\ \text{p.16 Mechanics} \\ \text{Data book} \end{array} \right)$$

$$\text{Hence } m.r = \frac{\rho \pi D^3}{64} \cdot \frac{D}{3\pi} = \underline{\underline{\frac{\rho D^4}{192}}}$$

(b) For a steel mass where $D = 0.2 \text{ m}$; $\rho = 7800 \text{ kg/m}^3$

$$\text{Hence } m.r = \frac{7800 \times 0.2^4}{192} = \underline{\underline{0.065 \text{ kg m}}}$$

(c) The speed fluctuation may be derived from consideration of the energy balance of the rotating system.

i.e. change in potential energy when centre of mass moves from highest to lowest position equals the change in kinetic energy.

Q4 (c) cont.

Change in potential energy = mgh , where h is the distance moved vertically by the centre of mass of the eccentric semi-cylinder. The cylindrical part of the mass shows no change in energy.

$$\begin{aligned} \therefore \Delta \text{ potential energy} &= mgh \\ &= \underbrace{\left(\rho \cdot \frac{\pi}{2} \cdot \left(\frac{D}{4}\right)^2 \cdot \frac{D}{2} \right)}_m \cdot g \cdot \underbrace{2 \cdot \frac{D}{3\pi}}_h \end{aligned}$$

note m is the mass of the semi-cylinder.

$$\text{Change in kinetic energy} = \frac{1}{2} I (\omega_{\max}^2 - \omega_{\min}^2)$$

$$\text{If } \omega_{\max} = \omega + \Delta\omega \text{ and } \omega_{\min} = \omega - \Delta\omega$$

$$\begin{aligned} \text{then } \Delta \text{ kinetic energy} &= \frac{1}{2} I ((\omega + \Delta\omega)^2 - (\omega - \Delta\omega)^2) \\ &= 2 I \omega \Delta\omega \end{aligned}$$

where I is the inertia of the whole eccentric mass

Now $I = 3 m k^2$ where m is the mass of a semi-cylinder and k its radius of gyration

$$\text{Hence } I = 3 m \cdot \left(\frac{1}{2} \cdot \left(\frac{D}{4}\right)^2 \right) \quad \left(\text{p 17 } \overset{\text{S.S.1}}{\text{Mechanics Data book}} \right)$$

$$\text{and } \Delta \text{ kinetic energy} = 3 m \left(\frac{D}{4}\right)^2 \omega \Delta\omega$$

Q4 (c) cont.

Δ kinetic energy = Δ potential energy

$$\therefore 3m\left(\frac{D}{4}\right)^2 \omega \Delta\omega = m.g. 2 \frac{\Delta}{3\pi}$$

rearrangement gives $\frac{\Delta\omega}{\omega} = \underline{\underline{\frac{32g}{9\pi\omega^2 D}}}$

(d) For the steel mass, assuming a 1% fluctuation is acceptable:

$$\frac{\Delta\omega}{\omega} = 0.01 = \frac{32g}{9\pi\omega^2 \times 2}$$

Hence $\omega = 74.5 \text{ radsec}^{-1}$

or $f = \underline{\underline{11.9 \text{ Hz}}}$

This is a lower limit, since a higher frequency would decrease $\Delta\omega/\omega$.

Q5. (a) Accelerometer noise typically arises from vibration of the pig moving along the pipe (high frequencies). Movement of spars due to ocean currents may introduce low frequency noise. There may also be electrical or mechanical noise generated within the vibrating pig.

(b) The voltage gain for the first amplifier is given by:

$$\text{gain} = -\frac{1}{1 + j\omega RC_1}$$

which represents a low-pass filter with a turning frequency of $1/RC_1$ radsec⁻¹.

The voltage gain for the second amplifier is given by:

$$\text{gain} = \frac{1}{1 + 1/j\omega RC_2}$$

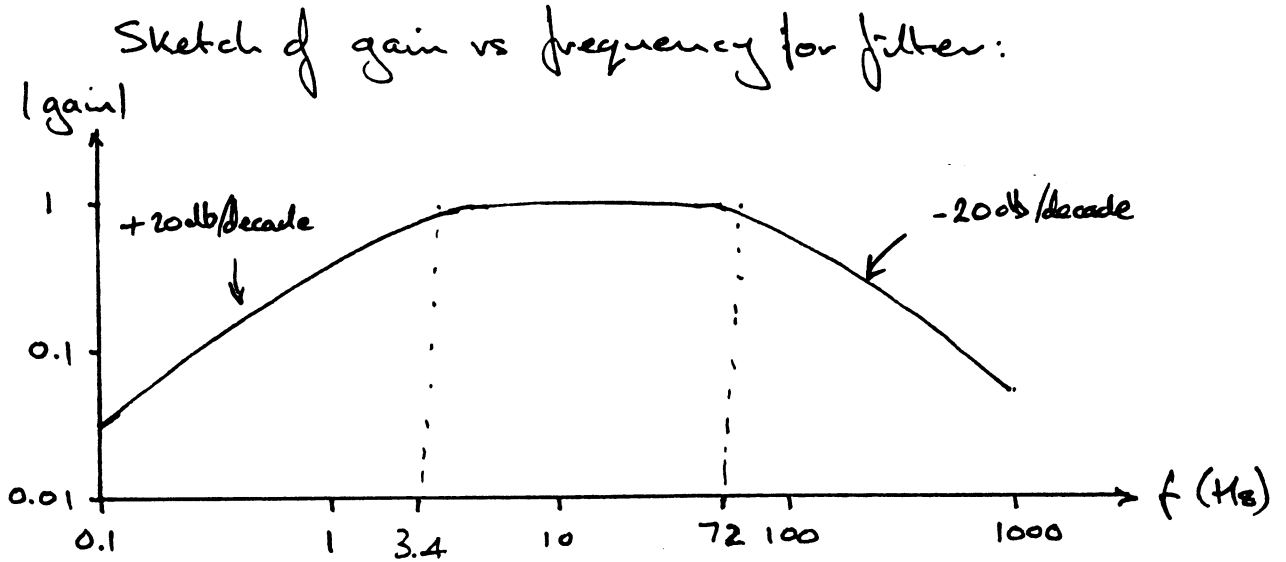
which represents a high-pass filter with a turning frequency of $1/RC_2$ radsec⁻¹

For $R = 10 \text{ k}\Omega$, $C_1 = 0.22 \mu\text{F}$ and $C_2 = 4.7 \mu\text{F}$:

$$\text{low-pass turning frequency} = \frac{1}{2\pi} \cdot \frac{1}{RC_1} = \underline{\underline{72 \text{ Hz}}}$$

$$\text{high-pass turning frequency} = \frac{1}{2\pi} \cdot \frac{1}{RC_2} = \underline{\underline{3.4 \text{ Hz}}}$$

Q5. (b) cont.



(c) The equation of motion for the mass is given by:

$$m\ddot{y} = k(x - y) + \lambda(\dot{x} - \dot{y})$$

$$\therefore \frac{m}{k}\ddot{y} + \frac{\lambda}{k}\dot{y} + y = \frac{\lambda}{k}\dot{x} + x$$

$$\text{cf. } \frac{1}{\omega_n^2}\ddot{y} + \frac{2c}{\omega_n}\dot{y} + y = \frac{2c}{\omega_n}\dot{x} + x \quad \left(\begin{array}{l} 4.6.3 \\ \text{p 12 Mechanics} \\ \text{Data book} \\ \text{case (c)} \end{array} \right)$$

$$\text{Turning frequency } \approx \omega_n = \sqrt{\frac{k}{m}} = 72 \times 2\pi \text{ rad sec}^{-1}$$

$$\therefore k = (2\pi \times 72)^2 \times 0.01 = \underline{\underline{2.1 \text{ kN/m}}}$$

$$\text{Also } \frac{2c}{\omega_n} = \frac{\lambda}{k}$$

$$\therefore c = \frac{2\pi \times 72}{2} \cdot \frac{3}{2100} = 0.32$$

Q5 (c) cont.

The previous answer is an oversimplification of the problem, but would result in an adequate design even if the peak gain is a little high.

A better result would be obtained by choosing a sensible value for c for a moderate peak gain, eg. $c = 0.8$. This suggests that the 3db point would then be a little above $2\omega_n$. To approximate:

$$2\omega_n = 72 \times 2\pi \text{ rad sec}^{-1}$$

$$k = (\pi \times 72)^2 \times 0.01 = \underline{\underline{510 \text{ N/m}}}$$

A more accurate answer could be found by letting

$$\frac{Y}{X} = \frac{1}{\sqrt{2}} = \frac{\{1 + (2c\omega/\omega_n)^2\}^{1/2}}{\{[1 - (\omega/\omega_n)^2]^2 + (2c\omega/\omega_n)^2\}^{1/2}}$$

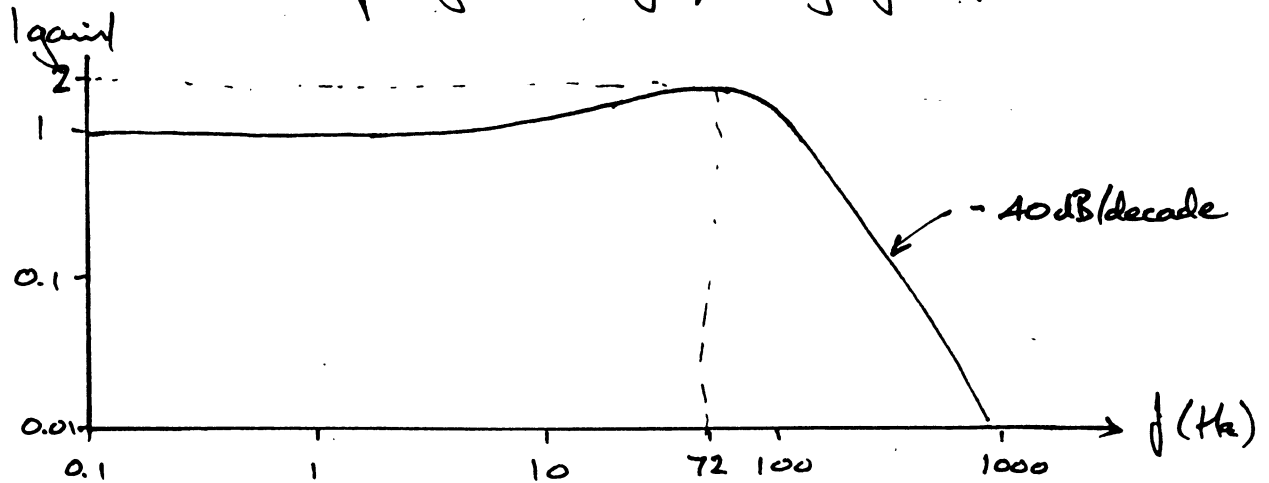
This is messy to solve, but possible. The graphical method is much easier.

In reality the 3dB frequency for the mechanical system would be increased to avoid the gain peak occurring at frequencies of interest.

A range of numeric answers would be acceptable.

Q5. (c) cont.

Sketch of gain vs frequency for filter:

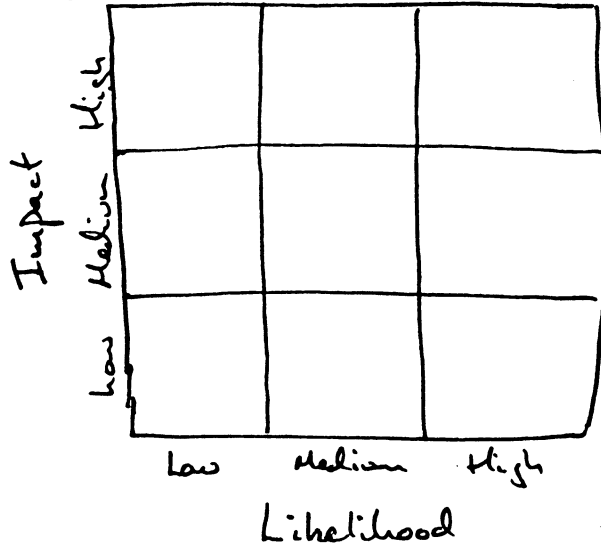


(d) The resonant frequency of the pipe is 15 Hz. Both filters will remove high-frequency noise above 72 Hz. Only the electronic filter will remove low frequency noise.

The electronic filter will most likely be most practical since it is easier to construct and tune to the correct frequencies and may be integrated with the charge amplifier which is already a part of the pig.

Q8. (a) (i) A likelihood / impact grid is a plot of known project risks against axes of likelihood of occurrence and impact.

e.g.



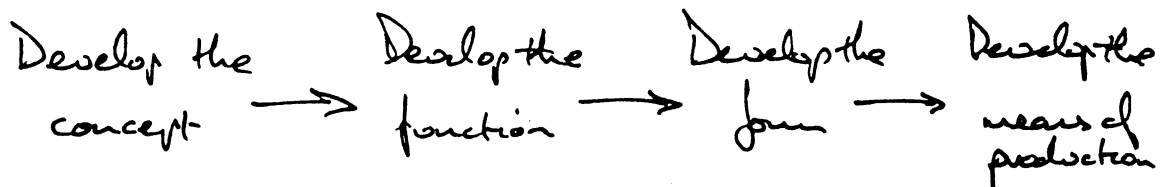
Project risk may be factors which effect development overrun, budget overspend, product or service performance, quality and safety.

Risks which appear in the upper-right sector of the plot must be actively reduced by either minimizing their impact or by reducing their likelihood of occurrence.

The impact and likelihood scales are qualitative and should be chosen to illustrate the full range of risks.

Q6 (a) (i) Phased development helps to reduce risk, especially if areas of high risk are investigated and tested early. Use of prototypes and models can be used to eliminate technical risks before committing to more detailed design.

In general:



The stages of BS 7000 are consistent with this, with different emphases for different types of product.

(ii) HAZOP takes a product or process or service which is assumed to be sound and systematically assesses the consequences and likely causes of abnormal behaviour.

For each design parameter in each part of the product the questions of NONE, MORE OF, LESS OF, FACTOR, MORE THAN and OTHER THAN are considered. If a deviation results in an undesirable effect (safety or operational) then the cause is investigated and the design modified to eliminate the effect. HAZOP is a team exercise.

Q8 (a) (iii) cont.

Results are usually tabulated:

Question	Effect	Cause	Action
MODE OF			

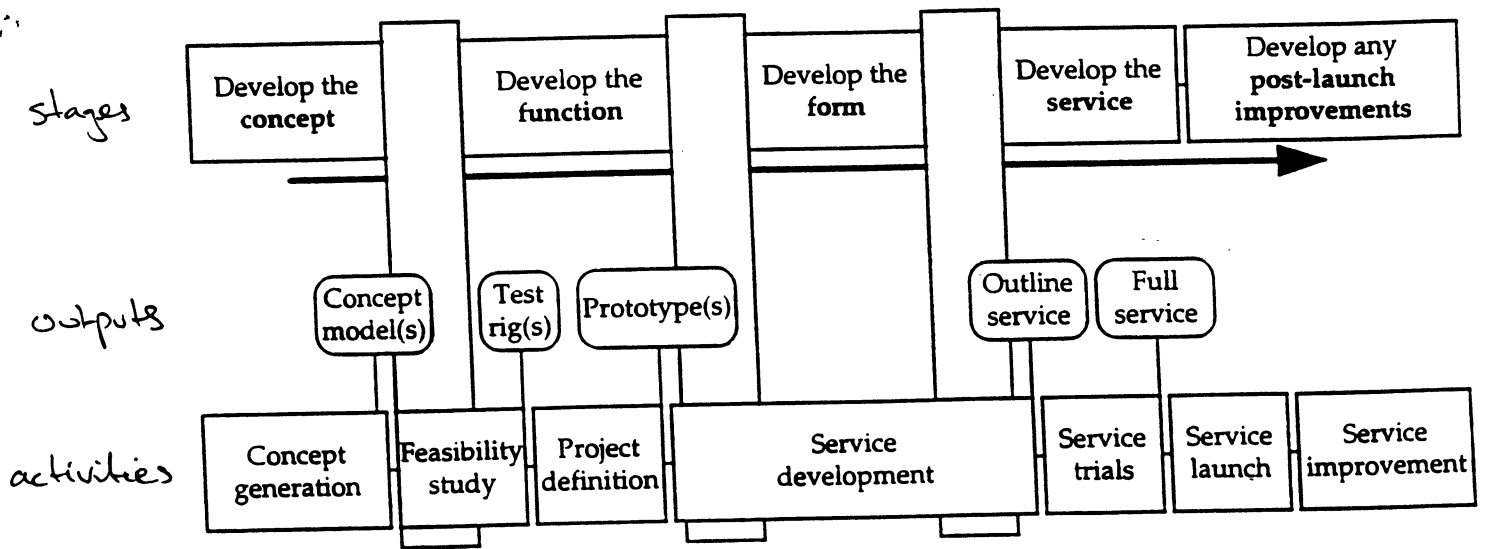
(iv) FMEA assesses the consequences of failure of a single component in a system. Failure may be total or partial and may effect more than one physical property. A component may be a stage in a service provision. FMEA may be a team or individual exercise.

Results are usually tabulated:

Failure mode	Effect	Action

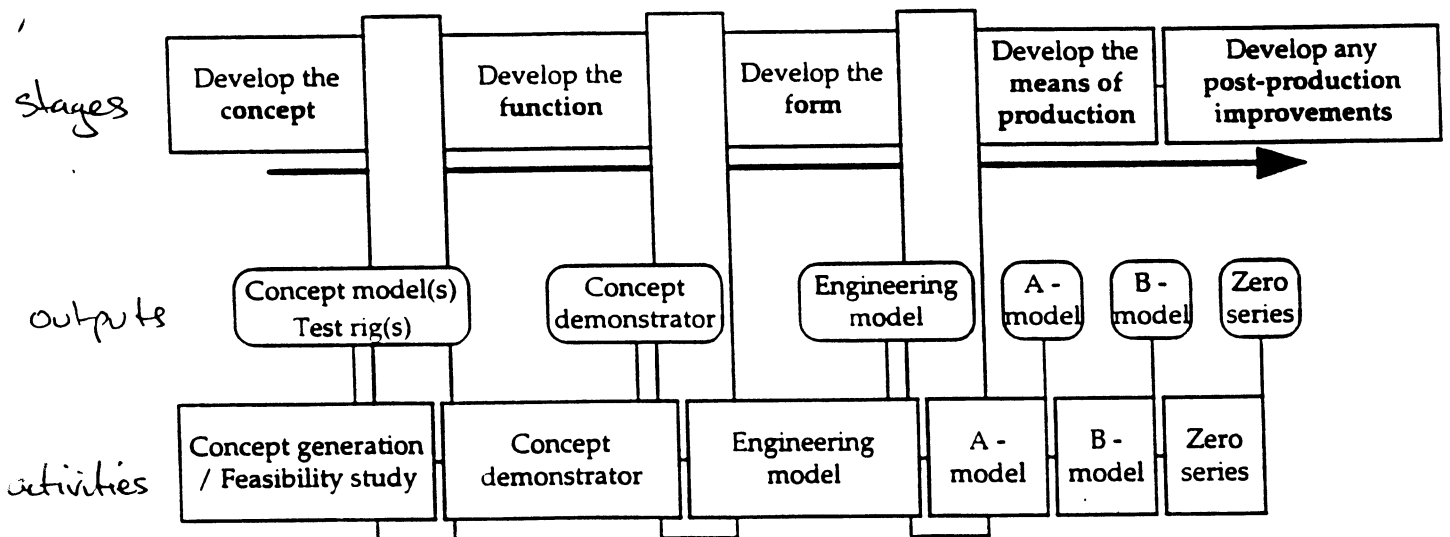
Failure modes are usually assessed for likelihood of occurrence before design actions are specified.

Q8 (b) Stages for service development are shown below:



Reference to BS 7000 is also appropriate.

(c) Stages for high-volume product development are:



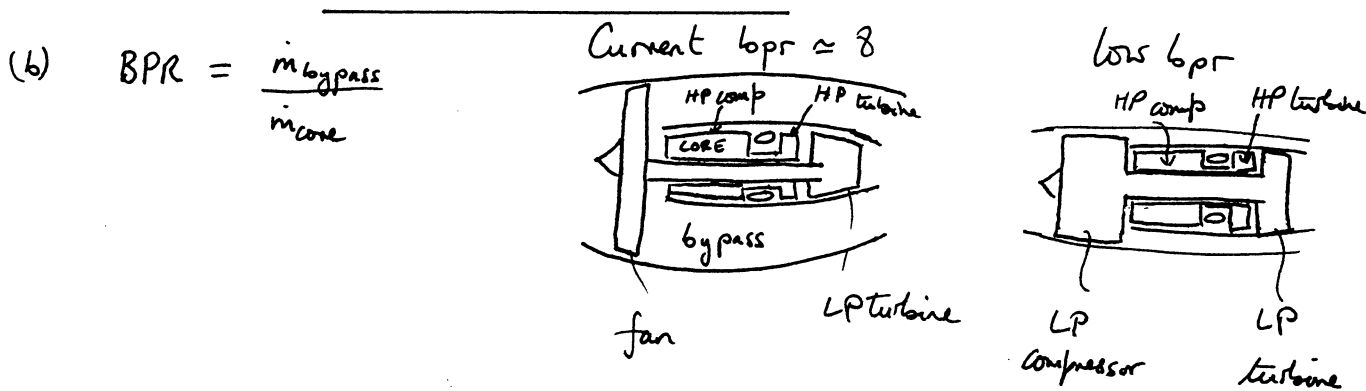
Key differences are the needs to develop production systems which may take place in parallel with product development. There may be a range of models and demonstrators to aid development of different parts of a product, and to assist in development of the production systems. Patents are important to review

$$7 (a) \quad \eta_p = \frac{\text{useful work}}{\Delta KE_{\text{gas}}} \quad \eta_{\text{cycle}} = \frac{\Delta KE_{\text{gas}}}{m_f \text{ LCV}} \quad \eta_{\text{ov}} = \frac{\text{useful work}}{m_f \text{ LCV}}$$

and useful work is the work done against the drag on the aircraft = $(m_{\text{air}} + m_{\text{fuel}}) V_f - m_{\text{air}} V_i \approx m_{\text{air}} (V_f - V_i)$
 ↖ flight speed

Flight speed does not scale directly with engine operating point.

∴ η_{cycle} same (same op. point), η_p & η_{ov} different



(c) Cycles which deliver high power are also efficient only if the pressure ratio is high. A typical value is 40:1.

Increasing the design value of turbine inlet temperature increases cycle efficiency to some extent (not by that much if today's typical values are taken as datum), but most importantly significantly increases the net power output from the ideal cycle. This power can be used to increase the power available to the fan and thus increase the propulsive efficiency through an increase in bypass ratio.

(d) In addition to low levels of propulsive efficiency and increased noise, a low bypass ratio engine has (for given core pressure ratio and turbine inlet temperature) a low ratio of top of climb thrust and of take off thrust to cruise thrust.

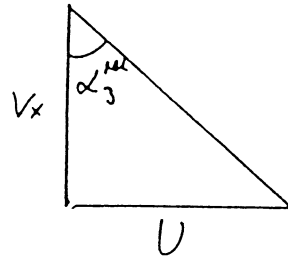
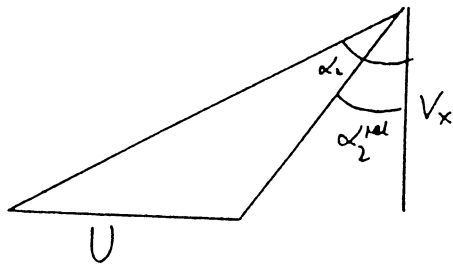
(e) $\eta_p \approx \frac{2v_1}{v_J + v_1}$ increases as v_J is reduced. High bypass ratios reduce v_J and thus have better η_p (& η_o) than low bypass ones.

Bypass ratios are limited by : ratio of LP turbine speed / fan speed needs to be kept reasonable to avoid too many turbine stages / a gearbox; weight of engine increases with bypass ratio; greater size of engine (which implies higher manufacturing costs, more nacelle drag, a high and expensive undercarriage and potential problems with existing airport facilities).

8. (a)

Stator Exit ②

Rotor Exit ③



$$(b) \quad \tan \alpha_3^{\text{rel}} = \frac{U}{V_x} = \frac{1}{.55} \Rightarrow \underline{\underline{61.2^\circ = \alpha_3^{\text{rel}}}}$$

$$(c) \quad W_x = U(V_{\theta 2} - V_{\theta 3}) \quad \& \quad V_{\theta 3} = 0$$

$$\therefore \frac{V_{\theta 2}}{U} = \frac{W_x}{U^2} = 1.5 \quad \Rightarrow \quad \frac{V_{\theta 2}}{V_{x2}} = \frac{V_{\theta 2}/U}{V_{x2}/U} = \frac{1.5}{.55} = \tan \alpha_2$$

$$\Rightarrow \underline{\underline{\alpha_2 = 69.9^\circ}}$$

(d)

Work output = Change of enthalpy (i.e. stagnation enthalpy)
(per unit mass) (per unit mass)

$$W_x = C_p \Delta T_o$$

$$\therefore \Delta T_o \text{ per stage} = \frac{W_x}{C_p} = \frac{1.5 U^2}{C_p} = \frac{1.5 \cdot 350^2}{1005} = 183 \text{ K}$$

ΔT_o is same for each stage

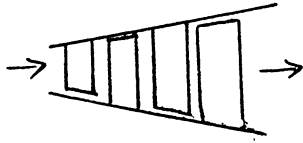
$$\therefore \Delta T_o \text{ turbine} = 366$$

$$T_{\text{in}} = 1450 \text{ K} \Rightarrow \underline{\underline{T_{\text{out}} = 1084 \text{ K}}}$$

$$\eta_{\text{is}} = \frac{T_{\text{in}} - T_{\text{out}}}{T_{\text{in}} - T_{\text{out}}^{\text{is}}} \Rightarrow \eta_{\text{is}} = \frac{1 - \frac{T_{\text{out}}}{T_{\text{in}}}}{1 - \left(\frac{T_{\text{out}}}{T_{\text{in}}}\right)^{\frac{\gamma-1}{\gamma}}} \quad \gamma = \text{pressure ratio}$$

$$\therefore \frac{1}{\gamma} = .316 \Rightarrow \underline{\underline{\gamma = 3.16}}$$

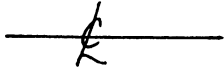
(e)



mass flow in = mass flow out

$$\text{i.e. } \rho_{in} V_{in} A_{in} = \rho_{out} V_{out} A_{out}$$

Axial velocity constant $\Rightarrow \frac{A_{out}}{A_{in}} = \frac{\rho_{in}}{\rho_{out}}$



$$\frac{\rho_{in}}{\rho_{out}} = \frac{\rho_{in}}{T_{in}} \frac{T_{out}}{\rho_{out}} \approx \frac{\rho_{in}}{\rho_{out}} \frac{T_{out}}{T_{in}} \text{ using stagnation values}$$

$$\therefore \frac{\rho_{in}}{\rho_{out}} = \frac{A_{out}}{A_{in}} = \frac{3.161084}{1450} \approx 2.4$$

9(a) For high power cycles typical of high bypass ratio engines T_{04} should be as high as possible. A typical cruise value might be 1450 K. In addition, high power settings - takeoff, top-of-climb, involve temperatures higher than this. Creep is a real problem and T_{04} is chosen as a compromise between high efficiency and adequate turbine life.

To achieve high T_{04} , turbine blades are cooled, using air bled from the compressor, both internally and by ejecting this air through holes in the turbine blade to form a layer of cool air surrounding the blade. Better life is obtained using cast blades which are directionally-solidified or single crystal.

$$(b) \quad p_{02} = p_a \left(1 + \frac{\gamma-1}{2} M^2\right)^{\frac{\gamma}{\gamma-1}} ; \quad p_{02} = 47.5, \quad p_a = 28.7 \Rightarrow \underline{M = .88}$$

$$T_{02} = \left(\frac{p_{02}}{p_a}\right)^{\frac{\gamma-1}{\gamma}} T_a, \quad T_a = 225 \Rightarrow \underline{T_{02} = 259.8 \text{ K}}$$

$$(c) \quad T_{023} = 300, \quad p_{023} = 75, \quad \dot{m} = 100 \text{ kg/s}$$

$$\frac{p_{03}}{p_{023}} = 25, \quad \frac{p_{04}}{p_{045}} = 5.5, \quad \eta_c = .88, \quad T_{04} = 1450.$$

$$(i) \quad \eta_c = \frac{\left(\frac{p_{03}}{p_{023}}\right)^{\frac{\gamma-1}{\gamma}} - 1}{\frac{T_{03}}{T_{023}} - 1} \Rightarrow \frac{T_{03}}{T_{023}} = 1 + \frac{1}{\eta_c} (25^{\frac{\gamma-1}{\gamma}} - 1) = 2.714$$

(& so $T_{03} = 814.3$)

Compressor $\Delta T = 514.3$, Turbine work = Compressor work
 \Rightarrow Turbine $\Delta T = \underline{514.3}$ ($\Rightarrow T_{045} = 935.7$)

$$\eta_T = \frac{T_{04} - T_{045}}{T_{04} - T_{045}^{IS}} = \frac{1 - \frac{935.7}{1450}}{1 - \left(\frac{1}{5.5}\right)^{\frac{\gamma-1}{\gamma}}} = \underline{92\%}$$

$$(ii) \quad \dot{m}_f \text{ LCV} = \dot{m}_{\text{air}} C_p (T_{04} - T_{03}) \Rightarrow \dot{m}_f = \frac{100 \times 1005 (1450 - 814.3)}{43 \times 10^6}$$

i.e. $\underline{\dot{m}_f = 1.49 \text{ kg/s}}$

(d) \dot{m}_f is seen by cycle as an energy flux \dot{m}_f LCV (and not as a mass flow).

Assuming choked nozzles, etc, suitable variables with which to non-dimensionalise \dot{m}_f LCV will be those appropriate to the flow at inlet to the engine.

Since energy flux = power = force \times velocity
 & force $\sim p_{02} D^2$, velocity $\sim \sqrt{\gamma R T_{02}}$

\therefore In non-dimensional terms

$$\frac{\dot{m}_f \text{ LCV}}{p_{02} D^2 \sqrt{\gamma R T_{02}}} = f_n(\text{op. point})$$

For the same operating point $\dot{m}_f \propto p_{02} \sqrt{T_{02}}$

For ground test $T_{02} = T_a$ $p_{02} = p_a$ (288K & 100 kPa)

$$\begin{aligned} \therefore \dot{m}_f &= \dot{m}_f^{\text{cruise}} \frac{p_a}{p_{02}^{\text{cruise}}} \sqrt{\frac{T_a}{T_{02}^{\text{cruise}}}} = 1.49 \frac{100}{47.5} \sqrt{\frac{288}{259.8}} \\ &= \underline{\underline{3.3 \text{ kg/s}}} \end{aligned}$$

10 Some points are as follows: -

H₂O no impact ∴ < natural emissions, condensing vapour

no control (generally)

CO_x CO₂ is a greenhouse gas, emissions may contribute to global warming, ~ 7% of natural emissions

- control by reducing combustion (turn down central heating, alternative energy sources, higher efficiency) or dark for gas

CO poison

- control by better design of combustion systems (mixing, lean burn), or catalytic oxidation

NO_x photochemical smog, acid rain, only ~ 5% of natural emissions

- control: non-catalytic reduction, selective catalytic reduction, design burner to reduce residence time at high temperature

SO_x

>> natural emissions. Acid rain

- deforestation (maybe), acidification of lakes (maybe), damage to buildings

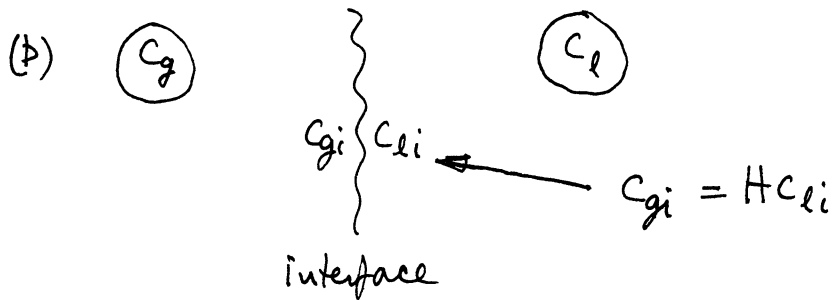
- control: switch fuel, flue gas desulfurisation, limestone in fluidised combustor, wash fuel

$$N_A = (N_A + N_B) x_A - D_{AB} \frac{dc_A}{dz}$$

$$N_B = (N_A + N_B) x_B - D_{BA} \frac{dc_B}{dz}$$

$$N_A + N_B = 0 \rightarrow D_{AB} \frac{dc_A}{dz} + D_{BA} \frac{dc_B}{dz} = 0$$

$$C_A + C_B = \text{uniform} \rightarrow \frac{dc_A}{dz} + \frac{dc_B}{dz} = 0 \rightarrow \underline{D_{AB} = D_{BA}}$$



$$N = k_g (C_g - C_{gi}) = k_l (C_{li} - C_l) = k_g (C_g - H C_l)$$

$$\left. \begin{aligned} \frac{N}{k_g} &= C_g - C_{gi} \\ \frac{H N}{k_l} &= H C_{li} - H C_l \end{aligned} \right\} N \left(\frac{1}{k_g} + \frac{H}{k_l} \right) = C_g - H C_l$$

$$\rightarrow \underline{\frac{1}{k_g} = \frac{1}{k_g} + \frac{H}{k_l}}$$

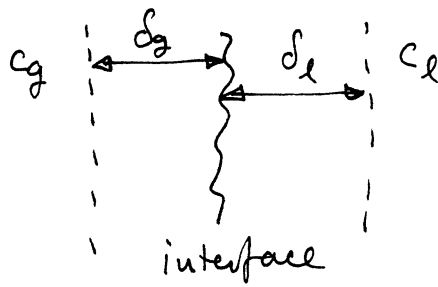
suspect: large H liquid side control

 small H gas side control

strictly: $\frac{H}{k_l} \gg \frac{1}{k_g}$ liquid side control

$\frac{H}{k_l} \ll \frac{1}{k_g}$ gas side control

(c) Whitman



$$N = k_g (c_g - c_{gi}) = D_g \frac{c_g - c_{gi}}{\delta_g}$$

$$\text{so } \underline{k_g = \frac{D_g}{\delta_g}}, \text{ and similarly } \underline{k_l = \frac{D_l}{\delta_l}}$$

$$\text{NH}_3 \quad k_g = K_g = 4.0 \times 10^{-2} \text{ m/s}$$

$$\delta_g = \frac{D_g}{k_g} = \frac{2.36 \times 10^{-5}}{4 \times 10^{-2}} = 5.9 \times 10^{-4} \text{ m}$$

$$\text{CO}_2 \quad k_l = K_l = H K_g = 1.05 \times 1.9 \times 10^{-4}$$

$$= 1.995 \times 10^{-4} \text{ m/s}$$

$$\delta_l = \frac{D_l}{k_l} = \frac{1.8 \times 10^{-9}}{1.995 \times 10^{-4}} = 9.0 \times 10^{-6} \text{ m}$$

$$\text{SO}_2 \quad \frac{1}{K_g} = \frac{\delta_g}{D_g} + \frac{\delta_l}{D_l} = \frac{5.9 \times 10^{-4}}{1.46 \times 10^{-5}} + \frac{9.03 \times 10^{-3} \times 9.0 \times 10^{-6}}{1.9 \times 10^{-9}}$$

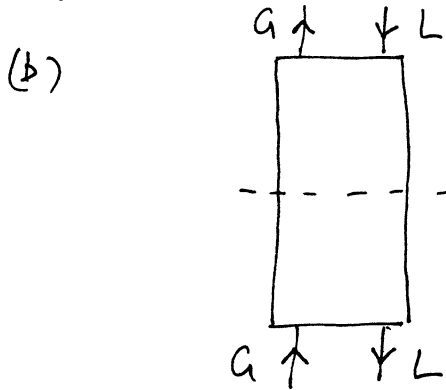
$$= 40.4 + 42.8 = 83.2$$

↑ ↑
 └────────────────┘ resistance from both
 sides

$$\underline{K_g = 1.2 \times 10^{-2} \text{ m/s}}$$

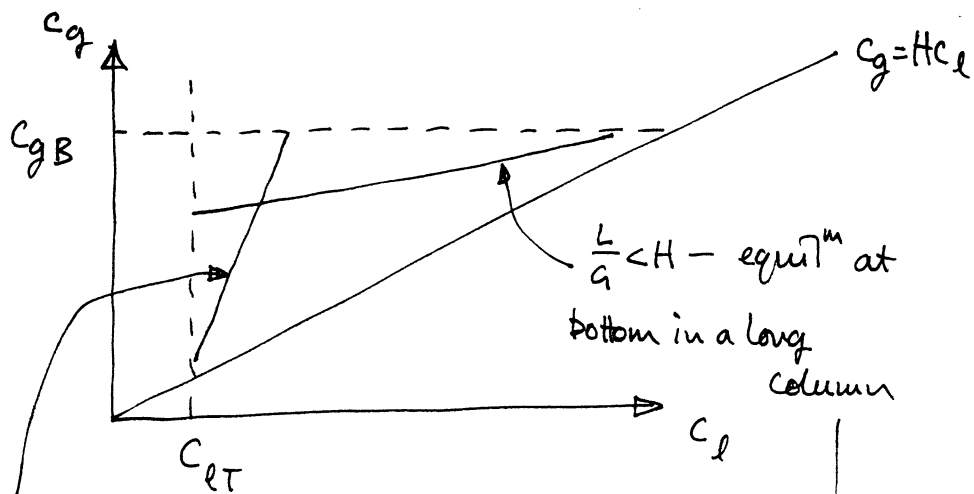
? Check that the ignored terms for NH_3 and CO_2 are negligible (they are)?

12 (a) Vertical cylindrical column is packed with solids eg raching rings to increase interfacial area. Gas passes up, liquid down, mass transfer from one stream to the other. Gas impedes the flow of liquid (loading), eventually, at high enough gas flowrate, giving flooding.



$$Gc_g + Lc_{lB} = Lc_l + Gc_{gB}$$

$$c_g = \frac{L}{G} c_l + c_{gB} - \frac{L}{G} c_{lB}$$



$\frac{L}{G} > H$ - equil^m at top in a long column,
 - then $c_{lT} = 0 \rightarrow c_{gT} \approx 0$, and
 all pollutant removed

so $c_{lB} \approx \frac{1}{H} c_{gB}$

$$c_{gT} = \frac{L}{G} c_{lT} + c_{gB} - \frac{L}{G} c_{lB}$$

$$\approx \underset{\substack{\uparrow \\ \text{maybe}}}{0} + c_{gB} - \frac{L}{GH} c_{gB} = \left(1 - \frac{L}{GH}\right) c_{gB}$$

$$(c) \quad \frac{HG}{L} = \frac{9.03 \times 10^{-3} \times 1.1}{0.012} = 0.828 < 1 \quad \text{can remove 'all' pollutant if column long enough}$$

$$G(c_{gB} - c_{gT}) = K_g A \left\{ \frac{(c_{gB} - Hc_{lB}) - c_{gT}}{\ln[(c_{gB} - Hc_{lB}) / c_{gT}]} \right\} \quad \left. \vphantom{\frac{(c_{gB} - Hc_{lB}) - c_{gT}}{\ln[(c_{gB} - Hc_{lB}) / c_{gT}]}} \right\}_{c_{lT}=0}$$

$$G(c_{gB} - c_{gT}) = L c_{lB}$$

$$\Rightarrow c_{lB} = \frac{G}{L} (c_{gB} - c_{gT})$$

$$G(c_{gB} - c_{gT}) = K_g A \frac{(1 - \frac{HG}{L})(c_{gB} - c_{gT})}{\ln \left[\frac{(1 - \frac{HG}{L})c_{gB} - \frac{HG}{L}c_{gT}}{c_{gT}} \right]}$$

$$\ln \left[\left(1 - \frac{HG}{L}\right) \frac{c_{gB}}{c_{gT}} + \frac{HG}{L} \right] = \frac{K_g A}{G} \left(1 - \frac{HG}{L}\right)$$

$$A = \frac{G}{K_g \left(1 - \frac{HG}{L}\right)} \ln \left[\left(1 - \frac{HG}{L}\right) \frac{c_{gB}}{c_{gT}} + \frac{HG}{L} \right]$$

$$= 712.1 \text{ m}^2 = \frac{\pi d^2}{4} a h$$

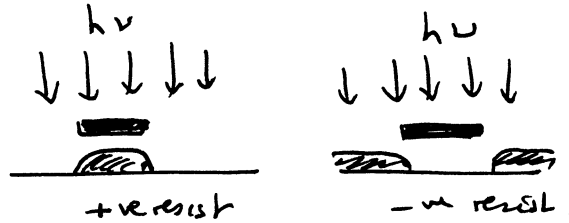
$$h = \frac{4 \times 712.1}{\pi \times 1.2^2 \times 194} = \underline{\underline{3.2 \text{ m}}}$$

To get the same reduction in SO_2 at higher G , you need to keep G/L constant ... but then you run the risk of flooding. At fixed L , you will not get so good SO_2 removal.

Cribs Paper 8 - Section E

13 (a) and (b)

Positive and Negative Resist



Photoresist is a radiation sensitive compound. They can be classified as positive or negative, depending on how they respond to radiation.

For positive resists the exposed regions become more soluble and thus more easily removed in the development process. The net result is that patterns formed in the positive resist are the same as those on the mask. For negative resists the exposed regions become less soluble and the patterns formed in the negative resist are the reverse of the mask patterns (see above figures).

Depth of field - the region over which the image remains in focus. If variation in height of the surface is > depth of focus then get degradation of image.

Resolution - minimum feature size that can be reproduced repeatedly.

(i) UV optical exposure is the standard industrial process. High throughput and whole wafer exposure is possible.

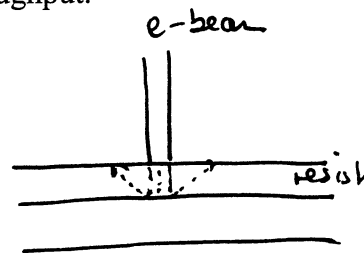
The spectrum from a typical mercury UV lamp shows the strongest emission at $\lambda \approx 365$ nm. The practical limit in UV lithography however is diffraction at the edges of the mask used for contact printing.

eg 0.5 microns for 365 nm radiation

(ii) E-beam exposure is a well developed but expensive process. Serial exposure using an e-beam focused to a small spot of typical diameter 0.1 micron. The desired pattern is traced using raster or vector scan. Low throughput.

eg For 30 keV $\lambda = \frac{h}{(2mE)^{\frac{1}{2}}}$

thus $\lambda = 7 \times 10^{-12} \text{ m}$



The practical limit in e-beam lithography is NOT the diffraction limit to the spot because the wavelength is so small but the scattering of secondary electrons which expose the resist when they are reflected as shown in the diagram.

(iii) X-ray lithography is new technology which can expose whole wafers and therefore is capable of high throughput BUT it is in its infancy and still numerous problems to be overcome.

The practical limit in X-ray lithography may be ≈ 0.1 micron determined by the range of secondary electrons produced in the resist but at present the main problem is how to make an effective, stable mask.

$$\text{Resolution} = \left(\frac{0.5\lambda}{NA} \right) \approx 0.4 \mu\text{m}$$

$$\text{Depth of focus} = \frac{\lambda}{(NA)^2} \approx 2.77 \mu\text{m}$$

Resolution not good enough therefore use shorter wavelength light if photolithography to be used or X-ray or

Depth of focus also not good enough therefore would use step and repeat or other lithography process.

14. a) Electron diffraction by crystal. Line spectrum of optical emission by atoms - Bohr atom.

b) $E = 0.1.1.6.10^{-19} \text{ J}$

$$E = (h/\lambda)^2/2m$$

$$\lambda = h/\sqrt{2m * E} = 6.6. 10^{-34}/\sqrt{(2.0.9.10^{-30}.0.1.0.1.1.6.10^{-19})} = 1.3.10^{-8} \text{ m}$$

c) classical - reflection by the barrier
 quantum - finite probability of tunnelling, because the KE is positive again on the RHS of barrier.

d) LHS $\psi = A \exp(jkx) + B \exp(-jkx)$
 (incoming + reflected wave)

inside barrier
 $\psi = C \exp(-k'x) + D \exp(k'x)$
 (right going evanescent wave + reflected evanescent wave)

RHS outside barrier
 $\psi = F \exp(jkx)$

A, B, C, D, F all real constants to be found by setting continuity conditions ψ and $d\psi/dx$ being continuous across each junction barrier.

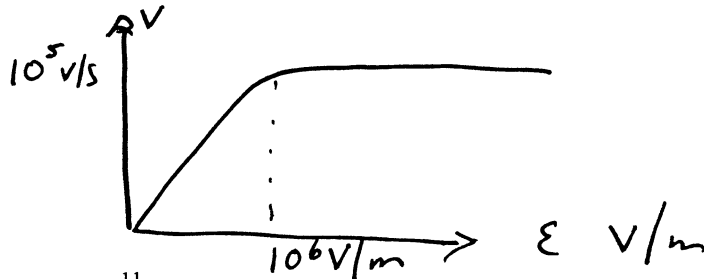
$$k = \sqrt{(2m^*E)/\hbar}, k' = \sqrt{(2m^*[V-E])/\hbar}$$

$$\text{Tunnelling probability} = F^2/A^2$$

Probability drops by factor e^2 when $Lk' = 1$. $k' \approx 10^8 \text{ m}^{-1}$, from above. So $L = 10^{-8} \text{ m}$.

15. $\mu = 0.1 \text{ m}^2/\text{V}\cdot\text{s}$
 $v_{\text{max}} = 10^5 \text{ m/s}$

a) Sketch shows linear increase of velocity with field E , until $v = 10^5 \text{ m/s}$ and $E = 10^6 \text{ V/m}$, above which v is flat, at v_{max} .



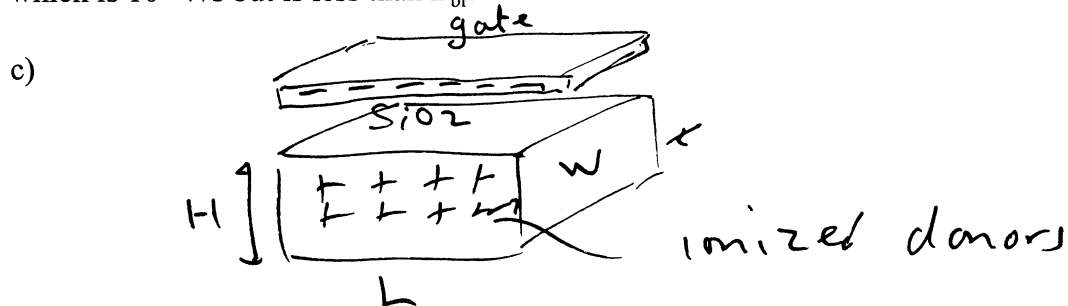
b) $\tau = 10^{-11} \text{ s}$

$$V = 2V$$

Assume velocity is in the scattering limited regime for the moment, then check this.

$$L = v_{\text{max}} \cdot \tau = 10^5 \cdot 10^{-11} = 10^{-6} \text{ m}$$

$E = V/L = 2/10^{-6} = 2 \times 10^6 \text{ V/s}$. This exceeds the threshold for scattering limited regime which is 10^6 V/s but is less than E_{br} .



Enclosed charge to a depth x is $Q = q \cdot A \cdot x$

electric displacement at bottom surface, $D = 0$

By Gauss' theorem, electric displacement at top surface is

$$A \cdot D = Q = q \cdot A \cdot x$$

$$\text{so } E = D / \epsilon = \frac{qAx}{A\epsilon} = \frac{qx}{\epsilon}$$

$$dV = -E \cdot dx$$

$$V = \int E \cdot dx = - \int (q \cdot x / \epsilon) \cdot dx$$

$$= q \cdot H^2 / 2\epsilon$$

capacitance per unit area $C' = Q / (V \cdot A)$; $Q = q \cdot A \cdot H$;

$$\text{so } C' = q \cdot A \cdot H / (q \cdot A \cdot H \cdot H / 2\epsilon) = 2\epsilon / H$$

$$C = L \cdot W \cdot C'$$

$L = 10^{-6} \text{m}$, $W/L = 10$,
so $W = 10^{-5} \text{m}$ and $C = 10^{-6} \cdot 10^{-5} \cdot 2 \cdot 10^{-10} / 10^{-6} = 2 \cdot 10^{-15} \text{F}$.

d) $R = L\rho / (W \cdot H)$, $\rho = 0.01 \text{ ohms.m}$

$$\text{so } R = 10^{-6} \cdot 10^{-2} / (10^{-5} \cdot 10^{-6}) = 10^3 \text{ ohms}$$

$$\text{and } \tau' = RC = 2 \cdot 10^{-15} \cdot 10^3 = 2 \text{ps.}$$

Thus the simplest RC time constant delay due to an internal device capacitance is less than the transit time, and the transit time dominates.

1B Paper 8 1997 - Exam Cribs

Question 16

1st Part: Description of hashing is bookwork. It is assumed that information is to be stored and retrieved based on a 'key' (eg a number, a name, etc). The basic principle is then to use a one-many mapping (called a hash function) to partition the space of these search keys into a number of subsets. Within each subset, the data is typically stored in a conventional list or array structure. Given a search key, an item is found by first locating the subset using the hash function and then making an exhaustive search for the specific item required (eg by a simple linear search). If each subset is small, then a linear search is still quite quick. A good answer might also mention the distinction between closed and open hashing schemes.

The hash function needs to be chosen such that it evenly distributes the data items amongst the subsets and that it is cheap to compute.

[5 marks /20]

The posed storage problem appears to be different to the examples given in lectures in that there are two lookup keys: SA and DA. Whilst this would severely complicate some schemes, using hashing, it only affects the hash function. It does not necessitate any modifications to the basic data structure.

2nd Part: Data Structure [6 marks /20]

CONST hashmax = ???; { a large prime number }

TYPE Link = ^Entry;

```
Entry = RECORD { linked list of equivalent hash entries }
    sa,da: Address;
    sc: SecurityCode;
    next: Link
END;
```

```
HashIndex = 1..hashmax;
HashTable = ARRAY [HashIndex] OF Link; { all NIL initially }
```

VAR htab: HashTable;

3rd Part: Function [6 marks /20]

```
FUNCTION LookUp(sa: Address; da: Address) : SecurityCode;
    VAR h:HashIndex;
        l:Link;
        Boolean found;
```



```
BEGIN
  h := (sa * da) MOD hashmax + 1; { or similar }
  l := htab[h]; found := FALSE;
  WHILE NOT found AND (l <> NIL) DO BEGIN
    found := (sa = l^.sa) AND (da = l^.da);
    IF NOT found l := l^.next
  END;
  IF found THEN
    LookUp := l^.sc
  ELSE
    LookUp := 0
END;
```

4th Part: Comments [3 marks /20]

The effects of correlation in SA and DA would depend on the design of the hash function but in general, it will make the allocation of SA/DA pairs to hash lists less evenly distributed thereby increasing retrieval time. Note however that the "mod primenumber" operation will minimise the effect of such correlations.

Question 17

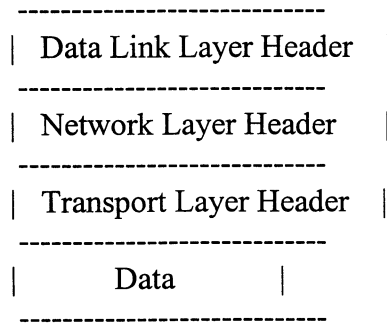
a) In a connection-oriented protocol, a connection is first established. This allows the two communicating systems to retain per-connection state information about one another thereby enabling error correction, ie requests for retransmission of lost or corrupted packets identified by a unique sequence number; and data flow control by indicating how many additional packets or bytes may be sent.

In a connectionless protocol, each packet is an independent event and no state information is retained about the other end.

The advantage of connection-oriented protocols is that they can provide flow control and error handling. The advantage of connectionless protocols is simplicity: the lack of the need to set up a connection and for either end to retain any state information. In the case of lower layer protocols such state information would need to be maintained by all the gateways along the route between two end systems. Also Broadcast schemes are inherently connectionless and are difficult and wasteful to implement in a connection-oriented framework. [6 marks /20]

b) A transport layer packet consists of the header for this layer, preceded by those for the lower layers and followed by the data being transferred. This data may in turn contain header information for higher layers and in the same way, the transport layer header plus the data is viewed by the layer below (the network layer) as simply data. The overall packet thus looks like:

| Physical Layer Header |



The physical layer header will typically at most amount to a unique pattern of ones and zeros used to synchronise the local oscillators at sender and receiver and to identify the start and end of packets. There may also be a checksum for low level error detection.

The main essential fields in the next three layers are those associated with addressing and in ISO/OSI reference model terms identify the Service Access Point (SAP) of the interface to the layer above. There may also typically be fields giving the length of the packet and/or header and a checksum for the header and/or complete packet. Looking at the address fields in more detail:

Data Link Layer - the source and destination MAC (or "hardware") address plus information to identify the Network Layer protocol.

Network Layer - the source and destination network address (used to route packets from end system to end system via gateways connecting different physical networks) plus information to identify the Transport Layer protocol.

Transport Layer - strictly the address information here identifies the Session Layer SAP but in practice the normally bears a one-to-one correspondance with the application and thus identifies a particular instance of a particular application, eg a particular login session. [7 marks /20]

c) The main function needed of the network analyser is similar to that of the traffic command described in lectures except that grouping other than by MAC source and MAC destination address will be required. The specific information needed in each case is:

i) very like that in the traffic command except that only the source address is of interest in grouping the packets. The hardware must be set to receive all packets.

ii) this is a similar problem except that we no longer require the hardware to receive every packet; indeed it may be used in its normal operational mode as it will receive all broadcasts.

iii) we now need first to use the technique in (i) to identify which computer is causing the problem and then, as a second separate investigation, group each packet originated by it by unique transpor layer address (ie unique combination of network layer and

transport layer header address information). It will not be possible with normal network interface hardware to do the filtering required in this second stage (this provides filtering on destination rather than source MAC address) so all packets will have to be received and then filtered using software. [7 marks /20]

Question 18

In polled mode, the cpu repeatedly tests a status bit to determine when it can write or read data. In interrupt-mode, the cpu can proceed with other tasks. When the device is ready for input or output, an interrupt is generated. This causes the current cpu state to be saved and an interrupt service routine executed. The latter normally terminates by restoring the cpu state to resume the interrupted program. Polling wastes cpu but is also fast since it avoids the need to save and restore state information. [3 marks /20]

The device registers are simply described by Pascal packed records

TYPE Resolution = 0..7;

```
StatusReg = PACKED RECORD
    dataReady: Boolean;
    errCode: 0..7;
    unused: 0..15
END;
```

```
ControlReg= PACKED RECORD
    reset: Boolean;
    intEnable: Boolean;
    xRes: 0..7;
    yRes: 0..7
END;
```

```
VAR data : Byte @ 0FE000H;
    stat : StatusReg @ 0FE001H;
    cont : ControlReg @ 0FE001H;
```

[6 marks /20]

The required interface module would be very simple since all it needs is an initialisation routine, a function to indicate how many screen presses are available and a procedure to get the next screen press in the queue, ie

```
MODULE TouchScreen;
EXPORT
    TYPE Position = RECORD
        x,y:Byte
    END;
```

```
PROCEDURE InitTS(xres, yres: Resolution);  
FUNCTION NumTouches: integer;  
PROCEDURE GetPosition(VAR pos: Position);  
END;
```

[4 marks /20]

The interrupt service routine simply reads the data register twice and stores a position in the buffer. Only real problem is what to do about overflow.

```
CONST buffersize = 100;  
VAR buffer : ARRAY[1..100]OF Position;  
    used,next : Integer;  
  
INTERRUPT PROCEDURE TSInt@0200H;  
    VAR p:Position;  
BEGIN  
    IF stat.errCode <> 0 THEN error ...;  
    p.x = data; p.y = data;  
    IF (used = buffersize) THEN error ...;  
    buffer[next] = p;  
    next = next MOD buffersize + 1;  
    used = used + 1  
END;
```

[5 marks /20]

In a multiprocess environment, an event would be defined for the buffer not empty condition. When a process calls GetPosition and NumTouches is zero, the process is sent to sleep. TSInt then sends the buffer not empty event every time it is executed in order to wakeup any sleeping processes.

[2 marks /20]

1997 Part IB Paper 8 ~~Section A~~
Solutions

1. (a) 1820 kg/m^3 , 0.99. (b) 39kPa, 133kPa (d) (i) 44.9° , 35.7°
(ii) 60kPa
(e) 10kPa, -7kPa
3. (a) $T = 360 \text{ kN}$ (b) 316mm
 ~~\neq~~ $M_{m..} = 596.5 \text{ kNm}$
4. (b) $\sim 0.065 \text{ kym}$ (d) $\sim 12 \text{ Hz}$ for a 1% speed fluctuation
5. (b) low-pass $\sim 72 \text{ Hz}$
high-pass $\sim 3.4 \text{ Hz}$
(c) range $\sim 0.5 - 2.5 \text{ kNm}^{-1}$ depending on choice of c and 3db frequency
6. -
7. (c) About 40:1
8. (b) 61.2° (c) 69.9° (d) 1084 K, 3.16 (e) 2.4
9. (b) .88, 259.8K (c) (i) 514.3 & 92% (ii) 1.49 kg/s (d) 3.3 kg/s
11. (c) $K_g = \frac{Dg}{\delta g}$, $K_e = \frac{Dl}{\delta l}$
(d) $9 \times 10^{-6} \text{ m}$, $1.2 \times 10^{-2} \text{ ms}^{-1}$
12. (c) $h = 3.2 \text{ m}$
13. Resolution $\sim 0.4 \mu \text{ m}$, Depth of focus $\sim 2.77 \mu \text{ m}$
14. $\approx 10^{-8} \text{ m}$
15. $L = 1 \mu$, $C = 2 \times 10^{-15} \text{ F}$, $R = 10^3 \Omega$