



## **Verification Manual**

LUSAS Version 21.0 : Issue 1

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## **Example 1.1.1**

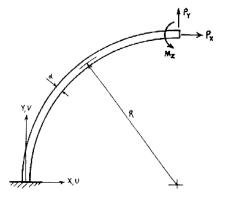
# **Linear Static Analysis Of A Curved Cantilever (using 4 elements)**

### **Keywords**

**Curved Beam** 

### **Description**

Determine the tip displacements for a quarter circle cantilever with the dimensions, properties and subject to the loading as given below. Neglect the effect of shear deformations but include the effect of axial deformations on the displacements.



### **Discretisation**

Model with four equal length curved BM3 beam elements which, being thin beam elements, explicitly exclude the effects of shear deformations.

### **Geometric properties**

Radius, mm	Cross-section breadth (b), mm	Cross-section depth (d), mm
200	10	10

## **Material properties**

Young's modulus	$200 \text{ x} 10^3 \text{ N/mm}^2$
	200 X10 1 WIIIII

### Loading

Point load at tip in X direction (P <sub>x</sub> ), N	Point load at tip in Y direction (P <sub>Y</sub> ), N	Concentrated moment at tip (M <sub>z</sub> ), Nm
200	200	20

### Theory

The following equations for the tip displacements may be derived using the unit load method.

	Point load at tip in X direction, P <sub>x</sub>	Point load at tip in Y direction, P <sub>Y</sub>	Concentrated moment at tip, M <sub>z</sub>
u	$\frac{P_x R^3 (3-8)}{4EI} + \frac{P_x R}{4EA}$	$-\frac{P_yR^3}{2EI} + \frac{P_yR}{2EA}$	$-\frac{M_zR^2(\pi-2)}{2EI}$
v	$-\frac{P_x R^3}{2EI} + \frac{P_x R}{2EA}$	$\pi \Bigg( \frac{P_y R^3}{4EI} + \frac{P_y R}{4EA} \Bigg)$	$\frac{\mathrm{M_zR^2}}{\mathrm{EI}}$
θ	$-\frac{P_xR^2\big(\pi\!-\!2\big)}{2\mathrm{EI}}$	$\frac{P_y R^2}{EI}$	$\pi \frac{M_z R}{2EI}$

## Comparison

Load	Result source	u	v	θ
Point load at tip in X direction, P <sub>X</sub>	LUSAS	3.42023	-4.79880	-0.0273976
	Theory	3.42104	-4.799	-0.0273982
Point load at tip in Y direction, P <sub>Y</sub>	LUSAS	-4.79880	7.54071	0.048
	Theory	-4.799	7.54139	0.048

Load	Result source	u	٧	θ
Concentrated	LUSAS	-2.73976	4.8	0.0376988
moment at tip, MZ	Theory	-2.73982	4.8	0.0376991

### References

. Timoshenko, S.P., and Gere, J.W., Mechanics of materials, Van Nostrand Reinhold, 1972.

### **Input data**

X01D11A.DAT

## **Example 1.1.2**

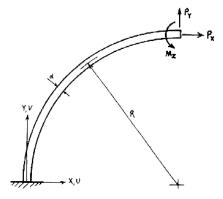
### **Linear Analysis Of A Curved Cantilever (using 16 Elements)**

### **Keywords**

**Curved Beam** 

### **Problem Description**

Determine the tip displacements for a quarter circle cantilever with the dimensions, properties and subject to the loading as given below. Neglect the effect of shear deformations but include the effect of axial deformations, on the displacements.



### **Discretisation**

Model with 16 equal length curved BEAM elements. Use a large shear area to ensure negligible shear deformations.

### **Geometric properties**

Radiu	s, mm	ross-section breadth (b), mm	Cross-section depth (d), mm
20	00	10	10

### **Material properties**

## Loading

Point load at tip in X direction (P <sub>x</sub> ), N	Point load at tip in Y direction (P <sub>Y</sub> ), N	Concentrated moment at tip (M <sub>z</sub> ), Nm
200	200	20

### **Theory**

The following equations for the tip displacements may be derived using the unit load method.

	Point load at tip in X direction, P <sub>x</sub>	Point load at tip in Y direction, P <sub>Y</sub>	Concentrated moment at tip, M <sub>z</sub>
u	$\frac{P_x R^3 (3-8)}{4EI} + \frac{P_x R}{4EA}$	$-\frac{P_yR^3}{2EI} + \frac{P_yR}{2EA}$	$-\frac{M_zR^2(\pi-2)}{2EI}$
v	$-\frac{P_x R^3}{2EI} + \frac{P_x R}{2EA}$	$\pi \Bigg( \frac{P_y R^3}{4 E I} + \frac{P_y R}{4 E A} \Bigg)$	$\frac{M_z R^2}{EI}$
θ	$-\frac{P_xR^2\left(\pi-2\right)}{2EI}$	$\frac{P_y R^2}{EI}$	$\pi \frac{M_z R}{2EI}$

## Comparison

Load	Result source	u	v	θ
Point load at tip in	LUSAS	3.42298	-4.79707	-0.0274258
X direction, P <sub>X</sub>	Theory	3.42104	-4.799	-0.0273982
Point load at tip in	LUSAS	-4.79707	7.52627	0.0479422
Y direction, P <sub>Y</sub>	Theory	-4.799	7.54139	0.048
Concentrated	LUSAS	-2.74258	4.79422	0.0376840
moment at tip, $M_Z$	Theory	-2.73982	4.8	0.0376991

### References

2. Timoshenko, S.P., and Gere, J.W., Mechanics of materials, Van Nostrand Reinhold, 1972.

### **Input data**

X01D12A.DAT

## **Example 1.1.3**

## **Linear Static Stress Analysis Of A Plane Frame Using Beam Elements**

### **Keywords**

Plane Frame

### **Problem Description**

To determine the bending moments and shear forces when the plane frame shown in Figure 1 is subjected to the following load cases;

- ☐ Three horizontal point loads of 25 KN (Figure 2).
- ☐ Three vertical point loads of 50, 60 and 70KN along with three udl loads of 10, 10 and 20 KN/m (Figure 3).
- ☐ A combination of load cases (1) and (2) with a scaling factor of 0.9 and 1.4 respectively (Figure 4).
- ☐ A combination of load cases (1) and (2) with a scaling factor of 1.4 (Figure 5).

#### **Finite Element Model**

The frame is made up of 11 BEAM elements with the element and nodal numbering detailed in Figure 6.

### **Geometric properties**

Members	Area (m²)	I (m⁴)	Shear area (m²)
1-6,10	0.1	0.00133	0.1
7,8	0.3	0.00399	0.3
9,11	0.2	0.00266	0.2

### **Material properties**

Young's modulus	$30 \text{ x} 10^6 \text{ kN/m}^2$
Poisson's ratio	0.3

### **Boundary conditions**

The left hand column (node 1) is fully restrained in both translation and rotation while the centre (node 2) and right hand columns (node 3) are only restrained in translation.

### **Theory**

A number of methods exist for the solution of plane frames, e.g. moment distribution and slope deflection methods of analysis, which are described in [1].

### **Modelling Hints**

The four load cases defined above may be specified in the LUSAS data file in the following way:

The horizontal and vertical load conditions (cases 1 and 2) are declared as two separate LOAD CASE's using CONCENTRATED LOAD (CL) and ELEMENT LOADS (ELDS) respectively. A number of combinations of the previously defined separate LOAD CASE's may be analysed by using the LOAD COMBINATION card. In addition the ENVELOPE facility may be used to compute the maximum loads of the combined load cases (i.e. worst conditions). The form in which the commands have been implemented is shown in the input data file listing.

#### References

1. Steel Designers' Manual (Constrado) Granada publishing 1983.

### Input data

X01D13A.DAT

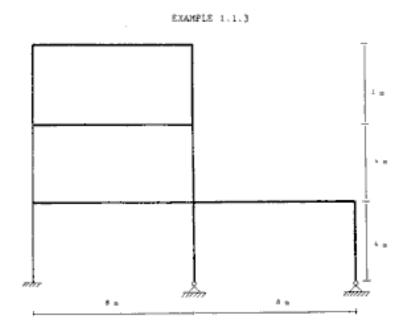
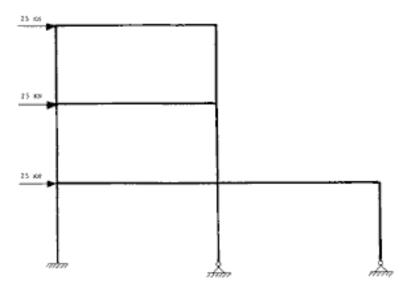


Figure 1. Dimensions of Frame



#### EXAMPLE 1.1.3

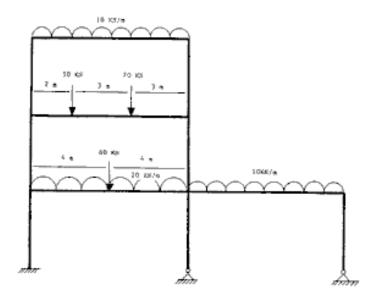
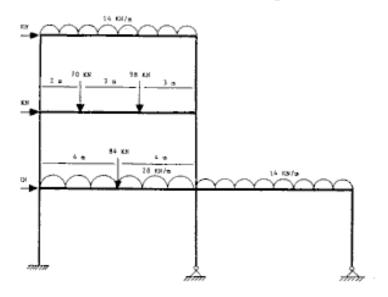


Figure 3. Load Case 2. Vertical loading condition



.gure 4. Load Case 3. A combination of Load Cases 1. and 2. with a scaling factor of 0.9 and 1.4 respectively

#### DXAMPLE 1.1.1.5

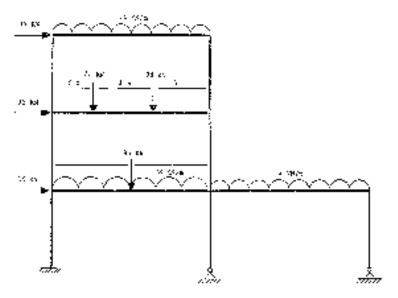


Figure 5. Spad Case 4. A combination of Load Cases 1. and 7. With a scaling factor of 1.4

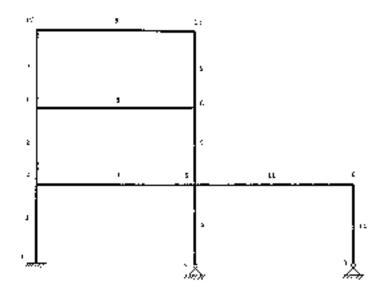


Figure 6. Finite element discretization of the structure

## Example 1.1.4

## **Temperature Dependent Properties Constant Strain Cantilever**

### **Keywords**

Constant Strain, Temperature Dependent Properties, Beam 3D

### **Problem Description**

A cantilever of length 10392mm, width 200mm and depth 100mm is subject to a variable temperature loading such that the product of the coefficient of thermal expansion and the temperature rise at any point is constant. Figures 1 and 2 show the geometry of the cantilever.

### **Discretisation**

The cantilever beam is modelled using two meshes, each of three elements. The finite element discretisation is shown in figure 3.

Mesh 1 - three BS4 elements

Mesh 2 - three BSX4 elements

### **Material properties**

The material properties assumed for the analysis are as follows:

Young's modulus  $= 200000 \text{ N/mm}^2$ 

Poisson's ratio = 0.3

Thermal expansivity (variable) \* Temperature

= Constant

= 2.4

### **Boundary conditions**

One end of the beam is fully restrained while the other is free.

### **Theory**

The structure is only restrained to prevent rigid body motions so that stress free, thermal expansion may occur. For load case 1, with constant temperature throughout the section, the strain at any point is constant and is the product of the coefficient of thermal expansion and the temperature at that point:-

$$Strain = \alpha * T = 2.4 \tag{0.0.1}$$

Where  $\alpha$  is the coefficient of thermal expansion and T is the temperature rise.

For load cases 2 and 3, a linear variation is used such that the top and bottom fibre temperature strains are 120 and -120 respectively. Therefore the resulting strain will be pure bending with a value of:

Strain = 
$$\frac{(120) - (-120)}{100} = 2.4$$
 (0.0.2)

### Comparison

A constant strain condition (of 2.4) is achieved for all the cases analysed.

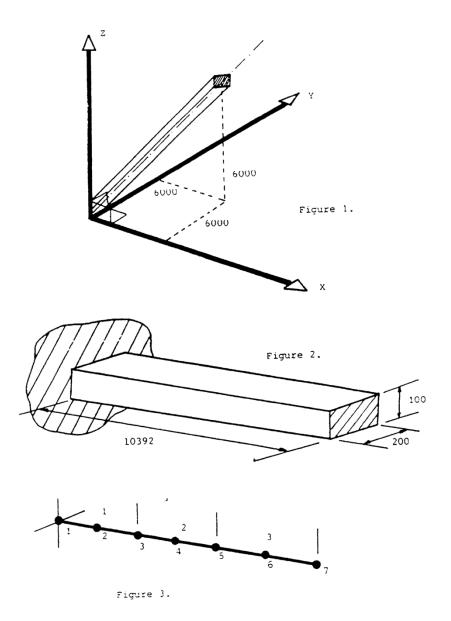
#### References

 Roark.R, Young.W 'Formulae for stress and strain: Fifth Edition', McGraw-Hill Publishing Company (1975).

### **Input Data**

X01D14A.DAT

X01D14B.DAT



## **Example 1.2.1**

# **Linear Elastic Static Fracture Analysis Of A Three-Point Bend Specimen**

### **Keywords**

Fracture, Stress Intensity

### **Problem Description**

In the specialised field of fracture mechanics a number of small scale tests are used extensively in determining the quality of a material. One such test is the three-point bend specimen. A typical arrangement is shown in Figure 1.

The test specimen is supported at positions A and B while a CONCENTRATED LOAD is applied at point C. The stress intensity factor may then be evaluated for the initial crack length  $a_0$ .

#### **Discretisation**

As the test piece and loading conditions are symmetric, only half the specimen need be analysed. The finite element discretisation is shown in Figure 2, consisting of 126 eight-noded plane strain membrane elements (QPN8) and two crack tip elements (QNK8).

### **Material properties**

Young's modulus, E	214x10 <sup>6</sup> Pa
Poisson's ratio, v	0.3

### **Loading condition**

A concentrated load (P) of 2000 N applied at position C.

### **Theory**

Griffith's criterion 
$$G = \frac{1}{t} \frac{\partial U}{\partial a}$$
 (0.0.3)

Irwin's relation: 
$$K_I^2 = \frac{GE}{\left(1 - \upsilon^2\right)}$$
 (0.0.4)

where

a Crack length

E Young's modulus

G Energy release rate

K<sub>I</sub> Stress intensity factor (mode I)

t Thickness

U Strain energy

v Poisson's ratio

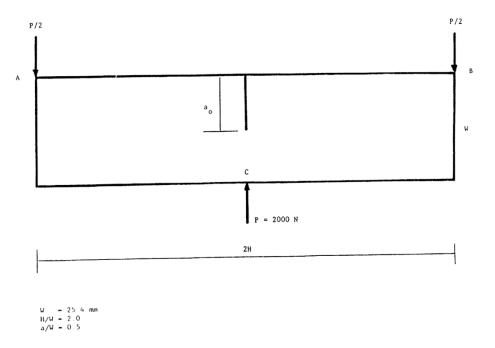


Figure 1. Three-point bend specimen.

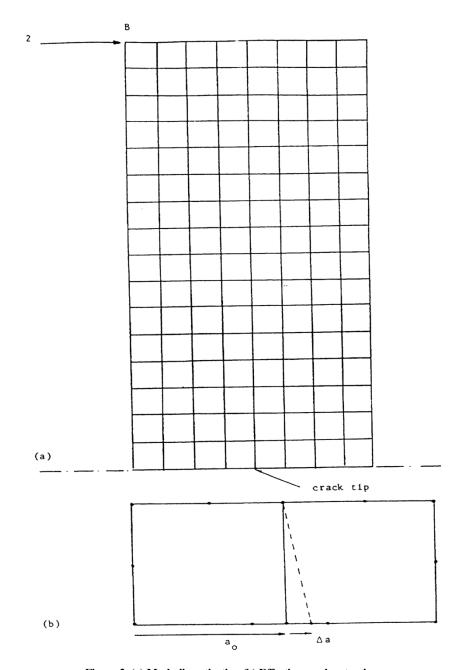


Figure 2. (a) Mesh discretisation (b) Effective crack extension.

### **Modelling Hints**

In this example the load is applied at position A in the form of a reaction P/2. There exist a number of methods for determining the value of the stress intensity factor, the method adopted here is the energy balance approach.

The user conducts a linear elastic static analysis with an initial crack length  $a_0$ . A second analysis may then be done after repositioning the crack tip to give a new crack length,  $a_0 + a$ . The change in energy may then be used to compute the value of  $K_I$ .

### **Solution Comparison**

(1) Crack length a

Displacement at the point of application of the reaction

$$u_B = 0.513712 \text{ mm}$$

Strain energy

$$U_1 = P/2 * u_B = 0.513712 \text{ Nm}$$

(2) Crack length  $a_0 + a$ 

Displacement at the point of application of the reaction

$$u_B = 0.553229 \text{ mm}$$

Strain energy

$$U_2 = P/2 * u_B = 0.553229 \text{ Nm}$$

Energy release rate

$$G = (0.553229 - 0.513712)/0.0005 = 79.036 \text{ Nm/m}^2$$

As symmetry has been considered this value of G must be multiplied by a factor of 2. Therefore:

$$G = 158.072 \text{ Nm/m}^2$$

This value of G, the energy release rate, may now be inserted into Irwin's relation (0.0.4) to give the stress intensity factor:

$$K_r^2 = 158.072 \times 214 \times 10^9 / 0.91 = 3.717 \times 10^{13}$$

$$K_I = 6.097 \text{ MN/m}^{3/2}$$

An alternative method to the energy balance approach is the displacement extrapolation technique [1]. The analytical expressions for the displacement variations along radial lines emanating from the crack tip, in terms of the stress intensity factor are as follows:

KI 
$$[(2k-1)\cos\theta/2 - \cos 3\theta/2] = 4\mu(2\Pi/r)1/2$$
 (u)

KI 
$$[(2k+1)\sin\theta/2 - \sin 3\theta/2] = 4\mu(2\Pi/r)1/2$$
 (v)

#### where:

- u x displacement component
- v y displacement component
- r radial distance from nodal point to crack tip.
  - shear modulus
- k 3-4y for plane strain
- $\theta$  angle between radial path chosen and the crack path ahead of the crack tip.

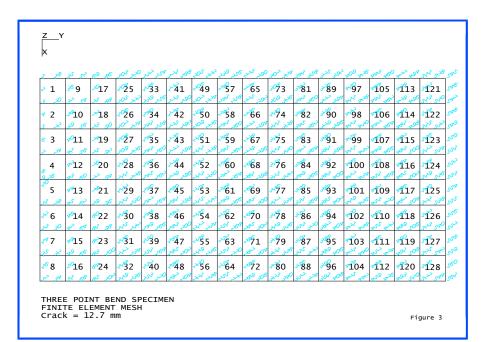
If we assume  $\theta = 90$ , then we are considering the radial path of nodes vertically from the crack tip i.e. nodes 9 to 553 (Figure 3). In this particular instant we shall only consider nodes 26, 43, 77, 11, 145,179, 213, 247, 281, 315, 349, 383, 417, 451, 485, 519, and 553. The corresponding nodal displacements in the x-direction have been substituted into the above equation and the values of K computed. The results have been plotted and are shown in Figure 7. By discarding points close to the crack tip the solutions can be extrapolated to r=0[1]. In this case the result is approximately that obtained by the energy balance method.

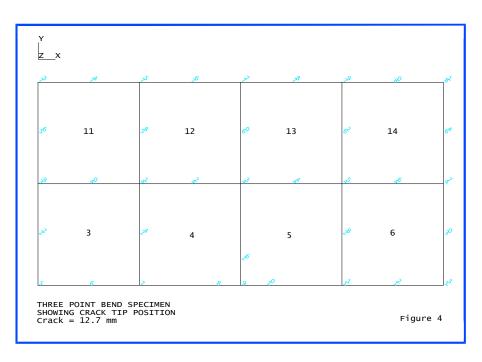
#### References

 Owen, D.R.J and Fawkes, A.J. Engineering Fracture Mechanics Numerical Methods and Applications. Pineridge Press Ltd. 1983. p43

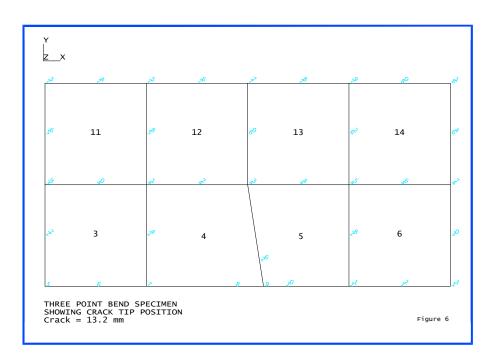
### **Input Data**

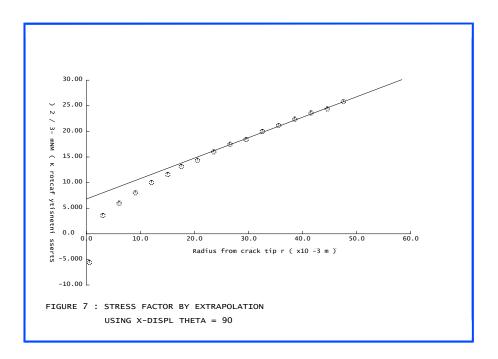
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<b>≠</b> 2	<b>%</b> 10	<b>≈</b> 18	<mark>₹%</mark> 6	<b>√</b> 34	<b>4</b> 2	<del>20</del> 5°0	<b>7</b> 58	<del>≥</del> 66	<del>₹7</del> 4	<b>₹8</b> 2	<b>~</b> 90	<b>√</b> 98	106		122
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5	<b>1</b> 3	<b>~</b> 21	<b>₹</b> 29	<b>₹</b> 37	<b>₹</b> 45	₹53	<b>₹6</b> 1	<del>~6</del> 9	<del>₹9</del> 7	<del>3</del> 585	<b>×</b> 93	101	109		125
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₹ <sup>6</sup> 8	<b>∽</b> 16	<b>€</b> 24	₹32	<b>₹</b> 40	<b>₹</b> 48	₹6	<b>≈</b> 64	<mark>≈</mark> 72	×80	<del>3∕8</del> 8	<sup>36</sup> 96	104	<b>∕1</b> 12	120	<b>1</b> 28
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## **Example 1.2.3**

## **Connecting Incompatible Models Using Constraint Equations**

### **Keywords**

**Incompatible Models, Curvilinear Coordinates, Shape Functions** 

### **Problem Description**

A study of the localised stresses at the root and tip of a cantilever beam, details of which are shown in Figure 1, is to be carried out. To save on computer resources, the central portion of the cantilever is to be modelled with a coarse mesh while finer meshes are to be used at the tip and the root. Derive the constraint equations relating the nodal variables at the interfaces between the fine meshes at the root and tip and the coarse mesh of the central portion. Analyse the cantilever when subject to a shear load of 135 MN at the tip. Compare the results for the tip displacements and the stress distributions for each Cartesian stress component with the results obtained using a fine mesh over the whole cantilever.

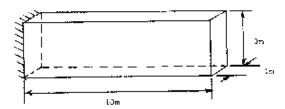


Figure 1 - Problem geometry and material properties

### **Discretisation**

As the problem is essentially two dimensional, a finite element model using plane membrane elements is appropriate. The higher order, 8-noded plane membrane element, QPM8, is used and the mesh adopted for the analysis is shown in Figure 2. The full, fine mesh used for comparison purposes is shown in Figure 3.

### **Material properties**

Young's modulus 200.0x10 <sup>9</sup> Pa
--

Poisson's ratio 0.3

### **Theory**

The variables at nodes 22, 39, 73, 90, 30, 47, 81 and 98 are required to be constrained to the edges of element 8, see Figure 1c. The displacement variation at any point in element 8 is given by:

$$u = \sum_{i=1}^{m} N_i u_i \tag{0.0.5}$$

where u is the displacement vector at any point, N<sub>i</sub> are the shape or interpolation functions for nodal point i, u the displacement vector for nodal point i and m the total number of nodes for the element.

Using the plane membrane finite element QPM8, with m = 8, the interpolation functions are given by:

For corner nodes:-

$$N_{i} = \frac{1}{4} (1 + \xi_{o}) (1 + \eta_{o}) (\xi_{o} + \eta_{o} - 1)$$
(0.0.6)

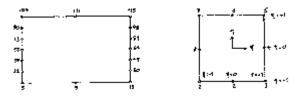
For midside nodes:-

when 
$$\xi_i = 0 \;, \quad N_i = \frac{1}{2} \Big( 1 - \xi^2 \Big) \Big( 1 + \eta_o \, \Big) \eqno(0.0.7)$$

when 
$$\eta_i = 0$$
,  $N_i = \frac{1}{2} (1 - \eta^2) (1 + \xi_o)$  (0.0.8)

$$\xi_0 = \xi \xi_i \; ; \; \eta_0 = \eta \eta_i$$
 (0.0.9)

The  $\xi$ ,  $\eta$  curvilinear co-ordinate system for an 8-noded element is shown in Figure 4b. Comparing this with element 8, see Figure 4a, enables the co-ordinates of nodes 22, 39, 73, 90, 30, 47, 81 and 98 in the curvilinear co-ordinate system of element 8 to be determined. These are given in columns 2 and 3 of Table 1.



a) Element 8

b) Mapping domain

Figure 4

The required constraint equations are of the form:

$$u^{8}(\xi,\eta) = u_{n}$$
 (0.0.10)

where the superscript 8 refers to element 8, the subscript n refers to the node (22, 39, 73, 90, 30, 47, 81 or 98) and  $\xi$ , $\eta$  are the curvilinear co-ordinates of node  $\eta$  with respect to element 8. Using (0.0.5), equations (0.0.6) to (0.0.9) may be written as:

$$\sum_{i=1}^{8} N_i u_i^8 - u_n = 0 \tag{0.0.11}$$

Substituting the  $\xi$ , $\eta$  co-ordinates of each node in turn into equations (0.0.6)-(0.0.9) yields the  $N_i$  values, which are tabulated in cols 4 - 11 in Table 1.

Table 1: Interface node curvilinear co-ordinates and shape functions

Node	ξ	η	N <sub>1</sub>	$N_2$	$N_3$	$N_4$	$N_5$	$N_6$	$N_7$	N <sub>8</sub>
22	-1	-2/3	+5/9	0	0	0	0	0	-1/9	+5/9
39	-1	-1/3	+2/9	0	0	0	0	0	-1/9	+8/9
73	-1	+1/3	-1/9	0	0	0	0	0	+2/9	+8/9
90	-1	+2/3	-1/9	0	0	0	0	0	+5/9	+5/9
30	+1	-2/3	0	0	+5/9	+5/9	-1/9	0	0	0
47	+1	-1/3	0	0	+2/9	+8/9	-1/9	0	0	0
81	+1	+1/3	0	0	-1/9	+8/9	+2/9	0	0	0
98	+1	+2/3	0	0	-1/9	+5/9	+5/9	0	0	0

The displacement vector at any point for the plane membrane family of elements is given by:

$$u = \begin{cases} U \\ V \end{cases} \tag{0.0.12}$$

where U and V are the displacement components in the global X and Y directions respectively. Using a subscript to denote the node number and noting the correspondence between the node numbering of the mapping domain, Figure 4b, and element 8, Figure 4a, the constraint conditions of (0.0.11) may be written explicitly as:

$$\begin{split} &\frac{5}{9}U_5 - \frac{1}{9}U_{107} + \frac{5}{9}U_{56} - U_{22} = 0 \\ &\frac{5}{9}V_5 - \frac{1}{9}V_{107} + \frac{5}{9}V_{56} - V_{22} = 0 \\ &\frac{2}{9}U_5 - \frac{1}{9}U_{107} + \frac{8}{9}U_{56} - U_{39} = 0 \\ &\frac{5}{9}V_5 - \frac{1}{9}V_{107} + \frac{8}{9}U_{56} - U_{39} = 0 \\ &-\frac{1}{9}U_5 + \frac{2}{9}U_{107} + \frac{8}{9}U_{56} - U_{73} = 0 \\ &-\frac{1}{9}V_5 + \frac{2}{9}V_{107} + \frac{8}{9}V_{56} - V_{73} = 0 \\ &-\frac{1}{9}V_5 + \frac{5}{9}U_{107} + \frac{5}{9}U_{56} - U_{90} = 0 \\ &-\frac{1}{9}V_5 + \frac{5}{9}V_{107} + \frac{5}{9}V_{56} - V_{90} = 0 \\ &\frac{5}{9}U_{13} + \frac{5}{9}U_{64} - \frac{1}{9}U_{115} - U_{30} = 0 \\ &\frac{5}{9}V_{13} + \frac{8}{9}U_{64} - \frac{1}{9}V_{115} - V_{30} = 0 \\ &\frac{2}{9}V_{13} + \frac{8}{9}U_{64} - \frac{1}{9}V_{115} - V_{47} = 0 \\ &-\frac{1}{9}U_{13} + \frac{8}{9}U_{64} + \frac{2}{9}U_{115} - U_{81} = 0 \\ &-\frac{1}{9}U_{13} + \frac{8}{9}V_{64} + \frac{2}{9}V_{115} - V_{81} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}U_{64} + \frac{5}{9}U_{115} - U_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{64} + \frac{5}{9}V_{115} - V_{98} = 0 \\ &-\frac{1}{9}V_{13} + \frac{5}{9}V_{13} + \frac{5}{9}V_{13} + \frac{5}{9}V_{13} + \frac{5}{9}V_{13} + \frac{5}{9}$$

Equations (0.0.13) are now in the form required for input to LUSAS.

### Comparison

1. Tip displacements at node 17.

	Fine mesh	Coarse mesh with Constraint Equations
u	0.0228175	0.0228031
v	-0.107897	-0.107364

#### Stress distributions.

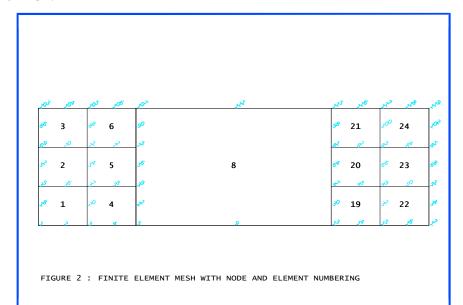
Figures 5 to 10 show the stress distributions as plotted with MYSTRO.

#### Note

The displacement field for the structure is relatively unaffected by the use of a single element to model the central half of the cantilever. Stress distributions at the tip and the root are also relatively unaffected by the use of a single element to model the central half of the cantilever.

## Input data

X01D23A.DAT X01D23B.DAT



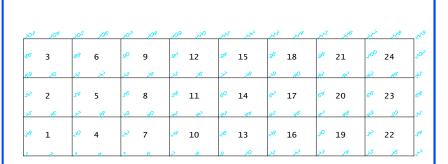
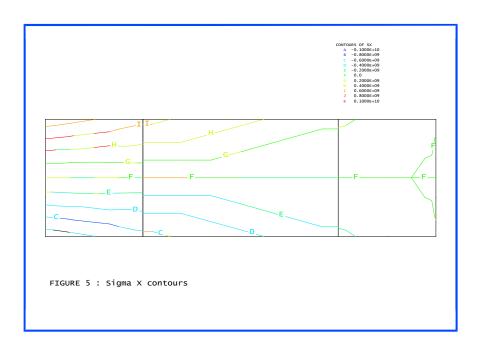
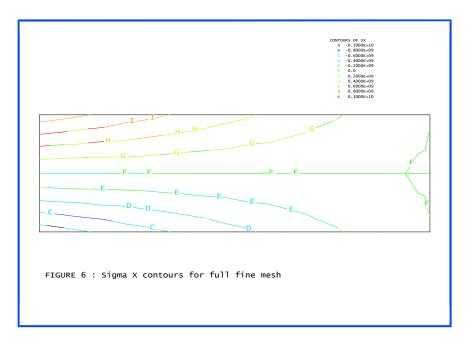
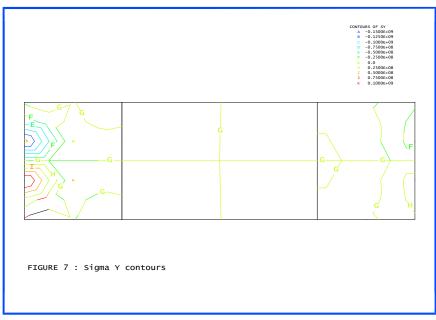
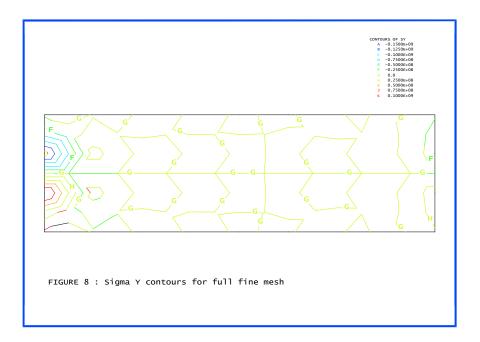


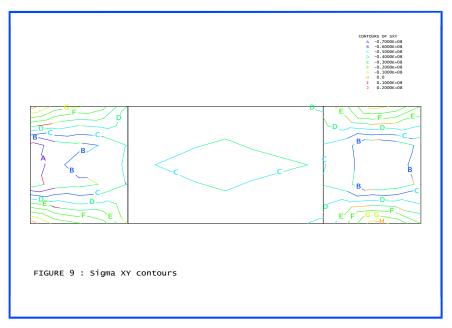
FIGURE 3 : FULL FINE MESH WITH NODE AND ELEMENT NUMBERING

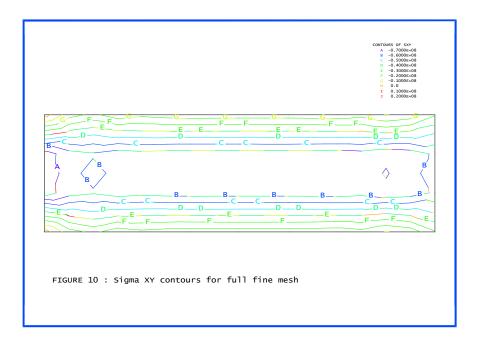












# Example 1.2.4

## Plane Membrane Analysis - Perforated Sheet Under Pure Tension

#### **Keywords**

Plane Stress, Stress Concentration Factor

#### **Problem Description**

A thin rectangular sheet with a central perforation is subjected to a tensile loading. The objective is to investigate the longitudinal stress distribution and to evaluate the stress intensity factor.

The geometry of the structure is as follows:

Length, 1	240 mm
Width, w	180 mm
Thickness, t	5 mm
Radius of central opening, r	30 m

#### **Discretisation**

One quarter of the structure is discretised using 27 isoparametric plane stress elements (QPM8). The mesh is graded towards the central opening in order gain a better approximation to the stress concentration expected in this region of the structure. The internal boundaries of the mesh are restrained according to the double symmetry conditions assumed. The external boundaries of the mesh are unrestrained.

The uniform tensile load is modelled by the application of a series of nodal point loads acting in the Y direction along the bottom edge of the structure (note that in order to obtain a uniform distribution the nodal point loads are applied in the ratio 1:4:1 to the first corner, midside and second corner nodes of the element face respectively). The total load applied to the bottom edge of the structure is 36,000N.

#### **Material properties**

The material properties assumed in the analysis are as follows:

Young's modulus	207x10 <sup>3</sup> N/mm <sup>2</sup>
Poisson's ratio	0.3

### Theory

The uniform stress distribution in the absence of the central opening may be calculated from elastic theory.

Total applied load F = 36,000 N

Total section area  $A_y = 5 \text{mm} * 180 \text{mm} = 900 \text{mm}$ 

Stress  $S_v = F/A = (36,000/900) = 40 \text{ N/mm}^2 (0.0.14)$ 

The presence of the central opening will change this uniform stress distribution to form a concentration of tensile stress in the proximity of the opening [1]. From Saint-Venant's principal it may be concluded that at distances which are large compared to the dimensions of the opening its effect on the stress distribution will be negligible. The expected distribution of longitudinal stress is therefore in the form of a decay from high tensile stresses at the opening towards the normal stress level at the edge of the plate. The concentration factor may be obtained from the mean stress (equation (0.0.14)) and the computed stress (at the opening) as

stress concentration factor = computed stress / mean stress

#### Comparison

The LUSAS results for the stress in the Y direction are given below. ((\*) denotes the stress value is the average from the contributing nodes):

Node	LUSAS
13	143.431
26	111.922
39(*)	81.6375
52	69.7183
65(*)	57.2874
78	51.7408
91(*)	46.0919
104	38.1439
117	29.5542

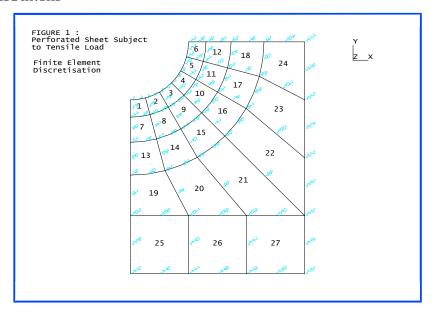
Stress concentration factor = 143.4/40.0 = 3.585

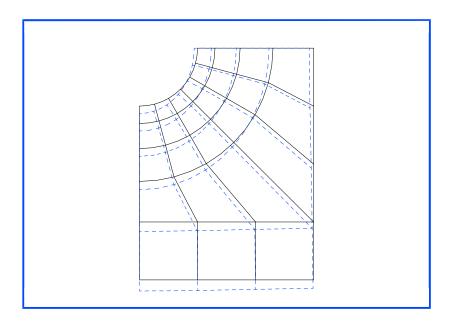
#### References

 'Theory of Elasticity', Second edition, S.Timoshenko, J.N.Goodier, Publisher. McGraw-Hill Book Co.Ltd. (1951)

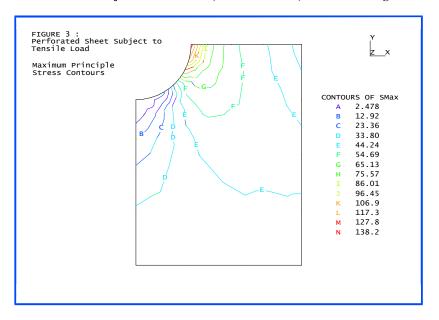
## **Input data**

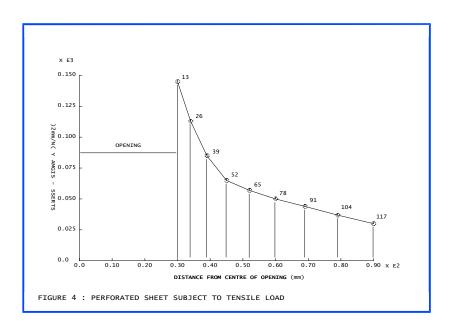
X01D24A.DAT





#### Perforated Sheet Subject to Tensile Load (Plane Membrane) Deformed Configuration





# Example 1.3.1

## **Orthotropic Plate Under Pressure Load**

## **Keywords**

Plate, Orthotropic, Anisotropic

#### **Problem Description**

A simply supported square plate is analysed to verify the plate behaviour when the material is anisotropic and the plate is subjected to a uniform pressure load.

#### **Finite Element Model**

Allowing for symmetry only one quarter of the plate need be analysed. The model consists of a 4\*4 element mesh with 8-noded elements.

The problem is solved using both data cards, MATERIAL PROPERTIES, ORTHOTROPIC and ANISOTROPIC.

#### **Geometric Properties**

Thickness	0.015 m
Length	0.3 m

#### **Material properties**

Anisc	otropic	Ortho	tropic	Isotr	opic
$E_{\boldsymbol{x}\boldsymbol{x}}$	$1.38x10^{10}\;N/m^2$	$E_{\boldsymbol{x}}$	$1.3555 \ x10^{10} \ N/m^2$	E	$2.0 \text{ x} 10^{11} \text{ N/m}^2$
$E_{xy} \\$	$5.31 \ x10^8 \ N/m^2$	$E_{\boldsymbol{y}}$	$1.1290\;x10^9\;N/m^2$	ν	0.3
$E_{yy} \\$	$1.15 \text{ x} 10^9 \text{ N/m}^2$	$\nu_{xy}$	0.4619		
$G_{xy}$	$1.17 \text{ x} 10^9 \text{ N/m}^2$	$\nu_{yx}$	0.03848		
		$G_{xy}$	$1.1700\;x10^9\;N/m^2$		

### **Boundary Conditions**

Simply supported around the plate edges.

#### **Loading Conditions**

A uniformly distributed pressure load  $p = 1000 \text{ N/m}^2$ 

#### **Theory**

The expression for the deflection w in the z-direction takes the following form;

$$w = \sum_{m=1,3,5} \sum_{n=1,3,5} a_{mn} \sin\left(\frac{\pi mx}{a}\right) \sin\left(\frac{\pi ny}{b}\right)$$
 (0.0.15)

where the coefficient  $\,a_{mn}\,$  and a full description of the theory is given in [1].

### **Solution Comparison**

The solutions obtained from the LUSAS orthotropic and anisotropic analysis are presented below, and are shown in the following figures. A principal stress plot for the isotropic case is included for comparison.

Vertical Deflection w (m)

Location	Theory	LUSAS (Anisotropic)	LUSAS (Orthotropic)
C	3.543E-04	3.545E-04	3.545E-04
E	2.529E-04	2.531E-04	2.531E-04
F	1.925E-04	1.939E-04	1.937E-04

Bending Moments (Nm/m)

Location	Theory	LUSAS (Anisotropic)	LUSAS (Orthotropic)
C	$M_X = 37.49$	$M_X = 38.21$	$M_X = 38.21$
	$M_Y = 3.685$	$M_Y = 3.710$	$M_Y = 3.710$
E	$M_X = 28.44$	$M_X = 29.27$	$M_X = 29.27$
	$M_Y = 2.680$	$M_Y = 2.690$	M <sub>Y</sub> =2.690

#### References

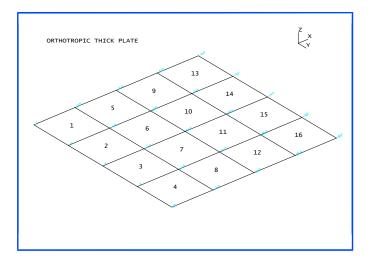
 Timoshenko, S.P. and Woinowsky-Krieger, S., Theory of Plates and Shells, Second Edition, McGraw-Hill, 1959.

#### **Input data**

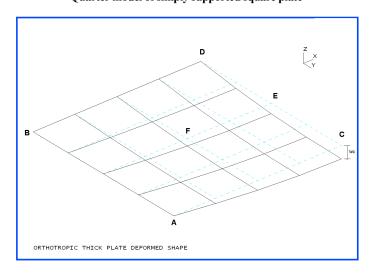
X01D31A.DAT

X01D31B.DAT

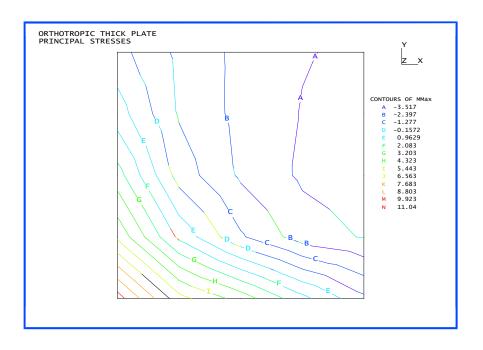
#### X01D31C.DAT

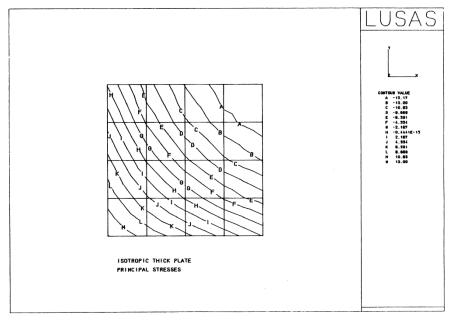


#### Quarter model of simply supported square plate



Vertical deflection under loading





# **Example 1.3.2**

## **Static Linear Analysis Of A Thick Circular Plate**

## **Keywords**

Plate, Linear

#### **Problem Description**

A clamped circular plate is subjected to a transverse uniformly distributed load. The dimensions of the plate are as follows:

Radius, a	0.5 inches
Thickness, h	0.1 inches

#### **Discretisation**

The structure is idealised using the thick plate flexure elements QTF8 and TTF6. By using the CARTESIAN SETS and TRANSFORMED FREEDOM facilities it is only necessary to idealise a small segment of the total structure. In this example a 30 degree segment is analysed. The outer boundaries of the circular plate are assumed to be fully restrained.

The material properties are as follows

Young's modulus	10.92 lb/in <sup>2</sup>
Poisson's ratio	0.3

### **Theory**

The linear elastic thick plate solution may be obtained by combining the 'thin' plate solution with a correction for out of plane transverse shearing effects.

The thin plate solution is obtained by the solution of the biharmonic plate equation [1]

$$r\frac{d}{dr}\left(\frac{1}{r}\frac{d}{dr}\left(r\frac{dw}{dr}\right)\right) = \frac{1}{D}\int_{0}^{r}q\,r\,dr \tag{0.0.16}$$

Differentiating with respect to r and dividing by r yields

$$\frac{1}{r}\frac{d}{dr}\left(r\frac{d}{dr}\left(\frac{1}{r}\frac{d}{dr}\left(r\frac{dw}{dr}\right)\right)\right) = \frac{q}{D}$$
(0.0.17)

where

$$D = \frac{Eh^3}{12(1-v^2)}$$
 (0.0.18)

Triple integration of (0.0.17) enables the deflection profile w = f(r) to be obtained as:

$$w = \frac{qr^4}{64D} + \frac{Ar^2}{4} + B\log\frac{r}{a} + C$$
 (0.0.19)

where A, B and C are the constants of integration. By applying the deflection and slope conditions associated with the clamped periphery of the plate, (0.0.19) may be simplified as:

$$w = \frac{q}{64D} \left( a^2 - r^2 \right)^2 \tag{0.0.20}$$

The maximum displacement occurs at the centre of the plate (r=0) and is given by the expression –

$$w_{\text{max}} = \frac{qa^4}{64D} \tag{0.0.21}$$

The deflection profile defined by (0.0.20) was derived under the conditions of pure bending. The effects of transverse shear may be included by the addition of a shear correction term to yield the deflection profile [1]

$$w = \frac{q}{64D} \left( (a^2 - r^2) + \frac{4h^2}{(1 - v)} (a^2 - r^2) \right)$$
 (0.0.22)

and the maximum deflection relationship (at r=0)

$$w = \frac{q}{64D} \left( a^4 + \frac{4h^2a^2}{(1-v)} \right)$$
 (0.0.23)

#### Comparison

The LUSAS results for the maximum deflection are compared to the thin and thick plate solutions below:

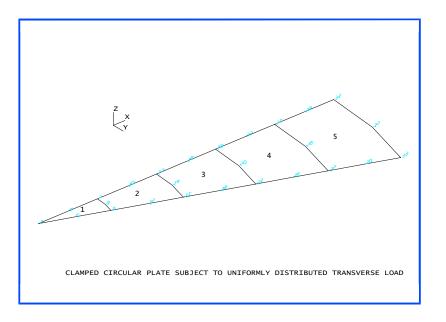
 h	h/2a	Thin Plate Theory	Thick Plate Theory	LUSAS results
 0.1	0.1	-0.97656	-1.19966	-1.15480

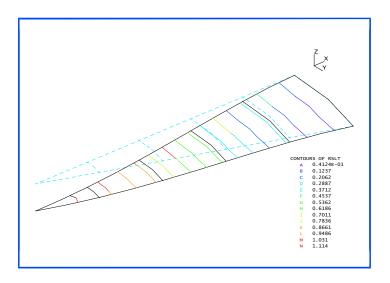
### References

1. S.P.Timoshenko, S.Woinowsky-Kreiger, 'Theory of Plates and Shells', Second edition, Publisher. McGraw-Hill Book Co. Ltd (1959).

### **Input data**

X01D32A.DAT





Clamped Circular Plate subject to uniform load deformed configuration and Displacement Contours

## Example 1.3.3

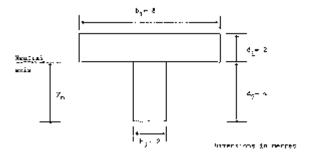
#### **Eccentric Ribbed Plate**

#### **Keywords**

Plate, Linear, Eccentric

#### **Problem Description**

A cantilevered eccentric ribbed plate is subjected to an end moment. The dimensions of the plate are as follows:



#### **Discretisation**

The structure is idealised using 4 thick shell elements QTS4 modelling the plate and 2 eccentric stiffeners BMS3 modelling the web. The plate is supported as a cantilever fully restrained at one end with a constant moment of 1000 KNm applied to the free end.

## **Material properties**

Young's modulus	100.0 x 10 <sup>6</sup> KN/m2
Poisson's ratio	0.0

#### **Theory**

The position of the neutral axis from the bottom of the section is calculated as:

$$y_n = \frac{A_1 y_1 + A_2 y_2}{A_1 + A_2} = 4m \tag{0.0.24}$$

The moment of inertia about the neutral axis is:

$$I_{yy} = \frac{b_1 d_1^3}{12} + A_1 y_1^2 + \frac{b_2 d_2^3}{12} + A_2 y_2^2 = 64m^4$$
 (0.0.25)

Stress on the top surface:

$$\sigma_{\rm t} = 2 \frac{M_{\rm y}}{I_{\rm yy}} = 31.25 \text{ KN/m}^2$$
 (0.0.26)

Stress on the bottom surface:

$$\sigma_{b} = 4 \frac{M_{y}}{I_{yy}} = 62.5 \text{ KN/m}^{2}$$
 (0.0.27)

#### Comparison

The LUSAS results for the stresses compared to theoretical results. Plate output for nodes at fixed end:

Top stress = 
$$31.267 \text{ KN/m}$$
 (average)

Beam output for nodes at fixed end:

Axial force = 
$$250.018 \text{ KN}$$

Moment of inertia for beam:

$$I_{yy} = \frac{bd^3}{12} = 10.667 \text{m}^4 \tag{0.0.28}$$

**Bottom Stress:** 

$$\sigma_b = \frac{F}{A} + \frac{M_y y}{Ivv} = 62.64 \text{ KN/m}^2$$
 (0.0.29)

## **Input Data**

X01D33B.DAT

# **Example 1.4.1**

# **Static Stress Analysis Of A Cantilever Subjected To Multiple Load Cases**

## **Keywords**

Three Dimensional Beam, Loading

#### **Problem description**

Determine the tip displacements, moments and flexural strains for a straight cantilever. The geometry of the cantilever is as follows:

Length, l	5.0
Breadth, b	0.25
Depth, d	1.0

#### **Discretisation**

The cantilever is modelled using one or two beam BMI21 (with Residual Bending Flexibility correction), BMI31 or BS4 elements. The material properties are as follows:

Young's modulus	30000.0
Poisson's ratio	0.3
Density	0.283
Coefficient of thermal expansion	0.0003

The fixed end of the cantilever is assumed to be fully restrained.

#### **Theory**

The cantilever is subjected to the following load cases:

Case	Loading	Description of load case
1	CL	End concentrated load -30 in Y direction
2	CL	End concentrated load 30 in X direction
3	CL	End anticlockwise moment (positive) 30
4	UDL	Uniformly distributed load 15 in local x direction
5	UDL	Uniformly distributed load -16 in local y direction
6	CBF	Constant body forces 15×4 in global X direction
7	CBF	Constant body forces -16×4 in global Y direction
8	CBF	Centrifugal forces 15 about global Y axis
9	CBF	Centrifugal forces 15 about global X axis
10	TEMP	Uniform temperature rise 50 at nodes
11	TEMP	Flexural temperature gradient $\partial T/\partial y = 10$ at nodes
12	BFP	Constant nodal body force 15*4 and -16*4 at nodes in global X and Y directions
13	SSI	Initial stress resultant 15 at nodes in local x direction
14	SSI	Initial strains -0.002 at nodes in local x direction
15	SSIG	Initial stress resultant 15 at Gauss points in local x direction
16	SSIG	Initial strains -0.002 at Gauss points in local x direction
17	BFP	Body force potential 15 in local x direction
18	BFP	Body force potential -16 in local y direction
19	COMB	Combination of $2 * case (1) + 2 * case (2)$

## **Theory and Solution Comparison**

Load case	Quantity	Theoretical solution	LUSAS
	Displacement	$v = -PI^3/3EI = -2.00$	-2.0 (-2.062*)
1	Moment	M = P(x-1) = 150.0	150.0
	Strain	= M/EI = 0.24	$0.24~(0.12^{\dagger})$
	Displacement	u = PI/EA = 0.02	0.02
2	Axial Force	$F_X P = 30.0$	30.0
	Strain	= P/EA = 0.004	0.004
	Displacement	$v = M1^2/2EI = 0.60$	0.60
3	Moment	M = M = -30.0	-30.0
	Strain	= M/EI = -0.048	-0.048
· · · · · · · · · · · · · · · · · · ·	·	· · · · · · · · · · · · · · · · · · ·	·

Load case	Quantity	Theoretical solution	on	LUSAS
	Displacement	$d = WI^2/2EA$	= 0.025	0.025
4	Moment	$F_x = W(1-x)$	= 75.0	75.0
	Strain	$= F_x/EA$	= 0.01	$0.01~(0.005^{\dagger})$
	Displacement	$d = WI^3/8EI$	= -2.0	-2.0 (-2.083*)
5	Moment	$M = WI^2/2$	= 200.0	191.67 (200†)
	Strain	= M/EI	= 0.32	$0.3067 \ (0.1067^{\dagger})$
6		See load case (4)		
7		See load case (5)		
8	Displacement	$u = w^2 l^3 / 3E$	= 0.08843	0.088437
	Axial force	$Fx = w^2 A 1^2 / 2$	= 198.98	207.3 (199.0†)
	Strain	$w^2l^2/2E$	= 0.02653	0.02764 (0.01769†)
	Displacement	$v = Fyl^4/8EI$	= 0.99487	0.9949 (1.036*)
9	Moment	$M = Fyl^2/2$	= -99.492	-95.347 (-99.492 <sup>†</sup> )
	Strain	= M/EI	= 0.1592	-0.1526 (-0.0531 <sup>†</sup> )
	Displacement	u = TI	=0.075	0.075
10	Strain	= u/l	= 0.015	0.015
	Displacement	v = (dT/dy) 12/2	= 0.0375	-0.0375
11	Strain	= (dT/dy)	= 0.003	0.003
12		See load case (4) & (5) combined		
	Displacement	u = Fl / EA	= -0.01	-0.01
13	Strain	= u/1	= -0.002	-0.002
	Displacement	u = 1	= -0.01	-0.01
14	Strain	=	= -0.002	-0.002
15		See load case (13)		
16		See load case (14)		
17		See load case (4)		
18		See load case (5)		
	Displacement	See load case (2) =	= 0.06	0.06
19	Axial Force	See load case (2) =	90.0	90.0
	Strain	See load case (2) =	-0.012	-0.012

Displacements in brackets are results from thick beam elements BMI21/BMI31 if different from thin beam element BS4.

† Moments and strains in brackets are results from BMI21 if different from BMI31 and BS4.

### References

 Roark, R.J. Young, C.T. 'Formulas for stress and strain: Fifth edition', McGraw-Hill Publishing Company.

### **Input Data**

X01D41A.DAT

X01D41B.DAT

X01D41C.DAT

# **Example 1.5.1**

# **Linear Elastic Stress Analysis Of A Compact Tension Fracture Specimen**

#### **Keywords**

Compact Tension Specimen, Three Dimensional Continuum, Solid

#### **Problem Description**

Determine the opening displacement and linear elastic stress distribution in a compact tension fracture specimen.

The compact tension test consists of a plane sheet loaded at either side of the crack by two loading pins (figure 1). The dimensions of the test specimen are as follows:

Height, h	120.0 mm
Width, w	100.0 mm
Thickness, t	3.0 mm

The loading pins are positioned at 55 mm centres and have a diameter of 25 mm. The stress concentration required for crack propagation is achieved by a pointed notch 50 mm long and 6 mm wide cut into the specimen. At the point of the notch a cut is made into the specimen to a further depth of 21.6mm.

The crack opening displacement is measured by a clip gauge mounted across the notch directly in line with the applied loading.

#### **Discretisation**

The full test specimen is modelled using three dimensional continuum (HX20 and PN15) elements.

The loading pins are included in the finite element model and assumed to behave as a rigid bodies compared to the test specimen. This is in an attempt to reduce localised effects around the loading pins and hence more accurately model the physical test conditions.

The following material properties are assumed in the analysis:

Young's modulus	210.915 kN/mm <sup>2</sup>
Poisson's ratio	0.33

#### **Theory and Loading**

Previous experimental and numerical investigations of this specimen have established the stress concentration around the notch, and have indicated that localised plastification occurs around this point prior to the unstable propagation of the crack throughout the specimen [1-3]. The structure is known to behave linearly until a load of approximately 10 KN. In the finite element model the loading is applied to the structure via point loads acting at the centre of the rigid pins.

#### Solution

The results obtained from the LUSAS three-dimensional analysis are presented in figures 2-4.

#### References

- Bleackley, M.H. Luxmoore, A.R. 'Comparison of finite element solutions with analytical and experimental data for elastic-plastic cracked problems'. International Journal of Fracture (in press)
- 2. Bleackley, M.H. 'A numerical study of energy criteria in fracture mechanics'. PhD. Thesis, University of Wales, (1981).
- 3. Owen, D.R.J., Fawkes, A.J., 'Engineering fracture mechanics: Numerical methods and applications'. Publisher. Pineridge Press Ltd, Swansea, U.K. (1983).

#### Input data

X01D51A.DAT

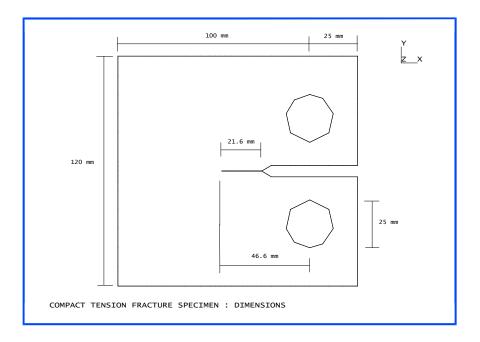


Figure 1: Dimensions of compact tension fracture specimen.

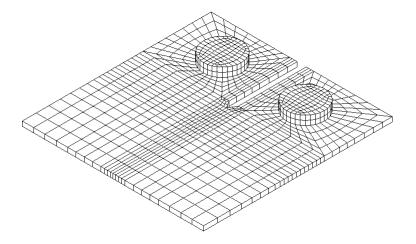


Figure 2: Finite element mesh for compact tension fracture specimen.

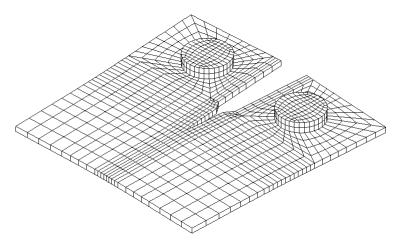


Figure 3: Defomed mesh under loading.

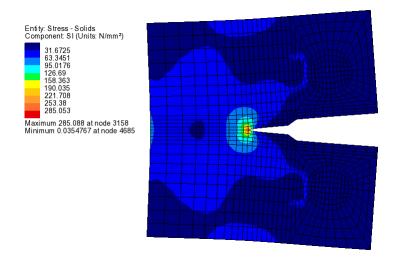


Figure 4: Contours of principal stress.

# **Example 1.6.1**

## A Simply Supported Twin Box Beam Under Concentrated Loads

#### **Keywords**

**Box Beam** 

#### **Problem description**

A simply supported twin box beam with trapezoidal cross-section is subjected to two symmetrical point loads near the centre span (Figure 1). The beam is made of thin mild steel plates with the following thicknesses:

Top flange thickness	0.5 cm
Web thickness	0.3 cm
Bottom flange thickness	0.3 cm

#### **Loading condition**

Two point loads of 20KN are applied at a position 7/16 of the span section over the inner webs (Figure 1).

#### **Finite Element Model**

The discretisation consists of 264 SHI4 elements. Due to the non-symmetric loading case the whole beam must be included in the discretisation.

### **Material properties**

Young's modulus, E	19620 KN/cm2
Poisson's ratio, v	0.27

#### **Boundary conditions**

The beam is simply supported at each end along the bottom flange.

#### **Theory**

Structural design of spine-beam bridges presents many difficulties because of the complex nature of the interaction of individual elements. A number of analysis methods exist; however, the reader is referred to publications by Maisel and Roll [1,2] for further information.

#### **Solution Comparison**

The problem investigated in this example is one that was performed experimentally by Zhang [3] at The City University London. The experimental results obtained have been compared to those of LUSAS in Figures 3 and 4.

#### References

- Maisel, B.I. Review of Literature Related to the Analysis and Design of Thin-Walled Beams, Technical Report TRA 440, Cement and Concrete Association London, July 1970.
- Maisel, B.I. and Roll, F. Methods of Analysis and Design of Concrete Boxbeams with Side Cantilevers, Technical Report 42.494, Cement and Concrete Association London, November 1974.
- Zhang, S.H. The Finite Element Analysis of Thin-Walled Box Spine Beam Bridges, PhD thesis, The City University, London February 1982.

#### Input data

#### X01D61A.DAT

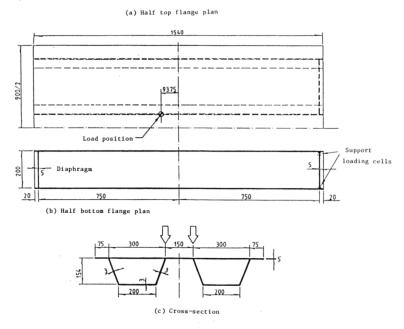
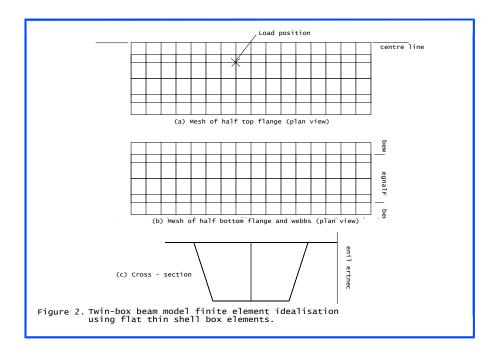
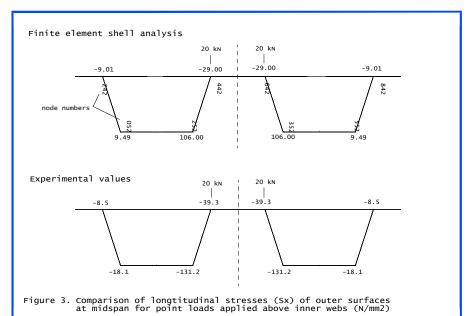
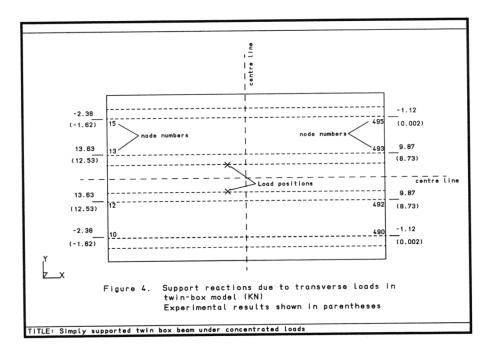


Figure 1. Simply supported twin-box beam model (in mm)







# **Example 1.6.2**

## **Static Stress Analysis Of A Shallow Spherical Shell**

## **Keywords**

**Shallow Spherical Shell** 

#### **Problem Description**

Determine the central deflection of a shallow spherical shell under a central point load and an eccentric patch load (see figure 1).

#### **Discretisation**

The analysis is performed with two different element meshes. The first model uses 6 BXS3 axisymmetric shell elements (see figure 2). This models one radian of the structure and consequently is only applicable to the central point load case. The second model uses 5 X 12 QSI4 and 1 X 12 TS3 incompatible flat shell elements (see figure 3). This models half of the structure and can therefore be applied to both load cases. It should be noted that a one radian section of 12 QSI4 and 1 TS3 flat shell elements would have been sufficient to model the central point load case.

### **Geometry**

The geometry of the shallow spherical shell structure is shown in figure 1.

#### **Material properties**

Young's modulus	70000 N/mm2
Poisson's ratio	0.3

#### **Boundary conditions**

The axisymmetric model is fully restrained at the base, node 1, and restrained from translation or rotation across the line of symmetry, node 13. The thin shell model is fully restrained at the base, nodes 7 to 91 in steps of 7, and restrained from translation or rotation across the line of symmetry, nodes 1 to 6 and nodes 86 to 90.

### **Theory**

The fundamental theory for the central deflection of a shallow spherical shell under central point load is given in page 477 of [1].

#### **Results**

Table 1 gives the results for the central deflection due to a central point load for theory, axisymmetric model and thin shell model.

Analysis	Central deflection (mm)
Theory	-0.9514
BXS3	-0.8961
QSI4/TS3	-0.9495 (figure 4)

The central deflection due to the eccentric patch load is +0.0235 mm and the deflections at nodes 3 and 4 under the patch load are 0.473 mm and 0.458 mm (figure 5).

#### References

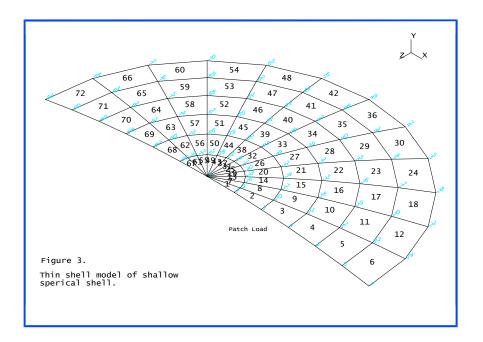
 R. J. Roark and W. C. Young, "Formulas for Stress and Strain", Fifth edition. McGraw-Hill, 1975.

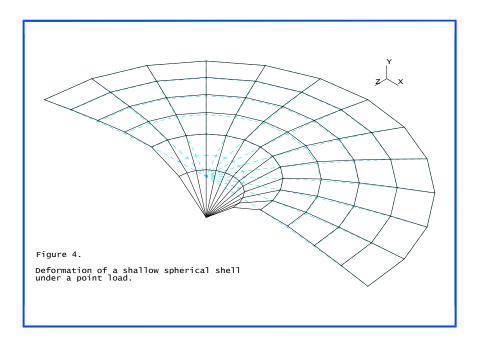
## **Input data**

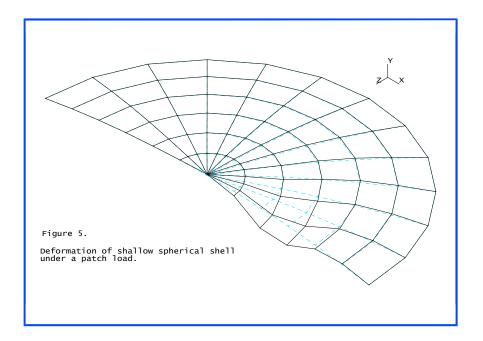
X01D62A.DAT

X01D62B.DAT

- 1. Axisymmetric shell analysis, central point load
- 2. Thin shell analysis, central point load and patch load







# **Example 1.6.3**

# **Graphite/Epoxy Resin Laminated Orthotropic Square Plate**

## **Keywords**

Composite Analysis, Laminated Plate

### **Problem Description**

The composite/laminated material comprises 9 graphite/epoxy resin material layers. Each layer is arranged so that its principal directions of orthotropy form a 0/90/0/90/0/90/0/90/0 sequence with respect to the global reference axes. The layup sequence is shown in figure 1. Owing to the different thicknesses of each of the layers in the parallel and perpendicular pairs, the plate is orthotropic with respect to both the material laminates and the resulting composite material. The plate is loaded with a udl and the central deflection of the plate is computed.

### **Discretisation**

A symmetric quarter of the plate is modelled using a fine mesh of 6\*6 Semiloof shell elements, figure 2. The composite construction of the material is modelled using 9 orthotropic material layers.

## Geometry

The geometry of the plate is as follows:

Side length (a) = 10cm

Thickness (t) = 0.1cm

Span/Thickness = 100

Load Intensity (q) =  $100 \text{ N/mm}^2$ 

## **Material properties**

The orthotropic material properties are as follows:

Young's modulus  $E_1 = 40E6 \text{ N/cm}^2$ 

Young's modulus  $E_2 = 1E6 \text{ N/cm}^2$ 

Poisson's ratio  $v_{12} = 0.25$ 

Shear Modulus  $G_{12} = 0.6 E6 \text{ N/cm}$ 

Orthotropy  $E_1/E_2 = 40$ 

The material lay-up sequence for the laminated plate is as follows (note that the LUSAS convention is sequential from the bottom to the top of the material, and that the orientations are defined as that of the principal direction of orthotropy to the global x-axis:

Position	Layer	Thickness	Orientation
Bottom	1	0.01	0
	2	0.0125	90
	3	0.01	0
	4	0.0125	90
	5	0.01	0
	6	0.0125	90
	7	0.01	0
	8	0.0125	90
Top	9	0.01	0

## **Boundary Conditions**

The external and internal plate boundary conditions are specified simply supported and symmetric respectively.

## **Theory**

The non-dimensional central deflection of the plate may be obtained from the expression (see first reference):

$$\overline{\mathbf{w}} = \mathbf{w} \left( 1000 \frac{\mathbf{E}_2 \mathbf{t}^3}{\mathbf{Pa}^4} \right) \tag{0.0.30}$$

### **LUSAS** results

Analysis method	Reference	Finite element mesh	Deflection w
Theory	(2)	N/A	-4.486
LUSAS		6*6 QSl8	-4.526
FE2000	(1)	Laminated solid 6*6*1	-4.480

LUSAS results for on-axis lamina stresses in layers 7, 8 & 9 are displayed in figure 3.

## References

- 1. Razzaque, A. Mathers, M.D., 'Layered solid elements for non-linear analysis of composite structure', Quality assurance in Finite Element Technology (1988)
- Noor,A.K., Mathers,M.D., 'Shear flexible models of laminated composite plates', NASA-TN D-8044 (1975)

## **Input data**

X01D63A.DAT

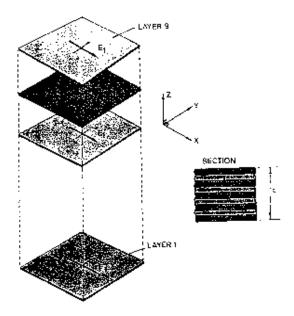
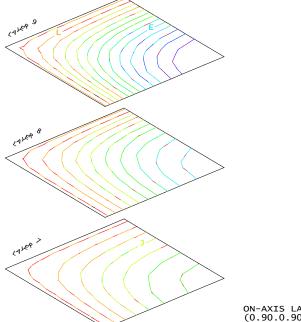


Figure 1 Layup Sequence for the Laminated Graphic/Epoxy Resin Orthotropic Plate

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CONTOURS OF SX
A -0.7000e+06
B -0.6434e+06
C -0.5868E+06
D -0.5303e+06
F -0.4171E+06
G -0.3605e+06
H -0.3040E+06
I -0.2474E+06
J -0.1908E+06
K -0.1342E+06
L -0.7764E+05
M -0.2106E+05
N 0.3552E+05

ON-AXIS LAMINA STRESSES FOR (0.90.0.90.0)s GRAPHITE/EPOXY PLATE

# **Example 1.7.1**

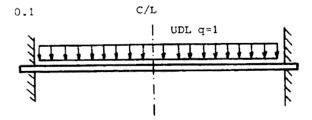
# **Axisymmetric Analysis Of A Clamped Circular Plate**

## **Keywords**

Circular Plate, Axisymmetric Solid, Axisymmetric Sheet

## **Problem Description**

The problem considered in this example is a clamped circular plate subject to a uniform pressure load. Figure 1 shows the dimensions and material properties of the problem under consideration.



### **Discretisation**

Four axisymmetric elements are used to model the circular plate. Figures 2 & 3 show the finite element mesh for QAX8 and BXS3 elements respectively.

## **Material properties**

Young's modulus	$10.0x10^6 \text{ N/m}^2$
Poisson's ratio, v	0.3

## **Boundary conditions**

The edge of the plate is fully fixed against translation or rotation.

## **Theory**

$$D = \frac{Et^3}{12(1-v^2)} \tag{0.0.31}$$

$$w = \frac{qa^4(a^2 - r^2)^2}{64D} \tag{0.0.32}$$

where

w = downward deflection

q = uniform pressure

r = distance measured radially from centre of plate

a = radius of plate

D = flexural rigidity of plate

E = Young's Modulus

v = Poisson's Ratio

t = thickness of plate

For further information see [2].

## **Solution comparison**

Radius, m	Displacement Theory, m	Rotation Theory, radians	Displacement BXS3, m	Rotation BXS3, radians	Displacement QAX8, m
0	-0.17063	0	-0.17080	0	-0.16237
2.5	0.14996	0.01599	-0.15004	0.01597	-0.14212
5.0	-0.09598	0.02559	-0.09603	0.02558	-0.08970
7.5	-0.03266	0.02239	-0.03269	0.02239	-0.02891

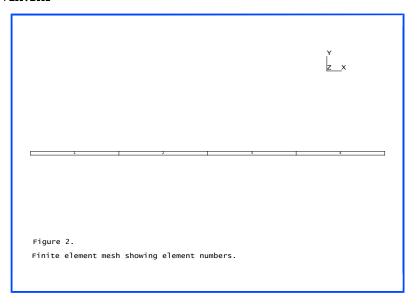
Although only one element is modelled through the thickness the LUSAS results compare favourably with the theory.

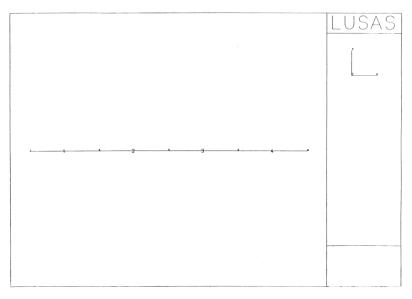
### References

- M.E. Honnor, 'Axisymmetric thin shell element', FEAL internal report FEAL503, 10th December 1985.
- S.P. Timoshenko and S.Woinowsky-Krieger, 'Theory of plates and shells', pp. 55-56, McGraw-Hill, 1970.

# **Input data**

X01D71A.DAT





# **Example 1.7.2**

# **Axisymmetric Thin-Walled Pressure Vessel**

## **Keywords**

Pressure Vessel, Axisymmetric Solid, Axisymmetric Shell, Axisymmetric Sheet

## **Problem Description**

The analysis of a thin-walled, cylindrical and hollow pressure vessel is considered in this example. Three different element types are utilised together with three different load cases. Axisymmetry about the X & Y axes is also considered for one set of the meshes. The element types are utilised as follows:-

- a) Axisymmetric eight-noded solid elements QAX8
- b) Axisymmetric thin shell element BXS3
- c) Axisymmetric sheet element BXM3

The three load cases considered are:

- Case 1) Uniform axial load per unit length of 6 N/m
- Case 2) Uniform radial pressure of 10 Pa
- Case 3) Linearly varying radial pressure from zero at the bottom to 100Pa at the top.

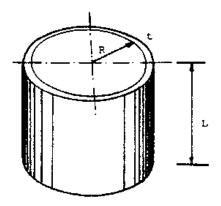
Figure 1 shows the dimensions and material properties of the problem under consideration.

### **Discretisation**

Four elements are used along the length of the wall. The finite element mesh for QAX8 is shown in Figure 2. Figure 3 shows the mesh for BXS3 and BXM3.

## **Material properties**

Young's modulus	$200x10^3 Pa$
Poisson's ratio	0.3



 $L=100m,\ R=50m,\ t=1.25m,\ E=0.2E6\ Pa\ \upsilon=0.3$ 

## **Boundary Conditions**

The bottom end of the cylinder is restrained against rotation and movement in the meridional direction but free to move radially.

## **Theory**

Notation:

E	Young's Modulus
L	Length of cylinder
p	Axial load per unit length
q	Uniform pressure per unit area
r	Radial displacement at top end
R	Radius of cylinder
t	Thickness of cylinder
у	Meridional displacement at top end
υ	Poisson's Ratio
$s_{\rm m}$	Meridional stress at top end
S <sub>e</sub>	Circumferential stress top end
τ	Rotation of a meridian

	Case 1	Case 2	Case 3
$\sigma_{\text{m}}$	$\frac{p}{t}$	0	0
σe	0	$\frac{qR}{t}$	qR t
r	$-\frac{pR}{Et}$	$\frac{qR^2}{Et}$	$\frac{qR^2}{Et}$
у	pL Et	$-qR\frac{L}{Et}$	$-qR\frac{L}{2Et}$
τ	0	0	$\frac{qR^2}{EtL}$

For more information see [1].

## **LUSAS Results**

Load case 1	Theory	QAX8	BXS3	BXM3
Meridional stress (Pa)	4.8	4.8	4.8	4.8
Circumferential stress (Pa)	0.0	0.0104	0.0	0.0
Radial displacement at end (m)	-3.5x10 <sup>-4</sup>	-3.6x10 <sup>-4</sup>	-3.6x10 <sup>-4</sup>	-3.6x10 <sup>-4</sup>
Meridional displacement at end (m)	2.4x10 <sup>-3</sup>	2.4x10 <sup>-3</sup>	2.4x10 <sup>-3</sup>	2.4x10 <sup>-3</sup>
Rotation (radians)	0.0	-	0.0	-

Load case 1	Theory	QAX8	BXS3	BXM3
Meridional stress (Pa)	0.0	0.0	0.0	0.0
Circumferential stress (Pa)	400.0	400.0	400.0	400.0
Radial displacement at end (m)	0.1	0.09998	0.1	0.1
Meridional displacement at end (m)	-0.06	-0.06	-0.06	-0.06
Rotation (radians)	0.0	-	0.0	-

Load case 1	Theory	QAX8	BXS3	BXM3
Meridional stress (Pa)	0.0	-0.0058	0.0	
Circumferential stress (Pa)	4000.0	4000.2	4000.0	4000.0

Load case 1	Theory	QAX8	BXS3	BXM3
Radial displacement at end (m)	1.0	0.9999	0.9999	1.0
Meridional displacement at end (m)	-0.3	-0.3	-0.3	-0.3
Rotation (radians)	-0.01	-	-0.0099	-

## References

1. R.Roark and W.Young, 'Formulas for stresses and strains', pp.448. LUSAS Results

## **Input data**

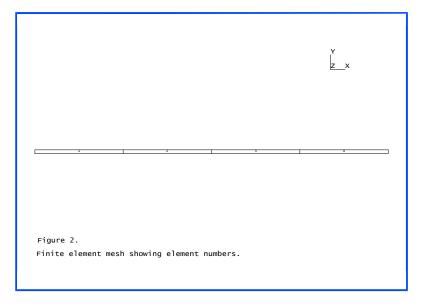
X01D72A.DAT

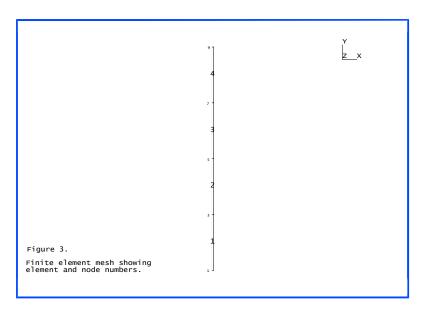
X01D72B.DAT

X01D72C.DAT

X01D72D.DAT

(axisymmetry about the Y-axis)





# **Example 1.7.3**

# **Cooling Tower Subject To Wind Loading**

## **Keywords**

Fourier, Semiloof Shell, Face Loading, Curve Definitions, Curve Assignment, Cartesian Set Cylindrical, Wind Loading

## **Problem Description**

The effect of wind load on a cooling tower is modelled by a static analysis. For comparison, the analysis is performed using both the solid Fourier ring element QAX8F and the Semiloof shell element QSL8.

The pressure of the wind varies circumferentially, but is taken to be independent of the height above ground. The distribution of the load is symmetric about the diameter 0-180R. This simplifies the solution since only symmetric harmonics need be considered for the Fourier analysis, and only half the cooling tower need be meshed by the QSL8 elements. The load variation around the circumference is given in Figure 3.

#### **Discretisation**

The Fourier analysis utilises 10 elements to model the cooling tower length with 5 and then 10 harmonics taken to expand the variation of the load circumferentially. 10 QSL8 elements are also used to model the tower length, whilst 6 elements are used to model the circumferential section 0 to 180R.

## **Geometry**

The shell mid-surface is defined by the parabolic equation:

$$\left[\frac{x}{84}\right]^2 - \left[\frac{y - 270}{209.661}\right]^2 = 1\tag{0.0.33}$$

The overall height of the cooling tower is 330ft (100.584m) and has a constant wall thickness of 0.5833ft (0.1778m). The outside radii at the bottom and top of the tower are 136.96ft (41.745m) and 87.4ft (26.64m) respectively.

## **Boundary Conditions**

The cooling tower is encastre at the base. Symmetry boundary conditions are applied to the QSL8 mesh in cylindrical co-ordinates using transformed freedoms. Tangential displacements and loof rotations are restrained.

## **Material properties**

Young's modulus	4.320x10 <sup>8</sup> psf, 20.68x10 <sup>9</sup> Pa
Poisson's ratio	0.15

## **Theory**

The original problem was solved using a finite difference solution of the shell differential equations, with the circumferential variations modelled using a Fourier expansion. A fuller description is given in [1,2].

**Note:** the load was applied as a USER curve rather than using the Fourier coefficients given in [2] to allow a fairer comparison between the two analysis types.

#### **LUSAS Results**

Figures 1 & 2 show both the initial mesh and the final deformed shape from perspective and elevation views respectively. Figure 3 depicts the radial displacements for a node at the top of the tower; the crosses represent results from the Fourier analysis, the solid line corresponds to the shell analysis. The QSL8 radial displacements (Table 1) were obtained using option 115 to output displacements and reactions in transformed directions. Figure 3 also details the variation of wind load with circumference. Note that a positive face load acts in towards the surface.

Solution type	Radial displacement at the top edge
QAX8F (5 terms)	-0.2564 (Node 42, θ=0°)
QAX8F (10 terms)	-0.2565 (Node 42, θ=0°)
QSL8	-0.2493 (Node 21)

## References

- Program Verification and Qualification Library, ASME Pressure Vessel and Piping Division, Committee on Computer Technology, 1972.
- Albasiny, E.L. and Martin, D.W., Bending and Membrane Equilibrium in Cooling Towers, Jour. Eng. Mech. Div., ASCE, EM3, 1967.

## **Input Data**

X01D73A.DAT X01D73B.DAT

#### X01D73C.DAT

**Note:** The different results for taking 5, 10 terms of the Fourier expansion can be obtained by changing the range of the harmonic series defined in the FOURIER CONTROL data chapter.

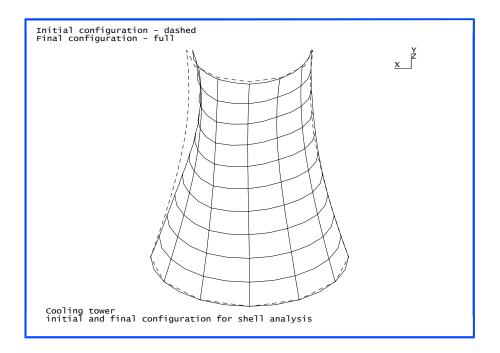


Figure 1

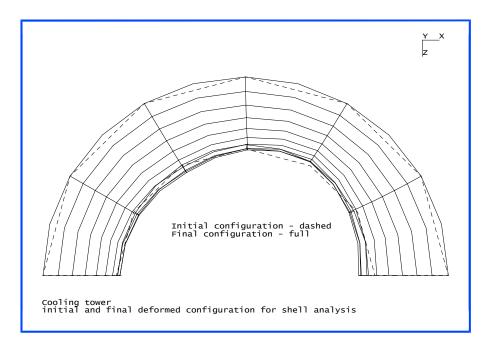


Figure 2 0.200 Circumferention angle (deg) 0.108 )ni( tnemecalpsiD 0.37 0 0 Ø.05 0.09 0.19 0.28 -0.075 Final deformed shape of top of cooloing tower -0.258 Final deformed shape of top of cooling tower -0.350 0.160 0.117 0.073 -6 0.030 -da o - 0.0130 - 0 0.073 Circumferention angle (deg) 0.19 0.37 0.09 0.14 0.23 0.28 0.32 -0.057 Load variation around circumference -0.100

Figure 3

Cooling tower: Final deformed configuration and load variation

# **Example 2.2.1**

# **Groundwater Seepage Problem**

## **Keywords**

Groundwater, Seepage, Cofferdam

## **Problem Description**

The problem considered in this example is the flow of groundwater beneath a cofferdam using a potential formulation approach. Figure 1 shows the dimensions of the problem under consideration.

#### **Finite Element Model**

The finite element mesh is shown in Figure 2, where 180 regular quadrilateral elements are used. The thickness of the sheet pile wall has not been modelled as this is assumed negligible compared to the overall dimensions of the problem. The sheet pile wall is represented by a boundary where no flow occurs.

## **Material properties**

Permeability  $K_X = K_Y = 0.864$  m/day

## **Boundary conditions**

It is assumed that the boundary ABC (see Figure 1) is impermeable as is the sheet pile wall EFG. The pressure head in this case is the difference in height between AG and DF which is 3.0 m. Arbitrarily setting  $\phi$  = zero along DF, since the flow velocities depend only on the gradient of  $\phi$  then  $\phi$  = 3.0 along AG. Along ABC and on either side of the sheet pile wall it is required that  $\partial \phi/\partial n = 0$  (see theory section). Also symmetry conditions along CD require  $\partial \phi/\partial n = 0$ .

The total flow rate may be obtained by summing the nodal reactions across section AG or DF on which the value of is prescribed.

## **Theory**

The steady state behaviour of many physical phenomena can be described in 2-dimensions by the following quasi-harmonic equation:

$$\frac{\partial}{\partial_{\mathbf{X}}} (\mathbf{K}_{\mathbf{X}} \frac{\partial \phi}{\partial_{\mathbf{X}}}) + \frac{\partial}{\partial_{\mathbf{Y}}} (\mathbf{K}_{\mathbf{Y}} \frac{\partial \phi}{\partial_{\mathbf{Y}}}) + \mathbf{Q} = \mathbf{0}$$
(0.0.34)

where  $\phi$  is the unknown function and Kx, Ky and Q are material parameters which can be functions of x and y. A number of field problems are governed by (0.0.34) and the physical interpretations of Kx, Ky and Q are listed on page 2.31 of the LUSAS User Manual.

Two main types of boundary condition are of interest;

(a) The value of the unknown to be specified at the nodal points on the boundary,

$$\phi = \phi_{\mathbf{p}} \tag{0.0.35}$$

(b) That a boundary loading exists of the form,

$$K_{x}\frac{\partial\phi}{\partial x}L_{x}+K_{y}\frac{\partial\phi}{\partial y}L_{y}+q+\alpha(\phi-\phi_{a})=0 \tag{0.0.36}$$

in which q, and  $f_a$  are constants and Lx, Ly are the direction cosines between the outward normal, n, and the x and y axes respectively.

Three different cases are now defined for boundary condition (b) assuming Kx = Ky = K.

Case 1. 
$$q = \alpha = 0$$
 
$$\frac{\P f}{\P n} = 0$$
 (0.0.37)

which implies the temp (/flow etc) gradient in a direction normal to the surface is zero i.e. this portion of the surface is perfectly insulated (/impermeable).

Case 2. 
$$\alpha = 0$$
  $K \frac{\P f}{\P n} = -q$  (0.0.38)

This states that a specified quantity of heat(/fluid), q, flows into the body per unit area of the surface. Also known as the 'flux' boundary condition.

Case 3. 
$$q = 0$$
  $K \frac{\P f}{\P n} = -a(f - f_a)$  (0.0.39)

The flow of heat (/fluid) from any point on the surface is directly proportional to the difference in the temperature (/potential) and the ambient temperature (/potential)  $\phi_a$ . Also known as the convection boundary condition.

Groundwater flow is given by:

$$\frac{\partial \phi}{\partial x} (Kx \frac{\partial \phi}{\partial x}) + \frac{\partial \phi}{\partial y} (Ky \frac{\partial \phi}{\partial y}) + S - \frac{dV}{dt} = 0$$
 (0.0.40)

where S is the rate fluid is injected into an element per unit volume and dV/dt is the rate of change in volume/unit volume in the element during flow. As stated previously (Case 1) the flow velocity in the direction normal to an impermeable boundary is equal to zero. From (0.0.34) and (0.0.40) we see a potential analysis problem reduces to a solution of the quasi-harmonic equation with:

$$Q = S - dV/dt \tag{0.0.41}$$

although S and dV/dt have no physical interpretation in ideal fluid flow.

## **Solution comparison**

	Flow rate value qm <sup>3</sup> /day				
Theory	8.6				
LUSAS	8.4				

Figure 3 shows a contour plot of the pressure head variation, thus each contour line is a line of equal potential or equipotential line.

#### References

 Hinton, E. and Owen, D.R.J. 'An introduction to Finite Element Computations', Pineridge Press, 1981.

## **Input Data**

X02D21A.DAT

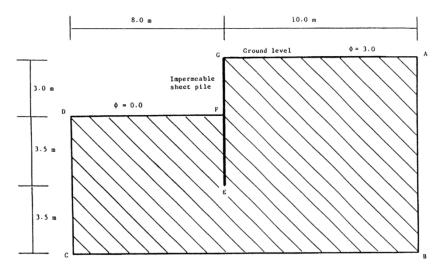


Figure 1. Groundwater flow beneath a coffer dam.

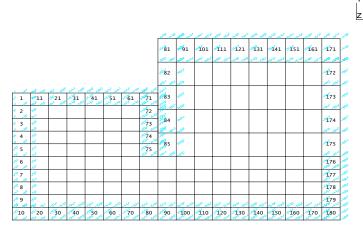


Figure 2. Figu. Finite element mesh with partial element and node numbering

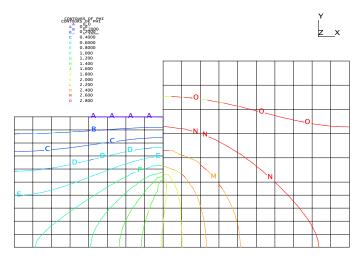


Figure 3.
Finite element mesh with lines of equal potential.

# **Example 2.2.2**

# **Steady State Thermal Analysis Of An Underground Tunnel**

## **Keywords**

Field, Steady State, Thermal Conduction, Tunnel

## **Problem Description**

To determine the steady state temperature distribution in the rock strata surrounding an operational underground rail tunnel.

Figure (1) shows a section of the Channel Tunnel at a distance of 32 km from the Dover entrance. The surrounding rock mass consists of chalk, chalk/marl and clay, the tunnel itself being confined by the chalk/marl stratum.

The heat generated within the tunnel is assumed to act as a line heat source to the rock mass. The origins of this heat source are considered to be the result of turbulent frictional contact at the train/air and air/tunnel boundaries, increased air pressure within the tunnel, and the operating temperature of the train itself.

In addition it is assumed that the sea bed is environmentally controlled at a temperature of 10.1 degrees, and that a geothermal gradient of 0.036 degrees per metre exists at this latitude.

#### **Discretisation**

The section through the rock mass is discretised using quadrilateral field elements (QDF8). The finite element mesh is graded towards the tunnel opening in order to accurately model the local temperature gradient anticipated in this region. The boundaries of the mesh are considered to be sufficiently removed from the tunnel so as to be subject to environmental conditions only.

The material properties for the rock strata are as follows:

-	K <sub>x</sub>	$K_{v}$	С
Chalk	10.58	7.958	2178
Chalk/Marl	7.642	7.642	2267
Clay	7.988	7.988	2353

## **Boundary Conditions**

Prescribed temperature conditions are imposed on the upper (sea bed) and lower mesh limits. Free conditions are imposed at the vertical mesh boundaries in order to impose a zero temperature gradient at these limits.

## **Theory**

The heat flux input resulting from an operational tunnel is calculated on the basis of twelve trains per 24 hours [1]. The total heat generated per metre of tunnel in this time is 10,500 KJ.

Two heat flux input conditions are considered. The first (gross heat flux) assumes that all the heat generated within the tunnel is available to increase the temperature of the surrounding rock. The second (net heat flux) assumes that a percentage of the total heat generated will be responsible for an increase in the train operating temperature. The net heat flux condition is calculated as a function of the number, type, and mass of each carriage, and the initial train and tunnel lining temperatures [1].

Gross heat flux condition q = 19.89 KJ per hr, per sq.m.

Nett heat flux condition q = 8.999 KJ per hr, per sq.m.

The steady state temperature distribution is governed by the two dimensional quasi-harmonic equation [2]

$$0 = \frac{d}{dx} \left( K_x \frac{dT}{dx} \right) + \frac{d}{dy} \left( K_y \frac{dT}{dy} \right) + Q \tag{0.0.42}$$

where

K<sub>x</sub>, K<sub>y</sub> are the orthogonal thermal conductivities

C is the material heat capacity

T is the local temperature variable

A simplified analytical solution may be obtained by the 'mirror image method' [1]. The steady state temperature at a point P which is at distances R1 and R2 from the heat sink and source respectively may be calculated from the expression:

$$T_{\rm P} = \frac{Q}{2\pi k} \left( \ln \left( \frac{R1}{R2} \right) \right) + T_{\rm O} \tag{0.0.43}$$

where

T<sub>P</sub> is the steady state temperature at point P

T<sub>O</sub> is the steady state temperature at the boundary

## **Comparison**

The temperature distributions (in degrees Celsius) obtained from the LUSAS steady state thermal analyses (gross and net heat flux input in the absence of a geothermal gradient) are compared to the analytical solution below.

		Gross Heat Flux		Nett H	eat Flux
Node	Distance(X)	Analytical temp	LUSAS temp	Analytical temp	LUSAS temp
42	20.0	24.118	24.997	16.487	17.791
44	60.0	15.606	17.179	12.592	14.252
46	160.0	11.401	13.231	10.688	12.465
48	280.0	10.556	12.313	10.310	12.050
50	400.0	10.334	12.074	10.206	11.942
52	700.0	10.177	11.956	10.131	11.888

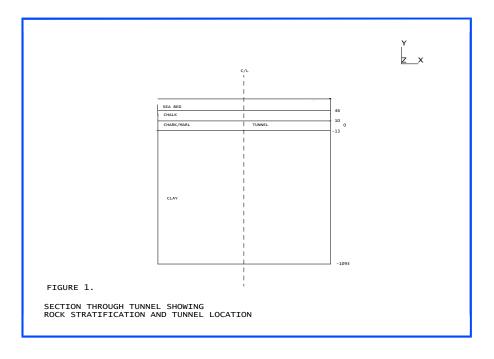
Note: The analytical solution does not account for any stratification in the surrounding rock mass.

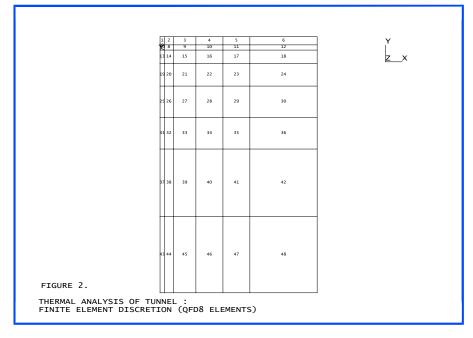
#### References

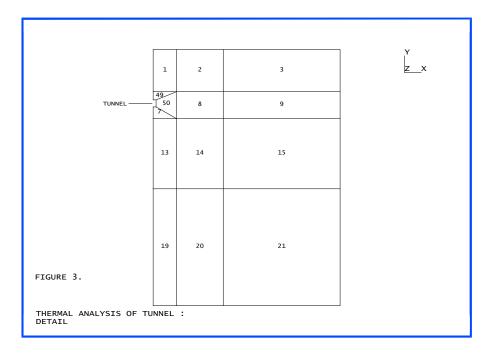
- Kent,A.J. 'The English Channel Tunnel: An investigation of rock and lining temperatures', Internal thesis CP/838/83, Dept. Civil Engineering, University, College of Swansea, U.K. (1983).
- Hinton, E. Owen, D.R.J. A simple guide to finite elements' Publisher. Pineridge press, Swansea. U.K.

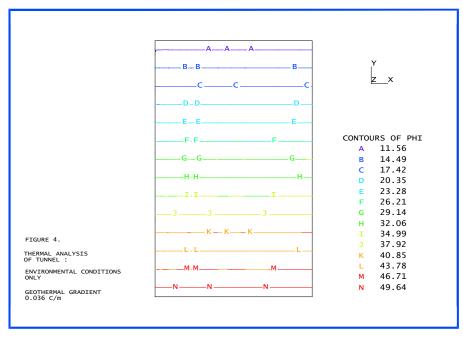
## **Input Data**

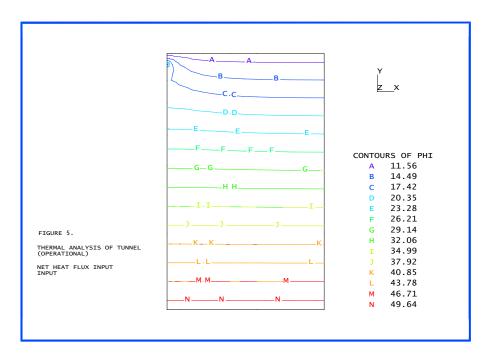
X02D22A.DAT

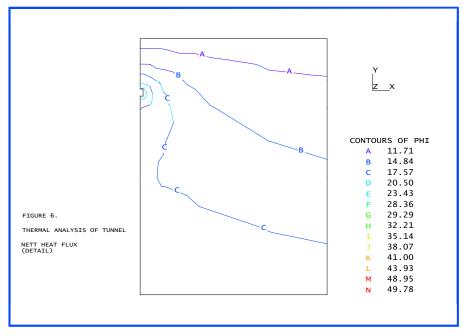


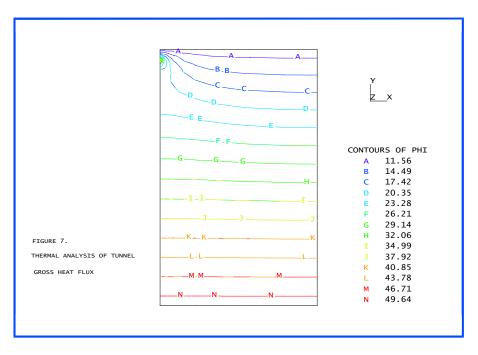


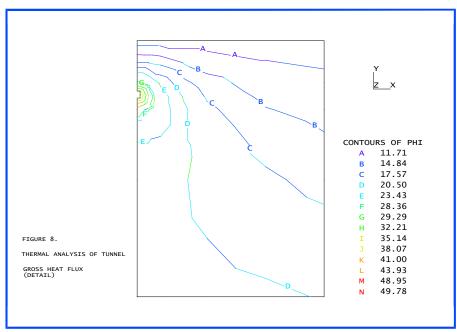


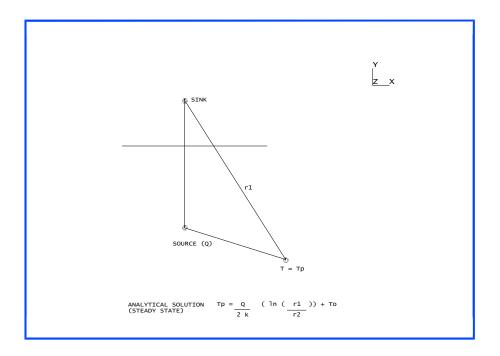












# **Example 2.2.3**

# **Conductance Between A Plate And Half Cylinder**

## **Keywords**

Conductance, Thermal Surfaces, Environmental Nodes, Field

## **Description**

Heat flows across the gap between a half cylinder and a plate, figure 1. The gap conductance is a linear function of gap opening and heat is allowed to flow to the environment. The problem is analysed with and without the use of environmental nodes for comparison. The first case corresponds to an infinite heat sink surrounding the structures and the second to an enclosure.

#### **Discretisation**

The plate and cylinder are modelled with QFD4 plane field elements and the element mesh is shown in figure 2.

## **Geometry**

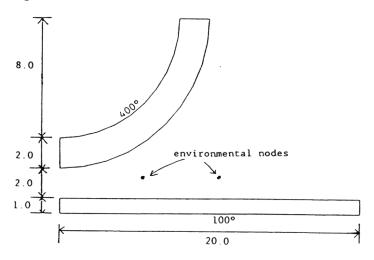


Figure 1 - Initial geometry

## **Boundary conditions**

The internal surface of the cylinder is at 400 degrees and the bottom of the plate at 100 degrees.

## **Material properties**

Solid conductivity = 80 W/m/degrees

Gap conductance

Conductance for closed gap ko = 100 W/m/degrees

Variation of conductance with gap opening, dk/dL = -12.5 W/m/m/degrees

Maximum limit for thermal link length, Lmax = 8 m

Surface properties

Conductance to environment = 2.0 W/m/degrees

Environmental temperature = 0.0 degrees

#### **LUSAS** results

Smooth temperature profiles should result with hot spots occurring at the point of closest approach of the plate to the half cylinder.

Figures 3 and 4 show the temperature distribution of the analysis with and without the environmental nodes. It is noticeable that the point at which the links from the plate intersect with the cylinder results in a fluctuation in the isotherm.

Figure 5 illustrates the change in element connectivity when environmental nodes are included in the solution. Introducing these environmental nodes has the effect of increasing the surface area of the cylinder which is at the temperature of the inner surface. That is, they have the effect of insulating the surface. This is expected since heat cannot be lost from the solution, but must flow from the cylinder to the plate via the environmental nodes (the temperature distribution is seen to be slightly different in Figure 4 which shows that more of the cylinder remains at a higher temperature). The heat flows through the plate for the different analyses are:

Without environmental nodes = 39418 W

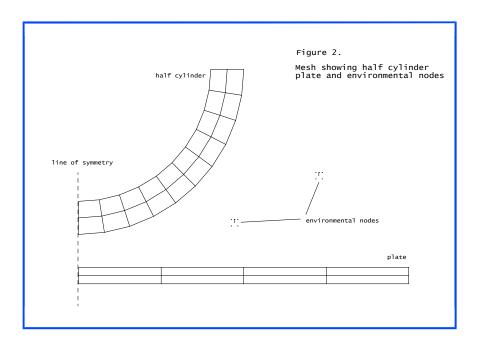
With environmental nodes = 43967 W

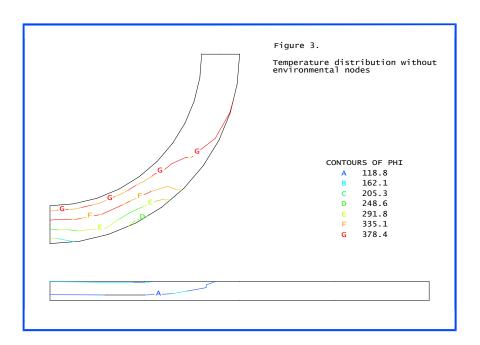
For this example, the introduction of environmental nodes increases the heat flow by approximately 10%

## **Input data**

X02D23A.DAT

X02D23B.DAT





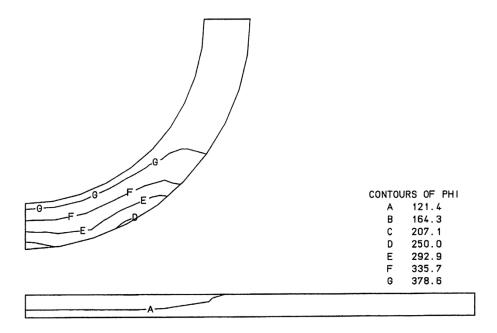
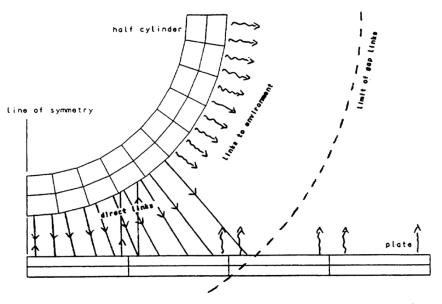


Figure 4 Temperature distribution with 2 environmental nodes



(a) links formed without environmental nodes

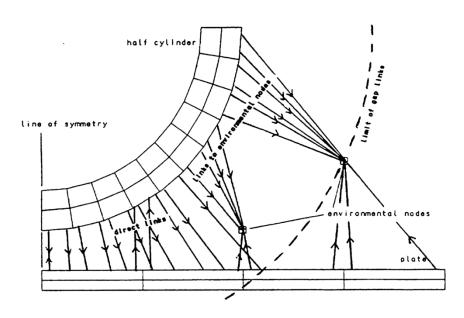


Figure 5 Thermal link formation

# **Example 2.3.1**

# **Cylinder With Temperature Dependent Conductivity**

## **Keywords**

Steady State, Conductivity

## **Description**

This problem involves a hollow cylinder with prescribed temperatures around the inner and outer circumferences. It is required to find the steady state temperatures at intervals of 1/5 of the cylinder thickness. The convergence rate and CPU time for HF8 and QXF8 element discretisations using both MNR and NR solution procedures are also compared. This problem is an application of LUSAS to solve steady state heat conduction problems with temperature dependent materials.

#### **Discretisation**

15 Solid HF8 elements are used to model a 15R slice of the cylinder, figure 3. 5 QXF8 elements are used for the axisymmetric analysis, figure 2.

## **Geometry**

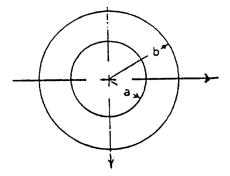


Figure 1

Inner radius, a 0.041666 ft
Outer radius, b 0.083333 ft

## **Material properties**

$C_0$	50.0 BTU/hr.ft. F
$\mathbf{C}_1$	0.5 BTU/hr.ft. F
K(t)	$C_0 + C_1 t$

## **Boundary conditions**

Prescribed temperature T=100RF around inner circumference r=a, and T=0RF, around outer circumference r=b.

#### Reference

 Schneider, P. J., Conduction Heat Transfer, Addison-Wesley Publishing Co., Inc., Reading, Mass., USA, 2nd Printing, 1957.

### **LUSAS** results

	Point 1	Point 2	Point 3	Point 4	Point 5	Point 6
REFERENCE	100.0	79.2	59.6	40.2	20.8	0.0
HF8	100.0	79.222	59.567	40.298	20.720	0.0
QXF8	100.0	79.190	59.490	40.206	20.665	0.0

Comparison of NR and MNR

	NR	CPU	MNR	CPU
HF8	4	67.46	10	88.02
QXF8	4	20.70		

The QFX8 element solution failed to converge using MNR after 20 iterations. At the time of ceasing computation the MNR solution appeared to be exhibiting a very slow rate of convergence.

## **Input Data**

X02D31A.DAT

X02D31B.DAT

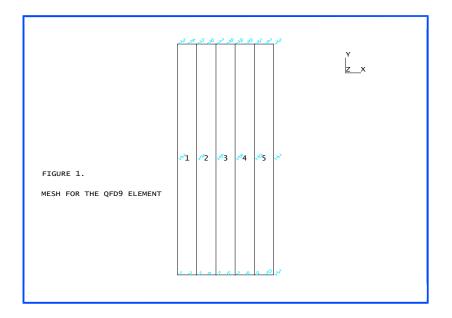


Figure 2

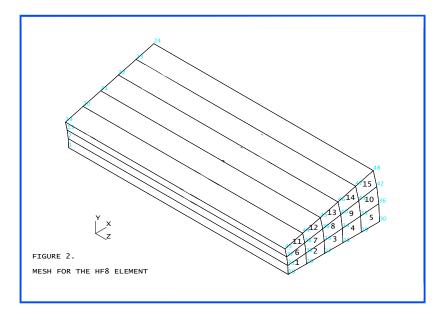


Figure 3

# Example 3.1.1

# **Frequency Analysis Of A Simply Supported Beam**

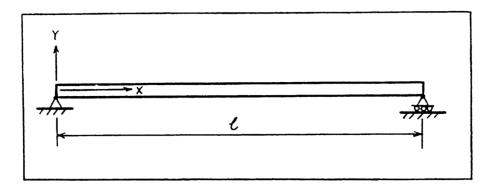
# **Keywords**

Eigenvalue

#### **Description**

Determine the first three fundamental frequencies of the following simply supported beam.

Length, l	2.032 m
Area, A	$2.58x10^{-3} \text{ m}^2$



#### **Discretisation**

Model the complete beam to include both symmetric and non-symmetric modes.

## **Geometric properties**

## **Material properties**

Young's modulus, E	$30 \text{ x} 10^6 \text{ kN/m}^2$
Poisson's ratio, v	0.3
Density, p	$7780 \text{ kg/m}^3$

## **Theory**

The fundamental frequencies may be determined from the following relationships;

$$\omega_{n} = (K_{n})^{2} \sqrt{(EI/ml^{4})}$$
 
$$f_{n} = \frac{\omega_{n}}{2\pi}$$
 (0.0.44) 
$$I_{n} = w_{n}^{2} = eigenvalue$$
 
$$K_{n} = np$$
 
$$m = mass/unit length = r A$$

# Comparison

Mode	Exact	LUSAS
$\mathbf{f}_1$	28.763	28.7663
$f_2$	115.05	115.400
$f_3$	258.86	263.016

#### References

 W.C. Hurty and M.F. Rubinstein, Dynamics of structures, Prentice-Hall, 1964. Chap. 5, p 203.

# **Input data**

X03D11A.DAT

# **Example 3.1.2**

# **Frequency Analysis Of A Sprung Mass**

### **Keywords**

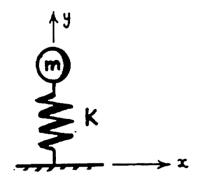
Eigenvalue

### **Description**

Determine the natural frequency of vibration (f) of a mass, m, set on a rubber mount system with a stiffness K.

Mass (m) = 0.00647668 lb-sec<sup>2</sup>/in.

Stiffness of mount (K) = 48.0 lb/in.



#### **Discretisation**

Model with single joint element

# Theory

The natural frequency may be determined from the following relationships;

$$\omega = \sqrt{\frac{K}{m}} \qquad \qquad \text{frequency} = \frac{\omega}{2\pi} \qquad (0.0.45)$$

 $I = w^2 = eigenvalue$ 

# Comparison

	Exact	LUSAS	
Frequency (Hz)	13.7014	13.7014	

#### References

1. W.C. Hurty and M.F. Rubinstein, Dynamics of structures, Prentice-Hall, 1964.

# Input data

X03D12A.DAT

# **Example 3.1.3**

# **Linear Buckling Analysis Of A Simple Portal Frame**

### **Keywords**

**Linear Buckling** 

### **Problem Description**

Determine the first two critical buckling loads for the rigid jointed portal frame shown in Figure 1. Compare the first in-plane critical load with a theoretical solution.

#### **Discretisation**

Model the rigid jointed portal frame using a coarse mesh of three BM3 beam elements and a finer mesh of nine BM3 or 18 BMI21 elements.

### Geometry

The vertical members of the portal frame are 10m in length while the horizontal member is 8m in length (see Figure 1).

## **Geometric properties**

All members have a 0.1m square, solid section. The sections properties of the members are listed in the following table.

Cross-sectional area	0.01 m <sup>2</sup>
Second moments of area $(I_{yy}, I_{zz})$	8.33333x10 <sup>-6</sup> m <sup>4</sup>
Torsional constant	14.0833x10 <sup>-6</sup> m <sup>4</sup>
Effective shear areas (A <sub>sy</sub> , A <sub>sz</sub> )	$8.333333x10^{-3} m^2$

# **Material properties**

All members have steel material properties (UK, BD2101), as listed below.

Young's modulus	$20.0x10^9 \text{ N/m}^2$
Poisson's ratio	0.3
Density	$7.85 x 10^3 \text{ Kg/m}^3$

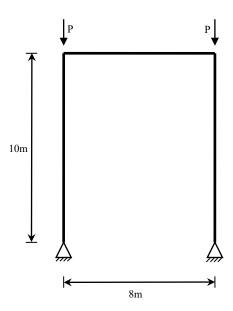


Figure 1: Rigid jointed portal frame used in linear buckling analysis. Loads P are unit loads.

### Loading

The portal frame is subjected to unit loads applied to the ends of the vertical members in the vertical direction (Figure 1).

## **Boundary conditions**

The portal frame is rigidly supported at the base nodes (Figure 1).

## **Theory**

The theoretical solution for the in-plane critical buckling load, P<sub>cr</sub>, is [1]

$$P_{cr} = k^2 E I_1 \tag{0.0.46}$$

where  $I_1$  is the second moment of area of the horizontal member. The term k is found by solving the following equation.

$$\frac{k1}{\tan(k1)} = -6\frac{1I_1}{bI_2} \tag{0.0.47}$$

where l is the length of vertical member, b the length of horizontal member and  $I_2$  the second moment of area of the vertical member.

With the material and geometric properties used in this example, the critical buckling load using equation (0.0.47) is

$$P_{cr} = 132.6 \times 10^3 \text{ N}$$

# **Comparison**

The results from the theory, the coarse mesh and the fine meshes are listed in the table below

Analysis		P <sub>cr</sub> (N)
Theory		$132.6 \times 10^3$
Coarse mesh	BM3	$147.9 \times 10^3$
Fine mesh	BM3	$132.8 \times 10^3$
	BMI21	$135.3 \times 10^3$

#### References

 S.P. Timoshenko and J. M. Gere, "Theory of Elastic Stability."2nd ed. McGraw-Hill Kogakusha Ltd., 1961, pp 149-150

# **Input data**

 Coarse mesh (BM3)
 : X03D13A.DAT

 Fine mesh (BM3)
 : X03D13B.DAT

 Fine mesh (BMI21)
 : X03D13C.DAT

# **Example 3.3.1**

# **Cantilevered Thin Square Plate**

# **Keywords**

Eigenvalue, Guyan Reduction

### **Description**

A square plate with side length 10m and uniform thickness 0.05m is restrained against translation and rotation along the X=0 edge, figure 1. It is required to find the lowest 6 eigenmodes of the plate (shown in figures 2 to 7) using:

- i) Subspace iteration
- ii) Guyan reduction with 12 automatic master freedoms
- iii) Guyan reduction with 20 automatic master freedoms
- iv) Guyan reduction (with automatic masters) as a first approximation to the subspace iteration vectors
- v) Guyan reduction with 20 specified master freedoms
- vi) Guyan reduction (with specified masters) as a first approximation to the subspace iteration vectors

This example compares the convergence of Guyan reduced solutions with that of full subspace iteration analysis as master density is increased. Further, it illustrates the use of a Guyan reduced solution as a starting point for subspace iteration analyses.

#### **Discretisation**

The whole plate is discretised using a regular mesh of 16 (4\*4) quadrilateral 8-noded Semiloof shell elements (QSL8).

## **Material properties**

Young's modulus	$200.0x10^6N/m^2$
Poisson's ratio, v	0.3
Mass density, $\rho$	$8000 \text{ Kg/m}^3$

## **Theory**

Thin plate predictions of the natural frequencies of a cantilevered square plate of side length (a) and thickness (t) are given by the following relationship:

$$f_{ij} = \frac{\lambda_{ij}^2}{2\pi a^2} \left[ \frac{Et^2}{12\rho(1-\upsilon^2)} \right]^{1/2}$$
 (0.0.48)

For a Poisson's ratio of 0.3, the values tabulated below may be obtained:

Mode	l² <sub>ij</sub>	f <sub>ij</sub> (Hz)
1	3.494	0.421
2	8.546	1.029
3	21.440	2.582
4	27.460	3.306
5	31.169	3.753
6	54.440	6.555

#### **Lusas results**

The results obtained from the LUSAS analyses are presented below, and are compared with those given in [2]:

Mode Number	1	2	3	4	5	6
Reference solution	0.421	1.029	2.582	3.306	3.753	6.555
Subspace iteration eigenvalue evaluation	0.415	1.013	2.502	3.233	3.684	6.422
Guyan reduction 12 automatic masters	0.416	1.025	2.797	3.552	4.121	9.399
Guyan reduction 20 automatic masters	0.415	1.014	2.583	3.293	3.908	7.336
Guyan reduction with automatic master approximation	0.415	1.013	2.501	3.233	3.684	6.422
Guyan reduction 20 specified asters	0.415	1.014	2.519	3.274	3.746	6.762
Guyan reduction with specified master approximation	0.415	1.013	2.502	3.233	3.684	6.422

#### Note

- 1. The relatively poor solutions obtained using automatic master selection with Guyan reduction are due to the fact that the K/M ratios are constant for all internal nodes.
- The subspace iteration procedure converges in less iterations when the eigenvectors of the Guyan reduction analysis are used as starting iteration vectors.

#### References

- Abbassian, F., Dawswell, D.J., Knowles, N.C. "Free Vibration Benchmarks", Tests 73 and 74, NAFEMS Report FVB, EAD 6.11.87, (November 1987), Report prepared by W.S. Atkins Engineering Science for National Agency for Finite Element Methods and Standards.
- Blevins, R.D., "Formulas for Natural Frequency and mode shape", Publishers Van-Nostrand, (1979).
- LEISSA,A.W., "Vibration of plates", National Aeronautical and Space Administration, N.A.S.A Report SP-160, (1969)

#### **Input Data**

X03D31A.DAT

X03D31B.DAT

X03D31C.DAT

X03D31D.DAT

X03D31E.DAT

X03D31F.DAT

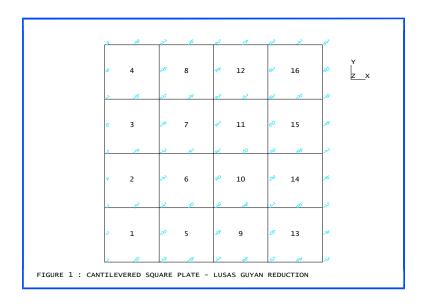


Figure 1

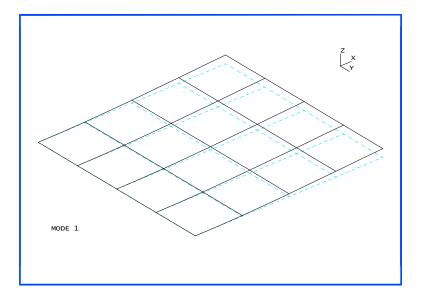


Figure 2

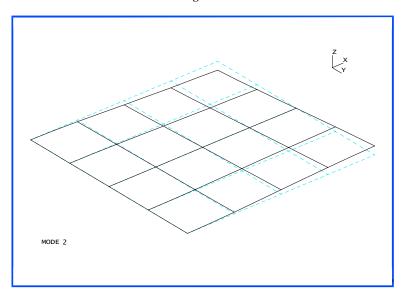


Figure 3

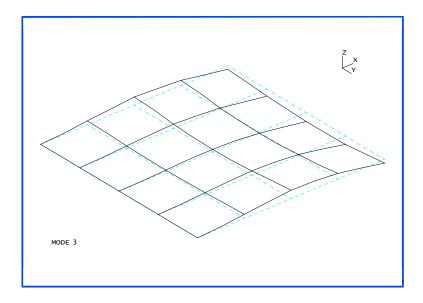


Figure 4

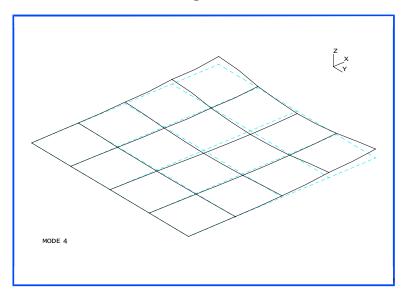


Figure 5

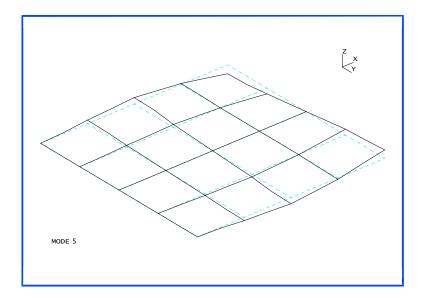


Figure 6

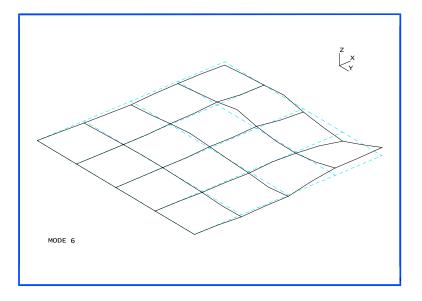


Figure 7

# **Example 3.4.1**

#### **Bifurcation Of A Double Arch**

# **Keywords**

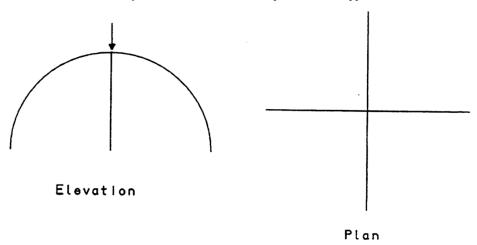
Total Lagrangian, Large Rotation, Eigenvectors Of Stiffness Matrix

### **Description**

Two orthogonal arches, joined at their crowns, are subject to a point load applied in the vertical direction. The structure is loaded to close to its critical load and the eigenvectors corresponding to the buckling modes are evaluated.

#### **Finite Element Model**

The arches are modelled by 60 BTS3 beams, which are pinned at the supports.



## Geometry

Radius of the arch = 5

Angle subtended by arch = 180 degrees

Beam geometric properties

Area	= 0.1
2nd moment of area about y-axis I <sub>yy</sub>	= 0.4E-5
2nd moment of area about z-axis $I_{zz}$	= 0.8E-5
Torsion stiffness K <sub>T</sub>	$= 0.1406 \times 10^{-4}$
Shear area As <sub>y</sub>	= 0.1
Neutral axis eccentricity e <sub>z</sub>	= 0.0

# **Material properties**

Young's modulus	210.0x10 <sup>9</sup>
Poisson's ratio	0.0

#### Load

Load applied in (downwards) vertical direction at joint of arches in three

## **Steps**

step 1 - 1.5\*10<sup>5</sup> step 2 - 3.0\*10<sup>5</sup> step 3 - 4.042\*10<sup>5</sup>

# **Boundary Conditions**

Beam ends are free to rotate but are restrained against translations in the x, y and z directions.

## **Solution Comparison**

The eigenvalues extracted at near the critical load are:

Mode	Eigenvalue
1	-0. 398217
2	-0. 398217
3	1025.41
4	13100.2
5	13347.1
6	14828.8

At a critical load corresponding to a bifurcation or limit point, the determinant of the stiffness matrix is zero or equivalently one or more of the eigenvalues of the stiffness matrix are zero. The negative eigenvalues indicate that the load applied is slightly above the critical buckling load but these numbers are close enough to zero to assume that the buckling load is given by the load applied. The eigenvector corresponding to the lowest eigenvalue is the buckling mode. In this instance, the double symmetry of the combined arches leads to a compound bifurcation involving two buckling modes as shown in figure 1. Further higher modes are also presented in figure 1.

#### **Input Data**

X03D41A.DAT

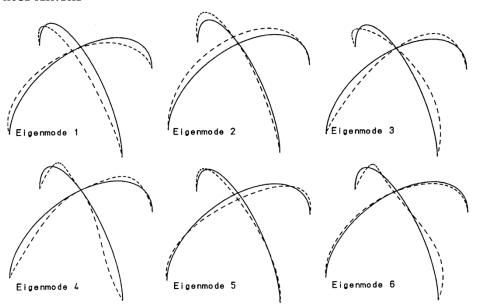


Figure 1 Eigenmodes of Double Arch

# **Example 3.5.1**

# **Natural Frequency Analysis Of A Solid Cantilever**

# **Keywords**

Eigenvalue, Natural Frequency

### **Description**

To determine the first four natural frequencies of the cantilevered beam shown in Figure 1.

#### **Finite Element Model**

The cantilevered beam consists of four HX20 3-D solid elements with the element and node numbering detailed in Figure 2.

## **Material properties**

Young's modulus	$30.0x10^3 \text{ N/m}$
Mass density	5.0 kg/m

# **Boundary conditions**

All nodes in the y-z plane at x=0 are fully restrained resulting in an encastre support condition.

## **Theory**

The natural frequencies of beams with various end constraints can be found in [1]. The natural frequencies, w, of a cantilevered beam may be determined from the following equation:

$$\mathbf{w}_{\mathbf{n}} = \left(\mathbf{B}\mathbf{I}\right)_{\mathbf{n}}^{2} \left(\frac{\mathbf{E}\mathbf{I}}{\mathbf{m}\mathbf{I}^{4}}\right) \tag{0.0.49}$$

where the Bl factors for the first four natural frequencies are given by

 $B1_1 = 1.872$ 

 $Bl_2 = 4.694$ 

 $Bl_3 = 7.855$ 

 $Bl_4 = 10.9955$ 

The other terms appearing in the equation for the natural frequencies are, for this particular example

 $E = Young's modulus = 30,000 N/m^2$ 

I = second moment of inertia = 8.33333E-02 m<sup>4</sup>

m = mass per unit length = 5.00 Kg/m

1 = beam length = 10 m

## **Modelling Hints**

This simple example has been chosen for two main reasons. The first is to enable a comparison to be made with theoretical values, and the second is to highlight LOCAL CARTESIAN COORDINATES, COPY NODES and EIGENVALUE CONTROL facilities. Comments on the use of these commands have been included in the data file printout.

The first side of the beam is defined by using QUADRILATERAL SPACING and SIDE POINTS. The mid-height and full height nodes are described by defining a LOCAL CARTESIAN COORDINATE system about these heights and using COPY NODES to copy a repetitive pattern of nodes to a new position and orientation. In this case from the lower surface to the mid-height and full height. These coordinates are then transformed to the global system by using the command GLOBAL CARTESIAN COORDINATES.

It must be noted that because of the regular geometry, a more convenient method of generating the nodal coordinates, would be by using the INCREMENTAL line generation facility.

#### e.g. NODE COORDINATES

FIRST	1	0	0	0	
INC	1	0	0	0.5	3
INC	3	0	0.5		(3)
INC	9	0.125			(9)

### **Solution Comparison**

Mode	1	2	3	4
Theory	0.6179	0.6179	24.27	24.27
LUSAS	0.6168	0.6168	24.00	24.00

#### Note:

The coefficients (Bl) and (Bl) defined in the theory section refer to modes in 2-D, and are repeated in the 3-D case. So modes 1 and 2 in 2-D correspond to modes 1, 2, 3 and 4 respectively in 3-D.

# References

1. Hurty, W.C. and Rubinstein, M.F., 'Dynamics of Structures', Prentice-Hall Inc.

## **Input data**

X03D51A.DAT

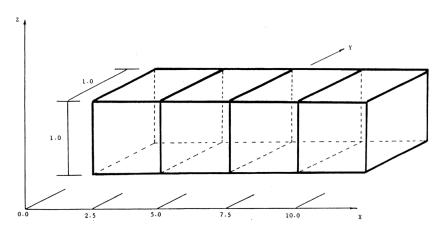
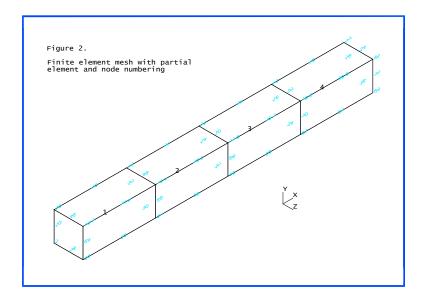


Figure 1. Cantilever beam made up of 4 HX20 type elements.



# **Example 3.6.1**

# **Buckling Analysis Of A Rectangular Panel**

# **Keywords**

**Buckling, Eigenvalue, Semiloof** 

#### Description

Determine the critical buckling load for a rectangular panel subject to in-plane compressive loading. The dimensions of the panel are as follows:

Length, a	2 m
Width, b	0.5 m
Thickness, t	0.001 m

The panel is subjected to an in-plane compressive load acting along the short edges.

#### **Discretisation**

The whole structure is discretised using 16 Semiloof shell elements (QSL8) and is simply supported on all sides. The in-plane compressive loading is applied as a series of concentrated loads acting in the plane of the structure parallel to its long sides. A total load of 24 N is applied to each loaded edge.

### Material properties

Young's modulus	70x10 <sup>9</sup>
Poisson's ratio	0.3

### **Theory**

For a given integer side length ratio (a/b) and number of half waves in the buckled configuration (m), the critical buckling load is given by the expression

$$N_{xcrit} = \frac{\pi^2 D}{b^2} \left( m \frac{b}{a} + \frac{1}{m} \frac{a}{b} \right)^2$$
 (0.0.50)

For 4 half waves and a side length ratio of 4, (0.0.50) therefore reduces to

$$N_{xcrit} = \frac{4^2 \pi^2 D}{h^2}$$
 (0.0.51)

Where

$$D = \frac{Et^3}{12(1-v^2)}$$
 (0.0.52)

The critical buckling stress is given by the expression

$$\sigma_{\text{xcrit}} = \frac{N_{\text{xcrit}}}{t} \tag{0.0.53}$$

or, combining (0.0.51)-(0.0.53) by

$$\sigma_{\text{xcrit}} = \frac{4\pi^2 \text{Et}^3}{12b^2 (1 - v^2)t}$$
 (0.0.54)

# Comparison

The results obtained from the LUSAS analysis are compared to the theoretical results. From (0.0.54) the theoretical critical buckling stress is given by the expression

$$\sigma_{\text{xcrit}} = 3.62 \text{ E} \left(\frac{\text{t}}{\text{b}}\right)^2$$

or

$$s_{xerit} = 1.013 \text{ MN} / \text{m}^2$$

Results from the LUSAS analysis were as follows:

#### Eigenvalues after 11 iterations

Mode	Eigenvalue	Load factor	Error norm
1	19.1316	19.1316	0.678980E-01

The critical buckling load and corresponding stress are

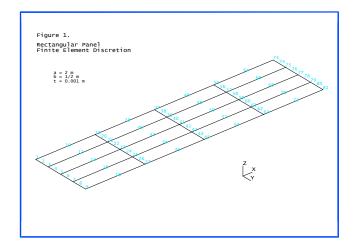
$$N_{xcrit} = 19.1316*24 = 459.184 \text{ N}$$
  
 $s_{xcrit} = 459.1584/(Area) = 459.1584/(0.5*0.001) = 0.918316 \text{ MN/m}^2$ 

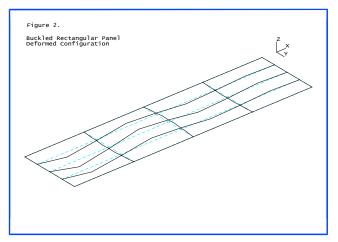
#### References

1. 'Formulas for stress and strain', Fourth Edition, R.J.Roark, Publisher. McGraw-Hill Book Co. Kogakusha Co. Ltd. Tokyo (1965).

## **Input data**

X03D61A.DAT





# **Example 3.7.1**

# **Eigen-Analysis Of A Cooling Tower**

#### **Keywords**

Fourier, Semiloof Shell, Eigenvalue

#### **Description**

The first four modes of a cooling tower are extracted. Both Fourier and Semiloof elements are used and the results compared.

#### **Discretisation**

The cooling tower is modelled initially by 10 QAX8F Fourier elements. The circumferential number of harmonics is varied from 0 to 6 with an eigenvalue analysis performed for each harmonic.

The QSL8 elements are utilised with two meshes; 10 elements are used in both to model the length of the cooling tower, whilst 12 and 48 elements respectively are used to represent coarse and fine modelling of the circumference.

### Geometry

The shell mid-surface is defined by the parabolic equation,

$$\left[\frac{x}{84}\right]^2 - \left[\frac{y - 270}{209.661}\right]^2 - 1\tag{0.0.55}$$

The overall height of the cooling tower is 330ft (100.584m) and has a constant wall thickness of 0.5833ft (0.1778m). The outside radii at the bottom and top of the tower are 136.96ft (41.745m) and 87.4ft (26.64m) respectively.

### **Boundary Conditions**

The cooling tower is assumed to be encastre at the base.

### **Material properties**

Young's modulus	432.0 x10 <sup>6</sup> psf, 20.68x10 <sup>9</sup> Pa
Poisson's ratio	0.15

### **Theory**

The eigenvalues obtained from the two analysis types are compared.

#### **LUSAS** results

Figures 1 and 2 show the first four eigenvectors from the coarse and fine shell analyses respectively. The extracted frequencies from the Fourier analysis are detailed in Table 1 below.

**Table 1: Fourier frequencies** 

Eigenvalue	Harmonic Number						
Number	0	1	2	3	4	5	6
1	0.1855	0.0788	0.0423	0.0332	0.0296	0.0283	0.0333
2	0.2805	0.1633	0.0889	0.0483	0.0360	0.0378	0.0387
3	0.2978	0.2544	0.1683	0.1052	0.0679	0.0514	0.0530

The eigenvalues obtained from the Fourier analyses indicate that the lowest four modes correspond to the n=5,4,3,6 harmonics. However, the frequencies obtained from the coarse shell analysis (see Table 2) produce higher frequencies and ordered the harmonics as n=4,3,5,6. The inability of the structure to deform into its true eigenmodes is due to the coarse mesh discretisation and the resulting stiffer structural response. A refined QSL8 mesh produces answers that are in closer agreement with the Fourier analyses and predict a harmonic order of n=5,4,3,6.

Table 2: Frequency comparison

Harmonic Number	Fourier	Course shell	Fine shell
n=3	0.03318	0.03423	0.03356
n=4	0.02958	0.03319	0.02994
n=5	0.02829	0.03637	0.02871
n=6	0.03333	0.03788	0.03394

Using the QSL8 elements, two repeated eigenvalues are obtained for each mode which are outof-phase with each other (not always by 90 degrees). For a cylindrical structure there are an infinite number of eigenvectors for each harmonic, but all can be modelled by two out-of-phase eigenvectors. Relative solution times for the analyses are detailed in Table 3, from which it may be seen that the Fourier analysis compares favourably with the shell analyses.

**Table 3: Solution times** 

Fourier (QAX8F)	QSL8(coarse)	QSL8(fine)
1	7	52

The mode number n can sometimes be very large for thin shell structures, and in these instances, the Fourier solution quickly provides a sweep of selected harmonics n. If the precise eigenvalues from a shell analysis are required, the knowledge provided from the Fourier analysis can be used to decide on the mesh discretisation for a complete model of the structure or on the selection of a segment of the shell to which symmetric and asymmetric boundary conditions can be applied in turn.

#### References

- Program Verification and Qualification Library, ASME Pressure Vessel and Piping Division, Committee on Computer Technology, 1972.
- 2. Albasiny, E.L. and Martin, D.W., Bending and Membrane Equilibrium in Cooling Towers, Jour. Eng. Mech. Div., ASCE, EM3, 1967.

### **Input Data**

X03D71A.DAT

X03D71B.DAT

X03D71C.DAT

X03D71D.DAT

X03D71E.DAT

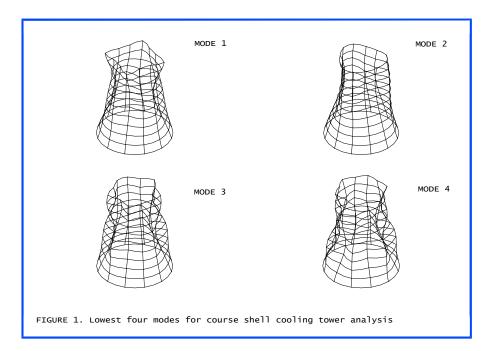
X03D71F.DAT

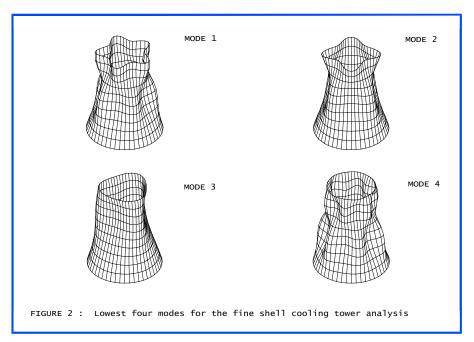
X03D71G.DAT

X03D71H.DAT

X03D71I.DAT

Note: The different results for taking 0,1,2... terms of the Fourier expansion can be obtained by changing the range of the harmonic series defined in the FOURIER CONTROL data chapter.





# Example 4.1.1

# **Geometrically Nonlinear Analysis Of A Cantilevered Beam**

# **Keywords**

**Updated Lagrangian, Geometric Nonlinearity** 

### **Description**

Determine the large deformation response of the thin, cantilevered strip, shown in Figure 1, subject to a point load at the tip.

#### **Discretisation**

Model the cantilever using eight, equal length, two dimensional BM3 beam elements (see Figure 1).

# **Geometry**

The cantilever has a width of 1 inch, a depth of 0.1 inches and the nodal line is assumed to coincide with the centroid of the cross section. Thus, each element has the following geometric properties;

Cross-sectional area  $= 0.1 \text{ inch}^2$ 

Second moment of area about the local z axis = 0.8333E-4 inch<sup>4</sup>

First moment of area about the local z axis  $= 0.0 \text{ inch}^3$ 

## **Material properties**

Young's modulus	120.0x10 <sup>6</sup> lb/in <sup>2</sup>
Poisson's ratio	0.3

## **Boundary conditions**

The cantilever is rigidly supported at node 1 (see Figure 1).

# **Theory**

This problem is one of the classical elastic problems which have analytical solutions based on elliptical integrals. Reference 1 gives further details on this method of solution and also presents a table of tip displacements and rotations for various load levels.

### **Comparison**

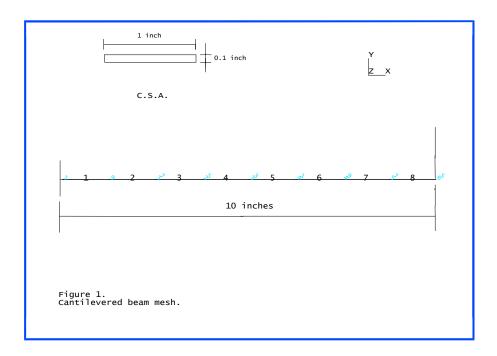
A comparison of the theoretical and LUSAS results is given in Figure 2. Theta is the rotation at the tip in radians. Figure 3 shows the deformed configurations for various PL\*\*2/EI values.

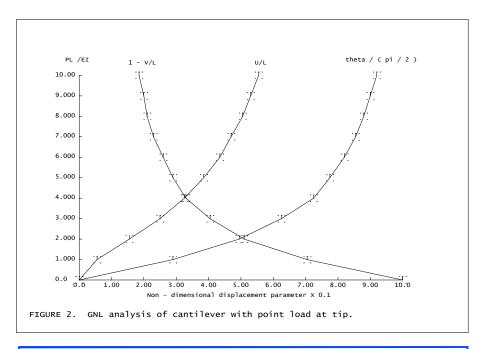
#### References

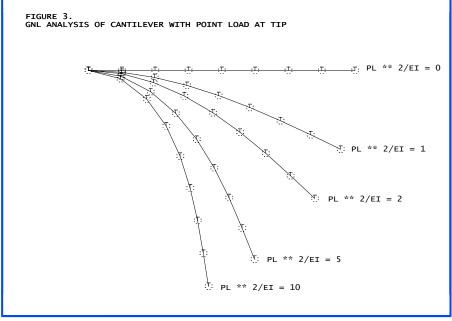
 TIMOSHENKO, S.P. and GERE, J.M. Mechanics of materials. Van Nostrand Reinhold Co., S.I. ed., 1973.

#### **Input Data**

X04D11A.DAT







# **Example 4.1.2**

# **Bifurcation Of Simple Bar-Spring System**

# **Keywords**

Bar, Spring, Bifurcation

### **Description**

The bar-spring system shown in Figure 1 is subjected to an axial load. The system is stable until the horizontal load reaches the buckling load. At this point, the solution is forced to branch on to an unstable path and the end of the bar that is attached to the spring begins to move vertically. By continuing the solution further, the vertical displacement of the end of the bar reaches a peak and then reduces towards zero. The bifurcation path is followed until a second bifurcation point is encountered, Figure 2.

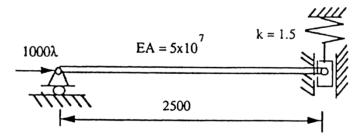


Figure 1: Bar-spring system

#### **Discretisation**

The system consists of a horizontal bar element BAR2 with area A=1.0 connected to a vertical spring element JNT3 at one end.

## **Material properties**

BAR3: Young's Modulus =  $.5E8 \text{ KN/m}^2$ 

Poisson's ratio = 0.0

JNT3: Spring stiffnesses = 0.0 (x direction)

= 1.5 (y direction)

### **Theory**

The equilibrium equations for the system can be written as:

$$\begin{cases}
F \\
0
\end{cases} = \begin{bmatrix}
EA/L & 0 \\
0 & k+N/L
\end{bmatrix} \begin{cases}
\Delta u \\
\Delta w
\end{cases}$$
(0.0.56)

where L is the length of the bar and N the axial force in the bar; the vertical spring stiffness is k.  $\Delta u$  and  $\Delta w$  are increments in the displacements shown in Figure 1. The bar-spring system is stable until the stiffness matrix becomes singular. This occurs when N = -Lk, which is equal to 3750 for the parameters used in this problem. As the axial force in the bar is given by F=EA/L\*u, the axial displacement at the loaded end at the bifurcation point can be computed as u=0.1875.

### Comparison

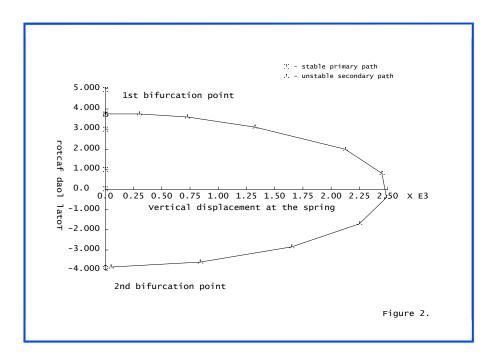
Figure 2 shows the stable primary solution path which was obtained by carrying out a nonlinear analysis with bracketing. It can be seen that the vertical displacement was zero along this path. The bifurcation point that was bracketed is shown to be at the point F=3749.7, w=0, u=0.187506 which agrees very closely with the theoretical values. The unstable secondary (or bifurcated) path is also shown in the figure and it can be seen that the vertical displacement peaks at a value of just under 2500 while the horizontal load varies between values of +/- 3750. This path was obtained after a restart analysis by utilising the branching option within the nonlinear control data section of the data file. The solution is continued along the unstable path until a second bifurcation point is encountered, Figure 2. It is only possible to bracket one bifurcation point therefore the solution along this unstable path cannot progress any further.

#### References

1. Non-linear Finite Element Analysis of Solids and Structures (volume 1), M.A. Crisfield. John Wiley & Sons Ltd., England (1991).

## **Input Data**

X04D12A.DAT



# Example 4.3.1

# Free Vibration Analysis Of A Rotating Blade.

### **Keywords**

Centripetal Stiffening, Rotating Machinery, Total Lagrangian, Geometric Nonlinearity

### **Description**

Determine the free vibration response of a cantilevered plate rotating at 20000 rpm. The dimensions and details of the blade are shown in Figure 1.

#### **Discretisation**

Model the blade structure using four BSL4 Semiloof beam elements (see Figure 2).

### **Geometry**

The blade structure has the following geometric properties for each element

Cross-sectional area  $= 6.4516E-5 \text{ m}^2$ Second moment of area about the local y axis  $= 3.46859E-11 \text{ m}^4$ Second moment of area about the local z axis  $= 3.46859E-9 \text{ m}^4$ Torsional constant  $= 3.50327E-9 \text{ m}^4$ First moment of area about the local y axis  $= 0.0 \text{ m}^3$ First moment of area about the local z axis  $= 0.0 \text{ m}^3$ Product moment of area  $= 0.0 \text{ m}^4$ 

# **Material properties**

Young's modulus	$72.4072x10^9\ N/m^2$
Poisson's ratio	0.3
Density	$2726.0\;kg/m^3$

### **Boundary conditions**

The blade is rigidly supported at the root, node 1 (see Figure 2).

### **Theory**

To analyse this problem correctly a nonlinear solution procedure is adopted to account for the large displacements [1]. These geometric nonlinearities are handled via the Total Lagrangian formulation. In addition to the usual large displacement and initial stress stiffness matrices a further contribution to the tangential stiffness due to centripetal acceleration is included. This additional stiffness matrix is called a load correction stiffness matrix.

## Comparison

Table 1 compares the natural frequencies of the fundamental mode for various element meshes for an angular velocity of zero, 20000 rpm and 20000 rpm with the load correction stiffness matrix due to centripetal acceleration neglected (Lusas Solver option 102). As can be seen from Table 1, the effect of neglecting the load correction stiffness matrix due to centripetal acceleration results in errors of approximately 20% in the calculation of the natural frequency of the fundamental mode of vibration.

Frequency, Hz Load Load correction correction stiffness stiffness Element Integration rule included neglected Mass Stiffness Mesh Type matrix matrix ω=0 ω=20,000 rpm ω=20,000 rpm HX16 4 x 1  $2 \times 2 \times 2$ 2 x 2 x 2 225 380 458 4 x 1 OSL8 208 358 5 point 5 point 440 5 x 2 HX20  $3 \times 3 \times 3$  $3 \times 3 \times 3$ 216 345 454 5 x 2 QSL8 5 point 5 point 211 366 447 5 x 2 HX20  $2 \times 2 \times 2$ 2 x 2 x 2 216 375 454 4 BSL4 3 3 209 367 448 10 BSL4 3 3 209 448 367 5 x 2 HX20  $3 \times 3 \times 3$  $3 \times 3 \times 3$ 219 347 456

**Table 1: Natural Frequencies of vibration** 

#### References

 M. E. Honnor and P. Lyons, "Free Vibrations in Rotating Machinery.", Finite Element Analysis Ltd Internal Report No. FEAL401, 1984.

# Input data

X04D31.DAT

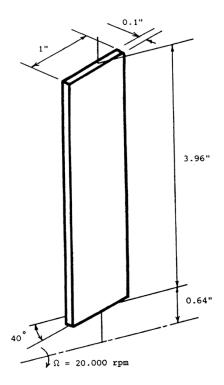


Figure 1 Problem Geometry and Material properties

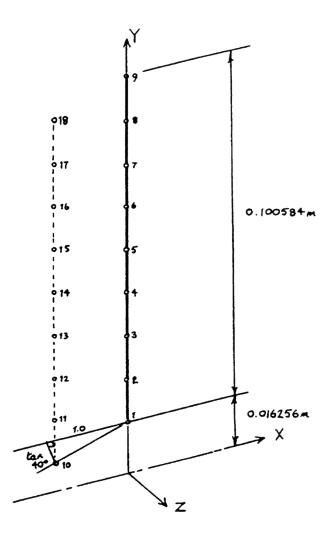


Figure 2 Finite Element Mesh

# **Example 4.3.2**

## **Curved Cantilever Under Concentrated End Load**

## **Keywords**

Nonlinear, Co-Rotational, Total Lagrangian

### **Description**

A curved cantilever is subjected to a transverse concentrated load applied at the free end. This problem involves a three dimensional response to a geometrically nonlinear problem. Large displacements and rotations occur in this example and the beam element used, BTS3, is based on a co-rotational formulation. The beam is also modelled with continuum elements where a Total Lagrangian formulation is used. Linear material properties are assumed.

### **Discretisation**

The curved cantilever is modelled with:

- i) 8 straight linear BTS3 elements of equal length.
- ii) 16 HX8M solid elements of equal length and unit square cross section.

### **Geometry**

Figure 1 shows the initial geometry of the beam P=0.0

Radius of curvature = 100.0

Co-ordinates for centre of curvature = (0.0, 0.0, 100.0)

The co-ordinates at the ends of the beam centre line are:

Fixed end=
$$(0.0, 0.0, 0.0)$$

Free end 
$$= (70.71, 0.0, 29.29)$$

Sectional properties for beam elements:

Axx = 1.0

Asy = 1.0

Asz = 1.0

Iyy = 8.3333E-2

Izz = 8.3333E-2

Kt = 14.06E-2

The cantilever is of unit square cross section and Kt is evaluated from:

$$Kt = 0.1406a^4$$

where a is the length of the sides of the cross section.

### **Boundary conditions**

Fully fixed at one end.

### **Material properties**

Young's modulus	$10x10^6$
Poisson's ratio	0.0

### Loading

For the beam elements, total concentrated loads of 300, 450 and 600 are applied incrementally at the free end. For the solid elements, the total load of 600 is applied in 12 equal steps. The loads are applied in the vertical (Y) direction.

#### **Lusas Results**

Figure 1 shows the deformation of the beam elements for the loading considered. A table of tip displacements is given below including a comparison with the reference solutions. For the HX8M elements the average of the 4 end node displacements is quoted.

Tip displacement (u,v,w) for curved cantilever

		Load level	
	300	450	600
Simo and Vu-Quoc	-11.87,40.08,-6.96	-18.39,48.39,-10.67	-23.48,53.37,-13.50
Cardona and Geradin	-12.07,40.35,-7.15	-18.60,48.59,-10.91	-23.67,53.50,-13.74
BTS3	-12.16,40.47,-7.16	-18.78,48.76,-10.93	-23.91,53.71,-13.77
HX8M	-11.83,40.05,-7.06	-18.29,48.29,-10.79	-23.30,53.20,-13.60

# Comparison

The results obtained for this example are very sensitive to the value taken for the torsional constant Kt. The actual value used for executing this test is seldom given in published results.

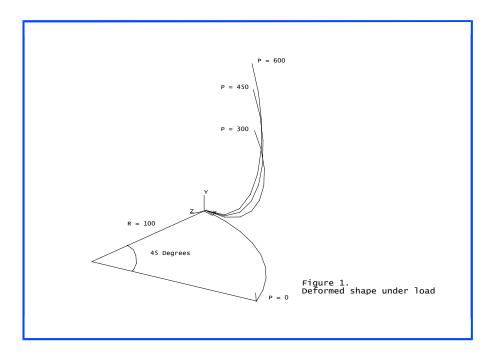
The effective value of Kt for the solid elements is slightly higher than that used for the beam elements; this results in a stiffer response when using HX8M elements. For the analysis involving BTS3 elements, large load steps were taken to demonstrate the robustness of the consistent co-rotational formulation. Load steps of this size can cause negative pivots to occur during the iterative solution procedure. However, these negative pivots are not present when a solution has converged. In practise, smaller load steps should be used to ensure a more stable sequence of iterations.

#### References

- SIMO,J.C. and Vu-Quoc, L., A Three Dimensional Finite Strain Rod Model. Part II: Computational Aspects, Comp. Meth. in Appl. Mech. and Engineering, Vol 58, pp 79-116, 1986.
- CARDONA,A., GERADIN, M. A Beam Finite Element Non-Linear Theory with Finite Rotations. Int. Journ. for Num. Meth. in Eng., Vol 26, pp 2403-2438, 1988.

### **Input data**

X04D32A.DAT



# **Example 4.5.1**

## **Hinged Cylindrical Shell Under Central Point Load.**

## **Keywords**

Total Lagrangian, Geometric Nonlinearity, Snap Through

### **Description**

Determine the nonlinear load/displacement response of a cylindrical shell of the form shown in Figure 1, when it is subjected to a point load acting vertically downwards at its centre.

#### **Discretisation**

As the loading and boundary conditions are symmetric, only 1/4 of the shell is analysed, using a 4x4 mesh of QSL8 Semiloof shell elements (see Figure 1).

### Geometry

The geometry of the hinged cylindrical shell is shown in Figure 1.

### **Material properties**

Young's modulus	3103 N/mm <sup>2</sup>
Poisson's ratio	0.3

### **Boundary conditions**

The shell structure is hinged along the left-hand side, nodes 1 to 9, restrained from crossing and rotating about the yz plane, nodes 73 to 80, restrained from crossing and rotating about the xz plane, nodes 18, 27, 36, 45, 54, 63 and 72, and totally restrained at its centre, node 81. The z direction is restrained to permit the application of prescribed displacements.

### **Theory**

This example is commonly referred to as a snap through problem which has been studied by Sabir and Lock [1]. Due to the large displacements and this snap through effect the problem must be analysed nonlinearly using the Total Lagrangian geometric nonlinearity formulation.

## Comparison

A Comparison of the results obtained by LUSAS with those of [1] are shown in Figure 2.

#### References

 Sabir, A.B., Lock, A.C.'The Application of Finite Elements to the Large Deflection Geometrically Nonlinear Behaviour of Cylindrical Shells'. Variational Methods in Engineering. Ed. Brebbia et al. Pub. Soton. Univ. Press. 1972.

### **Input data**

X04D51A.DAT

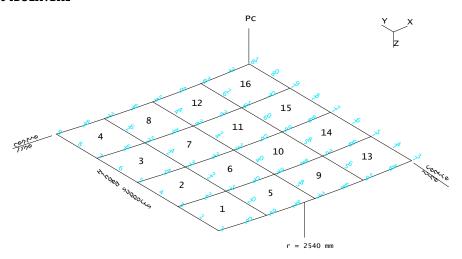
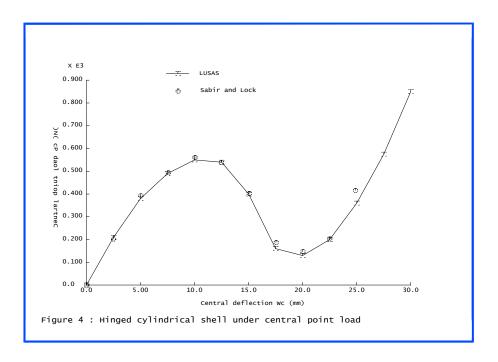


Figure 1.
Finite element mesh for cylindrical shell problem



# **Example 4.5.2**

## **Clamped Spherical Cap Subject To A Point Load**

## **Keywords**

Total And Updated Lagrangian, Axisymmetric Solid, Axisymmetric Shell

### **Description**

This geometrically non-linear example consists of a clamped spherical cap subject to a point load at the apex. Figure 1 shows the geometry and loading of the problem under consideration. For each mesh the analysis is performed using a total Lagrangian formulation and an updated Lagrangian formulation.

Internal Radius	R	22.27 inches
Thickness of shell	t	0.41 inches
Semi-angle	θ	26.67 degrees

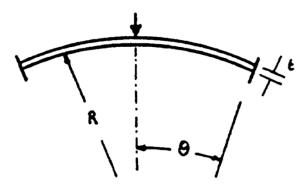


Figure 1.

#### **Discretisation**

The spherical cap is modelled with five QAX8 elements for the first mesh and twelve BXS3 elements for the second mesh. Figures 2 and 3 show the mesh for QAX8 and BXS3 elements respectively.

## **Material properties**

Young's modulus	$10.0x10^6\ N/m^2$
Poisson's ratio	0.3

## **Boundary conditions**

The edge of the cap is fully fixed against translation and rotation.

### **Theory**

See [2].

## **Solution comparison**

Apex Deflection	Load P QAX8 Updated Lagrangian (N)	Load P BXS3 Total Lagrangian (N)
0.01	9.576	9.689
0.02	14.00	14.28
0.03	16.75	17.06
0.04	19.29	19.53
0.05	22.02	22.17
0.06	25.01	25.09
0.07	28.18	28.20
0.08	31.34	31.37
0.09	34.28	34.42
0.10	36.83	37.16
0.11	38.94	39.46
0.12	40.81	41.32
0.13	43.30	43.21
0.14	48.40	46.56
0.15	59.18	54.13
0.16	78.58	69.32

### References

- 1. M.E. Honnor, 'Axisymmetric thin shell element', FEAL internal report FEAL503.
- 2. W.E. Haisler, J.A. Stricklin and F.J. Stebbins, 'Development and evaluation of solution procedures for geometrically nonlinear structural analysis by the direct stiffness method',

AIAA/ASME 12th Structures Struct. Dynamics and Materials Conference, Anahein, California.

## **Input Data**

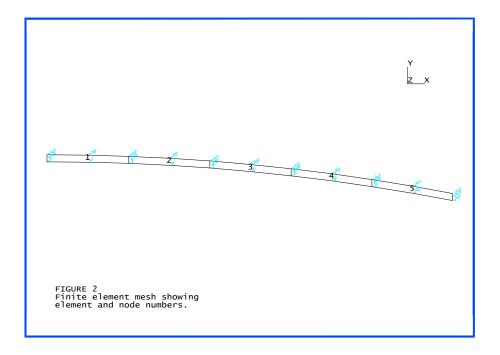
X04D52A.DAT

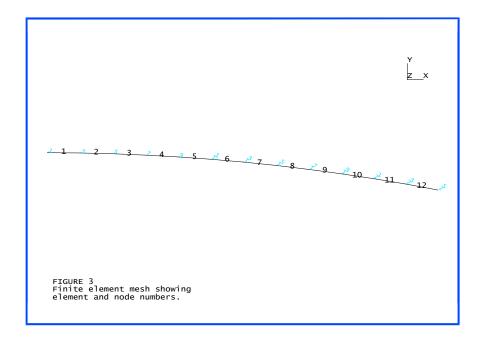
X04D52B.DAT

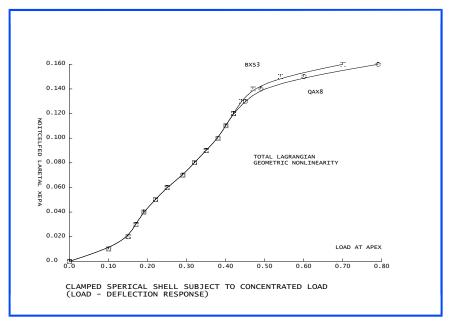
Input data for axisymmetry about the Y-axis:

Note: Include OPTION 47 for axisymmetry about the X-axis.

Note: Include OPTION 47 for axisymmetry about the X-axis







# **Example 4.5.3**

## **Large Displacement Of A Hyperbolic Paraboloid**

### **Keywords**

Total Lagrangian, Large Rotation, Warped Elements

### **Description**

The hyperbolic paraboloid tests the large rotational and translational behaviour of a structure with an initial mesh of distorted and warped elements. The paraboloid is supported on two inwardly sloping rollers about which concentrated moments are applied. The sloping supports provide membrane restraint only, but allow large rigid body motion under the moment loading.

#### **Finite Element Model**

Taking advantage of symmetry only one quarter of the paraboloid is modelled. Symmetry conditions are imposed along two of the edges while the third edge is free. The paraboloid is supported on inwardly sloping hinged rollers as shown below. A 12x4 graded mesh (see Figure 1) comprising QTS4 elements compares directly with the results presented in [1].

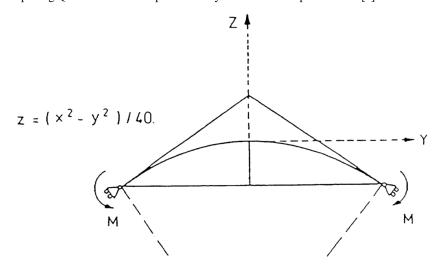
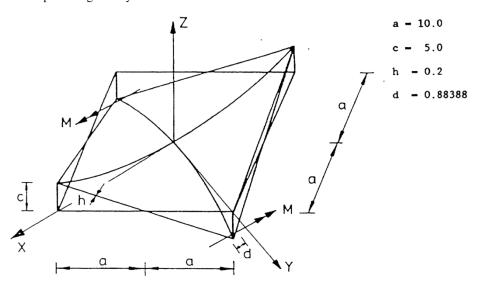


Figure 1

## **Geometry**

The full problem geometry is:



## **Material properties**

Young's modulus	$10.0 \times 10^3$
Poisson's ratio	0.0

## **Boundary Conditions**

Roller supports allowing rotation about and translation along the transformed axes.

# **Solution Comparison**

LUSAS results are detailed for node 13. The deformed shape is shown in Figure 2.

#### Support displacement along transformed freedom

Load	0.5	1.0	1.5	2.0	2.5	3.0	3.5	4.0
Ref.	1.8567	5.1118	9.0081	12.8471	15.9420	17.8904	19.4014	20.2423
LUSAS	1.8574	5.1081	8.9973	12.8238	15.9011	17.8400	19.3280	20.1348

Support rotation
------------------

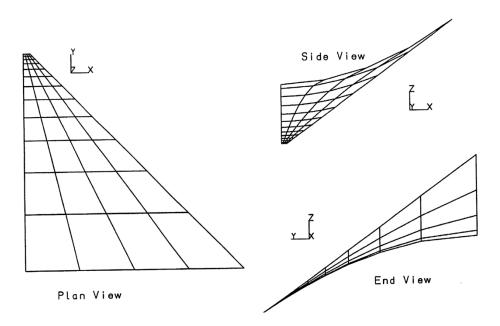
Load	0.5	1.0	1.5	2.0	2.5	3.0	3.5	4.0
Ref.	34.32	74.49	116.85	160	207.55	244.00	292.43	352.79
LUSAS	34.36	74.54	116.43	160.83	207.23	243.54	291.02	346.95

## References

1. Stander N, Matzenmiller A, and Ramm E. An assessment of assumed strain methods in finite rotation shell analysis. Engineering Computations, Vol. 6, pp58-66, 1989.

## **Input Data**

X04D53A.DAT



Grading of Mesh for Hyperbolic Parabaloid

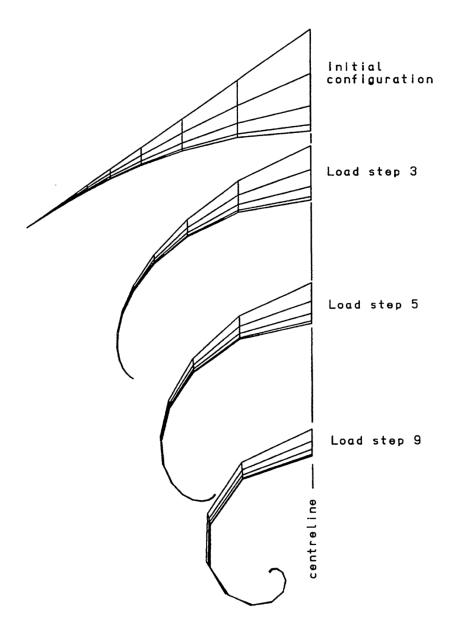


Figure 2 Deformation of hyperbolic paraboloid

# **Example 4.5.4**

## **Large Displacement Of A Hemispherical Shell**

### **Keywords**

**Co-Rotational Geometric Nonlinearity, Large Rotations** 

### **Description**

A hemispherical shell is subjected to point loads as shown in Figures 1 and 2. In addition to rigid body rotations, both membrane and bending strains contribute significantly to the radial displacement at the loading points.

#### **Finite Element Model**

Taking advantage of symmetry only one quarter of the hemisphere is modelled. Symmetry conditions are imposed along two of the edges while the other edges are free. The hemisphere is discretised with TSR6 facet shell elements as shown in Figure 1.

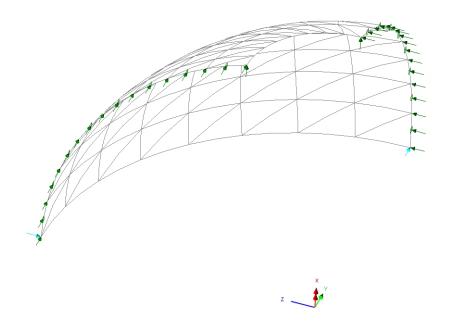


Figure 1

### **Geometry**

The full problem geometry is:

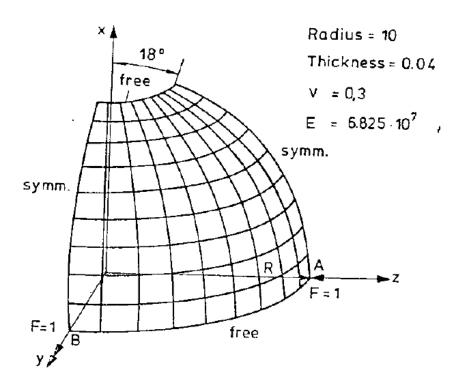


Figure 2

# **Material properties**

Young's modulus, E	68.25x10 <sup>6</sup>
Poisson's ratio, v	0.3

## **Boundary Conditions**

Due to the symmetry of the problem only  $\frac{1}{4}$  of the hemisphere is modelled. Two of the edges are free while symmetry is imposed at the other two edges.

## **Solution Comparison**

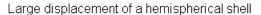
LUSAS results are compared with [1] in the following figures; the final deformed mesh is shown in Figure 6.

### References

1. Stander N, Matzenmiller A, and Ramm E. An assessment of assumed strain methods in finite rotation shell analysis. Engineering Computations, Vol. 6, pp58-66, 1989.

### **Input Data**

X04D54.DAT



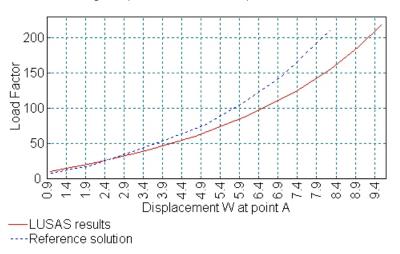
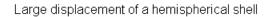


Figure 3



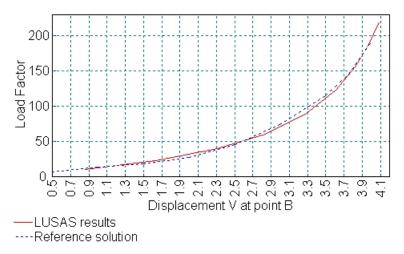


Figure 4

# Large displacement of a hemispherical shell

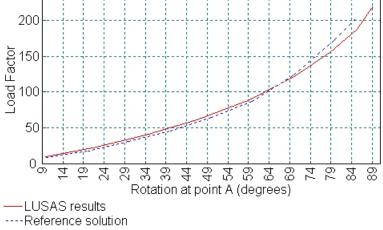


Figure 5

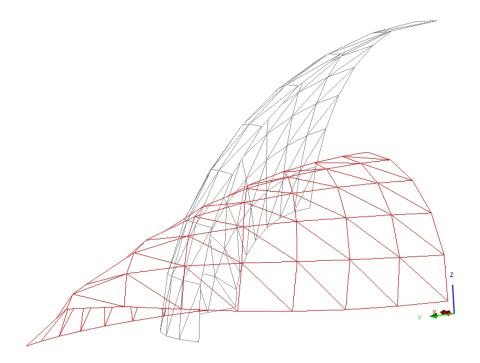


Figure 6

# **Example 4.5.5**

# Optimisation of cable forces in cable stayed bridge

## **Keywords**

Geometric Nonlinearity, Optimisation, Target value

## **Description**

A geometrically nonlinear cable stayed bridge is optimised to minimise the cable forces subject to limits in the tower sway and deck displacement under self weight loading. Additionally, The force in a pair of cables is specified using the target value loading type.

The horizontal movement at the top of the tower  $U_A$  is limited to a displacement of 1cm whilst the vertical displacements at  $W_B$  and  $W_C$  must be greater than or equal to zero. Finally, the cable forces in the pair of cables labelled  $T_I$  is set to 38,000kN.

### **Finite Element Model**

The simple cable stayed bridge is shown in figure 1.

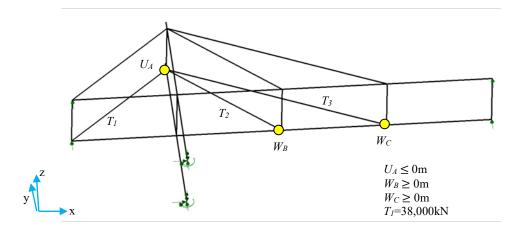


Figure 1

## Geometry

The problem geometry is:

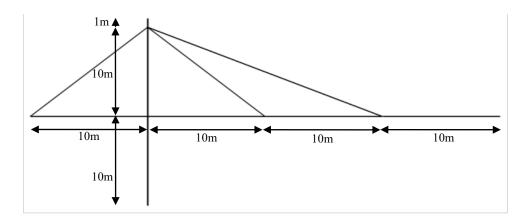


Figure 2

## **Material properties**

Young's modulus, E	$210 \times 10^6 \text{ kN}$
Poisson's ratio, v	0.3
Density, ρ	$7850 \text{ kg/m}^3$

## **Boundary Conditions**

The base of the towers are fully restrained and the ends of the spans are supported in the veritical direction.

## **Solution Comparison**

Value	Target	LUSAS
$U_A$	≤0	-0.0013
$W_B$	≥0	0.0041
$W_C$	≥0	-0.66x10 <sup>-18</sup>
$T_{I}$	38,000	38,000

## References

# **Input Data**

X04D55.DAT

# **Example 5.1.1**

## **Elastoplastic Analysis Of A Cantilever Bar**

### **Keywords**

Elastoplasticity, Strain Hardening, Load Cycling

### **Description**

The cantilever bar shown in Figure 1a is subjected to a uniaxial load which alternates from tension to compression. The nonlinear behaviour of the material is given in Figure 1b as the stress/strain relationship with strain hardening. The required output is the load/displacement history.

#### Finite element model

A single BAR2 element is used to model the complete cantilever. The bar is assumed to be perfectly straight and is not subject to buckling or any other geometrically nonlinear effects.

## **Geometric properties**

Length	10 mm
Cross-sectional area	$1 \text{ mm}^2$

## **Material properties**

Young's modulus, E	$1.0x10^5 \text{ N/mm}^2$
Plasticity modulus, E <sub>p</sub>	$2.0x10^4  N/mm^2$
Yield stress	$400\ N/mm^2$

### **Loading condition**

A displacement condition at end 2 of  $u_2$  (Figure 1a) is prescribed using load type PDSP. This displacement varies from +0.14 mm to -0.08 mm.

### **Solution Comparison**

The results in the form of the load/displacement curve are given in figure 2, where  $R_{x_1}$  is the reaction at node 1 and the positions marked on the graph are the displacement increments with the corresponding values of  $R_{x_1}$  and u given in Table 1. This example of a uniaxial load shows

how the solution converges to the exact value instantaneously. It should also be noted how if the loading/ unloading procedure were continued within the limits stated above, the model would converge to its linear elastic form, represented by the line 9-13 of the graph in Figure 2, as further strain hardening occurs.

### **Input Data**

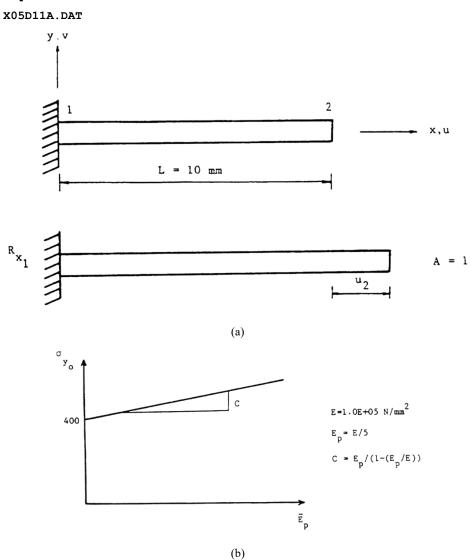


Figure 1. (a) Cantilever beam subjected to an extension (b) Nonlinear stress/strain relation for material model.

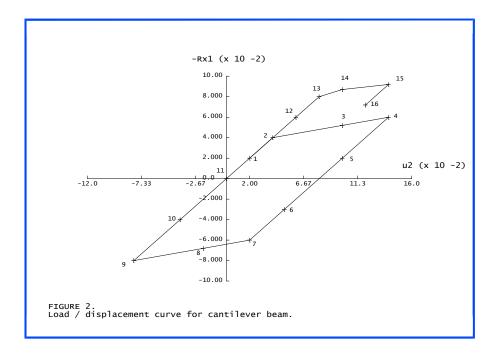


Figure 2. Load/displacement curve for cantilever beam.

# **Example 5.1.2**

# **Uniaxial Tension/ Compression Cycling**

### **Keywords**

**Modified Von Mises** 

## **Description**

A bar is subjected to cycling under uniaxial stress conditions, which causes yield in both tension and compression. The material is modelled by a Modified von Mises yield criterion which allows for different yield properties in tension and compression. Additionally, both the compressive and tensile strain hardening characteristics may be different.

P

### **Finite Element Model**

One plane stress element.

### **Geometry**

Length = 10

Width = 1

Depth = 1

### **Material properties**

Young's modulus = 2000

Poisson's ratio = 0.0

Yield stress  $\sigma_{y,0}^c = 30$ ,  $\sigma_{y,0}^t = 20$ 

Hardening  $H^c = 150$   $H^t = 100$ 

### **Boundary Conditions**

Free to pull-in at laterally supported end

### **Solution Comparison**

The load/displacement relationship is calculated by prescribing the load which uniquely determines the plastic strain developed to sustain it. As the plastic strain is accumulated, the yield stress in both tension and compression is changed according to the hardening characteristics of the material in each sector.

The plastic stress multiplier,  $\lambda$ , is given by:

$$\lambda = \frac{\Delta \overline{\varepsilon}^{p}}{\left\{ \left[ \underline{\underline{P}}\underline{\sigma} + \gamma \underline{\hat{p}} \right]^{T} \underline{\underline{M}} \left[ \underline{\underline{P}}\underline{\sigma} + \gamma \underline{\hat{p}} \right] \right\}^{1/2}}$$
(0.0.57)

where:

 $\Delta \overline{\epsilon}^p \qquad \text{effective plastic strain} = \left( \textbf{S}_{y,t}^{\textbf{a}} - \textbf{S}_{y,0}^{\textbf{a}} \right) / \left. H^{\textbf{a}} \right., \text{ where } H^{\textbf{a}} \text{ is the hardening gradient and } \\ \textbf{S}_{y,t}^{\textbf{a}} \text{ the current yield stress and } \alpha \text{ may be the tension or compression value.}$ 

P deviatoric components of the yield function

 $\hat{\mathbf{p}}$  pressure dependent components of the yield function

$$\hat{\mathbf{p}} = \begin{bmatrix} \frac{1}{3} \\ \frac{1}{3} \\ 0 \\ 0 \\ 0 \end{bmatrix} \tag{0.0.59}$$

σ current stress vector

and

$$\gamma = \sigma_{y,t}^c - \sigma_{y,t}^t, \quad \underline{\underline{\mathbf{M}}} = \begin{bmatrix} 1 & & & & \\ & 1 & & & \\ & & 1 & & \\ & & & \frac{1}{2} & & \\ & & & & \frac{1}{2} & \\ & & & & \frac{1}{2} \end{bmatrix}$$
 (0.0.60)

When considering the  $S_{xx}$  component of the stress vector to be the only non-zero value, the denominator of (0.0.57) reduces to:

$$\left[\underline{\underline{\mathbf{P}}}\underline{\boldsymbol{\sigma}} + \gamma \underline{\hat{\mathbf{p}}}\right]^{\mathrm{T}}\underline{\underline{\mathbf{M}}}\left[\underline{\underline{\mathbf{P}}}\underline{\boldsymbol{\sigma}} + \gamma \underline{\hat{\mathbf{p}}}\right] = \frac{2}{3}\sigma_{xx}^{2} + \frac{1}{3}\left(\sigma_{y,t}^{c} - \sigma_{y,t}^{t}\right)^{2}$$
(0.0.61)

The incremental plastic strain vector is then computed from:

$$\Delta \underline{\boldsymbol{\varepsilon}}^{\mathbf{p}} = \lambda \left[ \underbrace{\underline{\mathbf{P}}\underline{\boldsymbol{\sigma}}}_{\underline{\underline{\mathbf{q}}}} + \gamma \hat{\underline{\boldsymbol{p}}}_{\underline{\underline{\mathbf{p}}}} \right] \tag{0.0.62}$$

Using equations (0.0.57), (0.0.61) and (0.0.62) the computation of the  $\Delta \varepsilon_{xx}$  component of  $\Delta \overline{\varepsilon}$  reduces to:

$$\Delta \varepsilon_{xx}^{p} = \frac{\Delta \varepsilon^{-p} \left( \sigma_{xx} + \frac{1}{2} \left( \sigma_{y,t}^{c} - \sigma_{y,t}^{t} \right) \right)}{\left[ \sigma_{xx}^{2} + \frac{1}{2} \left( \sigma_{y,t}^{c} - \sigma_{y,t}^{t} \right)^{2} \right]^{1/2}}$$

$$(0.0.63)$$

The compressive and tensile yield stresses are a function of the plastic strain, which may be calculated for one dimensional problems from

$$\overline{\varepsilon}^{p} = \frac{\left|\sigma_{xx}\right| - \sigma_{y,0}^{t}}{H^{t}} \qquad \text{if} \qquad \sigma_{xx} \text{ is tensile} \qquad (0.0.64)$$

$$\overline{\epsilon}^{p} = \frac{\left|\sigma_{xx}\right| - \sigma_{y,0}^{t}}{H^{t}} \qquad \text{if} \qquad \sigma_{xx} \text{ is tensile} \qquad (0.0.65)$$

The yield stresses are then computed from:

$$\sigma_{\mathbf{v},\mathbf{t}}^{\mathbf{c}} = \sigma_{\mathbf{v},0}^{\mathbf{c}} + \mathbf{H}^{\mathbf{c}} \overline{\boldsymbol{\varepsilon}}^{\mathbf{p}} \tag{0.0.66}$$

$$\sigma_{y,t}^{t} = \sigma_{y,0}^{t} + H^{t} \overline{\epsilon}^{p}$$
 (0.0.67)

Summing the elastic and plastic strain components, the total axial strain is,

$$\varepsilon_{xx}^{T} = \varepsilon_{xx}^{e} + \sum_{i=1}^{n} \Delta \varepsilon_{xx,i}^{p} = \frac{\sigma_{xx}}{E} + \sum_{i=1}^{n} \Delta \varepsilon_{xx,i}^{p}$$
(0.0.68)

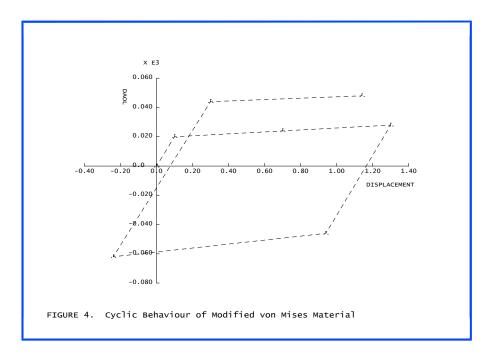
The results can be calculated by using (0.0.64) or (0.0.65) to calculate the total plastic strain. Once the plastic strain is known, the yield stresses are defined using (0.0.66) and (0.0.67). The axial plastic strain is then given by (0.0.63) and the total strain by (0.0.68).

$\begin{array}{c} \textbf{Load} \\ \sigma_{xx} \end{array}$	$\overline{\epsilon}^p$	$\Delta \overline{\epsilon}^p$	$\sigma_{y,t}^{c}$	$\sigma_{y,t}^t$	$\Delta\epsilon_{xx}^p$	$\Sigma \epsilon_{xx}^p$	$\epsilon_{xx}^T$
20.000	0.0000	0.0000	30.000	20.000	0.0000	0.0000	0.0100
25.000	0.0500	0.0500	37.500	25.000	0.0589	0.0589	0.0714
30.000	0.1000	0.0500	45.000	30.000	0.0589	0.1178	0.1329
29.000	0.1000	0.0000	45.000	30.000	0.0000	0.1178	0.1324
-45.000	0.1000	0.0000	45.000	30.000	0.0000	0.1178	0.0954
-65.000	0.2333	0.1333	65.000	43.333	-0.1081	0.0097	-0.0228
-64.000	0.2333	0.0000	65.000	43.333	0.0000	0.0097	-0.0223
43.333	0.2333	0.0000	65.000	43.333	0.0000	0.0097	0.0314
50.000	0.3000	0.0667	75.000	50.000	0.0786	0.0883	0.1133

Exact agreement is obtained from LUSAS. The load/displacement history is shown on figure 1. To prevent convergence difficulties, the stress is relaxed from the yield surface prior to applying a large stress reversal.

### **Input Data**

X05D12A.DAT



# **Example 5.1.3**

## **Uniaxial Cycling Elastic-Damage Analysis**

## **Keywords**

Damage, Cyclic Loading

### **Description**

A bar of elastic material is subjected to uniaxial compressive load cycling with an increasing stress level. A damage criterion for the material is specified so that progressive degradation of the elastic Young's modulus takes place as the analysis proceeds. As an elastic material is defined, zero strains are produced when the bar is stress free. Under this stress free state the stress/strain curve returns to the origin.

#### **Discretisation**

A single BAR2 element of length 10mm and unit cross section is used to represent the bar.

### **Material properties**

Young's modulus  $= 2E3 \text{ N/mm}^2$ 

Poisson's ratio = 0.0

Damage property data (Simo [1,2]):

Initial damage threshold,  $r_0 = 0.44$ 

Constants, A = B = 1.0

### **Boundary conditions**

Fully fixed at one end, lateral displacements restrained at the other.

### Loading

Cyclic vertical compressive load at the free end.

### **Theory**

The damage parameter, d, for the Simo [1,2] model is given by:

$$d = 1 - \frac{r_0 (1 - A)}{r_t} - Ae^{B(r_0 - r_t)}$$
(0.0.69)

Using the specified damage properties this reduces to:

$$d = 1 - e^{(0.44 - r_t)}$$

$$(0.0.70)$$

For the uniaxial case, the complementary energy norm, r<sub>t</sub>, reduces to:

$$\mathbf{r}_{t} = \left[\frac{\sigma^{2}}{E}\right]^{1/2} \tag{0.0.71}$$

The axial strain is then given by:

$$\epsilon = \frac{\sigma}{E'} \qquad \qquad \text{where} \qquad E' = \left(1 - d\right) E \qquad \qquad (0.0.72)$$

More information on the damage models available in LUSAS can be found in the LUSAS Theory Manual.

### **Solution comparison**

Using the above equations strains can be computed for each step of the analysis:

	Load	σ	$\mathbf{r}_{t}$	d	ε
Loading	20	20	0.447214	0.007188	0.010072
	30	30	0.670820	0.206118	0.018894
	40	40	0.894427	0.365189	0.031505
Unloading	30	30	0.894427	0.365189	0.023629
	20	20	0.894427	0.365189	0.015753
	0	0	0.894427	0.365189	0.000000
Loading	40	40	0.894427	0.365189	0.031505
	60	60	1.341641	0.594097	0.073909
Unloading	50	50	1.341641	0.594097	0.061591
	0	0	1.341641	0.594097	0.000000
Loading	60	60	1.341641	0.594097	0.073909
	80	80	1.788854	0.740463	0.154120
	85	85	1. 900658	0.767916	0. 183124
	90	90	2. 012461	0. 792466	0. 216832
Unloading	85	85	2. 012461	0. 792466	0. 204786

				_
0	0	2. 012461	0. 792466	0.000000

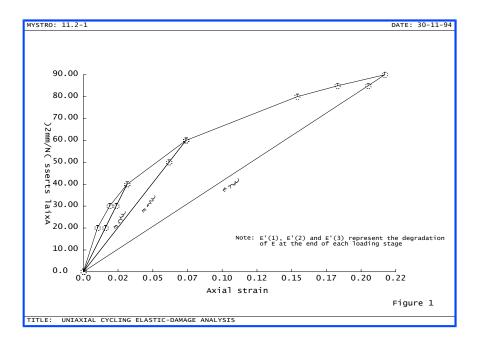
The theoretical assumptions for this damage model are such that, after the onset of damage, no further damage can occur until the maximum stress previously attained is exceeded. Hence the table shows that the damage parameter remains constant during unloading and reloading until a stress higher than the previous maximum is attained. The results for the LUSAS analysis reproduce these values exactly and are plotted in Figures 1 and 2.

#### References

- 1. Simo, J.C. and JU, J.W. Strain- and stress-based continuum damage models I. Formulations. Int. Journ. Solids and Structures, Vol. 23, No. 7, pp821-840, 1987.
- Simo, J.C. and JU, J.W. Strain- and stress-based continuum damage models II. Computational aspects. Int. Journ. Solids and Structures, Vol. 23, No. 7, pp841-869, 1987.

### **Input Data**

X05D13A.DAT



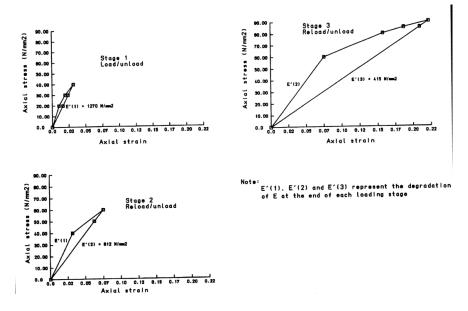


Figure 2. Loading/unloading stages plotted separately.

# **Example 5.1.4**

# **Uniaxial Cycling of a Piecewise Multi-Linear Elastic Material**

### **Keywords**

Multi-linear Material, Cyclic Loading

### **Description**

A bar comprising a multi-linear elastic material is subjected to three different loading sequences:

- 1. Increasing monotonic
- 2. Decreasing monotonic
- 3. Random cyclic sequence

The material model is such that stress/strain behaviour is completely recoverable.

#### **Discretisation**

A single BAR2 element of length 100 and unit cross section is used to represent the bar.

### **Material properties**

The piecewise linear material properties are defined by a series of stress/strain points which results in a segmental curve defining a seris of Young's moduli, Table 1.

Strain	-0.01	0.00	0.02	0.025	0.065
Stress	-15.00	-10.00	10.00	20.00	30.00

Table 1 Stress/strain input data defining piecewise E values

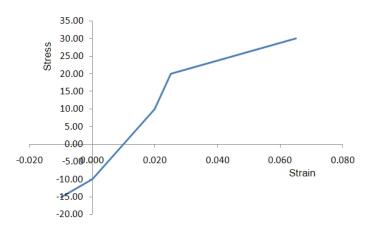


Figure 1 Stress/strain input data defining piecewise E values

### **Boundary conditions**

Fully fixed at one end, lateral displacements restrained at the other.

## Loading

The three loading sequences described above are applied via an axial concentrated load at the free end as shown in Table 2:

	Increment number							
Loading sequence	1	2	3	4	5	6	7	
1	-17.50	-15.00	-12.50	-10.00	-6.25	-1.25	0.00	
2	35.00	30.00	25.00	20.00	15.00	10.00	5.00	
3	35.00	-10.00	30.00	25.00	-15.00	20.00	15.00	

	Increment number (cont'd)							
Loading sequence	8	9	10	11	12	13	14	
1	5.00	10.00	15.00	20.00	25.00	30.00	35.00	
2	0.00	-1.25	-6.25	-10.00	-12.50	-15.00	-17.50	
3	-17.50	-6.25	5.00	-12.50	-1.25	10.00	0.00	

Table 2 Load sequences for applied concentrated load

### **Results**

	Load ascending order						
Results	1	2	3	4	5	6	7
Stress	-17.50	-15.00	-12.50	-10.00	-6.25	-1.25	0.00
Strain	-0.01500	-0.01000	-0.00500	0.00000	0.00375	0.00875	0.01000

	Load ascen	Load ascending order (cont'd)					
Results	8	9	10	11	12	13	14
Stress	5.00	10.00	15.00	20.00	25.00	30.00	35.00
Strain	0.01500	0.02000	0.02250	0.02500	0.04500	0.06500	0.08500

Table 3 Stress/strain results for all tests

Each test yielded identical results for stress/strain, convergence was achieved in 1 or 2 iterations, results are displayed in Table 3 and have been verified via hand calculations.

## **Input Data**

X05D14A.DAT

X05D14B.DAT

X05D14C.DAT

## **Buried Pipe With Soil-Pipe Interface Modelled Using 2d Interface Model**

## Keywords

**Elasto-Plastic, Friction** 

## **Description**

This example (figure 1) consists of an elastic pipe buried in a homogeneous and elastic soil. The interface between the pipe and soil is modelled using an elasto-plastic interface material. Three cases of interface friction are considered:-

- (i) frictionless slip ( $tan \phi = 0.001$ ),
- (ii) stick ( $tan \phi = 2$ ),
- (iii) frictional slip ( $tan \phi = 0.25$ ).

#### **Discretisation**

The discretisation (figure 2) is similar to that used in [1]. QPM8 elements are used to discretise the pipe, soil and interface.

## **Theory**

The interaction between the soil and pipe is governed by a frictional type law i.e. bond between the pipe and soil is maintained by a frictional force and once this is overcome debonding and slip will occur. This friction/slip behaviour is modelled using an elasto-plastic interface model, with yielding (slip) defined using the Mohr-Coulomb criterion [2] defined as

	$\tau =$	$c + \sigma tan \phi$	(0.0.73)
where	τ	maximum shear stress,	
	σ	stress normal to the soil/pipe interface,	
	c	cohesion,	
	ф	friction angle	

## Comparison

Figure 3 is the deformed mesh for the frictionless slip analysis. A large amount of sliding is visible in the interface. Figure 4 shows the shear stress distribution along the pipe-soil interface.

### References

- 1. DESAI,C.S., ZAMAN,M.M., LIGHTNER,J.G. and SIRIWARDANE,H.J. Thin-layer element for interfaces and joints. Int. J. Num. Anal. Meth. Geomech., Vol.8, 19-43, 1984.
- OWEN,D.R.J. and HINTON,E. Finite elements in plasticity: theory and practice. Pineridge Press, Swansea, UK, 1980.
- 3. KATONA,M.G. A simple contact-friction interface element with applications to buried culverts. Int. J. Num. Anal. Meth. Geomech., Vol.7, 371-384, 1983.

## **Input data**

X05D21A.DAT

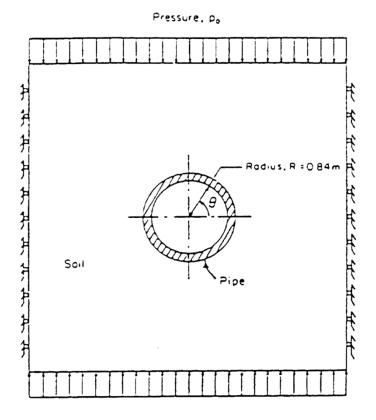
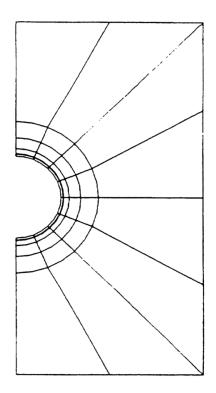
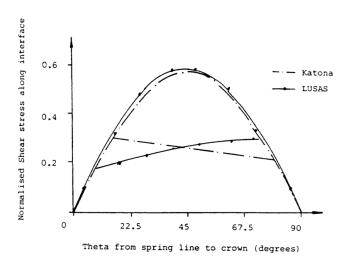


Figure 1 Buried Pipe Example



**Figure 2 Finite Element Mesh** 

Figure 3 Deformed Mesh Showing Sliding on Interface



## Elasto-Plastic Analysis Of A Thick Cylinder Under Internal Pressure

### **Keywords**

Plane Strain, Cylinder, Elasto-Plastic, Pressure

## **Description**

An infinitely long thick cylinder of internal and external radii 100 mm and 200 mm respectively subject to an increasing internal pressure

#### **Discretisation**

Plane strain conditions are idealised, the global Z axis being coincident with the axis of the cylinder. One symmetric quarter of the structure is idealised using 30 quadrilateral plane strain (QPN8) elements. Symmetric conditions are specified on the straight line boundaries, and the curved internal and external surfaces are unrestrained. The finite element discretisation is shown in Figure (1).

The nonlinear material behaviour is modelled using a Von-Mises yield criterion and a perfectly plastic post-yield response. The material properties are as follows:

Young's Modulus =  $21000 \text{dN/mm}^2$ 

Poisson's ratio = 0.25

Yield stress =  $24 \text{ dN/mm}^2$ 

## **Theory**

An increasing internal pressure is applied to the structure. The cylinder deforms elastically until an internal pressure of approximately 12dN/mm<sup>2</sup> is reached. Beyond this pressure level the cylinder behaves elasto-plastically. The spread of plastic deformation through the thickness of the cylinder can be seen from the distribution of the hoop stress component with increasing load.

#### **LUSAS Results**

The progressive deformation of the cylinder is shown in Figure (2). Figure (3) shows the von Mises stress distribution obtained from LUSAS as the internal pressure is increased.

#### References

- Owen, D.R.J., Hinton, E. Finite Elements in Plasticity: Theory and Practice Publisher. Pineridge Press Ltd. Swansea, U.K. 1980. ISBN 0-906674-05-2
- 2. Hinton, E., Owen, D.R.J. An Introduction to Finite Element Computations. Publisher. Pineridge Press Ltd. Swansea, U.K. 1979. ISBN 0-906674-06-9
- Zienkiewicz, O.C.Z. The Finite Element Method, 3rd Edition. Publisher. McGraw Hill Ltd, London, U.K. 1977. ISBN 0-07-084072-5
- 4. FEAL internal Report 'Plane Strain', FEAL708, Finite Element Analysis Ltd. Forge House, 66 High Street, Kingston Upon Thames, Surrey KT1 1HN. 1986.
- Hodge, P.G., White, G.N. A quantitative comparison of flow and deformation theories of plasticity. J.Applied Engineering Mechanics, Vol.17, 180-184, 1950.

### **Input Data**

#### X05D22A.DAT

Note: The load incrementation may alternatively be defined automatically using the following NONLINEAR CONTROL INCREMENTATION and TERMINATION cards:

C Ref.factor Max.Factor change Iterative variation

INCREMENTATION 1 0.25 0

C Max.Factor Max.No.Increments

TERMINATION 0 5

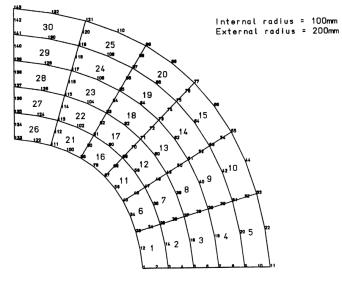


Figure 1

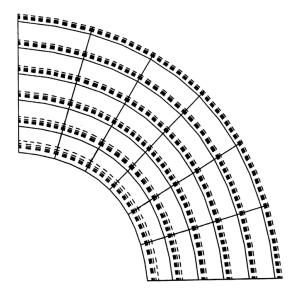


Figure 2

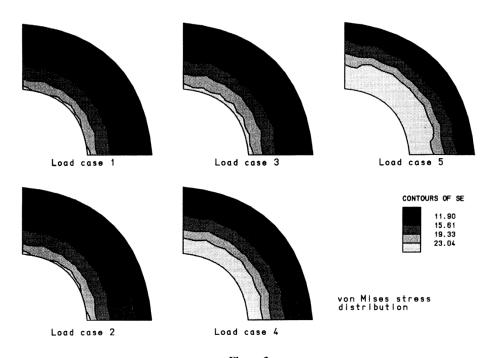


Figure 3

## Nonlinear Stress Analysis Of A Two Span Reinforced Concrete Beam

## **Keywords**

Plane Stress, Concrete, Reinforcement, Cracking

## **Description**

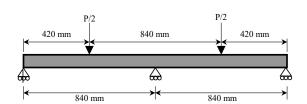
Nonlinear plane stress analysis of a reinforced concrete beam. The dimensions of the beam are as follows:

Total Length	1800 mm
Effective Length	1680 mm
Span	840 mm
Depth	80 mm
Breadth	60 mm

Reinforcement is provided at the upper and lower faces of the beam with an average cover of 3 mm. Both the upper and lower reinforcement arrangements consist of 4 bars with a total cross-sectional area of 50.4 sq.mm.

The beam is simply supported at its ends and centre, and a concentrated vertical load is applied at each midspan point.

Details of the geometry and steel arrangement are shown in Figures (1) and (2).



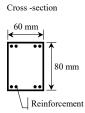


Figure 1: Geometry of reinforced concrete two span beam, also showing beam cross-section.

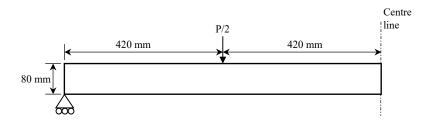


Figure 2: Dimensions of left hand span of reinforced concrete beam. Symmetry means that only one span of the beam needs to be analysed.

#### **Discretisation**

Owing to its symmetrical nature only one span of the beam is discretised. The finite element mesh consists of 112 elements; 84 plane stress (QPM8) elements to represent the concrete, and 28 bar (BAR3) elements to represent the two layers of steel reinforcement.



The plane stress elements are integrated using a 9 point Gaussian integration scheme.

## **Material properties**

The constitutive behaviour of the reinforced concrete beam is modelled using the LUSAS nonlinear von-Mises material model (model 72) for the reinforcement and the Multi-crack concrete material model (model 82) or the Multi-crack concrete with crushing material model (model 84) for the concrete [1,2].

Reinforcement: Nonlinear von-Mises material model, model 71.

Young's modulus	196.0 kN/mm <sup>2</sup>	Heat fraction	0.0
Poisson's ratio	0.3	Reference temperature	0.0
Density	0.0	Yied stress	$579.0\ N/mm^2$
Thermal Expansion	0.0	Hardening gradient	1.0
Mass damping	0.0	Plastic strain limit	100.0
Stiffness samping	0.0		

Concrete material 1: Nonlinear multi-crack concrete material model, model 82.

Young's modulus	16.68 kN/mm <sup>2</sup>	Tensile strength	1.67 N/mm <sup>2</sup>
Poisson's ratio	0.0	End of tensile softening	0.0035

		curve	
Density	0.0	Fracture energy	0.0
Thermal Expansion	0.0	Post fracture friction coefficient	1.5
Mass damping	0.0	Post fracture dilatancy coefficient	0.0
Stiffness damping	0.0	Shear/tensile strength ratio	2.5
Heat fraction	0.0	Reduction factor for shear plastic strain	0.5
Reference temperature	0.0	-	

Concrete material 2: Nonlinear multi-crack concrete with crushing material model, model 84.

Young's modulus	16.68 kN/mm <sup>2</sup>	Compressive strength	32.0 N/mm <sup>2</sup>
Poisson's ratio	0.0	Tensile strength	1.67 N/mm <sup>2</sup>
Density	0.0	Peak compressive strain	0.0027
Thermal Expansion	0.0	End of compressive softening curve	0.0035
Mass damping	0.0	End of tensile softening curve	0.0035
Stiffness damping	0.0	Fracture energy	0.0
Heat fraction	0.0	Yield/peak compressive stress ratio	1.0
Reference temperature	0.0		

## **Support Conditions**

Node	Support Type	Prescribed Boundary Conditions
1	Roller	FR
365	Restrained	RR
366-377	Symmetric (C/L)	RF

#### Notes:

The flexural steel reinforcement is represented by bar elements of equivalent cross sectional areas. It should be noted that the superposition of the nodal degrees of freedom at the steel/concrete interface assumes that a perfect bond exists between the two components.

Any deformation resulting from the self weight of the beam is neglected as being small compared to that resulting from the applied load.

The effects of the shear reinforcement are not considered in the analysis.

## **Theory**

The reinforced concrete beam analysed here formed part of an experimental investigation undertaken by Duddeck et al [3]. Under load the beam exhibits both flexural and shearing deformation, and progressses from a well distributed cracking pattern to a local material failure under the loads and at the middle support. The beam has been used as a benchmark test by many research workers in the field of nonlinear concrete analysis [4].

#### **LUSAS** results

Figure (4) shows the load deflection responses obtained from linear and nonlinear analyses of the beam. Figure (5) shows contours of principal stress at the initial loading stage and clearly demonstrates the areas of tensile stress. Figure (6) shows the progression of the cracks through the section of the beam as the load is increased.

#### References

- The LUSAS User's Manual Finite Element Analysis Ltd. Forge House, 66 High Street, Kingston Upon Thames, London, U.K.
- 2. LUSAS Internal Report, 'Nonlinear Concrete Model 24', FEAL703, Finite Element Analysis Ltd. Forge House, 66 High Street, Kingston Upon Thames, London, U.K.
- 3. Duddeck, H. Griebenow, Schaper, G. 'Material and time dependent nonlinear behaviour of cracked reinforced concrete slabs' IASS symposium, Nonlinear behaviour of reinforced spatial structures, Vol.1, 103-133, (July 1978).
- 4. Abdel-Rahman, H.H. 'Computational models for the nonlinear analysis of reinforced concrete flexural slab systems'. Ph.D. thesis, University of Wales, Department of Civil Engineering, Swansea, (1982).

## **Input data**

X05D23B.DAT

X05D23C.DAT

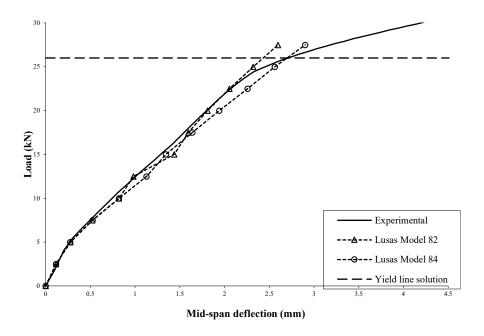


Figure 3: LOAD-DEFLECTION CURVE FOR DUDDECK'S REINFORCED CONCRETE BEAM

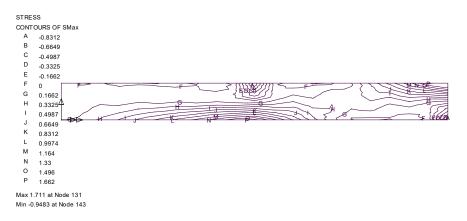


Figure 4: DUDDECK'S REINFORCED CONCRETE TWO SPAN BEAM. PRINCIPLE STRESS CONTOURS AT FIRST INCREMENT.



Figure 5: Crack pattern across one span of reinforced concrete beam. Load P=2.5kN. Crack pattern from multi-crack concrete material mode 82.



Figure 6: Crack pattern across one span of reinforced concrete beam. Load P=5.0kN. Crack pattern from multi-crack concrete material mode 82.

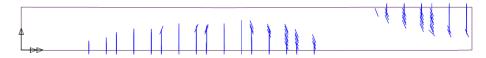


Figure 7: Crack pattern across one span of reinforced concrete beam. Load P=10.0kN. Crack pattern from multi-crack concrete material mode 82.

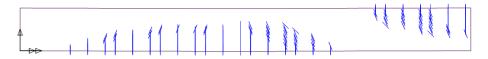


Figure 8: Crack pattern across one span of reinforced concrete beam. Load P=12.5kN. Crack pattern from multi-crack concrete material mode 82.

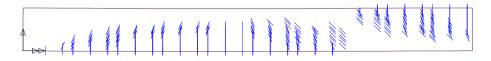


Figure 9: Crack pattern across one span of reinforced concrete beam. Load P=17.5kN. Crack pattern from multi-crack concrete material mode 82.

## **Uniaxial Cycling Elasto-Plastic Damage Analysis**

## Keywords

Damage, Cyclic Loading, Plasticity

## **Description**

A bar of elasto-plastic material is subjected to uniaxial compressive/ tensile load cycling with an increasing stress level. A damage criterion for the material is specified so that progressive degradation of the elastic Young's modulus takes place as the analysis proceeds. The damage model due to Oliver [1] is invoked which allows different damage thresholds to be specified for tension and compression.

#### **Discretisation**

A single QPM4 element of length 10mm and unit width and depth is used to represent the bar.

## **Material properties**

Young's modulus,  $E = 2E3 \text{ N/mm}^2$ 

Poisson's ratio, v = 0.0

Initial yield stress,  $\sigma_{yo} = 22 \text{ N/mm}^2$ Hardening gradient,  $h = 100 \text{ N/mm}^2$ 

Damage property data (Oliver):

Initial damage threshold,  $r_0 = 0.44$ Constant, A = 0.44

Ratio of initial stresses that cause damage in compression/tension =  $\frac{s_d^c}{s_d^t} = n = 2$ 

## **Boundary conditions**

The boundary conditions are such that lateral displacements are unrestrained while longitudinal displacements are fixed at one end, Figure 1.

## Loading

Cyclic vertical compressive/tensile load at the free end, Figure 1.

## **Theory**

The damage parameter, d, for the Oliver [1] model is given by:

$$d = 1 - \frac{r_0}{r_t} \exp\left[A\left(1 - \frac{r_t}{r_0}\right)\right] \tag{0.0.74}$$

Using the specified damage properties this equation becomes:

$$d = 1 - \frac{0.44}{r_t} exp \left[ 0.44 \left( 1 - \frac{r_t}{0.44} \right) \right]$$
 (0.0.75)

For this example, the complementary energy norm, r<sub>t</sub>, reduces to:

$$\mathbf{r}_{t} = \beta \left\lceil \frac{\sigma_{yy}^{2}}{E} \right\rceil^{1/2} \tag{0.0.76}$$

The parameter,  $\beta$ , is defined as:

$$\beta = \left(\theta \frac{1-\theta}{n}\right) \tag{0.0.77}$$

where for this example:

$$\theta = \frac{\left\langle \sigma_{yy} \right\rangle}{\left| \sigma_{yy} \right|} \qquad \text{with} \qquad \left\langle \sigma_{yy} \right\rangle = \left\{ \begin{matrix} \sigma_{yy} & (\sigma_{yy} \ge 0) \\ 0 & (\sigma_{yy} < 0) \end{matrix} \right\} \tag{0.0.78}$$

The plastic strain,  $\varepsilon_{yy}^p$ , is computed using:

$$\sigma_{yy} = \sigma_{yo} + \varepsilon_{yy}^{p} h \tag{0.0.79}$$

The axial strain,  $\in_{yy}$ , is then given by:

$$\varepsilon_{yy} = \frac{\sigma_{yy}}{E'} + \varepsilon_{yy}^{p}$$
 where  $E' = (1 - d)E$ 

$$(0.0.80)$$

More information on the damage models available in LUSAS can be found in the LUSAS Theory Manual.

## **Solution comparison**

Using the above equations strains can be computed for each step of the analysis:

Load	$\sigma_{yy}$	β	r <sub>t</sub>	d	$\boldsymbol{\mathcal{E}}_{y}^{p}$	$\mathcal{E}_{yy}$
20	20	1.0	0.447213	0.023202	0.000	0.010238
22.5	22.5	1.0	0.503115	0.178941	0.005	0.018702
25	25	1.0	0.559017	0.301322	0.030	0.047888
27.5	27.5	1.0	0.614919	0.399285	0.055	0.077889
30	30	1.0	0.670820	0.479282	0.080	0.108806
32.5	32.5	1.0	0.726722	0.545470	0.105	0.140751
35	35	1.0	0.782624	0.600883	0.130	0.173847
37.5	37.5	1.0	0.838525	0.647744	0.155	0.208228
40	40	1.0	0.894427	0.687714	0.180	0.244044
42.5	42.5	1.0	0.950329	0.722063	0.205	0.281456
45	45	1.0	1.006230	0.751776	0.230	0.320644
44.9999	49.9999	1.0	1.006230	0.751776	0.230	0.320644
0	0	1.0	1.006230	0.751776	0.230	0.230000
-20	-20	0.5	1.006230	0.751776	0.230	0.189714
-50	-50	0.5	1.006230	0.751776	0.180	0.079285
-60	-60	0.5	1.006230	0.751776	0.080	-0.040859
-70	-70	0.5	1.006230	0.751776	-0.020	-0.161001
-80	-80	0.5	1.006230	0.751776	-0.120	-0.281145
-90	-90	0.5	1.006230	0.751776	-0.220	-0.401288
-100	-100	0.5	1.118030	0.800230	-0.320	-0.570288
-99.9999	-99.999	0.5	1.118030	0.800230	-0.320	-0.570288
0	0	1.0	1.118030	0.800230	-0.320	-0.320000

The theoretical assumptions for this damage model are such that after the onset of damage, no further damage can occur until the maximum stress previously attained is exceeded. The Oliver damage model also allows some control over the damage threshold (by specification of the n parameter), depending on whether the material is in tension or compression. For this example, it can be seen from the table that the damage parameter remains constant between loads of -20 and -90 as a result of specifying n=2. The results for the LUSAS analysis reproduce these values exactly and are plotted in Figure 2.

#### References

1. Oliver, J., Cervera, M., Oller, S. and Lubliner, J. Isotropic damage models and smeared crack analysis of concrete. Computer Aided Analysis and Design of Concrete Structures, Edited by, Bicanic, N. and Mang, H., Vol. 2, 4th-6th April 1990.

## **Input Data**

X05D24A.DAT

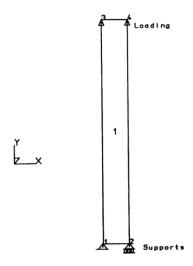


Figure 1. Supports and Loading condition

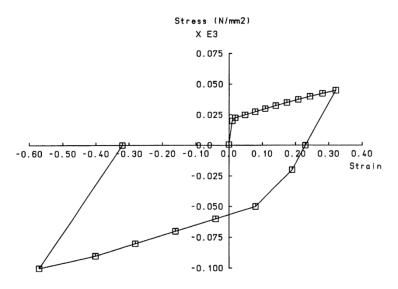


Figure 2. Stress v total strain - Sky components.

# Thermally Induced Creep Of Internally Pressurised Hollow Sphere

## **Keywords**

Implicit, Creep, Viscous

## **Description**

An internally pressurised sphere with a linear variation of temperature through its thickness is allowed to creep over a period of  $10^{10}$ . The creep is defined using a variant of the eigth parameter law.

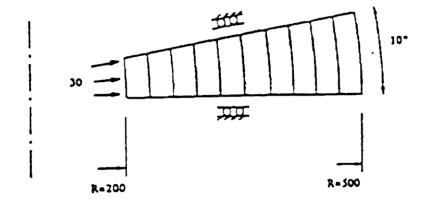
#### **Finite Element Model**

10 axisymmetric elements are used to model the through thickness variation.

## **Geometry and loads**

Internal radius 200

External radius 500



Face load of 30

Temperature gradient along sphere radius of T=333(1+100/R)

## **Material properties**

Young's modulus = 10000

Poisson's ratio = 0.25

Eight parameter law (specialised form of)

$$\varepsilon = Aq^n t^m e^{(-h/T)}$$

where A=3E-6, n=5.5, m=1, h=12500, T=temperature

## **Boundary conditions**

Free to displace along radius of centre of sphere

## **Solution Comparison**

The effective stresses at time  $10^{10}$  at various radii are compared below with the NAFEMs target solution. The solution is fully automatic with the initial step size evaluated from the first (static) stress solution. Further time steps are calculated using criterion which limit the change in creep strain across an increment. Figure 1 shows the variation of stresses with respect to time.

Radius	NAFEMS	LUSAS
205	11.47	11.54
350	17.61	17.81
495	21.10	21.07

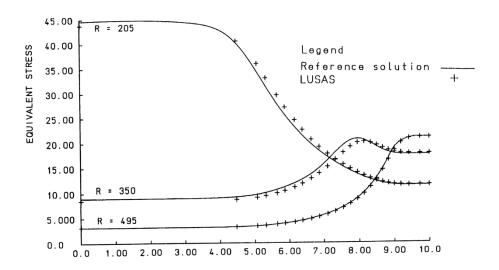
#### References

1. NAFEMS selected benchmarks for material non-linearity (to be published)

## **Input Data**

X05D25A.DAT





## **Combined Plasticity And Creep Of Bar**

## **Keywords**

Implicit, Creep, Viscous, Plastic

## **Description**

A bar is loaded from yield to a value of twice the initial yield stress and then unloaded back to the original yield stress. The unloading takes place in the elastic regime so that there is no additional plastic flow. However, the bar continues to creep since it is subject to stress.

#### **Finite Element Model**

A single plane stress element

## Geometry

Bar length = 10cross sectional area = 1

\_\_\_\_

P

## **Material properties**

Young's modulus  $= 2x10^5$ 

Poisson's ratio = 0.3

Hardening = 100

Power hardening law - time dependent form  $\, arepsilon = A \, \sigma^N t^m \,$ 

where A=1E-7, n=5, m=0.5

## **Boundary Conditions**

The element is restrained at one end in the axial direction, but it is allowed to pull-in to relieve stress in the orthogonal direction.

## **Solution Comparison**

Since the area of the bar is unity, the uniaxial stress s is equivalent to the applied load P. The stress varies linearly, initially increasing to twice the yield stress over a period of 1 second and then reducing back to the yield stress in 0.5 seconds.

$$P(t) = 20(t+1)$$
  $0 < t < 1$   $(0.0.81)$ 

$$P(t) = 40 - 40(t-1)$$
 1 < t < 1.5

and the stress history is therefore,

$$\sigma(t) = \sigma_0 + \dot{\sigma}_1 t \qquad 0 < t < 1$$

$$(0.0.82)$$

$$\sigma(t) = \sigma_1 + \dot{\sigma}_2 t \qquad 1 < t < 1.5 \qquad (0.0.83)$$

Where  $\sigma_1 = \sigma_0 + \dot{\sigma}_1$  since t = 1 and assuming  $\dot{\sigma}_1$  is constant.

Considering the period for 0<t<1, the rate of creep strain is

$$\dot{\varepsilon}^{c} = \operatorname{Am} \left( \sigma_{o} + \dot{\sigma}_{1} t \right)^{n} t^{m-1} \tag{0.0.84}$$

and integrating with respect to time, gives the creep strain as,

$$\epsilon^{c} = Am \int_{0}^{t} \left(\sigma_{o} + \dot{\sigma}_{l}t\right)^{n} t^{m-l} dt \tag{0.0.85}$$

The plastic strain is evaluated from the strain hardening definition as,

$$\varepsilon^{p} = \frac{\dot{\sigma}_{l}t}{h} \tag{0.0.86}$$

and the elastic strain is

$$\varepsilon^{e} = \frac{\sigma_{o} + \dot{\sigma}_{l}t}{F} \tag{0.0.87}$$

Summing the components, the total strain over the period 0<t<1 is then,

$$\begin{split} \epsilon^{T} &= \epsilon^{e} + \epsilon^{p} + \epsilon^{c} \\ &= \frac{\sigma_{0} + \dot{\sigma}_{1}t}{F} + \frac{\dot{\sigma}_{1}t}{h} + At^{1/2} \bigg[ \sigma_{0}^{5} + \frac{5}{3}\sigma_{0}^{4}\dot{\sigma}_{1}t + 2\sigma_{0}^{3}\sigma_{1}^{2}t^{2} + \frac{5}{9}\sigma_{0}\dot{\sigma}_{1}^{4}t^{4} + \frac{1}{11}\dot{\sigma}_{1}^{5}t^{5} \bigg] \end{split} \tag{0.0.88}$$

For the period 1<t<1.5 the prescribed stresses decrease. The new creep strain rate is then

$$\varepsilon^{c} = A \left[ \sigma_{1} + \dot{\sigma}_{2} (t - 1) \right]^{5} t^{-1/2}$$
(0.0.89)

For simplicity the rate of decrease of stress  $\dot{\sigma}_2$  has been set to the minus of the maximum attained stress  $\sigma_1$ . Integrating the creep strain over the period 1<t<1.5 yields,

$$\epsilon_2^c = A\sigma_1^5 t^{1/2} \left[ 32 - \frac{80}{3} t + 16t^2 - \frac{40}{7} t^3 + \frac{10}{9} t^4 - \frac{1}{11} t^5 \right] - \frac{11531}{693} A\sigma_1^5 (0.0.90)$$

and the total strain for the period 1<t<1.5 is,

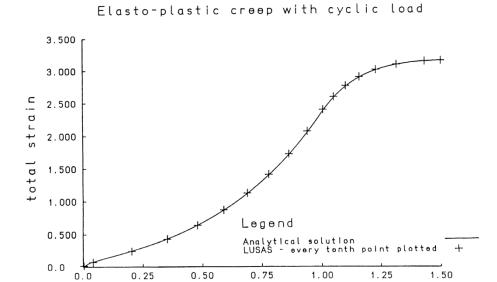
$$\varepsilon^{T} = \frac{\sigma_{1} + \dot{\sigma}_{2}(t - 1)}{E} + \frac{\dot{\sigma}_{1}}{h} + \frac{4672}{693} A \sigma_{o}^{2} + \varepsilon_{2}^{c}$$
 (0.0.91)

Note: the solution produces strains of over 3 which are invalid in the context of linear deformations. However, the purpose of the example is simply to demonstrate the integration of combined creep and plastic material behaviour and to compare with an analytical solution derived from the same assumptions as the implemented theory.

The LUSAS results are shown in Figure 1 and compare exactly with those computed by equations (0.0.88) and (0.0.91).

## **Input Data**

X05D26A.DAT



time

## **Extension Of A Double Notched Specimen**

## **Keywords**

**Incompatible Elements, Plasticity** 

## **Description**

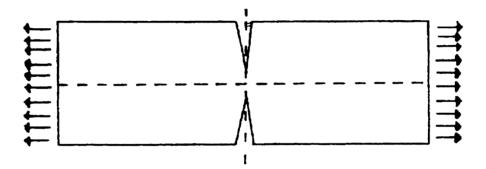
A notched bar is stretched until the limiting load, following yield, is attained. Standard (compatible) displacement based elements produce a load- deflection curve which increases monotonically beyond the limit load, whilst the incompatible elements are able to capture the true behaviour.

#### **Finite Element Model**

A rectangular mesh of 15\*5 elements models one quarter of the structure. The bar is considered to be in a state of plane strain.

## **Geometry**

Length	30.0
Width	10.0



## **Material properties**

Young's modulus	70.0
Poisson's ratio	0.3
Yield stress	0.243
Hardening	0.0

## **Boundary Conditions**

Symmetry boundary conditions are applied along the edges. The notch is modelled by freeing the displacements along its length.

## **Solution Comparison**

The theoretical limit load is 5.94 x yield stress = 1.4434. Two solutions are illustrated, one for the standard displacement elements and the other for the incompatible elements. The applied load is calculated by the summation of the reactions along the extended edge. From symmetry, the expected load is half that calculated for the full plate.

The initial mesh and support conditions are shown in the top diagram of figure 1. The two other diagrams of figure 1 shown the distribution of yield and the equivalent von Mises stress at the end of the final loadstep, for the incompatible elements. Figure 2 displays the corresponding yield and equivalent stress plots for the standard displacement elements. The yield zone for the standard elements is less localised than that of the incompatible elements and is capable of sustaining a larger load as is shown in figure 3.

#### References

1. Simo J.C., Rifai, M.S., A class of mixed assume strain methods and the method of incompatible modes, Int. J. Num. Meth. Eng., 31, 385-405, 1991.

## **Input Data**

#### X05D27A.DAT

The data file for the standard elements is obtained by substituting the element type QPN4 for OPN4M in the above data file.

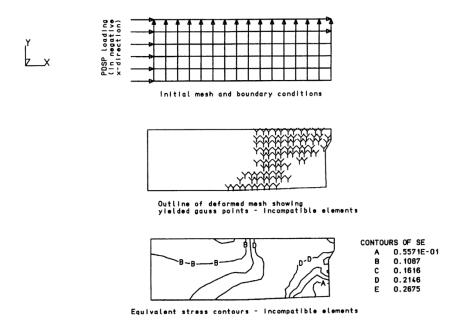


Figure 1 Failure of Notched Bar (1)

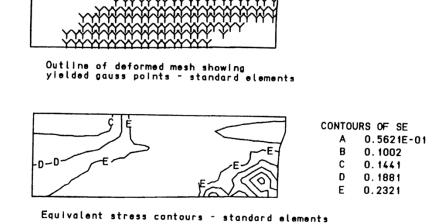


Figure 2 Failure of Notched Bar (2)

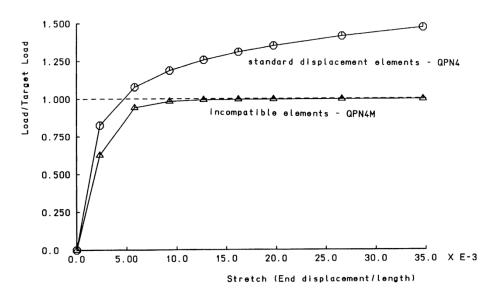


Figure 3 Load/Displacement Behaviour of Notched Bar

## **Plane Strain Limit Load Analysis of Granular Material**

## **Keywords**

Elasto-Plastic, Volumetric Crushing, Drucker-Prager, Mohr-Coulomb

## **Description**

A strip of granular material compressed by a rigid footing is analysed to determine its limit load. Details of the problem can be found in Figure 9.

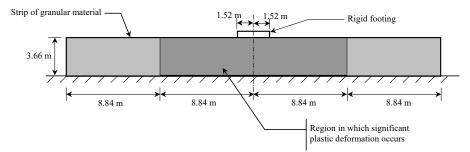


Figure 9: Dimensions of the granular material for which the limit load analysis is carried out. The dark shaded area is considered the region where significant plastic deformation takes place.

The strip of granular material is in fact considered to be infinitely long, so the vertical boundaries at the edge of the specimen have been positioned such that their presence has a minimal effect on the results.

The performance of three material models is examined – the Drucker-Prager material model, the Mohr-Coulomb material model and the Volumetric Crushing material model.

#### **Discretisation**

Utilising symmetry about the centre plane, only one half of the problem in Figure 9 is meshed. The strip itself is modelled using quadratic plane strain elements, QPN8. The finite element mesh can be found in Figure 10.

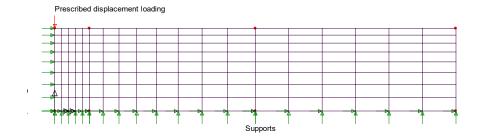


Figure 10: Finite element mesh for the granular material limit load analysis, showing supports and loading.

## **Material properties**

The material properties of the strip are defined using the Drucker-Prager material model in one test, the Mohr-Coulomb material model in a second test and the Volumetric Crushing material model in a third. Table 1 gives the Mohr-Coulomb material properties for this test.

Table 1: Mohr-Coulomb material properties for granular material limit load verification test.

Young's modulus, E	Poisson's ratio, ν	Angle of friction, φ	Cohesion, c
206.8 MPa	0.3	20o	68.94 KPa

#### **Mohr-Coulomb material properties**

The Mohr-Coulomb material properties can be found in Table 1.

### **Drucker-Prager material properties**

In order to determine the appropriate Drucker-Prager parameters to use in this test it is useful at this stage to probe the relationship between the Drucker-Prager and Mohr-Coulomb material models.

The general form of the Drucker-Prager yield function is given by

$$F(\underline{\boldsymbol{\sigma}}, \kappa) = \alpha I_1 + \sqrt{J_2} - k \tag{0.0.92}$$

where  $I_1$  is the first stress invariant,  $J_2$  is the second stress invariant and  $\alpha$  and k are constants that define the yield surface.

$$I_1 = \sigma_{xx} + \sigma_{yy} + \sigma_{zz} \tag{0.0.93}$$

$$J_{2} = \frac{1}{2} \left[ (\sigma_{xx} - \sigma_{m})^{2} + (\sigma_{yy} - \sigma_{m})^{2} + (\sigma_{zz} - \sigma_{m})^{2} \right] + \tau_{xy} + \tau_{yz} + \tau_{zx}$$
(0.0.94)

where 
$$\sigma_{\rm m} = \frac{1}{3} I_1$$

$$(0.0.95)$$

The LUSAS Drucker-Prager material model has its yield criteria coincide with the outer apices of the Mohr-Coulomb material model (see Figure 11). In this case, the yield function is given by

$$F(\underline{\boldsymbol{\sigma}},\kappa) = \frac{2\sin\phi}{\sqrt{3}(3-\sin\phi)}I_1 + \sqrt{J_2} - \frac{6\cos\phi}{\sqrt{3}(3-\sin\phi)}c \qquad (0.0.96)$$

where  $\phi$  is the angle of friction and c is the cohesion.

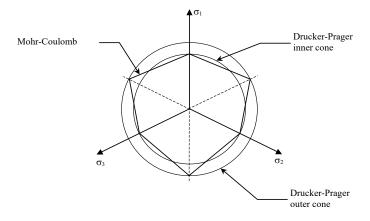


Figure 11: The two forms of the Drucker-Prager material model. The outer cone coincides with the outer apices of the Mohr-Coulomb model while the inner cone coincides with the inner apices of the Mohr-Coulomb model.

There is another form of the Drucker-Prager material model that coincides with the inner apices of the Mohr-Coulomb model (Figure 11). This given by

$$F(\underline{\boldsymbol{\sigma}},\kappa) = \frac{2\sin\phi}{\sqrt{3}(3+\sin\phi)}I_1 + \sqrt{J_2} - \frac{6\cos\phi}{\sqrt{3}(3+\sin\phi)}c \qquad (0.0.97)$$

It is possible to modify the outer Drucker-Prager cone in Figure 11 so that it coincides with the inner apices of the Mohr-Coulomb model. This is achieved by modifying the angle of friction and the cohesion used in (5) so that yield function for the outer cone form simulates the inner form.

The first step is to compute values for  $\alpha$  and k that represent the inner cone. Comparing the inner cone formulation in (6) with the more general form in (1), leads to

$$\alpha_{int} = \frac{2\sin\phi}{\sqrt{3}(3 + \sin\phi)}$$
 and  $k_{int} = \frac{6\cos\phi}{\sqrt{3}(3 + \sin\phi)}c$  (0.0.98)

Next, by comparing the outer cone formulation in (5) with (1),  $\alpha_{int}$  and  $k_{int}$  can then be used to derive an angle of friction and a cohesion that simulates the inner cone. They are given by

$$\sin\phi = \frac{3\alpha_{\rm int}\sqrt{3}}{2 + \alpha_{\rm int}\sqrt{3}} \tag{0.0.99}$$

$$c = \frac{k_{int}\sqrt{3}(3 - \sin\phi)}{6\cos\phi} \tag{0.0.100}$$

In the computation of the cohesion in (9) the value of  $\phi$  computed from (8) is used.

Applying this technique to the problem being investigated here, the angle of friction and the cohesion that will simulate the inner cone form of the Drucker-Prager material model, using the data in Table 1, are

$$\phi = 16.2$$
 and  $c = 54927 \text{ Pa}$  (0.0.101)

It can be seen that the difference between the material properties in Table 1 and those in equation (10) is quite significant.

Having shown how the LUSAS Drucker-Prager yield function can be adjusted to take different forms, we can proceed to make a further adjustment for plane strain. A proposed modification, corresponding to an adjustment of failure loads, is based on the following.

$$\alpha = \sin \frac{\phi}{3}$$
 and  $k = c \cos \phi$  (0.0.102)

Using (1), (5) and (11), the angle of friction and cohesion that should be used in the LUSAS Drucker-Prager model for plane strain are

$$\phi = 15.9^{\circ}$$
 and  $c = 53\ 007\ Pa$  (0.0.103)

A summary of material properties to use with the Drucker-Prager material model is given in Table 2.

Table 2: Material properties to be used with the LUSAS Drucker-Prager material model.

	Young's modulus, E	Poisson's ratio, ν	Angle of friction, φ	Cohesion, c
Outer cone form	206.8 MPa	0.3	20o	68 940 Pa
Inner cone form	206.8 MPa	0.3	16.2o	54 927 Pa
Proposed plane strain	206.8 MPa	0.3	15.9o	53 007 Pa

#### Volumetric crushing material properties

The Volumetric Crushing material properties are computed to match the Drucker-Prager properties. The yield function for the Volumetric Crushing material model is given by

$$\Phi = \frac{1}{2} \mathbf{S}^{\mathsf{T}} \mathbf{L} \mathbf{S} - \left( \mathbf{a}_0 + \mathbf{a}_1 \mathbf{p} + \mathbf{a}_2 \mathbf{p}^2 \right) \tag{0.0.104}$$

where  $\underline{\mathbf{S}}$  is the deviatoric stress vector,  $\underline{\underline{\mathbf{L}}}$  is given by

$$\mathbf{L} = \begin{bmatrix} 1 & 0 & 0 & 0 & 0 & 0 \\ 0 & 1 & 0 & 0 & 0 & 0 \\ 0 & 0 & 1 & 0 & 0 & 0 \\ 0 & 0 & 0 & 2 & 0 & 0 \\ 0 & 0 & 0 & 0 & 2 & 0 \\ 0 & 0 & 0 & 0 & 0 & 2 \end{bmatrix}$$
 (0.0.105)

 ${\bf p}$  is the volumetric pressure and  ${\bf a}_0$ ,  ${\bf a}_1$ ,  ${\bf a}_2$  are user defined constants. If the following applies

$$a_0 + a_1 p + a_2 p^2 = \sigma_o ag{0.0106}$$

then the yield function in (13) resembles the basic von Mises yield criterion, where  $\sigma_o$  is the uniaxial yield stress. If we also assume infinitesimal strains then the volumetric pressure,  $\,p\,$ , is defined as

$$p = -\frac{1}{3}(\sigma_{xx} + \sigma_{yy} + \sigma_{zz})$$
 (0.0.107)

When both (15) and (16) apply the Volumetric Crushing yield criteria can be re-expressed as

$$J_2 - \frac{1}{9} \left( 3a_0 - a_1 I_1 + a_2 I_1^2 \right) = 0 \tag{0.0.108}$$

Taking the square of (1) we have for the Drucker-Prager yield function

$$J_2 - (k^2 - 2\alpha k I_1 + \alpha^2 I_1^2) = 0 (0.0.109)$$

By equating the Volumetric Crushing yield function in (17) with the squared Drucker-Prager yield function in (18) the constants for the Volumetric Crushing model are given by

$$a_0 = k^2$$
,  $a_1 = 6\alpha k$ ,  $a_2 = 9\alpha^2$  (0.0.110)

If the inner cone form of the Drucker-Prager model is to be matched, then (6) should be used to compute  $\alpha$  and k while if the plane strain form is to be matched then (11) should be used.

The Volumetric Crushing constants can then be computed from (19) using these values of  $\,\alpha$  and  $\,k$  .

The remaining material properties for the Volumetric Crushing model are

Bulk modulus : 
$$K = \frac{E}{3(1-2v)} = 1.725 \times 10^8 \text{ Pa}$$

Shear modulus : 
$$G = \frac{E}{2(1+\nu)} = 7.962 \times 10^7 \text{ Pa}$$

The tension cut off,  $p_c$ , is defined as a large number,  $1 \times 10^9$  Pa.

A summary of the material properties for the Volumetric Crushing material model can be found in Table 3. To enable a comparison with the Drucker-Prager material model, the pressure-logarithm of relative volume curve is defined as linear, see Table 3: Material properties for the Volumetric Crushing material model that simulate the plane strain adjusted Drucker-Prager material model.

Table 3: Material properties for the Volumetric Crushing material model that simulate the plane strain adjusted Drucker-Prager material model.

Drucker- Prager match	Bulk Modulus, K	Shear Modulus, G	Constant a <sub>0</sub>	Constant a <sub>1</sub>	Constant a <sub>2</sub>	Tension cut off, p <sub>c</sub>
Inner cone	172.5 MPa	79.62 MPa	4.509 GPa	47.61 KPa	0.1257	1.0 GPa
Plane strain	172.5 MPa	79.62 MPa	4.192 GPa	45.10 KPa	0.1213	1.0 GPa

Table 4: Data for the pressure-logarithm of relative volume curve.

	In (V/V₀)	р
Start point	0	0
End point	-1 x 10 -4	17.25 KPa

## Loading

A compressive prescribed displacement is applied to the node at the centre of the rigid footing in the vertical direction. This is shown on the mesh in Figure 10. This vertical displacement is then linked to remainder of the nodes underneath the footing via constraint equations. In this way perfect roughness is ensured. A total prescribed displacement of 1.25m is applied.

#### **LUSAS Results**

Figure 12 shows results that are obtained from the Mohr-Coulomb, the Drucker-Prager and the Volumetric Crushing material models. On the same graph are limit analysis solutions from Terzaghi and Prandtl. The graph shows that all three material models have produced limit loads that lie within or are close to the boundaries of the two analytical solutions. The models have therefore returned acceptable limit loads.

In this test the outer cone form of the Drucker-Prager model produced a limit load of approximately 2.5 MPa. Both the inner cone form and the plane strain adjusted form produce results that are much closer to the analytical solutions. Since the outer cone form represents triaxial compression this illustrates that care is needed in identifying the dominant stress state and adjusting the Drucker-Prager material parameters accordingly.

Comparing the adjusted LUSAS Drucker-Prager results with those from the LUSAS Mohr-Coulomb model shows that the plane strain adjustment has brought the results closer to the Mohr-Coulomb results, but indicates that adjustment could be refined further for this particular problem, However, since the stress combination in this test is not simple, further refinement may not be feasible or straightforward.

Considering the Volumetric Crushing material model, results from both the inner cone and the plane strain matched models lie closer to the Mohr-Coulomb results than the Drucker-Prager results do. However, the Volumetric Crushing model produces results that are softer than both models. The latter is due to the non-associative flow rule that the Volumetric Crushing model uses.

## **Input Data**

Drucker-Prager : X05D28E.DAT, X05D28F.DAT, X05D28G.DAT

Mohr-Coulomb : X05D28H.DAT

Volumetric crushing : X05D28I.DAT, X05D28J.DAT

#### References

- 1. A. Mouazen, M. Nemenyi. 'Tillage tool design by the finite element method: Part1. Finite element modelling of soil plastic behaviour'. J. Agric. Engng. Res, Volume 72, pp.37-51 (1999).
- D.R.J. Owen, E. Hinton . 'Finite elements in plasticity: Theory and Practice'. Pineridge Press Limited, Swansea, UK. (1980).
- G. Duffett, G. Jelinc, P. Lyons. 'A study of the material models in LUSAS for the analysis
  of geotechnical materials'. In Computational Plasticity: Fundamentals and Applications,
  Barcelona 1995, Editors D.R.J. Owen, E. Onate. Pineridge Press Limited, Swansea, UK.
  (1995).
- 4. Theory Manual, LUSAS v13. LUSAS, Kingston-upon-Thames, UK (2001).
- LUSAS internal report SDR/LUSAS/061. 'Crushable foam material model', LUSAS, Kingston-upon-Thames, UK (1994).
- LUSAS internal report SDR/LUSAS/255a. 'Verification of the LUSAS Drucker-Prager material model', LUSAS, Kingston-upon-Thames, UK (2001).

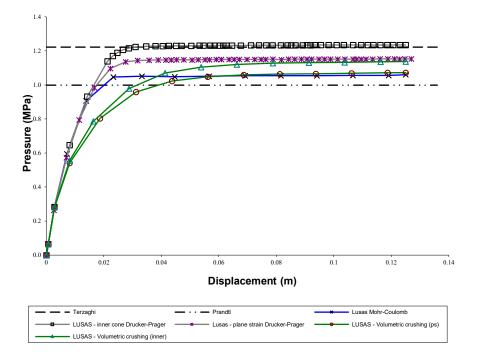


Figure 12: Variation of pressure underneath the footing against applied displacement. Volumetric crushing key: ps – matched with plane strain Drucker-Prager, inner – matched with Drucker-Prager inner cone.

# **Example 5.2.9**

## **Plane Strain Analysis Of Slope Stability Limit**

## **Keywords**

Elasto-Plastic, Mohr-Coulomb, Associative/non-associative Flow

### **Description**

In this analysis gravity is increased in value until the slope collapses. The larger the final value above the actual value of gravity the safer the slope. The problem is modelled using plane strain elements. The analysis is run twice, once using associative plastic flow properties and then for non-associative plastic flow. The non-symmetric solver is used for the non-associative flow solution.

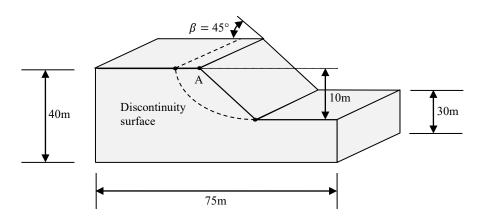


Figure 13: Dimensions of slope.

The strip of granular material is in fact considered to be infinitely long, so the vertical boundaries at the edge of the specimen have been positioned such that their presence has a minimal effect on the results.

#### **Discretisation**

The strip itself is modelled using quadratic plane strain elements, QPN8. The finite element mesh can be found in Figure 14.

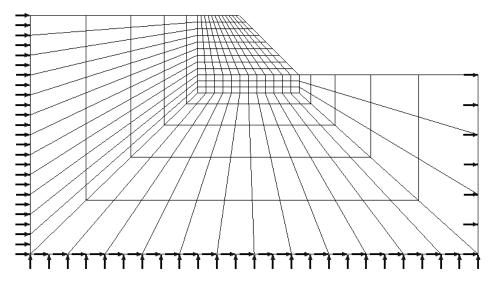


Figure 13: Finite element mesh showing supports.

## **Material properties**

The material properties of the strip are defined using the Modified Mohr-Coulomb material model. Table 1 gives the material properties for this test.

Table 5: Modified Mohr-Coulomb material properties for granular material limit load verification test.

Young's modulus, E	Poisson's ratio, ν	Angle of friction, Ø	Cohesion, c
$2x10^4$ KPa	0.49	20°	50 KPa

Anglee of dilation $\psi$	Density $ ho$	
Test 1. 20°	2039 Kg/m <sup>3</sup>	
Test 2. 10°	2037 <b>K</b> g/m	

### Loading

The soil weight due to gravity is increased using the arclength method until the soil fails.

#### **LUSAS Results**

If the slope angle  $\beta$  is greater than the angle of internal friction  $\phi$  the slope will collapse when the ratio N reaches a critical value

$$N = h \frac{\gamma}{c} \tag{5.2.39}$$

where h is the slope height and  $\gamma$  the specific weight of the soil

$$\gamma = \rho g \tag{5.2.40}$$

and  $\rho$  is the soil density, g the gravitational acceleration. The critical value of N, for the present geometry is tabulated [1] as

$$N_{\rm s} \approx 16.18$$
 (5.2.41)

The load factor  $\alpha_c$  applied to gravitational acceleration at collapse is therefore

$$a_c = \frac{N_s}{N} = \frac{N_s c}{h \rho g} = 4.045 \tag{5.2.42}$$

Figure 15 shows the increase in the gravity load factor until failure. The displacement plotted is for the total displacement at point A as shown on figure 13.

Both the associative and non-associative material models slightly overpredict the ultimate failure load.

The contours of effective stress at failure are shown in figure 16.

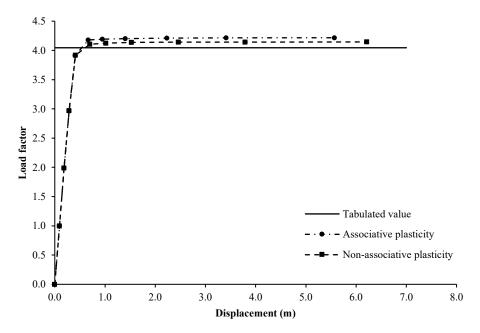


Figure 15 Load v. total displacement at point A

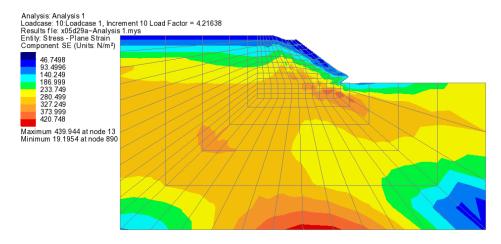


Figure 16 Contours of effective stress at failure (Associative flow)

### **Input Data**

Associative flow : X05D29A.DAT Non-associatve flow : X05D29B.DAT

#### References

1 de Souza Neto EA, Peric D. and Owen DRJ, Computational Methods for Plasticity, John Wiley and Sons Ltd, 2008.

# **Example 5.4.1**

# **Buried Pipe With Soil-Pipe Interface Modelled Using 3d Interface Model**

### Keywords

Soil Structure Interaction, Interface Model, Elasto-plastic, Friction

### **Description**

This example (figure 1) consists of an elastic pipe buried in a homogeneous and elastic soil. The interface between the pipe and soil is modelled using an elasto-plastic interface material. Three cases of interface friction are considered:-

- (i) frictionless slip (tano = 0.001),
- (ii) stick (tano = 2),
- (iii) frictional slip (tano = 0.25).

#### **Discretisation**

The discretisation (figure 2) is similar to that used in [1]. HX20 elements are used to discretise the pipe, soil and interface. The model is totally restrained from moving in the z-direction.

### **Theory**

The interaction between the soil and pipe is governed by a frictional type law i.e. bond between the pipe and soil is maintained by a frictional force and once this is overcome debonding and slip will occur. This friction/slip behaviour is modelled using an elasto-plastic interface model, with yielding (slip) defined using the Mohr-Coulomb criterion [2] defined as

 $\tau = c + \sigma \tan \phi \tag{0.0.111}$ 

where  $\tau$  maximum shear stress,

σ stress normal to the soil/pipe interface,

c cohesion,

friction angle.

## Comparison

Figure 3 is the deformed mesh for the frictionless slip analysis. A significant amount of sliding is visible in the interface. Figure 4 shows the shear stress distribution along the pipe-soil interface.

### References

- 1. DESAI,C.S., ZAMAN,M.M., LIGHTNER,J.G. and SIRIWARDANE,H.J. Thin-layer element for interfaces and joints. Int. J. Num. Anal. Meth. Geomech., Vol.8, 19-43, 1984.
- OWEN,D.R.J. and HINTON,E. Finite elements in plasticity: theory and practice. Pineridge Press, Swansea, UK, 1980.
- 3. KATONA,M.G. A simple contact-friction interface element with applications to buried culverts. Int. J. Num. Anal. Meth. Geomech., Vol.7, 371-384, 1983.

### **Input data**

X05D41A.DAT

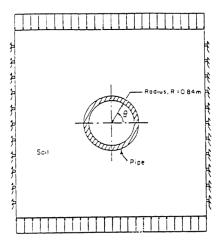
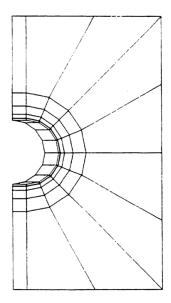


Figure 1 Buried Pipe Example



**Figure 2 Finite Element Mesh** 

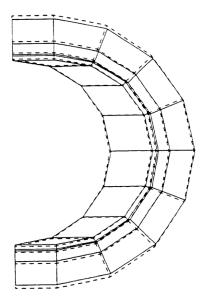
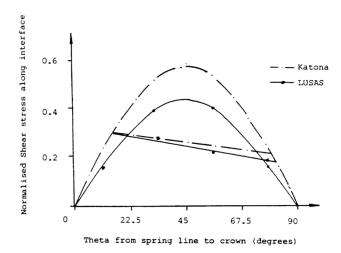


Figure 3 Deformed Mesh Showing Sliding on Interface



# **Example 5.4.2**

# Elasto-Plastic Analysis of a Thick Porous Cylinder Under Internal Pressure

### Keywords

Cylinder, Drained, Plasticity, Porous, Undrained

### **Description**

An infinitely long thick porous cylinder of internal and external radii 100 mm and 200 mm respectively is subjected to an increasing internal pressure.

### Geometry, boundary and loading conditions

The cylinder is idealised into both plane strain and 3D problems. The global Z axis is coincident with the axis of the cylinder. In the 3D analysis, one symmetric quarter of the structure with a length of 80mm is considered, as shown in Figure 1; boundary and loading conditions are also shown in the figure. Symmetric conditions are specified on the straight boundaries; the curved internal surface (which has a pressure loading) and external surface are unrestrained. The plane strain idealisation is a projection of the 3D idealisation in the Z-direction.

#### **Discretisation**

The 3D finite element discretisation with 120 hexahedral (HX20P) elements is shown in Figure 2. In the plane strain analysis, 30 quadrilateral (QPN8P) elements are used without discretisation in the Z-direction.

### **Material properties**

The nonlinear material behaviour is modelled using a von Mises yield criterion and a perfectly plastic post-yield response. The material properties are as follows:

Fully drained analysis:

Young's modulus  $E = 21000 \text{dN/mm}^2$ 

Poisson's ratio v = 0.3

Yield stress  $\sigma_{vs} = 24 \text{dN/mm}^2$ 

Undrained analysis:

Young's modulus of the soil skeleton  $E'= 17830.2 \text{ dN/mm}^2$ 

Poisson's ratio of the soil skeleton v = 0.1038

Bulk modulus of the pore fluid  $K_w = 10^4$ 

Porosity  $\eta = 1$ 

### **Theory**

Under fully drained conditions, geotechnical elements give identical results to their counterpart continuum elements using drained properties. Furthermore, following the principle of effective stress, if an undrained analysis is performed with soil and pore fluid properties being chosen so as to correspond to the drained material parameters, the obtained displacement and strain results (but not stresses or pore pressures) are comparable to those obtained by the fully drained analysis. Slight difference may occur due to a two-field (displacement and pressure) interpolation employed in the undrained analysis.

With the increase of the applied internal pressure, the cylinder deforms elastically until an internal pressure of approximately 12dN/mm<sup>2</sup> is reached; beyond this pressure level the cylinder behaves elasto-plastically. The spread of plastic deformation through the thickness of the cylinder can be seen from the distribution of the hoop stress component with increasing load in 5.2.2.

#### **LUSAS Results**

For the current materially non-linear, but geometrically linear analysis, the progressive radial displacements of the internal circumference against the internal pressure are:

	Internal pressure		8	10	12	14	16	18
	Т	QPN8	0.0721	0.0901	0.1116	0.1399	0.1823	0.2566
	Target	HX20	0.0726	0.0908	0.1115	0.1398	0.1828	0.2622
$u_r$	LUSAS	QPN8P	0.0721	0.0901	0.1116	0.1399	0.1823	0.2566
	fully drained	KX20P	0.0726	0.0908	0.1115	0.1398	0.1828	0.2622
	LUSAS	QPN8P	0.0721	0.0901	0.1116	0.1399	0.1821	0.2565
	undrained	KX20P	0.0726	0.0908	0.1115	0.1398	0.1826	0.2623

From the above table, 3D drained/undrained results agree favourably with 2D analysis. However, the 3D analysis reproduces better axisymmetry of the problem than the plane strain analysis. The following table gives the pore pressure on the inner and outer faces of the cylinder obtained by the 3D undrained analysis:

Internal pres	sure	8	10	12	14	16	18
	inner	-1.3206	-1.6508	-1.3847	-0.0732	0.9772	2.1293
Pore pressure	outer	-1.3206	-1.6508	-2.0188	-2.4936	-3.1749	-4.3960

As the deformation is small, similar results have been obtained with the consideration of geometric nonlinearity. A difference in pore pressure between the inner and outer surfaces can be seen at an internal pressure of 12 when the material has yielded.

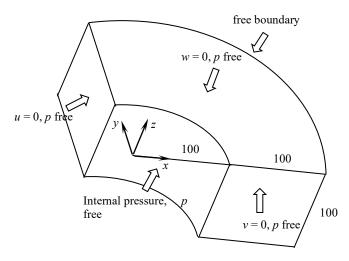
#### References

- Owen, D.R.J., Hinton, E. Finite Elements in Plasticity: Theory and Practice Publisher. Pineridge Press Ltd. Swansea, U.K. 1980. ISBN 0-906674-05-2
- Zienkiewicz, O.C... The Finite Element Method, 3rd Edition. Publisher. McGraw Hill Ltd, London, U.K. 1977. ISBN 0-07-084072-5
- 3. FEAL internal Report 'Plane Strain', FEAL708, Finite Element Analysis Ltd. Forge House, 66 High Street, Kingston Upon Thames, Surrey KT1 1HN. 1986.
- 4. Hodge, P.G., White, G.N. A quantitative comparison of flow and deformation theories of plasticity. J. Applied Engineering Mechanics, Vol.17, 180-184, 1950.

#### **Input Data**

X05D42A.DAT (Plane strain) X05D42B.DAT (3D continuum)

Note: The load incrementation may alternatively be defined automatically using NONLINEAR CONTROL INCREMENTATION and TERMINATION cards as in 5.2.2.



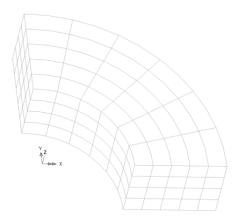


Figure 2

# **Example 5.5.1**

## **Cylindrical Shell Subject To Self Weight**

### **Keywords**

Anisotropic, Elasto-Plastic

### **Description**

This example (figure 1) consists of a cylindrical shell subjected to a uniformly distributed load over the outer surface. The shell has a length of 7.6m, a radius of 7.6m and a thickness of 0.076m. The angle subtended by the arc of the shell at the centre of the circle is 80 degrees.

#### **Discretisation**

Because of symmetry conditions only a quarter of the shell is modelled. A 4\*4 mesh of QSL8 (Semiloof shell) elements are used to discretise the shell; this is shown is figure 1.

### **Material properties**

This problem is geometrically linear but the non-linear Hill criterion for anisotropic plasticity is used to model the material. The following values are used to describe the yield surface for each of the anisotropic property definitions:

1. Isotropic

$$\overline{\sigma}_{11} = \overline{\sigma}_{22} = \overline{\sigma}_{33} = \overline{\sigma} \tag{0.0.112}$$

$$\overline{\sigma}_{12} = \overline{\sigma}_{23} = \overline{\sigma}_{13} = \frac{1}{\sqrt{3}}\overline{\sigma}$$
 (0.0.113)

2. Anisotropic

$$\overline{\sigma}_{11} = 2\overline{\sigma} \tag{0.0.114}$$

$$\overline{\sigma}_{22} = \overline{\sigma}_{33} = \overline{\sigma} \tag{0.0.115}$$

$$\overline{\sigma}_{12} = \overline{\sigma}_{23} = \overline{\sigma}_{13} = \frac{1}{\sqrt{3}}\overline{\sigma} \tag{0.0.116}$$

3. Anisotropic

$$\overline{\sigma}_{11} = \overline{\sigma} \qquad (0.0.117)$$

$$\overline{\sigma}_{22} = 2\overline{\sigma} \qquad (0.0.118)$$

$$\overline{\sigma}_{33} = \overline{\sigma} \qquad (0.0.119)$$

$$\overline{\sigma}_{12} = \overline{\sigma}_{23} = \overline{\sigma}_{13} = \frac{1}{\sqrt{3}}\overline{\sigma} \qquad (0.0.120)$$

Here  $\bar{\sigma}$  is the reference yield stress (=4200 kN/m<sup>2</sup>). Note that even though the shell formulation is based on the plane stress hypothesis, the complete yield surface must be defined for the material. No hardening is present in this problem. The Elastic modulus, E = 21E6 kN/m<sup>2</sup> and a Poisson's ratio of zero are used to define the elastic behaviour of the material.

### Comparison

Figure 2 shows the load-displacement curves at the point A. These results agree exactly with the results in the reference solution [1]. It can be seen that anisotropic properties allow a much higher limiting load to be sustained (an increase of approximately 25%).

#### References

- 1. De BORST, R. and FEENSTRA, P.H. Studies in Anisotropic Plasticity with Reference to the Hill Criterion. Int. J. Num. Meth. Eng., v29, pp315-336, 1990.
- LUSAS Theory Manual. Finite Element Analysis Ltd, Forge House, 66 High Street, Kingston-upon-Thames, Surrey, KT1 1HN.

### **Input Data**

X05D51A.DAT

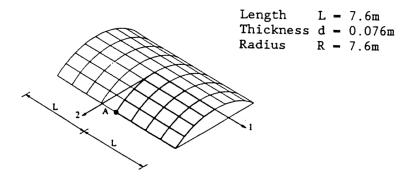


Figure 1 Geometry of cylindrical shell

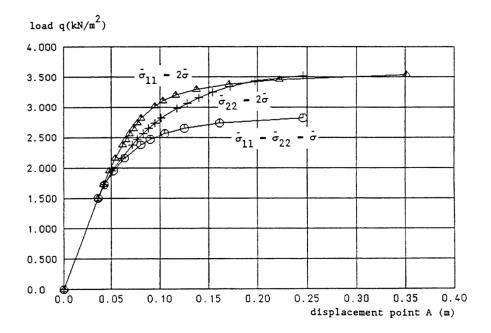


Figure 2 Load-displacement curve at point A of the cylindrical shell for isotropic and anisotropic material properties and using 4\*4 Semiloof elements.

# **Example 5.6.1**

## **Axisymmetric analysis of sand under compression**

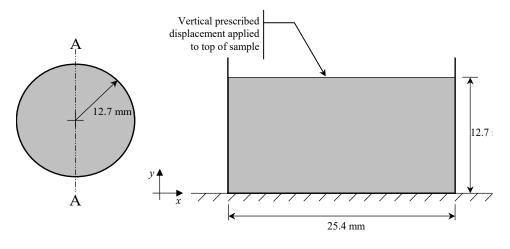
## **Keywords**

Elasto-Plastic, Drucker-Prager, Mohr-Coulomb

### **Description**

This example is a uniaxial strain test of McCormick Ranch sand, details of which can be found in Figure 13. A compressive load is applied to the sand sample and the resulting stress and strain characteristics are compared with experimental and other finite element solutions.

The axisymmetric nature of the model means that the example is analysed as an axisymmetric problem. The line of axisymmetry passes through the centre of the sand sample in the y direction.



Top view of sand sample

Section through A-A

Figure 14: The uniaxial McCormick Ranch sand test. Dimensions are not to scale.

#### **Discretisation**

The finite element mesh is shown in Figure 14, where only one half of the cross-section illustrated in Figure 13 has been modelled. The sand is modelled using four linear enhanced strain axisymmetric elements QAX4M.

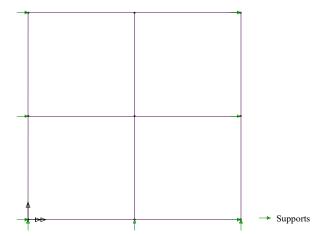


Figure 15: Finite element mesh for the uniaxial McCormick Ranch sand test. A vertical compressive prescribed displacement is applied to the top set of nodes.

## **Material properties**

The Drucker-Prager and Mohr-Coulomb material models are used to define the material properties of the sand. The material properties can be found in Table 5. For the Mohr-Coulomb material model, the dilation angle is set to the angle of friction.

Table 6: Material properties for uniaxial McCormick Ranch sand test.

Young's Modulus, E	Poisson's Ratio, v	Angle of friction, φ	Cohesion, c
689 MPa	0.25	7.15	0.577 MPa

## Loading

The top face of the sand sample is loaded by a compressive prescribed displacement in the y direction. The maximum vertical displacement is 0.8382 mm. In the full experimental test, the variation of the prescribed displacement is shown in Figure 15.

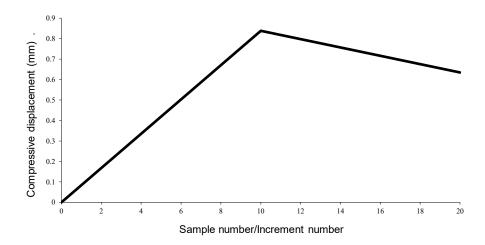


Figure 16: Variation of applied compressive prescribed displacement for the uniaxial McCormick Ranch sand test. The graph shows the full variation used for the experimental results. Only one set of results, from [1], is published for the reduction in displacement between increment numbers 10 and 20, with very specific cap and tension cut off definitions.

#### **LUSAS Results**

Results for the test can be found Figure 16, which shows the variation of compressive stress with compressive strain in the direction of the applied displacement. In addition to the results from the LUSAS Drucker-Prager model, the graph also shows experimental results [1,2], results from the LUSAS Mohr-Coulomb model and finite element results from [1]. Results in [1] are from a Drucker-Prager model that has a cap and a tensions cut off.

The graph in Figure 16 shows that the results from LUSAS are in good agreement with experimental results and results from other FE packages when compressing the sand specimen. Both LUSAS Drucker-Prager and Mohr-Coulomb material models produce identical results in this test.

Experimental results have also been generated for the case when the prescribed displacement loading is reversed (see Figure 15 between sample numbers 10 and 20), shown by the reduction in compressive stress after its peak value. Reference [1] has been able to approximate these experimental results with a very specific cap definition and tension cut off.

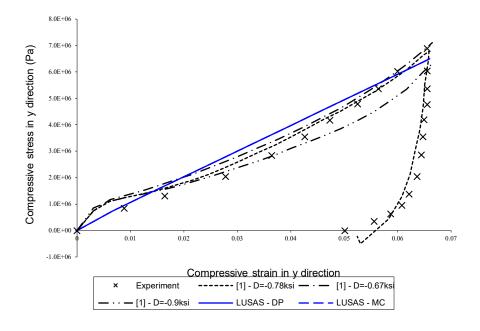


Figure 17: Variation of compressive stress with compressive strain in the direction of the applied displacement. Graphs with D values represent results from Drucker-Prager models that have caps and tension cut offs, where D is used to define the cap. See [1]. MC – LUSAS Mohr-Coulomb material model, DP – LUSAS Drucker-Prager material model.

## **Input Data**

Drucker-Prager : X05D61A.DAT Mohr-Coulomb : X05D61B.DAT

#### References

- 1. K.-J. Bathe, M.D. Snyder, A.P. Cimento, W.D. Rolph III. 'On some current procedures and difficulties in finite element analysis of elastic-plastic response'. Computers and Structures, Volume 12, pp. 607-624 (1980).
- 2. F.L. DiMaggio, I.S. Sandler. 'Material model for granular soils'. Journal of the Engineering Mechanics Division ASCE 97, pp. 935-949 (1971).

# **Example 5.7.1**

## Single joint test for lateral earth pressure

### Keywords

Lateral Earth Pressure, Multi-linear

### **Description**

This example models lateral earth pressures via a single joint. Displacements or forces are applied at one degree of freedom, while other degrees of freedom are fixed. The numerical response can then be compared with hand calculations.

## **Geometry, Boundary and Loading Conditions**

A single joint is defined by 3 or 4 nodes

node	1	2	3	4
Coordinates	2.0, 0.0, 0.0	1.0, 0.0, 0.0	2.0, 1.0, 0.0	2.0, 1.0, -1.0

For 2D joints, the 3rd node is used to define the local x-direction; for 3D joints, the 3rd and 4th nodes are used to define the local x-axis and local xy-plane respectively.

Node 2 is fixed; the prescribed displacement  $u_x$  at node 1 in the x-direction varies as

$t_i$	0.0	1.0	2.0	3.0	10.0	1.50
$f(t_i)$	0.0	0.016	0.1	0.016	-0.16	-0.2

where  $t_i$  and  $f(t_i)$  are controlling variable time and curve factor, respectively.

#### **Discretisation**

The joint is discretised into one 2D or 3D joint element.

### **Material properties**

The properties for active, at rest and passive lateral earth pressures are

$$\delta_a = -0.016$$
,  $f_a = 66.6$ ,  $\delta_0 = 0.0$ ,  $f_0 = 100.0$ ,  $\delta_p = 0.16$ ,  $f_p = 600.0$ 

in the lateral or horizontal direction; properties in all other directions are zero.

0

una 5	urc								
n	$f_1$	$\delta_1$	$f_2$	$\delta_2$	f <sub>3</sub>	δ3	f <sub>4</sub>	δ4	r
1	-600.0	-0.16	-66.6	0.016					0
2	-600.0	-0.16	-100.0	0.0	-66.6	0.016			0

The corresponding properties for multi-linear elastic models with number of segments n = 1, 2 and 3 are

IJS	ΔS	results	•

-600.0

-0.16

-350.0

-0.08

For all tests the obtained relationship between lateral earth pressure and relative joint deformation, as shown in Figure 1, was confirmed by hand calculations.

-100.0

0.0

-66.6

0.016

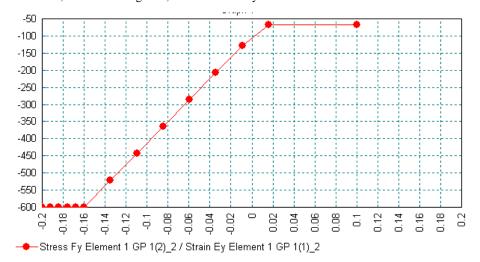


Figure 1

## **Input data**

x05d71a	(JNT3, coupled, $n = 3$ )
x05d71b	(JNT3, uncoupled, $n = 2$ )
x05d71c	(JNT3, coupled (USER), $n = 2$ )
x05d71d	(JNT3, uncoupled (USER), $n = 2$ )
x05d71e	(JNT4, coupled, $n = 2$ )
x05d71f	(JPH3, coupled, $n = 2$ )
x05d71g	(JAX3, coupled n = 2)
x05d71h	(JF3, coupled, $n = 2$ )
x05d71i	(JXS3, coupled, n = 2)

x05d71j	(JL43, coupled, n = 2)
x05d71k	(JSH4, coupled, $n = 2$ )
x05d711	(JL46, coupled, $n = 2$ )
x05d71m	(JSL4, coupled, $n = 2$ )

# **Example 5.7.2**

# Plastic hinge with axial force dependent moment-curvature relation

#### **Keywords**

Multi-linear, Plastic Hinge, Pushover

### **Description**

This example models a plastic hinge with axial force dependent moment-curvature relationship, which is widely used in pushover analysis. Rotations or moments are applied at one degree of freedom, while other degrees of freedom are fixed. The numerical response can then be compared with hand calculations.

### **Geometry, Boundary and Loading Conditions**

A single joint is defined by 3 or 4 nodes

node	1	2	3	4
Coordinates	2.0, 0.0, 0.0	1.0, 0.0, 0.0	2.0, 1.0, 0.0	2.0, 1.0, -1.0

For 2D joints, the 3rd node is used to define the local x-direction; for 3D joints, the 3rd and 4th nodes are used to define the local x-axis and local xy-plane respectively.

Node 2 is fixed; for the JPH3 element the prescribed rotation  $\theta_z$  at node 1 in the z-direction varies as:

step	θz
0	0.
1	-0.1
2	-5.0
3	-5.1
4	-10.
5	-10.1

For the JSH4 element, the same  $\theta z$  is applied at node 1 but in the positive direction.

#### **Discretisation**

The joint is discretised into one 2D or 3D joint element.

### **Material properties**

The following multi-linear moment-curvature relationship is defined in the local direction, which corresponds to the global z-direction

$$(0.0, 0.0), (0.0, 0.1), (10.0, 5.0), (0.0, 5.1), (5.0, 10.0), (0.0, 10.1), r = 0.$$
 at Fax = -10.0  $(0.0, 0.0), (20.0, 0.1), (20.0, 5.0), (10.0, 5.1), (5.0, 10.0), (0.0, 10.1), r = 0.$  at Fax = 10.0

Properties in all other directions are zero.

#### **LUSAS** results

The variation of joint moment with curvature is shown in Figure 1, which is in agreement with the prescribed moment-curvature relation at Fax = 0.

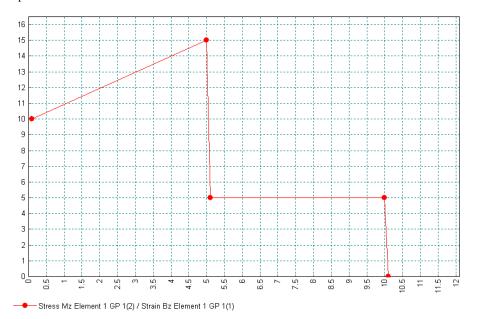


Figure 1

### Input data

**x05d72a** (JPH3) **x05d72b** (JSH4)

# **Example 5.8.1**

# **Definition of soil-structure interface material parameters** from experimental data

### **Keywords**

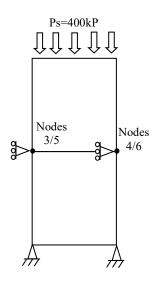
Interface, friction, dilation

## **Description**

This example shows how experimental data from a direct shear test may be input into the interface element material properties and how by including a variation of cohesion with tangential displacement the range of applicability of the interface under different normal stress may be extended. Experimental data is taken from tests by Hu and Pu [1].

### **Geometry, Boundary and Loading Conditions**

Two square plane strain QPN4 elements of unit dimension are connected by the interface element IPN4. A face load of 400kPa is applied to the upper surface and the blocks are then sheared by applying a horizontal displacement to the lower face of the upper block.



## **Material properties**

Block		Interface		
Е	1x10 <sup>6</sup> kPa	Kn	2x10 <sup>6</sup> kPa	
ν 0.0		Ps	400 kPa	
		С	0 kPa	
		t	1.0 m	

The variation of shear force and dilation with tangential displacement along interface under a normal pressure of  $400~\mathrm{kPa}$  is given in the table

Tangential Displacement (mm)	Shear force (kPa)	Dilation (mm)
0.0	0	0
0.1	150	_*_
0.25	225	_*_
0.5	289	0
0.75	330	_*_
1.0	352	0.12
1.25	365	_*_
1.5	370	0.25
2.0	367	0.38
2.5	352	0.51
3.0	337	0.61
3.5	319	0.69
4.0	301	0.76
4.5	285	0.8
5.0	272	0.84
5.5	260	0.85
6.0	254	0.86

6.5	252	0.87
10.0	252	0.87

A unit thickness of t is used so that the displacements input in the shear stress and dilation curves are equal to the physical displacements measured in the shear test.

#### **Discretisation**

2 QPN4 elements with 1 IPN4 interface element at their junction.

### **Experimental Results**

Data from experiments of Hu and Pu showing the shear stress and normal displacement with respect to relative tangential displacement are shown in figure 1.

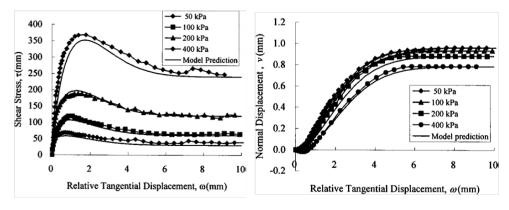


Figure 1. Experimetral results of direct shear test of Hu and Pu [1]

#### **LUSAS** results

Four runs are made with normal stresses of 50, 100, 200 and 400kPa. Results for each run are compared to the experimental data in figure 2. The data for the normal stress of 400kPa exactly reproduces the input curve. Peak shear stresses at the lower values of normal stress are not so well predicted.

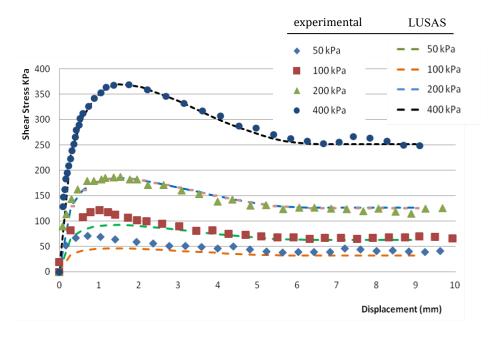


Figure 2. Shear stresses at different normal stresses compared with experimental data

We now modify the original solution data by including cohesion. At lower values of normal stress the effect of cohesion will have a greater effect on the results than at higher stress so we can improve the numerical model by adding a contribution from cohesion. To do this we plot the displacement and experimental results for the 100kPa normal pressure at a larger scale as shown in figure 3. We then attribute the difference between the experimental data and the numerical solution to cohesion which allows us to define a function to represent the variation of cohesion with tangential displacement.

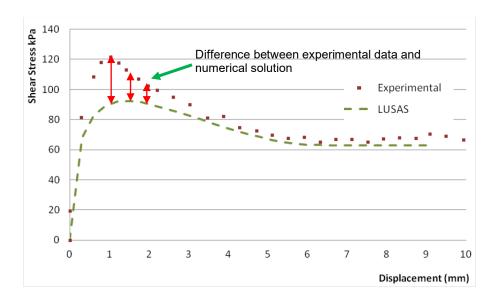


Figure 3. Evaluation of cohesion function from difference in numerical solution and experimental data

The resulting function defining the cohesion is shown in figure 4. It is assumed that the cohesion is constant until the peak stress is reached at which point it declines rapidly as the displacement increases.

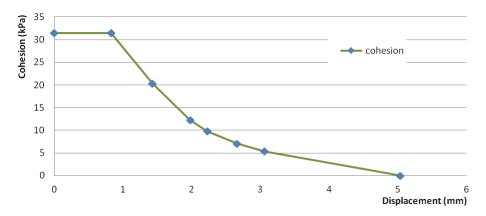


Figure 4. Variation of cohesion with tangential displacement

The numerical data are given in the following table.

Displacement (mm)	Cohesion (kPa)	Displacement (mm)	Cohesion (kPa)
0.0	31.42	2.66	7.08
0.83	31.42	3.06	5.33
1.43	20.25	3.47	4.1
1.98	12.22	5.04	0
2.23	9.82	10.0	0

The shear tests are re-run with the new cohesion data. This time there is much better agreement at the lower normal pressures as shown in figure 5.

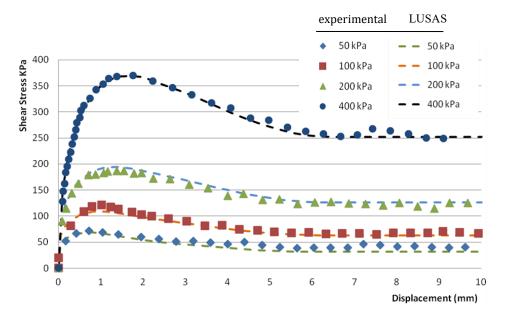


Figure 5. Shear stresses at different normal stresses compared with experimental data using improved material data

The dilation is not substantially affected by the normal pressure so the solution for only one normal pressure is shown below.

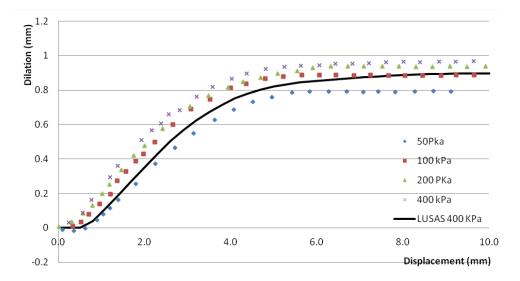


Figure 6. Variation of dilation at different normal stresses

#### References

3. Hu L. and Pu J., 'Testing and modeling of soil-structure interface', J.Geotech.Geoenv.Eng., ASCE, 851-860, (2004)

## **Input data**

X05D81.DAT

# **Example 6.3.1**

## **Materially And Geometrically Nonlinear Encastre Beam**

### **Keywords**

Nonlinear, Plasticity, Updated Lagrangian Description

### **Description**

An encastre beam of length 495mm, width 7.78mm, and depth 7.72mm is subjected to a central transverse concentrated load. The loading is applied in increments of 100N. Figure 1 shows the dimensions of the beam, the boundary conditions and the position of the load.

#### **Discretisation**

Due to its symmetrical nature only a half of the beam is discretised using six curved beam elements (BSX4). The nodal positions and element numbers are shown in figure 2. Symmetrical and fully restrained boundary conditions are imposed on the central and encastre ends of the mesh respectively.

### **Modelling**

The geometric nonlinearity is modelled using an updated Lagrangian description. The material nonlinear behaviour is modelled using a Von-Mises elasto-plastic model with the following material properties:

Young's modulus	197.3x10 <sup>3</sup> N/mm <sup>2</sup>
Poisson's ratio	0.3
Yield stress	248.0 N/mm <sup>2</sup>

### Comparison

The LUSAS results are presented in the table below, and are compared to the experimental data [1] in the following load deflection graph.

Applied Load, N	Central Deflection BSX4
100	1.059
200	2.044
300	2.921
400	3.758
500	4.609

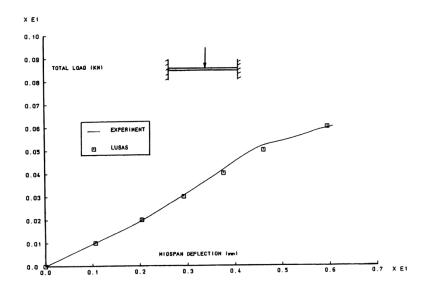
Applied Load, N	Central Deflection BSX4		
600	5.956		

## References

1. Campbell, T.I. Charlton, T.M. 'Finite defirmation of a fully fixed beam comprised of a nonlinear material'. Int.Jou.Mechanical Science, Vol.15, No.5, 415-428.

### **Input data**

X06D31A.DAT



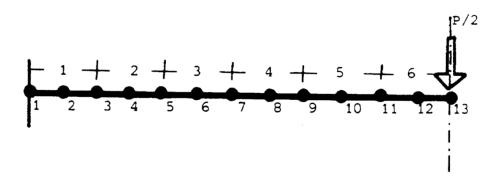
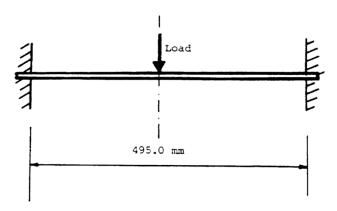


Figure 1.



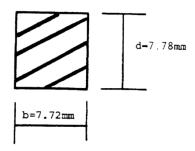


Figure 2.

# **Example 6.5.1**

# **Elasto-Plastic Buckling Analysis Of A An Imperfect Rectangular Panel**

#### **Keywords**

Buckling, Plasticity, Total Lagrangian, Nonlinear, Rectangular Panel

#### **Description**

Determine the buckling behaviour of an imperfect rectangular panel subject to an in-plane compressive loading.

The dimensions of the panel are as follows:

side lengtha = 875 mmside lengthb = 1000 mmthicknesst = 25 mmside length ratioa/b = 0.875aspect ratiob/t = 40

In order to initiate a buckling mode of failure the panel is given an out of plane initial imperfection in the form of half of a sine wave. The semi sine wave imperfection acts over the entire plate surface and has a maximum amplitude of 0.001\*b at the centre of the plate.

$$z = d_0 \sin\left(\frac{\pi x}{a}\right) \sin\left(\frac{\pi y}{b}\right) \tag{0.0.121}$$

where  $d_0$  is the maximum amplitude.

The panel is subjected to an in-plane compressive loading acting along its long edges.

#### **Discretisation**

A symmetric quarter of the panel is discretised using a uniform mesh of sixteen Semiloof shell (QSL8) elements. The nodal coordinates of the mesh are calculated so as to give an initial sine wave imperfection as described by (0.0.121).

The external edges of the panel are assumed to be simply supported and the internal edges are subject to symmetry enforcing boundary conditions.

The geometrically nonlinear behaviour of the panel is recorded using a total Lagrangian approach.

The material is assumed to behave in an elasto-plastic manner according to a Von Mises yield criterion. The assumed material properties of the panel are as follows:

Young's modulus	207.0x10 <sup>3</sup> N/mm <sup>2</sup>
Poisson's ratio	0.3
Yield stress	$247.0\ N/mm^2$

The in-plane compressive loading is applied via prescribed displacements along the long edges of the panel. The results quoted were obtained by using the standard frontal equation solver.

#### **Theory**

The behaviour of the panel is represented in the form of a load versus shortening curve. Previous studies [1-2] have established that the lowest critical buckling strength occurs at a side length ratio of approximately 0.875.

The squash load is calculated from the yield stress as

$$P_0 = \sigma_y A = \sigma_y \left(\frac{b}{2}\right) t = 247 * 500 * 25 = 3087.5 \text{ kN}$$

Similarly the strain at the squash load may be calculated as

$$\varepsilon_0 = \frac{\sigma_y}{E} = \frac{247.0}{207 \times 10^3} = 0.001932$$

## **Solution Comparison**

The results obtained from the nonlinear LUSAS analysis are compared to the linear solution and to the nonlinear solutions of [1].

Increment	Displacement in X direction along line Y=0	Average membrane strain in X direction (x10³)	Strain/squash strain (ε₀)	Load in X direction	Load/squash load (P₀)
1	0.261021	0.5966	0.5	1538.96	0.49891
2	0.417633	0.9548	0.8	2458.54	0.79628
3	0.469837	1.0739	0.9	2763.93	0.89519
4	0.522041	1.1932	1.0	3020.4	0.97826
5	0.574245	1.3124	1.1	3052.56	0.98868
6	0.626449	1.4317	1.2	3039.9	0.98458
7	0.678653	1.5512	1.3	3022.94	0.97908

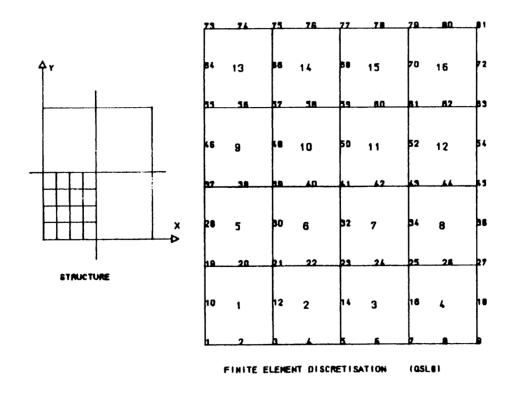
Increment	Displacement in X direction along line Y=0	Average membrane strain in X direction (x10³)	Strain/squash strain (ε₀)	Load in X direction	Load/squash load (P <sub>0</sub> )
8	0.730857	1.6706	1.4	3000.57	0.97184
9	0.783062	1.7898	1.5	2972.4	0.96272
10	0.835266	1.9092	1.6	2938.48	0.95173
11	0.887470	2.0285	1.7	2899.49	0.93911
12	0.939674	2.1411	1.8	2856.62	0.92522

#### References

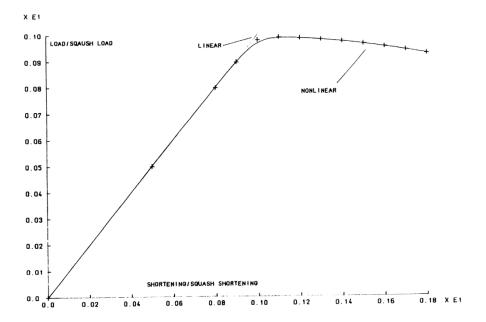
- 1. Crisfield, M.A. 'Full range analyses of steel plates and stiffened plating under uniaxial compression'. Proc.Inst.Civil Engineers, Part 2, Volume 59, 595-624, (1975)
- 2. Irving, D.J. 'Large deformation elasto-plastic finite element analysis of plates, shells and tubular joints using Semiloof shell elements', Ph.D. Thesis, Kingston Polytechnic, London, U.K. (1982).

## **Input data**

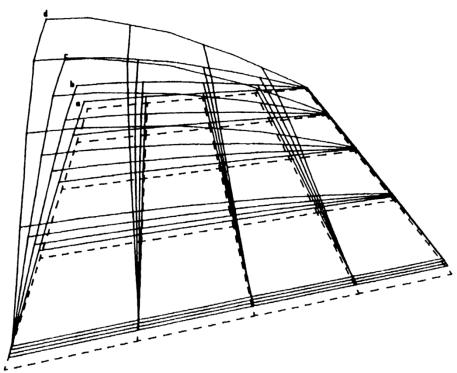
X06D51A.DAT



Imperfect rectangular panel subject to in-plane compressive load



Nonlinear buckling analysis of imperfect rectangular panel (QSL8 elements)



BUCKLED IMPERFECT RECTANGULAR PANEL DEFORMED CONFIGURATIONS

INITIAL CONFIGURATION

- a INCREMENT 3
- 1 INCREMENT 6
- c INCREMENT 9
- d INCREMENT 12

# **Example 6.5.2**

## **Elasto-Plastic Analysis Of A Clamped Spherical Cap**

## **Keywords**

Spherical Shell, Total Lagrangian, Nonlinearity, Elasto-Plastic,

#### **Description**

Determine the load-deflection response of an elasto-plastic spherical shell cap. The geometry of the shell is formed from a truncated sphere of radius 4.758 inches. The cap has a base radius of 0.9 inches, a central height of 0.085895 inches, and a thickness of 0.01576 inches. The shell is fully restrained at its base and is subject to an increasing concentrated transverse load at its apex.

#### **Discretisation**

Due to its doubly symmetric nature only a quarter of the structure is considered in the finite element model. The shell is discretised using 32 Semiloof shell (QSL8 and TSL6) elements. The initial geometry of the shell is formed using the LUSAS arc facility to generate the truncated portion of the sphere. Symmetric boundary conditions (restraint of the loof rotations and the appropriate lateral translations) are imposed on the internal boundaries of the mesh, and the nodes at the base of the structure are restrained against translation and rotation. The nonlinear material behaviour is modelled by a Von-Mises elasto-plastic model with the following material properties:

Young's modulus	10.0x10 <sup>6</sup> psi.
Poisson's ratio	0.3
Yield stress	$50.0x10^3$ psi.

## **Theory**

The maximum (apex) deflection for a fully restrained partial sphere is given by the expression [1]

$$y = \frac{-A_2}{16} \frac{PR^2}{\pi D}$$
 (0.0.122)

Where

$$D = \frac{Et^3}{12(1-v)^2}$$
 (0.0.123)

and in which  $A_2$  is tabulated according to the parameter  $\alpha$ , where

$$\alpha = 2 \frac{\left(3(1-v^2)h^2\right)^{1/4}}{t^2} \tag{0.0.124}$$

For the current sphere geometry the following values apply:

$$\alpha = 6.00$$
 A = 0.161 D = 3.584

Hence for a total load of 50 lbs the maximum deflection is

$$y = -0.036189$$
 inches

#### Comparison

The results obtained from the LUSAS analysis are compared to the linear solution [1] and the axisymmetric nonlinear analysis of Wood [2,3].

Increment Number	Displacement at Apex (ins)	Total Force (lb) (Elastic)	Total Force (lb) (Elastoplastic)
1	0.01	9.50	9.40
2	0.02	14.15	13.17
3	0.03	16.94	15.07
4	0.04	19.38	17.20
5	0.05	21.98	19.70
6	0.06	24.84	22.46
7	0.07	27.90	25.26
8	0.08	31.04	28.00
9	0.09	34.09	30.60
10	0.1	36.86	32.98
11	0.11	39.18	35.18
12	0.12	41.06	37.19
13	0.13	42.86	38.77
14	0.14	45.80	40.50

#### References

 Roark,R.J. Young,W.C. 'Formulas for stress and strain: Fifth edition', McGraw-Hill Publishing Co. (1975)

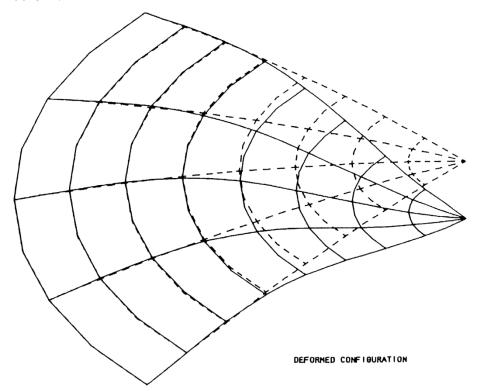
- 2. Oliver, J. Onate, E. A total Lagrangian formulation for the geometrically nonlinear analysis of structures using finite elements: Part 1: Two dimensional problems: Plate and shell structures'. Int. Journ. Num. Mthds. in Eng. Vol.20. 2253-2281, (1981).
- 3. Wood,R.D. 'The application of finite element methods to geometrically nonlinear structural analysis'. PhD.Thesis C/Ph/20/73, University of Wales, Swansea, U.K.

#### **Input Data**

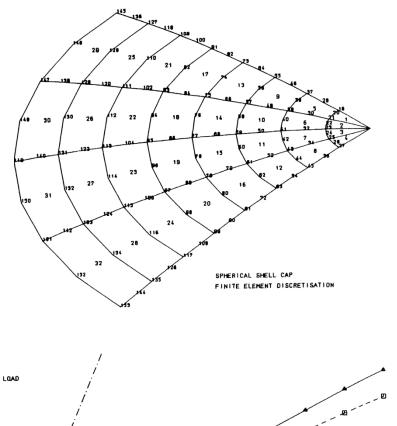
Note: Load incrementation may alternatively be defined using the nonlinear control INCREMENTATION and TERMINATION control cards.

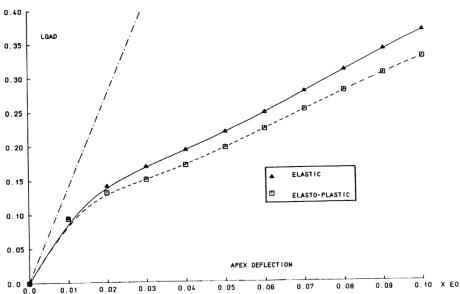
X06D52A.DAT

X06D52B.DAT



X E2





Load-deflection response for spherical shell under apex point load

# **Example 6.5.3**

## **Nonlinear Analysis Of A CHS Welded Tubular Joint**

#### Keywords

Tubular Joint, Elasto-Plastic, Total Lagrangian Nonlinearity, Semiloof Shell, Three Dimensional Solid

#### **Description**

Determine the relationship between the applied axial load and the penetration for the welded tubular joint.

The X-joint is formed by the right angled intersection of two circular section hollow tubes. The intersection is secured by a fillet weld. The vertical tube is subjected to an axial load such that the horizontal tube is progressively penetrated.

#### **Discretisation**

Owing to the symmetrical nature of the problem only one half and one quarter of the vertical and horizontal tubes are modelled respectively.

The hollow tubes are represented using 8 and 6 noded Semiloof shell (QSL8 and TSL6) elements, and the fillet weld by 15 noded solid wedge (PN15) elements.

The vertical tube is axially compressed by means of prescribed displacement of its top edge.

The nonlinear behaviour is modelled using a nonlinear Von-Mises elasto-plastic material model, and a total Lagrangian geometric description.

#### **Modelling Hints**

The solid wedge elements are numerically integrated using three pairs of two Gauss points located on the quadrilateral faces of the element. The Semiloof shell elements are integrated in plane using a 5 point integration rule, and through the thickness by a five point mid-ordinate rule.

The configuration of the loof node rotations [1] allows the Semiloof shell element to model branched or intersecting shell geometries. In this case the intersection of the shell and three dimensional elements is achieved by coupling the global translational degrees of freedom of the two element types. The loof rotations however are unrestrained. Although theoretically this permits a rotation at the tube-weld interface, in practice the small size of the weld and the relative stiffness of the connection prevents the occurrence of any serious discretisation error.

#### **Theory**

The welded tubular structure has been investigated experimentally and numerically by several authors [2-4].

The response of the structure is represented by a graph of applied axial load against the branch penetration (the branch penetration of the vertical tube is defined as the applied axial displacement minus any elastic compressional deformation). Although somewhat inconclusive, the experimental results clearly show a limit point in this response at a load level of approximately 90 kN.

#### Comparison

The solution obtained from the LUSAS analysis (tabulated below) is compared to the experimental results [2].

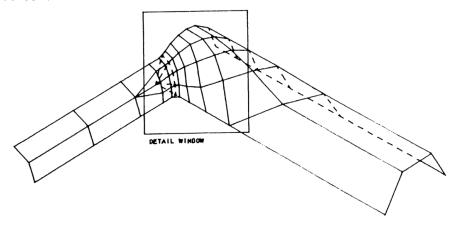
Increment Number	Prescribed Displacement	Branch Penetration	Axial Load (kN)
1	-1	-0.9491	-54.326
2	-2	-1.9086	-91.984
3	-3	-2.8934	-101.263
4	-4	-3.8908	-100.564
5	-5	-4.8972	-97.096
6	-6	-5.8944	-93.02
7	-7	-6.8987	-89.329
8	-8	-7.9019	-85.844
9	-9	-8.9053	-82.676
10	-10	-9.9084	-80.267
11	-11	-10.911	-78.134
12	-12	-11.9131	-76.631

#### References

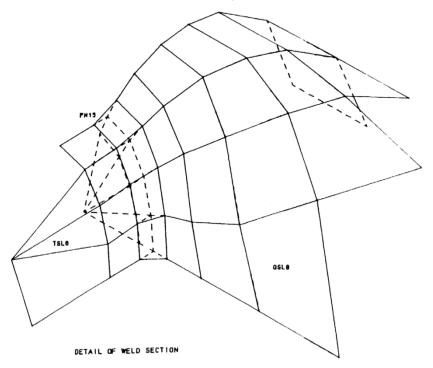
- 1. Irons, B.M. 'The Semiloof shell element', Finite Elements for thin shells and curved members, Eds. Ashwell and Gallagher, Wiley (1976).
- 2. Sparrow, K.D. 'Ultimate strength of welded joints in tubular steel structures', Ph.D. Thesis, Kingston Polytechnic, London (1979).
- 3. Hoadley, P.W. Yura, J.A. 'Ultimate strength of tubular joints subjected to combined loads', Department of Civil Engineering, University of Texas at Austin, U.S.A. (1982).
- Holsgrove, S.C. Lyons, L.P.R. 'Nonlinear FEM analysis of CHS T/X joints under axial load, in plane bending, and out of plane bending', Safety Criterion in the Design of Tubular Structures, (1987).

## Input data

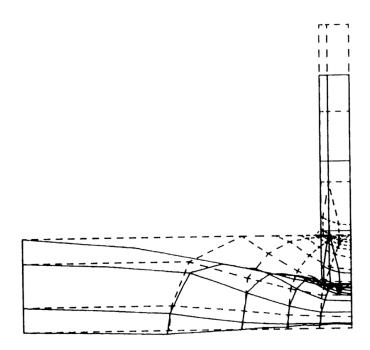
X06D53A.DAT



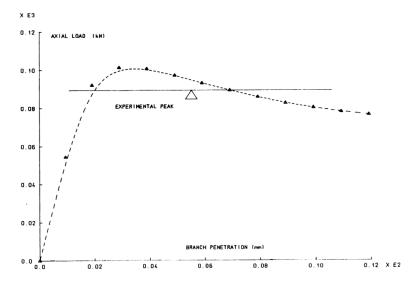
Circular hollow section welded tubular joint finite element discretisation



**Detail of weld section** 



Deformed configuration (increment 3) magnification 20



Load-penetration response for welded circular tube joint

# **Example 6.5.4**

## **Large Deflection Of Orthotropic Spherical Cap**

#### **Keywords**

Total Lagrangian, Large Displacement, Hill Yield Criterion

#### **Description**

A material which has yield stresses in the principle directions of 200, 200 and 100 is applied with the weak direction of the material aligning with the hoop, meridional and through thickness directions in turn. The orientation of the yield surface produces different responses as the cap snaps through under a point load applied at its vertex.

#### **Finite Element Model**

The cap is restrained to deform axisymmetrically and is modelled by a mesh of 8\*1 axisymmetric shell elements.

#### **Geometry**

Radius of sphere = 4.758

Angle subtended = 21.8 degrees

Shell thickness = 0.01576



## **Material properties**

Young's modulus in X direction =  $1x10^5$ 

```
Young's modulus in Y direction = 1x10^5
Poisson's ratio = 0.3
Case 1
 Yield stress in meridional direction
                                         = 200
 Yield stress in hoop direction
                                       = 100
 Yield stress in through-thickness direction = 200
 Shear yield stress (not used) = 1
Case 2
 Yield stress in meridional direction
                                         = 100
 Yield stress in hoop direction
                                       = 200
 Yield stress in through-thickness direction = 200
 Shear yield stress (not used) = 1
Case 3
 Yield stress in meridional direction
                                         = 200
 Yield stress in hoop direction
                                       = 200
 Yield stress in through-thickness direction = 100
 Shear yield stress (not used) = 1
```

## **Boundary Conditions**

Symmetry conditions are applied to the apex. The shell is hinged but fixed against translational motion.

#### **Solution Comparison**

Figure 1 illustrates the loading paths for the various material orientations. After buckling, three distinctly different responses occur. The apex displacement is detailed in figures 2 to 4, and figures 5 to 7 plot the development of the snap through for each case.

The material is considered to be in a state of plane stress. However, the through thickness yield stress is significant and does effect the results as the out-of-plane plastic strain is related to the inplane plastic strains, by the constraint that there must be no change of volume following inelastic straining.

It is important when considering orthotropic yield that all three yield stresses in the principle directions are known. On the other hand, the shear stresses are de-coupled, both from themselves and the direct stresses so only the relevant values need be input.

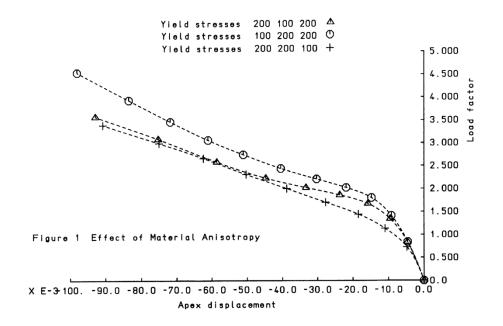
## **Input Data**

X06D54A.DAT

X06D54B.DAT

X06D54C.DAT

The data files for cases 2 and 3 are obtained by changing the yield values on the STRESS POTENTIAL data command.



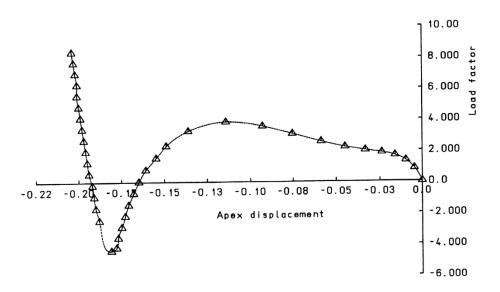


Figure 2 Yield stresses in Principle Directions 200 100 200

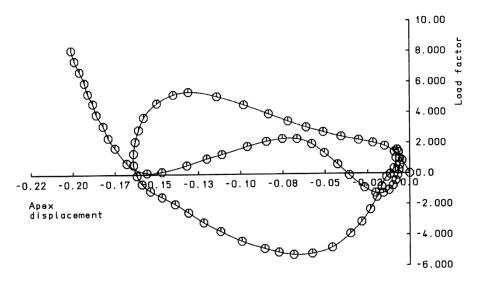


Figure 3 Yield stresses in Principle Directions 100 200 300

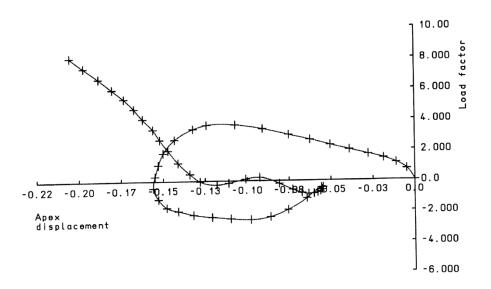


Figure 4 Yield stresses in Principle Directions 200 200 100

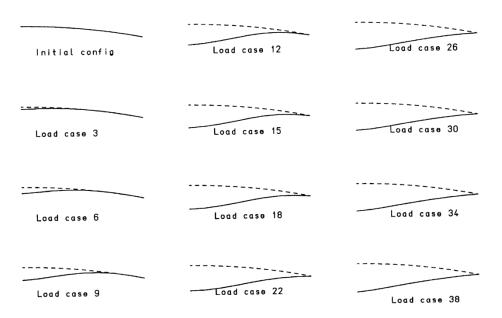


Figure 5 Deformation for yield stresses 200 100 200

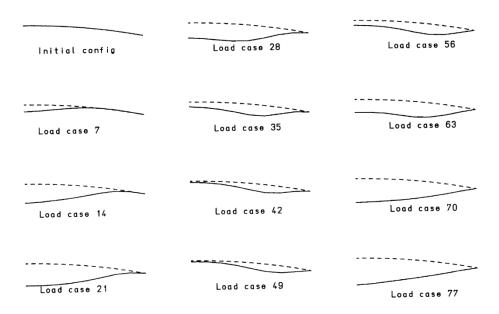


Figure 6 Deformation for yield stresses 100 200 200

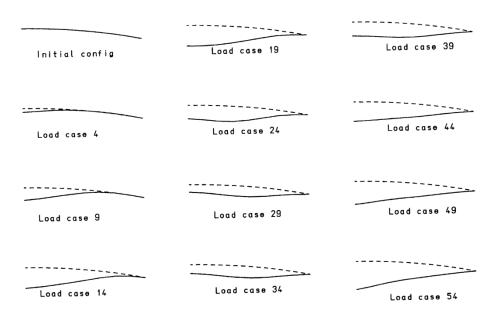


Figure 7 Deformation for yield stresses 200 200 100

# Example 7.1.1

## **Transient Analysis With Radiation And Convection**

## **Keywords**

Radiation, Convection, Transient Analysis

### **Description**

This problem involves an infinite slab subjected to combined radiation and convection and constitutes a two dimensional nonlinear transient analysis. The analysis involves computing temperature at the centre (Tc) and surface (Ts) of the slab as a function of time (t). This problem was chosen to demonstrate the iterative procedures available in LUSAS for solving problems involving nonlinear radiation and convection boundary conditions.

#### **Discretisation**

The analysis is performed using 20 QFD8 elements to model half the slab, figure 3.

#### **Geometry**

Infinite slab

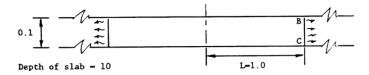


Figure 1

### **Material properties**

Thermal conductivity K=0.01 Btu/ F.hr.in.

Specific heat c=0.01 Btu/ F.in<sup>3</sup>

Convective heat transfer coefficient h<sub>c</sub> =0.04

Emissivity  $\in =1.0$ 

Stefan-Boltzmann's constant σ=0.118958E-10 Btu/in<sup>2</sup> .hr.R<sup>4</sup>

#### **Boundary conditions**

Initial temperature field Ti = 1498.1505 F

Radiation to environment at T = 0 F at points B and C

Convection to environment at T = 0 F at points B and C

#### **Lusas results**

	t=0.04 hr	t=0.66 hr	t=3.66 hr
$T_{s}$	681.448	191.611	1.899
$T_{c}$	1497.62	636.820	6.300

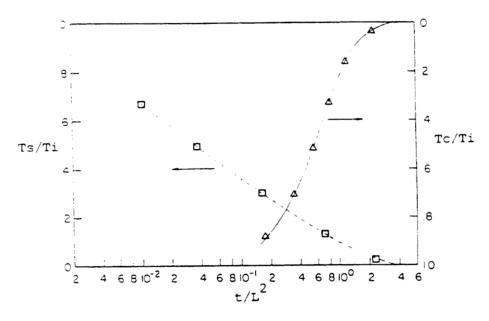


Figure 2

#### References

- BATHE,K.J. KHOSHGOFTAAR,M.R. Finite Element Formulation and Solution of Nonlinear Heat Transfer J. Nuclear Engineering and Design, Vol.51, 389-901
- 2. Haji-Sheikh, A. and Sparrow, E.M., The Solution of Heat Conduction Problems by Probability Methods, Trans. ASME, J. Heat Transfer, Vol. 39, pp. 121-131, 1967

## Input data

X07D11A.DAT

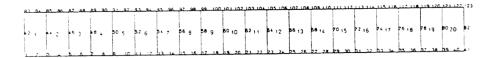


Figure 3

# **Example 7.2.1**

## **Transient Field Analysis Of Heat Conduction Problem**

## **Keywords**

Transient Field Analysis, Heat, Conduction

### **Description**

Determine the temperature distribution with time along an insulated bar, initially at zero degrees C, due to a unit temperature rise at one end (see Figure 1).

#### **Discretisation**

Model the bar using eight QFD4 plane field elements (see Figure 1).

#### Geometry

The geometry of the bar is shown in Figure 1.

## **Material properties**

Coefficient of thermal conductivity = 1.0 J/sec.m. C

Coefficient of specific heat  $= 1.0 \text{ J/m}^3 \cdot \text{C}$ 

This value is computed from the product of the specific heat and the density which are

Specific heat = 1.0 J/kg. CDensity =  $1.0 \text{ kg/m}^3$ 

Coefficient of convection heat transfer =  $0.0 \text{ J/m}^2$  .sec. C

## **Boundary conditions**

The left-hand end of the bar, nodes 1 and 10, are restrained to remain at unit temperature. The top, bottom and right-hand surfaces are insulated.

#### **Theory**

The exact solution for a uniform initial temperature of zero is given in [1] as

$$\phi = 1 - \sum_{k=1}^{\infty} \frac{4}{(2k-1)\pi} \exp\left\{ -\left[\frac{(2k-1)\pi}{8}\right]^2 t \right\} \sin\left[\frac{(2k-1)\pi x}{8}\right]$$
 (0.0.125)

The exponential factor decreases rapidly so that for practical purposes only the first few terms of the expansion need be considered.

#### **Comparison**

A comparison of the LUSAS results with the theoretical solution is presented in graphical form in Figure 2.

#### References

1. W. L. Wood and R. W. Lewis, "A Comparison of Time Marching Schemes for the Transient Heat Conduction Equations", Int. Jul for Num. Methods in Eng., Vol 9, 1975, p679

#### **Input Data**

X07D21A.BAT

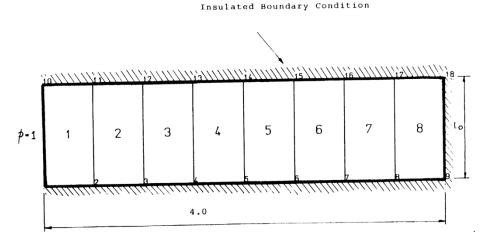


Figure 1 Finite element mesh for heat conduction problem.

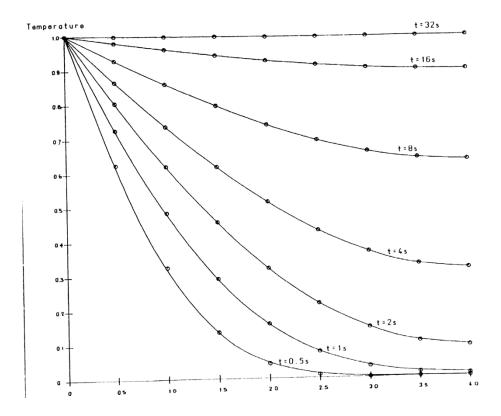


Figure 2 Temperature distribution in bar at times t=0.5s to 32s

# **Example 7.2.2**

## 2-D Solidification Of A Corner Region

## **Keywords**

Transient, Nonlinear, Phase Change, Enthalpy

### **Description**

This problem involves a transient nonlinear analysis with phase changes. A corner region, which is initially in a temperature field, has prescribed temperatures imposed along two of its boundaries. The temperatures at selected points in the region are computed as a function of time. This problem demonstrates facilities available in LUSAS for solving problems involving phase changes with the use of enthalpy averaging methods. The problem is solved for three different phase change intervals using a time step of  $\Delta t$ =0.05 sec.

#### **Discretisation**

The analysis is performed using 81 QFD4 elements. The element mesh is gradually refined towards the two sides of the corner where boundary temperatures are prescribed, Figure 2.

## **Geometry**

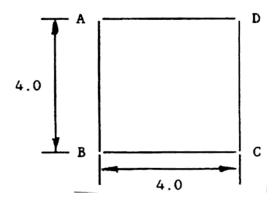


Figure 1

## **Material properties**

K = 1.08;  $\rho C = 1.0$ ;  $\rho L = 70.26$ 

Phase change temperature = -0.15

where:

K = Thermal conductivity

 $\rho = Density$ 

C = Specific heat

L = Latent heat

The analytical solution is evaluated for a pure substance i.e. it is assumed that the phase change occurs instantaneously at T = -0.15. With the enthalpy formulation a phase change zone must be defined. LUSAS analyses were carried out using three different phase change zones:

#### 1. Phase Change Zone = 0.5 C

Temperature	Conductivity	Specific Heat	Change of Total Enthalpy
-45.0	1.08	1.0	0.0
-0.40	1.08	1.0	44.6
0.10	1.08	1.0	115.36
10.1	1.08	1.0	125.36

#### 2. Phase Change Zone = 1.0 C

Temperature	Conductivity	Specific Heat	Change of Total Enthalpy
-45.0	1.08	1.0	0.0
-0.65	1.08	1.0	44.35
0.35	1.08	1.0	115.61
10.35	1.08	1.0	125.61

#### 3. Phase Change Zone = 1.5 C

Temperature	Conductivity	Specific Heat	Change of Total Enthalpy
-45.0	1.08	1.0	0.0
-0.90	1.08	1.0	44.10
0.60	1.08	1.0	115.86
10.60	1.08	1.0	125.86

### **Boundary conditions**

Initial Temperature field T = 0RF

Prescribed temperature T = -45RF along AB and BC

AD and CD insulated

#### **Theory**

The transformation of a material between phases is accompanied by either liberation or absorption of latent heat in the phase transition zone, i.e. the enthalpy gradient is discontinuous. The most suitable methods for incorporation in standard finite element codes are the enthalpy or specific heat formulations (see [1]).

#### Reference

1. Morgan, K., Lewis, R.W., Zienkiewicz, O.C., An Improved Algorithm for Heat Conduction Problems with Phase Change, Int. J. Num. Meth. Eng., Vol.12, 1191-1195, 1978

#### **LUSAS** results

Comparison between LUSAS results and those in [1] are shown for the time variation of the point x=1m, y=1m (node 45).

#### **Input data**

X07D22A.DAT

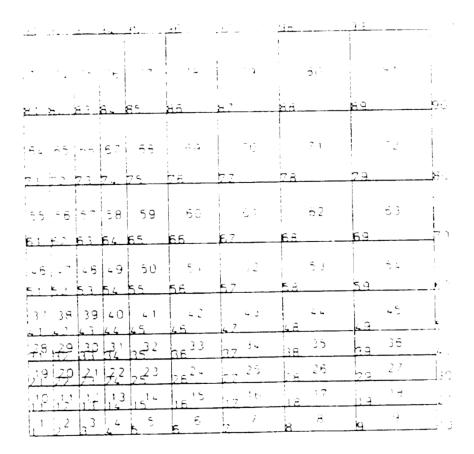


Figure 2

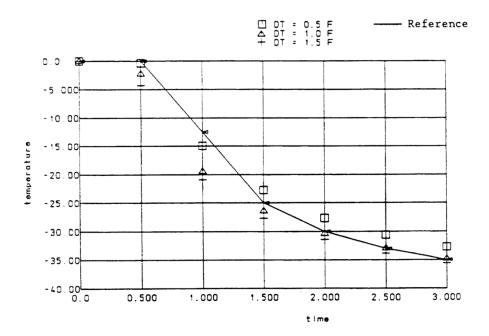


Figure 3 - Temperature at coordinates (1,1) Vs Time for phase change zones 0.5, 1.0 and 1.5

# **Example 7.2.3**

## **Heat Conducting Plate With Sudden Cooling**

## **Keywords**

Transient, Heat, Conduction

## **Description**

This example shows the LUSAS solution for heat transfer between two contacting surfaces of a structure. The problem is a nonlinear transient analysis with temperature dependant convective heat transfer coefficients. The problem is solved for  $\Delta t$ =0.35 hr using 20 steps.

#### **Discretisation**

Two analyses are performed using 10 QFD8 and 20 PF6 field elements respectively, figures 2 and 3.

#### **Geometry**

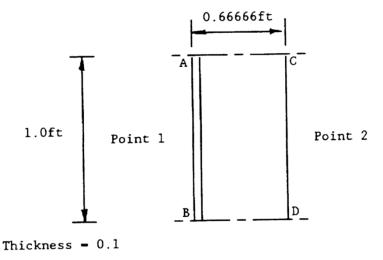


Figure 1

Note that the QFD8 element mesh forms an x-y plane between points 1 and 2, figure 2.

## **Material properties**

Density, r	800 lb/ft <sup>3</sup>
Specific heat, C	0.83333 BTU/lb F
Thermal conductivity, K	2.0 BTU/hr.ft F
Convective heat transfer coefficient, h	$0.04(T_{AB}\ +100)/22.0\ BTU/hr.ft}^{2}\ F$

#### **Boundary conditions**

- (i) Initial temperature along the surface AB is T=500 F and along CD is T=100 F.
- (ii) Step change in environmental temperature on AB to T=100 F.

#### Reference

1. Kreith, F., Principles of heat transfer, International Textbook Co., Scranton, Pennsylvania, USA, 2nd Printing, 1959.

#### **Lusas results**

	Point 1	Point 2
Reference	285.0	100.00
QFD8	287.679	100.00
*PF6	286.243	100.00

<sup>\*</sup> results for convection through triangular faces

LUSAS results for change in temperature at point 1 with respect to time are plotted in figure 4.

## **Input data**

X07D23A.DAT

LUSAS results for temperature at point 1 Vs time.

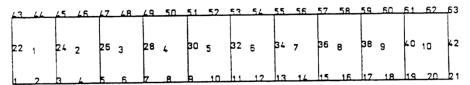


Figure 2

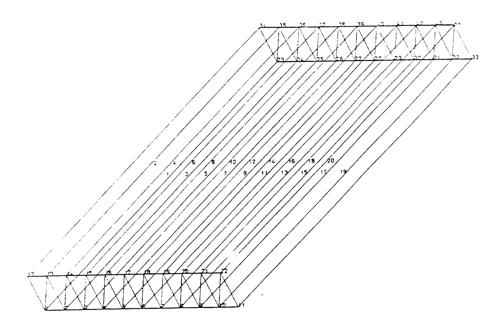


Figure 3

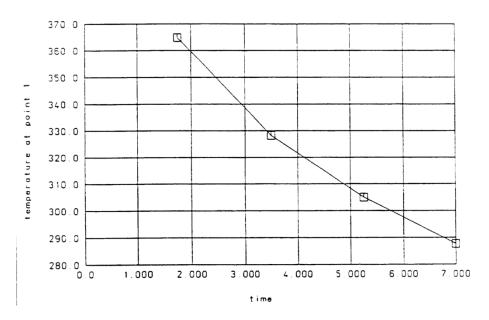


Figure 4 - LUSAS results for temperature at point 1 Vs time.

# **Example 8.1.1**

**Spectral Response Analysis Of A 2-D Frame Structure** 

This example has now been retired.

# **Example 8.1.2**

# **Linear Dynamic Analysis Of A Spring/Mass/ Damping System**

#### **Keywords**

**Linear Dynamics, Damping** 

#### **Problem Description**

Determine the damped natural period (Td) and the ratio (R) between any two succesive amplitudes of a freely vibrating spring/mass/damping system. The mass is initially displaced and subsequently allowed to vibrate freely.

#### **Discretisation**

The spring/mass/damping system is modelled using a single joint (JNT3) element. The material properties are as follows:

$$K_1 = 30 \text{ lb/in}$$
  $M_1 = 0.0259 \text{ lb sec}^2/\text{in}$   $C_1 = 0.12 \text{ lb sec}^2/\text{in}$ 

The system is fully restrained at node 1 and restrained in the Y direction at node 2, resulting in a single degree of freedom problem. An initial load of 30 lb is applied to the structure.

## **Theory**

The undamped natural frequency of the system in radians per second is

$$w_n = \sqrt{\frac{K}{M}} \tag{0.0.126}$$

so that 
$$w_n = \sqrt{\frac{30}{0.0259}} = 34.033 \text{ rads / sec}$$
 (0.0.127)

The critical damping coefficient and damping factor are

$$c_c = 2 \text{ M w}_n = 1.7629 \text{ lb/in/sec}$$
 (0.0.128)

$$T = \frac{c}{c_0} = \frac{0.12}{1.7629} = 0.06807 \tag{0.0.129}$$

The logarithmic decrement is

$$d = \frac{2T}{\sqrt{1 - T^2}} = 0.4286 \tag{0.0.130}$$

The amplitude ratio for any two consecutive cycles is

$$\frac{x_1}{x_2} = e^d = 1.535$$
(0.0.131)

The damped period is

$$T_{\rm d} = \frac{\rm d}{Tw_{\rm n}} = 0.185 \,{\rm secs}$$
 (0.0.132)

## **Comparison**

The LUSAS results are compared to the theoretical solution.

Peak	1	2	3	4
Maximum amplitude (in)	1.000	0.650	0.423	0.275
Time (secs)	0.000	0.186	0.372	0.558

Ratio	R(1-2)	R(2-3)	R(3-4)
Theory	1.535	1.535	1.535
LUSAS	1.538	1.538	1.538
Difference (percent)	0.195	0.195	0.195

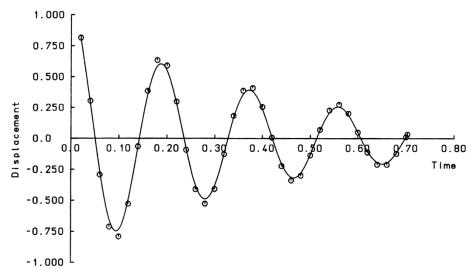
Damped natural Period	Td(1-2)	Td(2-3)	Td(3-4)
Theory	0.185	0.185	0.185
LUSAS	0.186	0.186	0.186
Difference (percent)	0.54	0.54	0.54

#### References

1. Thomson, W.T. 'Theory of vibration with applications', 2nd Edition, George Allen of Unwin.

#### **Input data**

X08D12A.DAT



Time v Displacement for damped spring mass system

# **Example 8.1.3**

# **Beam Subject To A Harmonic, Periodic And Step Load**

## **Keyword**

Linear Elastic, Periodic, Harmonic

## **Description**

The response of a deep, simply supported beam to a harmonic, periodic and direct uniformly distributed load is presented in this example. The objectives are to establish the displacement and stress amplitudes at the midpoint of the beam. An initial eigenvalue analysis of the system is also performed.

### **Discretisation**

The beam is modelled using 10 equal, two noded beam elements (BMS3). Boundary conditions are applied in the global X,Y,Z and jx directions at one of the beam extremities. The other extremity has Y and Z supports only.

## **Geometry**

Consider a beam (see figure 1) of length 10m, square cross-section (b) of 2m and aligned with the global X-axis, for which the following sectional values are used:

Iyy = Izz = 
$$(b^{**}4)/12$$
  
Kt =  $0.1406*(b^{**}4)$   
Asy = Asz = gA

where:

g = (10+10v)/(12+11 v) (v is Poisson's ratio)

## **Material properties**

Young's modulus	200x10 <sup>9</sup> N/m <sup>2</sup>
Poisson's ratio	0.3
Density	$8000~Kg/m^3$

## Loading

The forcing function for the steady state harmonic loading is:

$$F = 1E6 \sin(Wt) Nm^{-1}$$

where:

$$\Omega = 2\pi f \qquad (0 < f < 70 \text{ Hz})$$

Viscous damping of 0.02 is used with no structural damping. The first 16 modes are used to form the modal domain. This test is repeated with distributed viscous damping of approximately 0.02 for mode 8 and viscous damping of 0.02 for all other modes and no structural damping.

2. The forcing function for the steady state periodic loading is:

$$F = 1E6 \left( \sin(\Omega t) - \sin(3\Omega t) \right) Nm^{-1}$$

where:

$$\Omega = 2\pi f$$

Rayleigh damping is used with:  $\alpha = 5.36$ , and  $\beta = 7.46E-5$ 

3. The forcing function for the direct loading is:

$$F = 1E6 \text{ Nm}^{-1}$$

Rayleigh damping is used with: a = 5.36, and b = 7.46E-5

### **LUSAS** results

#### **Eigenvalue Analysis**

Figure 2 shows vibration modes 1, 6 and 9, whilst the table below gives the values obtained using Lusas:

	Frequency			
Mode	Lusas	Reference*		
1	42.5363	42.65		
2	42.5363	42.65		
3	71.2751	71.20		
4	125.129	125.0		
5	144.220	148.15		
6	144.220	148.15		
7	215.587	213.61		
8	260.624	283.47		
9	260.624	283.47		

<sup>\* :</sup> A closed form solution

#### **Harmonic Loading Response**

Figure 3 shows the response of the beam in the frequency domain, whilst the table below compares the values obtained at the beam midpoint for both the peak displacement and the peak stress.

	Lusas	Target*	Reference#
Frequency (Hz)	42.54	42.60	42.65
Peak displacement at beam midpoint	12.48	13.44	13.45
Peak stress at beam midpoint (Pa)	243.8E8	240.3E6	241.9E6

<sup>#:</sup> Closed form analytical solution

#### **Periodic Loading Response**

Figure 4 shows the response of the beam in the time domain as the transient components die out, whilst the table below compares the values obtained at the beam midpoint for both the peak displacement and the peak stress.

	Lusas	Target*	Reference#
Peak displacement (mm)	0.944	0.953	0.951
Peak stress (Pa)	17.19E6	17.33E6	17.1E6

<sup>\*:</sup> Value for engineer's beam (ignoring shear and rotary inertia)

# : Closed form analytical solution

#### **Direct Loading response**

Figure 4 shows the response of the beam in the time domain, whilst the table below compares the values obtained at the beam midpoint for both the peak displacement and the peak stress. Note that the peak occurs at t=0.0117 for both the Lusas and reference solutions.

	Lusas	Target*	Reference#
Peak displacement (mm)	1.044	1.057	1.043
Peak stress (Pa)	18.64E6	18.77E6	18.76E6
Time (s)	0.0117	0.0117	0.0117

# : Closed form analytical solution

- \* : Average of two codes:-
  - (i) Value for engineer's beam (ignoring shear and rotary inertia)
  - (ii) Timoshenko beam (with shear and rotary inertia)

### Reference

1. NAFEMS Report Number E1261/R002

## **Input Data**

X08D13A.DAT

X08D13B.DAT

X08D13C.DAT

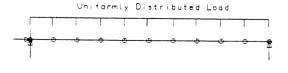
X08D13D.DAT

X08D13E.DAT

<sup>\*:</sup> Value for engineer's beam (ignoring shear and rotary inertia)

STEADY STATE HARMONIC

F = Fo sin wt
Fo = 1E6 N/m
w = 0 to 60 Hz
Damping 2% (every mode)



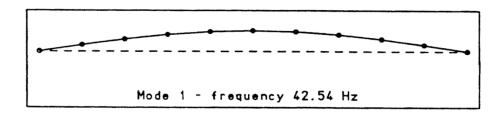
DISCRETISATION USING 10 BMS3 ELEMENTS

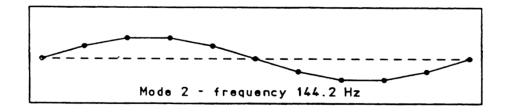
PERIODIC LOADING

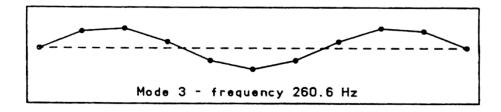
F = Fo(sin wt - sin 3wt)
Fa = 1E6 N/m
wt = 20 Hz
Damping
Ar = 5 36 Br = 7.46E-5

TRANSIENT STEP LOADING
Fa = 1E6 N/m
Damping
Ar = 5.36 Br = 7.46E-5

Figure 1

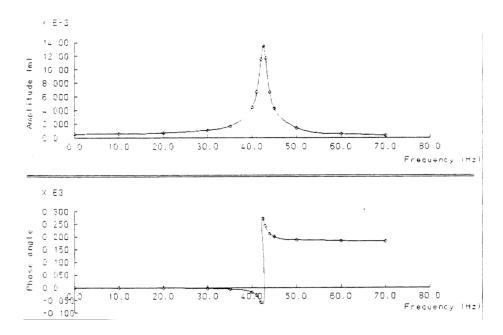






FLEXURAL MODE SHAPES AND FREQUENCIES

Figure 2



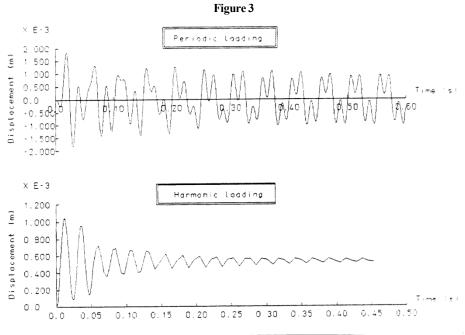


Figure 4

# **Example 8.1.4**

# **Large Deformation Dynamics of Stiff and Elastic Pendulum**

## **Keyword**

Large Deformation, Step-By-Step Dynamics

## **Description**

The rotation of a pendulum is simulated by the generalized  $\alpha$ -method together with a corotational formulation for large deformation. The pendulum can be stiff, undergoing approximate rigid-body rotation, or flexible, undergoing coupled rotation and axial vibration with medium strains.

#### Stiff pendulum

The geometry and material properties, boundary and initial conditions of the pendulum presented in Kuhl and Crisfield [2] and adopted by Bathe [1] are given in Figure 1. The pendulum is powered by the initial tangential velocity  $\dot{u}_0 = 7.72 \text{m/s}$ . In order to obtain an equilibrium state at the initial time  $t_0 = 0$  the initial radial acceleration is calculated to  $\ddot{u}_0 = 19.6 \text{m/s}^2$ .

#### Elastic pendulum

Following Kuhl and Crisfield [2] and Bathe [1], the stiffness of the truss element is reduced from  $EA_0 = 10^{10}$  N to  $EA_0 = 10^4$  N, hence the element is now flexible and shows an axial vibration. Also, we now use the initial radial acceleration  $\ddot{u}_0 = 0.0$  m/s<sup>2</sup>.

#### **Discretisation**

The pendulum is modelled by one two-node bar or truss BAR2 or BRS2 element with a consistent mass matrix. The time step  $\Delta t = 0.01$  s.

## **Theory**

#### Stiff pendulum

As a result of the large stiffness  $EA_0$  the motion of the truss is (approximately) a rigid-body rotation with the angular velocity  $\dot{\varphi} = 0.4036$  1/s (Figure 1). The analytical rotational period is 2.4777 s [1].

#### Elastic pendulum

The period of axial vibration is 0.28 s [1].

## Comparison

#### Stiff pendulum

Kuhl and Crisfield [2] already report that the trapezoidal rule is unstable; the generalized  $\alpha$ -method can be used to solve the problem with integration constant  $\alpha = -0.4/1.6$ . The obtained time history of displacement, velocity and acceleration shown in Figure 2 agree with the theoretical solution, and are very close to the results given by Bathe [1].

#### Elastic pendulum

The time step employed corresponds to 28 steps per cycle of the axial vibration, which is a reasonable time step size to capture this vibration. The obtained time history of displacement, velocity, acceleration and strain from BAR2 and BRS2 are the same (Figure 3), and close to the results given by Bathe [1]. The pendulum undergoes medium strains, which reach its maximum 2.56% at t = 0.14s.

#### References

- Bathe, K. J. 'Conserving energy and momentum in nonlinear dynamics: A simple implicit time integration scheme'. Comput. Struct. 85: 437–445 (2007).
- Kuhl, D., Crisfield, M. A. 'Energy-conserving and Decaying Algorithms in Non-linear Structural Dynamics'. Int. J. Numer. Meth. Engng. 45: 569-599 (1999).

## Input data

x08d14a

x08d14b

x08d14c

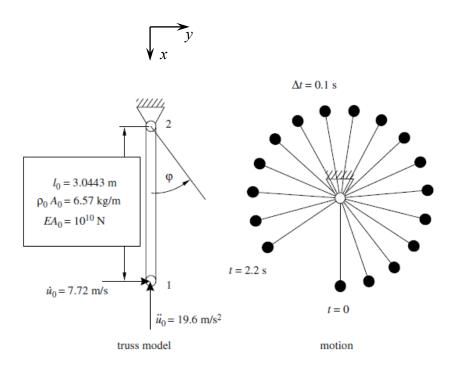
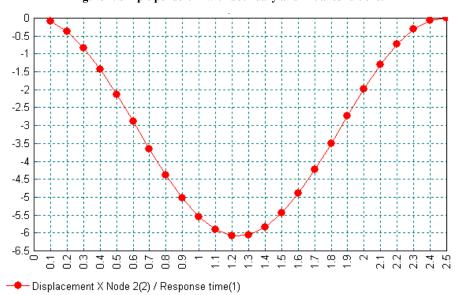
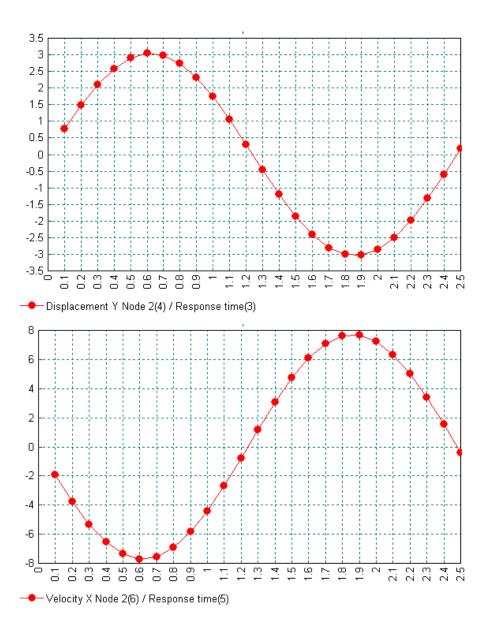
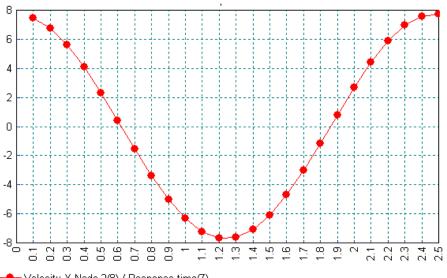
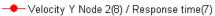


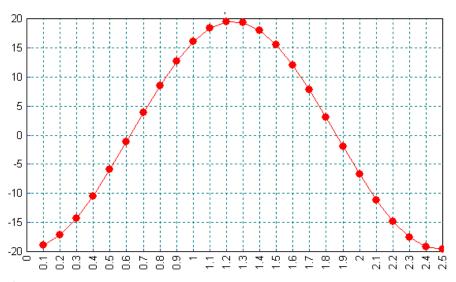
Figure 1. Simple pendulum with boundary and initial conditions.











--- Acceleration X Node 2(10) / Response time(9)

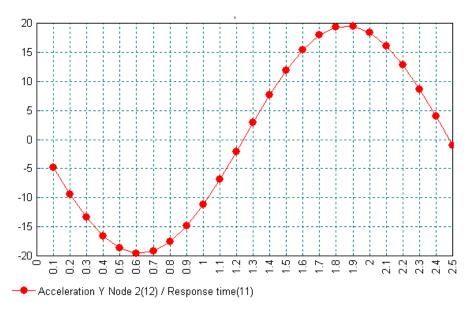
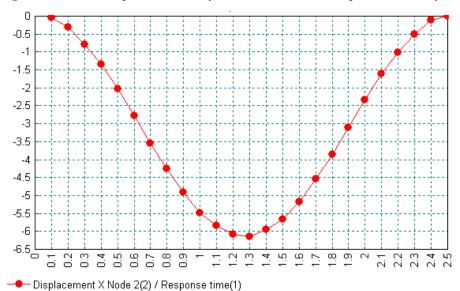
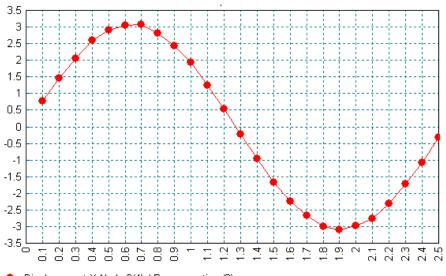
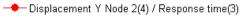
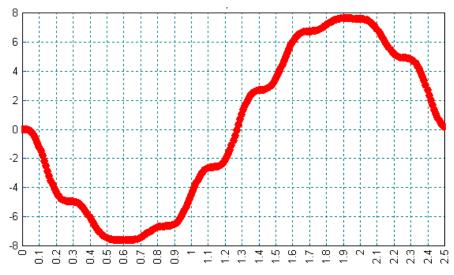


Figure 2. Variation of displacement, velocity and acceleration of the stiff pendulum in 1st cycle.

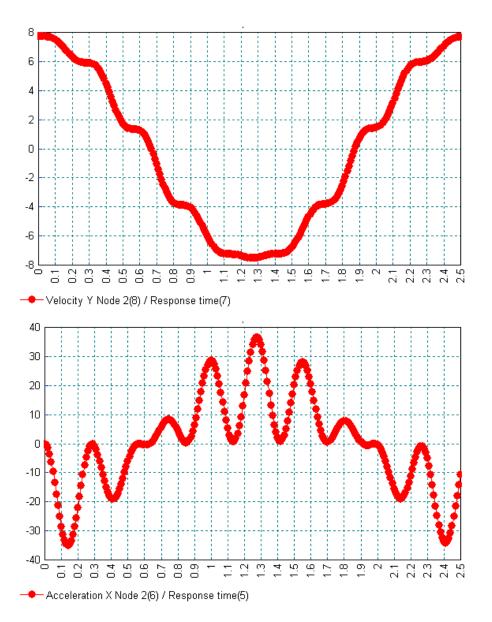








→ Velocity X Node 2(10) / Response time(9)



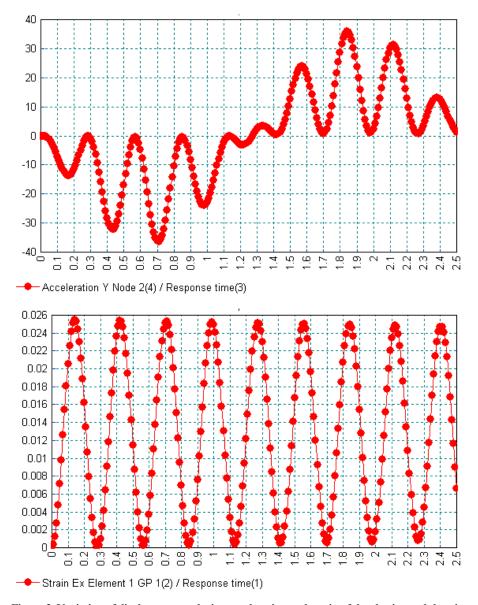


Figure 3. Variation of displacement, velocity, acceleration and strain of the elastic pendulum in 1st cycle.

# **Example 8.1.5**

# Free-Free Nearly Rigid Beam Undergoing Large Overall Motion

## **Keywords**

Large Deformation, Step-By-Step Dynamics

## **Description**

The free flight of a beam, which is affected by very small strains so that the motion is similar to a rigid body motion, is simulated by the generalized  $\alpha$ -method together with a co-rotational formulation for large deformation. The geometry, initial conditions and material properties, given in Figure 1, are the same as Crisfield et al. [1]. The principle axes of inertia of the cross-section are initially aligned with global coordinate axes X and Z.

### **Discretisation**

The beam is divided into 4 linear BMI21 or quadratic BMI31 elements with a consistent mass matrix. The initial time step size is  $\Delta t = 0.2$  s.

## Comparison

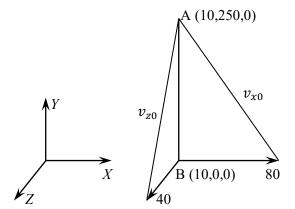
The integration constant  $\alpha = 0$  is adopted by the generalized  $\alpha$ -method. The displacements of the lower end B in the first revolution are given in Figure 2. BMI21 and BMI31 elements give results identical to the BMS3R element of Crisfield et al. [1].

#### References

1. Crisfield, M. A., Galvanetto, U. and Jelenić, G., 'Dynamics of 3-D co-rotational beams'. *Comput Mech* **20**: 507-519 (1997).

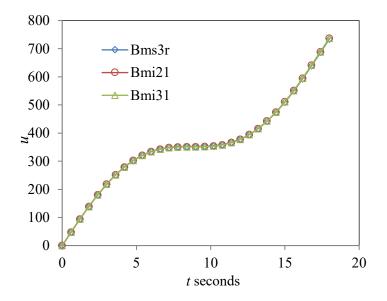
## Input data

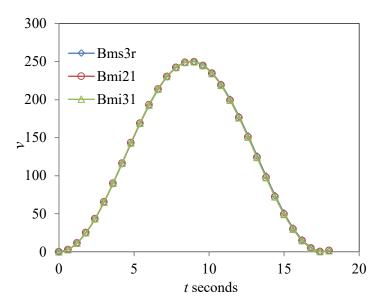
x08d15a x08d15b



Material properties:  $E = 10^{10}$ , v = 0,  $\rho = 1$ Geometric properties: A = 1,  $I_{yy} = I_{zz} = 1/12$ 

Figure 1. Free flight beam.





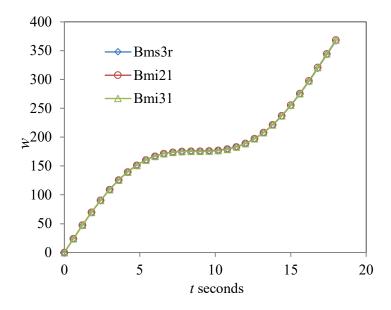


Figure 2. Time history of the displacements at the lower end B.

# **Example 8.2.1**

# **Linear Dynamic Analysis Of A Beam With Pressure Loading**

## **Keywords**

**Step-By-Step Dynamics** 

## **Description**

Determine the dynamic response of a cantilevered beam subject to a suddenly applied uniformly distributed load (see Figure 1).

#### **Discretisation**

Model the two-dimensional cantilevered structure using 8 QPM8 plane membrane elements (see Figure 1).

## Geometry

The cantilevered structure has unit depth and breadth and is ten inches in length (see Figure 1).

## **Material properties**

Young's modulus	12000 lb/in <sup>2</sup>
Poisson's ratio	0.3
Density	$10x10^{\text{-}6}\ lb.sec^2\ /in^2$

## **Boundary conditions**

The structure is fully restrained at nodes 1,18 and 35 forming a cantilevered condition. The pressure load is applied according to the load/time history graph in Figure 2. To allow for a zero initial displacement condition the structure is fully restrained at every node for the first load case and then freed accordingly for subsequent loads.

## **LUSAS Results**

From the results of an eigenvalue extraction of this structure the following parameters were determined:

Fundamental eigenvalue =  $0.1235E6 \text{ 1/sec}^2$ 

Natural frequency = 55.93 Hz

Period of response, T = 0.01788 sec

Increment of time for step-by-step, T/30 = 0.0006 sec

The time-displacement response curve derived from the dynamic analysis results is shown in figure 3.

### References

1. K.-J. BATHE, E. RAMM and E. L. WILSON, "Finite Element Formulations for Large Deformation Dynamic Analysis", Int. Jnl. Num. Mthd. Eng., VOL 9, pp 353-386, 1975.

### **Input Data**

X08D21A.DAT

X08D21B.DAT

1.425 lb/in applied to the top and bottom faces

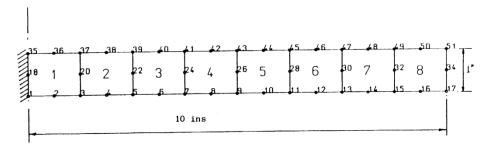


Figure 1 Linear dynamic analysis of cantilevered beam.

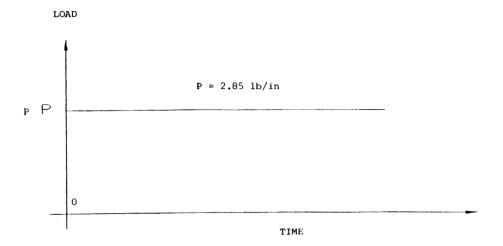


Figure 2 Load/time history graph for cantilevered beam.

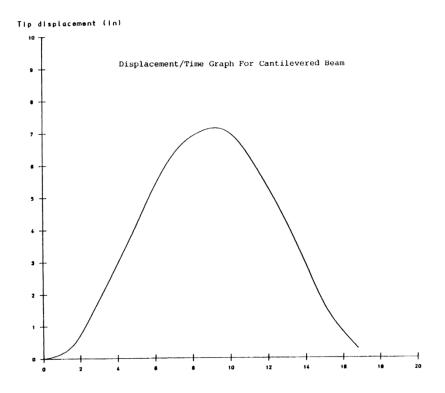


Figure 3 Displacement time graph for cantilevered beam.

# **Example 8.6.1**

# **Nonlinear Dynamic Analysis Of A Clamped Spherical Shell**

## **Keywords**

Nonlinear Dynamics, Elasto-Plastic, Natural Frequency, Spherical Shell

## **Description**

Determine the natural frequency and dynamic behaviour of a clamped spherical cap subject to a uniform lateral pressure over its outer surface. The shell geometry is in the form of a truncated sphere with a radius of 22.27 inches. The truncated spherical cap subtends a semi angle of 26.67 degrees at the centre of the sphere, and has a thickness of 0.41 inches. The shell is fully clamped around its periphery, and is subject to a uniform external pressure. The geometry and loading of the shell are shown in figure 1.

The finite element analysis is performed in four stages:

Stage 1 Eigenvalue extraction for natural frequency analysis

Stage 2 Linear elastic step-by-step dynamic stress analysis

Stage 3 Elasto-plastic step-by-step dynamic stress analysis

Stage 4 Geometrically nonlinear elasto-plastic step-by-step dynamic stress analysis

#### Discretisation

The shell structure is modelled using five curved axisymmetric shell elements (BXS3). The shell is assumed to be fully clamped at its supported base. The material properties used in the above analyses are as follows:

i) Eigenvalue extraction, and linear analysis

Young's modulus = 10.5E6 lb/in2

Poisson's ratio = 0.3

Mass density = 2.45E-4 lbsec2/in4

ii) Elasto-plastic analyses: Von-Mises elasto-plastic model

Young's modulus = 10.5E6 lb/in2

Poisson's ratio = 0.3

Mass density = 2.45E-4 lbsec2/in4

Yield stress = 0.024E6 lb/in2Hardening modulus = 2.142E5 lb/in2

The shell is assumed to be axisymmetric about its centre line and fully restrained against deflection and rotation at the nodes representing its clamped base. Geometrically nonlinear effects are modelled using a total Lagrangian description.

The dynamic analyses were performed using a time step of 0.000025 seconds.

## **Results Comparison**

Stage 1: Eigenvalue analysis for Natural Frequency

The natural frequency obtained from the LUSAS eigenvalue extraction analysis compares well with that obtained bt Bathe and Ozdenir in [1].

Lowest eigenvalue = 0.132202

Natural Frequency = 1829.95 Hz

Stages 2-4: Step-by-step Dynamic Analyses

Figure 2 presents results of the LUSAS step-by-step dynamic analyses of the pressure loaded shell and compares each of the following cases:

LE - Linear elastic step-by-step analysis

MNL - Materially nonlinear analysis (elastoplastic)

GMNL - Geometrically and materially nonlinear analysis

#### References

- Bathe, K.J. and Ozdemir, H. 'Elasto-plastic large deformation static and dynamic analyses', Computers and Structures, Volume 6, 81-92, 1976
- FEAL Internal Report FEAL503 'Axisymmetric thin shell element', Finite Element Analysis Ltd. Kingston upon Thames, (1985).

## **Input data**

Total Lagrangian geometric nonlinearity is invoked by including OPTION 87.

X08D61A.DAT

X08D61B.DAT

X08D61C.DAT

X08D61D.DAT

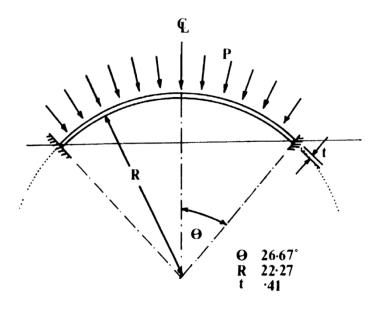


Figure 1

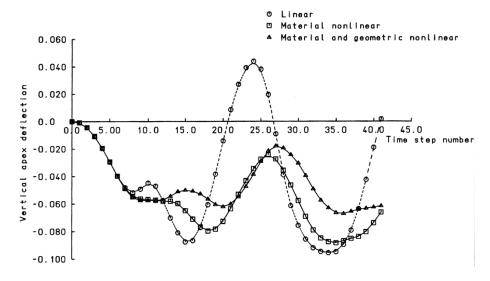


Figure 2 Clamped Spherical Cap

# **Example 9.1.1**

# **Quenching Of An Infinite Plate**

## **Keywords**

**Coupled Analysis, Thermal Stress** 

## **Description**

This example is an illustration of semi-coupled heat transfer and subsequent thermal stress analysis. Because a semi-analytic solution is available for the case (see [1]), the problem provides verification of this type of analysis. The purpose of the analysis is to predict the residual stresses caused by the quenching of a large, homogeneous plate in regions away from the edges of the plate. In this manner it can be treated as a plate of infinite extent in all but the thickness direction. The plate is made of an elastic, perfectly plastic material, with a yield stress that drops linearly with temperature above 121 C (250 F). Because the plate is assumed to be of infinite extent the problem is one-dimensional and the only gradient occurs through the thickness of the plate. The plate is initially at a uniform temperature near its melting point (when its yield stress is small). It is assumed to be stress free in this condition. The surface is then quenched by convective cooling to a medium at room temperature. Cooling is allowed to continue until all of the plate reaches room temperature. The analysis consists of a transient heat transfer analysis followed by a thermal stress analysis. During the heat transfer analysis the temperature-time history is used as input to the thermal stress analysis. The transient stresses are large enough to cause significant plastic flow, so that residual stresses will remain after the plate reaches room temperature.

### **Discretisation**

Ten elements are used through the half thickness of the plate. The heat transfer mesh uses QXF8 elements and the stress analysis QAX8 elements.

## **Geometry**

The plate is 914.4mm (36in) thick.

## **Material properties**

Young's modulus 206.8 GPa (30.0E6 lb/in<sup>2</sup>)

Poisson's ratio 0.3

Yield stress 248.2 MPa for j<121 C (36000 lb/in ,j<250 F)

248.2(1-(j-121)/1111.1) MPa, j>121 C

(36000(1-(j-250)/2000) lb/in<sup>2</sup>, j>250 F

Density  $7832 \text{ kg/m}^3 \text{ (0.283 lb/in}^3\text{)}$ 

Specific heat 0.6 kJ/kg C (0.1431 BThu/lb F)

Thermal conductivity 58.8 W/m C (7.872E-4 BThu/in sec F)

The film coefficient on the surface of the plate is 193.1 W/m. C (6.559E-5 BThu/in.sec F)

## Theory

Equations for the determination of transient and residual stresses in plates subject to transient temperature distributions, based on the assumption of a viscoelastic, perfectly plastic material obeying a von Mises temperature-dependent yield condition are given in [1]. The analytical values for residual stress for this problem are given in [1] and are compared with LUSAS values in figure 2.

### Reference

 Landau, H.G., J.H. Weiner, E.E Zwicky, Jr., "Thermal Stress in a Viscoelastic-Plastic Plate with Temperature Dependent Yield Stress", Journal of Applied Mechanics, Vol.27, pp.297-302 (1960).

## **Input Data**

X09D11A.DAT

X09D11B.DAT

X09D11C.DAT

X09D11D.DAT

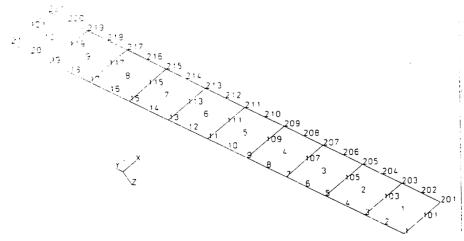


Figure 1

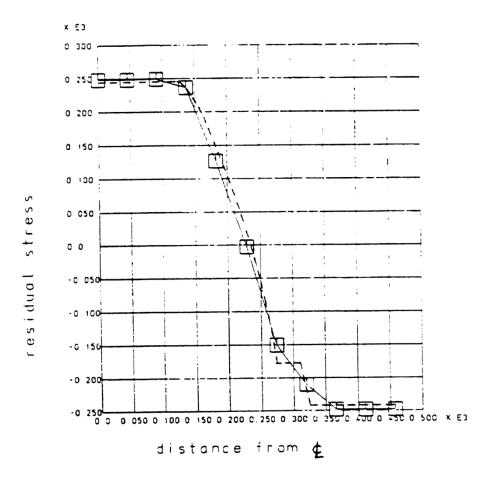


Figure 2 - Residual stresses in plate at room temperature. '\_' LUSAS results, '---' Reference.

# **Example 9.1.2**

# **Coupled Temperature Displacement Analysis**

## **Keywords**

**Coupled Analysis** 

## **Description**

This example is an elementary nonlinear case of one dimensional, fully coupled, heat transfer and stress analysis. The model is shown in figure 1. A conductive rod is fixed at one end, A, and free at the other end B. Between the free end and an adjacent body, C, there is a gap, across which heat will be conducted. The heat transfer coefficient for the gap drops linearly as the clearance increases. The fixed end of the rod, A, and the wall, C, (which is also fixed in position) are both held at fixed temperatures,  $\phi_A$  and  $\phi_C$ .

The objective is to predict the displacement,  $u_B$ , and temperature,  $\phi_B$ , of the free end of the rod. For simplicity, all temperatures and displacements are measured from a standard, constant temperature position in which the gap distance is  $d_o$ . It is assumed that the strains are small and the behaviour of the rod is linear elastic with constant modulus and thermal expansion coefficient. We also assume that the gap never closes, so that the rod is always stress free.

### **Discretisation**

The following element types are used:

- (i) Structural 1 HX8 element and 4 JNT4 elements.
- (ii) Thermal 1 HF8 element.
- (iii) 4 LFS2 link elements.

## **Geometry**

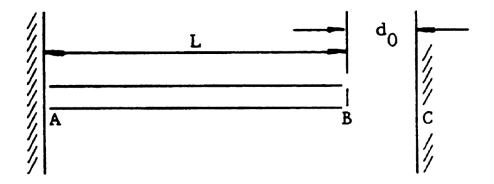


Figure 1

## **Material properties**

All specifications in consistent units

E = Young's modulus = 3.0E6 v = Poisson's ratio = 0.0  $\alpha = expansion$  coefficient = 1.0E-5 c = rod conductivity = 1.0

## **Boundary conditions**

- (i) Structural
  - at (a) restrained, at (b) free
- (ii) Field

at (a) temperature  $\phi_A = 400$ , at (c) temperature  $\phi_C = 200$ 

## **Theory**

The exact solution to the problem is developed using the equilibrium and compatibility equations with the appropriate boundary conditions. Heat conduction in the rod is assumed to be governed by Fourier's Law:

$$q = K_r \frac{d_{\phi}}{d_s} \tag{0.0.133}$$

where  $K_r$  is the thermal conductivity of the rod and distance s is measured from A.

Heat conduction across the gap is given by:

$$q = K_{g} \left( \phi_{B} - \phi_{C} \right) \tag{0.0.134}$$

and the thermal conductivity of the gap,  $K_{\rm g}$ , decreases linearly as the clearance increases:

$$K_{g} = K_{0} \left[ 1 + \frac{U_{B}}{d_{1} - d_{0}} \right] \tag{0.0.135}$$

where K<sub>0</sub> and d<sub>1</sub> are constants and U<sub>B</sub> the displacement of B.

For this problem the following values have been chosen:

$$K_r = K_0 - 1$$

$$K_g = 2.0$$
 when  $U_B = 0.0$ 

$$K_g = 0.0$$
 when  $U_B = 0.02$ 

$$d_1 - d_0 = 0.01$$

For thermal equilibrium:

$$\frac{\mathrm{dq}}{\mathrm{ds}} = 0 \tag{0.0.136}$$

Therefore, from (0.0.133) d $\phi$ /ds is constant in the rod, so that (0.0.133) may be expressed as:

$$q = K_r \left( \phi_\Delta - \phi_B \right) \tag{0.0.137}$$

The thermal expansion of the rod is given by:

$$U_{\rm B} = \frac{\alpha L}{2} (\phi_{\rm A} + \phi_{\rm B}) \tag{0.0.138}$$

For thermal equilibrium:

$$K_r(\phi_A - \phi_B) = K_g(\phi_B - \phi_C) \tag{0.0.139}$$

Substituting (0.0.135) and (0.0.138) in (0.0.139) yields a quadratic equation for  $\phi_B$ . The boundary values and chosen constants can be used in this equation to solve for  $\phi_B$ , where only one admissible positive root is obtained. Substitution of  $\phi_B$  in (0.0.139) gives  $U_B$ 

$$\phi_B = 285.4$$
  $U_B = 3.427E-3$ 

#### Results

Two approaches are implemented to allow a solution of this example. Firstly, if two analyses can be run concurrently, an iterative solution of the problem is possible. Secondly, if only one module can be run at once, a solution is obtained by the use of restart files. The temperature and displacement at point b computed by the two approaches are:

'	Iteration 0		Iteration 1		Iteration 2		Itn. 3
	Temp	Disp	Temp	Disp	Temp	Disp	Temp
Method 1 Thermal Structural	300.0	3.5x10 <sup>-3</sup>	285.084	3.425x10 <sup>-3</sup>	285.206	3.426x10 <sup>-3</sup>	285.208
Method 2 Thermal Structural	300.0	3.5x10 <sup>-3</sup>	284.936	3.425x10 <sup>-3</sup>	285.213	3.426x10 <sup>-3</sup>	

The slight differences with theoretical values are due to the use of a constant value for L in the manual calculations.

## **Input Data**

X09D11c.DAT

X09D11d.DAT

Note that restart file X09D12C.RST is renamed to Z09D12C.RST

Note that restart file X09D12D.RST is renamed to Z09D12D.RST

# **Example 9.2.1**

# **Compression Of An Angle Into A Corner**

## **Keywords**

Thermal Surfaces, Coupled Analysis, Nonlinear

## **Description**

A poorly formed angle bracket, subtending an angle greater than 90 degrees, is pushed against a second right angle bracket, figure 1. Contact along the faces is progressive as the angles deform. A semi-coupled static structural/thermal analysis is carried out using thermal surfaces. The temperature at the interface between the angles is a function of the pressure at the contact point.

### **Discretisation**

Each angle is meshed using the following two dimensional elements:

Structural analysis - 36 QPM4 plane stress continuum elements.

Thermal analysis - 36 QF4 plane field elements.

The mesh at the initial configuration is shown in figure 2.

## **Geometry**

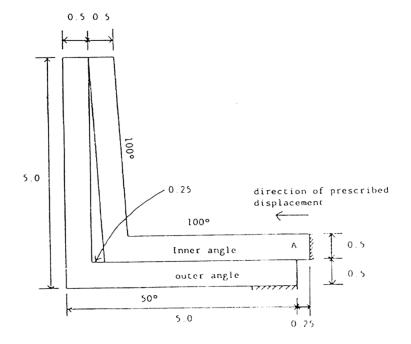


Figure 1 - Initial geometry

## **Boundary conditions**

Thermal:

Prescribed temperature

outer angle - outer surface temperature of 50

inner angle - outer surface temperature of 100

Structural:

Total prescribed displacements applied to end A

step 1 -0.12

step 2 -0.24

step 3 -0.25

step 4 -0.26

step 5 -0.27

## **Material properties**

```
Thermal: solid conductance = 101 thermal contact properties: gap conductance contact pressure 0 	 0 	 0 	 1000 	 0.03 Structural: E = 100 Poisson's ratio = 0.0 Master surface stiffness = 0.1 Slave surface stiffness = 0.1
```

## Loading

Structural:

Total prescribed displacements applied to end A

step 1 -0.12 step 2 -0.24 step 3 -0.25 step 4 -0.26 step 5 -0.27

## **LUSAS** results

The results should show a uniform progression of contact with the hottest point indicating the point of greatest contact pressure. If the surfaces are in contact, but there is no contact pressure, heat will not flow between the surfaces. Figure 3 details the progression of contact as the end displacement of the inner angle is increased. Initial contact is at the top of the angles as shown in load case 1. A second contact point is predicted at the point of prescribed displacement as the outer angle bends to accommodate the inner angle. The largest stresses arise from bending. As loading is increased, load cases 2 to 4, the point of contact moves down the angles. Above the point of contact the angles are stress free. Finally, in load case 5, the inner angle is in a state of almost pure compression and the outer angle is restraining this load by a combination of bending and tension. Figure 4 mirrors the progression of contact with the heat flowing across the contact as expected.

# Input data

X09D21A.DAT X09D21B.DAT

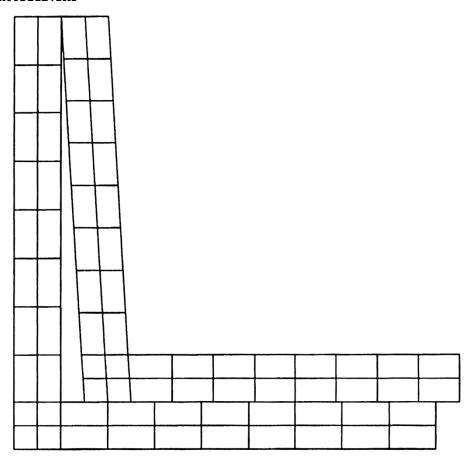


Figure 2 Mesh in initial configuration

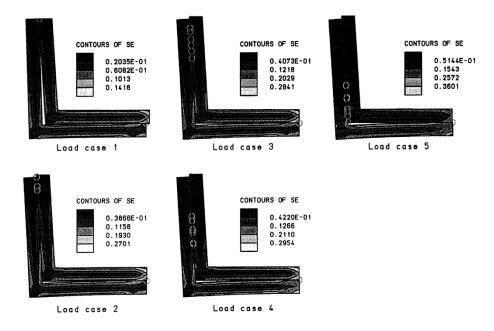


Figure 3 Stress contours and contact points (circled)

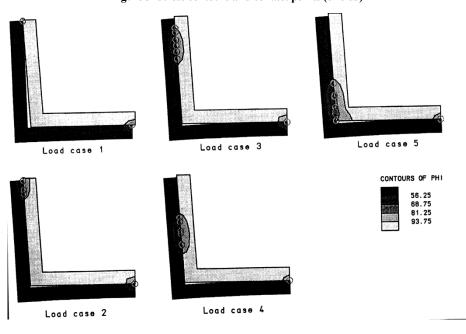


Figure 4 Temperature contours and contact points (circled)

# **Example 9.2.2**

# **Upsetting Of A Cylindrical Billet**

## **Keywords**

Plasticity, Coupled Analysis, Plastic Work

### **Description**

A cylindrical billet is compressed between two plattens so that its length is reduced by 52%. The rate at which the billet is compressed dictates the amount of heat developed as a result of plastic work done. This example compares the temperature distributions for a fast and slow rate of compression.

The billet is 30mm long and of 10mm radius with all surfaces fully insulated. The initial and final configurations of the billet are shown in Figure 1. Symmetry of the problem allows an axisymmetric analysis to be performed. Elasto-plastic material properties are assumed for this metal forming example. The effects of applying the prescribed displacement loading at a rate of 0.09 mm/sec and 90 mm/sec are compared. Note that inertia effects have been ignored and the time-step used in the thermal analysis dictates the rate at which the heat is developed. Data transfers are performed in parallel during the execution of this thermo- mechanical coupled analysis.

### **Discretisation**

A mesh of 119 axisymmetric solid elements (QAX4) was used in the structural analysis, while the same number of QXF4 elements was used in the thermal analysis. The initial mesh is shown in Figure 1.

### **Material properties**

Young's modulus =  $2E11 \text{ N/m}^2$ 

Poisson's ratio = 0.3

Thermal expansion coefficient = 1.2E-5 per °C

Initial yield stress =  $7E8 \text{ N/m}^2$ 

Work hardening rate =  $3E8 \text{ N/m}^2$ 

Specific heat =  $586J/(Kg \circ C)$ 

Density =  $7833 \text{ Kg/m}^3$ 

Conductivity =  $52J/(m-s-{}^{\circ}C)$ 

Heat fraction  $h_f = 0.9$ 

### **Boundary conditions**

Structural:

Symmetry conditions are applied along the axis and over the middle surface of the billet so that only half an axisymmetric segment is modelled. Nodes along the axis of symmetry are restrained horizontally but are free to move in the vertical direction. To simulate the loading platten, the nodes along the top surface are free to move in the horizontal direction but are restrained vertically solely to apply the prescribed displacements. Symmetry conditions for nodes on the middle surface require restraint in the vertical direction and freedom horizontally.

#### **Thermal**

All external surfaces are insulated. This is achieved by initialising the environmental temperature at the boundary to zero and defining a very low convective heat transfer coefficient at the external surface of the billet. An initial environmental load is applied to all element faces along the boundary to model this condition.

### **Solution comparison**

The general response observed for both the rapidly loaded test and the slower case agrees with what would be expected for such analyses. The slow loading test shows that the temperatures have time to diffuse leading to a uniform temperature distribution throughout the billet, Figure 3; the temperatures in this case range between 121 and 122.8 degrees. The opposite is true for the faster loading case where higher temperatures are computed with a large temperature gradient occurring across the billet. The highest temperatures are located at the point where most plastic deformation occurs. Results for each case are compared in Figures 2 and 3.

## **Input Data**

X09D22A.DAT

X09D22B.DAT

X09D22C.DAT

X09D22D.DAT

Note:

This data file can be used for both the fast and slow cases. However, different data transfer file names should be used to avoid confusion.

Note:

For the slow rate of loading, the incrementation line in TRANSIENT CONTROL must be changed from 0.01/9 to 10/9. The data transfer file should also be renamed to avoid confusion.

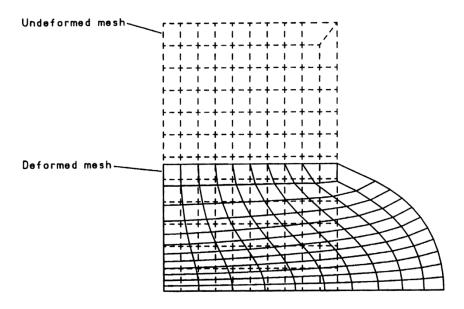


Figure 1. Deformed mesh plot at 52% upsetting

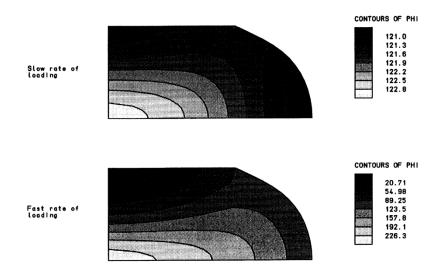


Figure 2. Temperature distributions for slow and fast rates of loading at 52% upsetting.

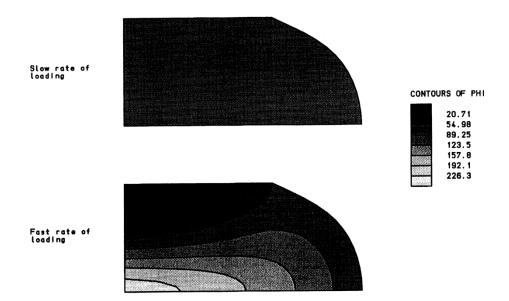


Figure 3. Comparison of temperature distribution for slow and fast rates of loading at 52% upsetting

# **Example 10.1.1**

# **Plastic Bar Impact Against A Rigid Wall.**

### **Keywords**

Impact, Slideline, Plastic Flow

### **Description**

This example consists of a cylindrical steel projectile striking perpendicular to a rigid boundary. Upon impact, the stress in the projectile at the impact point exceeds the elastic limit of the material, and a plastic front moves back through the projectile. Within the plastic zone the material flows radially and results in a shortening of the projectile.

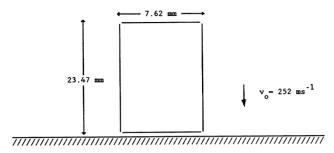
Wilkins [1] performed a series of experiments which illustrated that the final cylinder length is very sensitive to the yield strength of the material. Consequently, as the projectile undergoes finite strains, this example furnishes a worthwhile test for both elastoplastic algorithms and finite strain procedures in explicit dynamic analysis.

Figures 1 and 2 illustrate the final configuration of the bar for an initial velocity of 252 m/s. The accuracy of the bar shortening predicted in the analysis is compared with the shortening observed experimentally [1].

### **Discretisation**

The projectile is modelled using 5x21 4-noded one point quadrature axisymmetric elements (QAX4E), figure 1. The boundary between the projectile and the rigid wall is represented using a penalty based slideline procedure [2].

## **Geometry**



Young's modulus, E	211.0x10 <sup>9</sup> Pa
Poisson's ratio, v	0.3
Density, p	$7840.0 \text{ kg/m}^3$
Initial yield stress, $\sigma_0$	1.2x10 <sup>9</sup> Pa
Isotropic hardening parameter, H <sub>II</sub>	$1.0 \mathrm{x} 10^6  \mathrm{Pa}$

Default values for the hourglass viscosity and for the artificial viscosity coefficients are to be used.

### **LUSAS** results

(i) The final length of the bar can be approximated as 19.87mm by considering the net displacement of the top and bottom surfaces; this is in good agreement with the experimental value of 19.76mm given in [1], figure 1. The configuration at the point of maximum deformation is given in figure 2.

### References

- 1. WILKINS, M. L., GUINAN, M. W. Impact of Cylinders on a Rigid Boundary. J. Appl. Phys., Vol. 44, No. 3, pp1200-1207, March 1973.
- GOUDREAU, G. L., HALLQUIST, J. O. Recent Developments in Large-Scale Finite Element Lagrangian Hydrocode Technology. Comp. Meth. in Appl. Mech. and Eng., 33, pp725-757, 1982.

### **Input data**

X10D11A.DAT

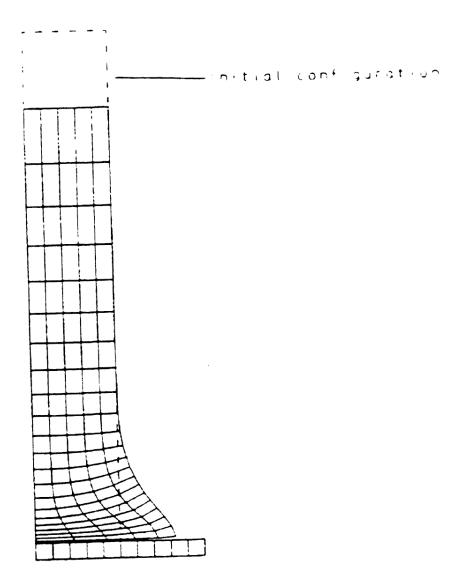


Figure 1

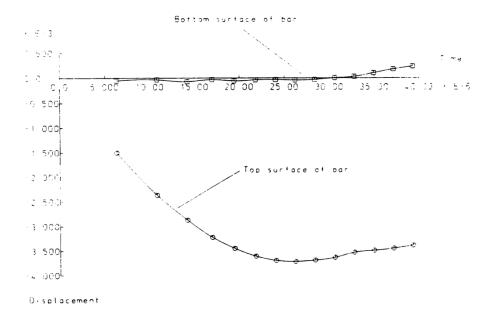


Figure 2

# **Example 10.1.2**

# **Cantilever Subject To Dynamic Loading.**

## **Keywords**

Impact, Tied Slideline

### **Description**

This example consists of an elastic cantilever which is subjected to an instantaneously applied uniformly distributed load. A single tied slideline is used to enable refinement of the mesh near the fixed end. The free end displacements computed by the dynamic finite element analysis are compared with those given by static linear and nonlinear analyses.

The free end displacements calculated by static analysis are:

Linear solution = 0.36

Geometrically nonlinear solution = 0.33

### **Discretisation**

The beam is modelled with 36 HX8E elements, figure 3, arranged in the following manner:

6 x 4 element mesh near the support

4 x 3 element mesh for remainder

### **Geometry**

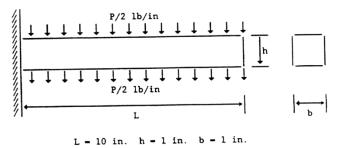


Figure 1

Young's modulus, E	12.0 x10 <sup>3</sup> lb/in <sup>2</sup>
Poisson's ratio, v	0.20
Density, p	$1.0x10^{-6} lb sec^2/in^4$

Default values for the hourglass viscosity and for the artificial bulk viscosity coefficients are to be used.

### Reference

1. BATHE, K-J., RAMM, E., WILSON, E. L. Finite Element Formulations for Large Deformation Dynamic Analysis. Int. J. Num. Meth. Eng., Vol. 9, pp353-386, 1975.

### **Lusas results**

Figure 2 shows the peak deflection computed by LUSAS compared with that given in [1]. The differences observed can be attributed to the use of different strain measures. In [1] a Total Lagrangian approach is used whereas the LUSAS solution is based on the Green-Naghdi stress rate and velocity strain.

Taking further time steps confirms oscillation about the nonlinear static solution. The deformed mesh at peak deflection is shown in figure 3.

### **Input data**

#### X10D12A.DAT

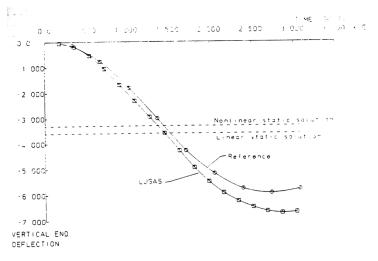


Figure 2

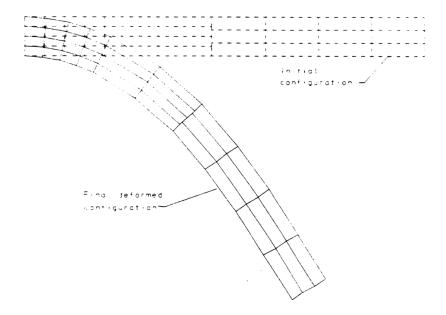


Figure 3

# **Example 10.1.3**

### **Interference Fit Test**

### Keywords

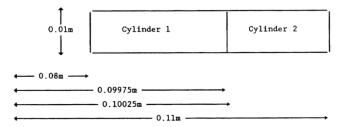
Axisymmetric Solid, Elastic, Slideline, Implicit

### **Description**

One range of problems which is of considerable interest to the Engineer is the assembly of components using interference fit techniques.

This example consists of two cylinders of length 0.01m and outer and inner diameters of (0.10025, 0.08m) and (0.11, 0.09975m) respectively. An initial interference of 0.0005m is thus considered between the cylinders. The geometry of the initial configuration of the cylinders is shown below. The first phase of the analysis is a linear elastic solution to obtain the initial configuration as a result of the interference fit. The response of the structure to subsequent loading on the innermost diameter corresponding to a suddenly applied constant radial load is also investigated. 2% damping of the two lowest modes is applied to the structure for this secondary phase utilising the Rayleigh damping facility. The cylinders are modelled using a 10x2 element mesh (see figure 1) with 4 noded axisymmetric solid elements (QAX4).

## **Geometry**



### **Boundary conditions**

Top and bottom surfaces fully restrained in global Y only

	Cylinder 1	Cylinder 2
Young's modulus, E	$150 \text{ x} 10^9 \text{ kN/m}^2$	$207 \ x 10^9 \ kN/m^2$
Poisson's ratio, v	0.25	0.3
Density, ρ	$7470 \text{ Kg/m}^3$	$7470 \text{ Kg/m}^3$

2% critical damping of the lowest two vibration modes is also required. This will give the Rayleigh damping factors as (according to the primary eigen-frequencies of 8213 and 8383 Hz obtained from an initial eigenvalue analysis on the structure):-

Rayleigh damping factor α	104265
Rayleigh damping factor β	3.83598x10 <sup>-7</sup>

### Loading

A compressive radial loading is applied over the inner diameter of the inner cylinder as a concentrated step load of magnitude 2.5 GPa.

### **Slideline Properties**

Stiffness scale factor, master surface	Stiffness scale factor, slave surface	Zonal contact detection parameter	Slideline extension
1.0	1.0	100.0	1.0

## Theory

The Rayleigh damping factors may be computed from the following equation:-

$$2\varsigma = \frac{\alpha}{\omega} + \beta\omega \tag{0.0.140}$$

where z is the known modal damping factor, a and b are the Rayleigh damping factors and w is the circular frequency at which the modal damping factor is operative. The Rayleigh damping factors are thus computed as:

$$\alpha = \frac{2\varsigma(1/\omega_2 - \omega_1)}{(1/\omega_2^2 - 1/\omega_1^2)} = 104265 \tag{0.0.141}$$

And

$$\beta = \frac{2\varsigma(\omega_2 - \omega_1)}{(\omega_2^2 - \omega_1^2)} = 3.83598 \times 10^{-7}$$
(0.0.142)

The approximate analytical solution for the interface pressure (P) due to an initial interference fit and an internal pressure  $(P_i)$  acting on the inside surface of the inner cylinder is given by the following:

$$P_{i} = \frac{\frac{\gamma}{b} + \frac{2Pa^{2}}{E_{1}(b^{2} - a^{2})}}{\frac{1}{E_{1}} \left[ \frac{a^{2} + b^{2}}{b^{2} - a^{2}} - \upsilon_{1} \right] + \frac{1}{E_{2}} \left[ \frac{b^{2} + c^{2}}{c^{2} - b^{2}} \right] + \upsilon_{2}}$$
(0.0.143)

where:

γ is the initial interference

a is the inner diameter of the inner cylinder (0.08)

b is the diameter of the interference fit (0.10)

c is the outer diameter of the outer cylinder (0.11)

For the initial interference configuration without internal pressure loading, the analytical solution yields:-

$$P = 61.7 \text{ MPa}$$

whilst for the secondary case of superimposed internal loading of 2.5 GPa, the interface pressure is given as:-

$$P = 793.4 \text{ MPa}$$

#### **Lusas Results**

The initial radial stress distribution over the cylinder cross-section is shown in Figure 3, from which the normal interface pressure obtained is 62.8 MPa. The effect of damping on the normal interface displacement for node 111, subsequent to the application of the internal pressure may be seen in Figure 2.

The final radial stress distribution over the cylinder cross-sections for the damped response is shown in Figure 4. It may be seen that the final value for the interface pressure of 818 MPa represents a difference of 3% when compared with the approximate analytical solution of 793.4 MPa. This analysis was also carried out using a statically applied load, figure 6, where the final interface pressure was found to be 820 MPa. A comparison of radial stresses at each side of the interface for the damped system is given in figure 5. This figure shows that as the analysis progresses a state is reached where the effects of loading the structure dynamically have effectively been damped. This leads to a final stress distribution very similar to that for the static analysis.

### Reference

1. Reismann, H., Pawlik, P., Elasticity: Theory and Applications, J. Wiley and Sons, 1980

### **Input Data**

X10D13A.DAT X10D13B.DAT X10D13C.DAT

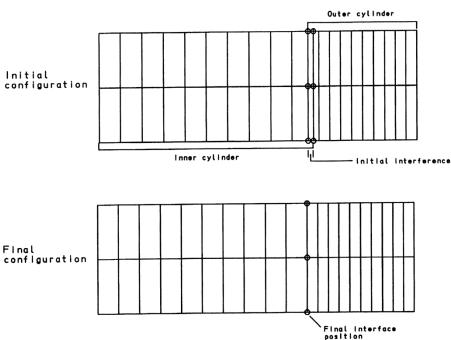


Figure 1 Initial and final configuration of cylinders

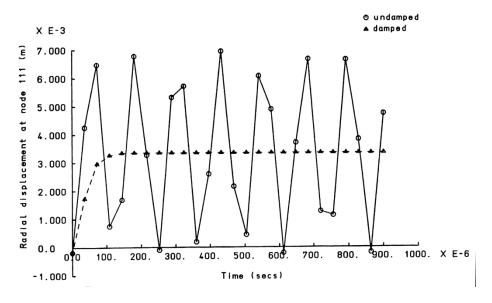


Figure 2 Comparison of damped and undamped response for Interface node 111

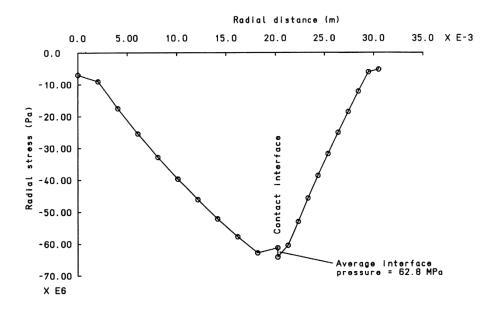


Figure 3 Initial radial stress distribution for cylinders.

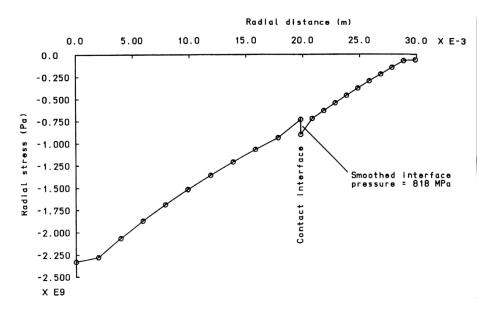


Figure 4 Final radial stress distribution for cylinders.

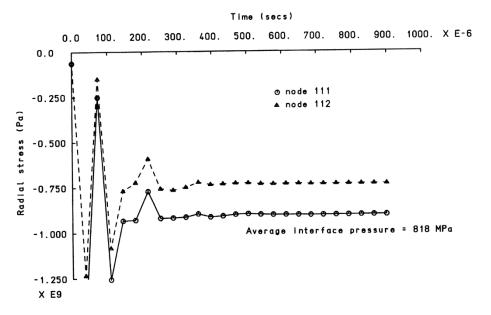


Figure 5 Radial stresses on each side of interface.

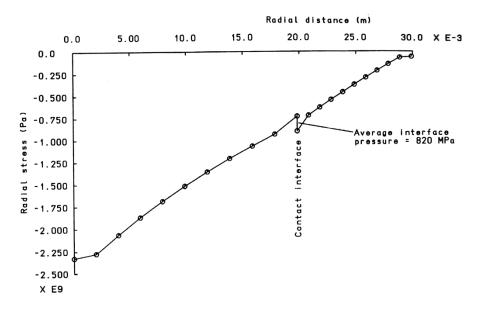


Figure 6 Final radial stress distribution - static analysis

# **Example 10.1.4**

# **Cylinder Compression Between Platens**

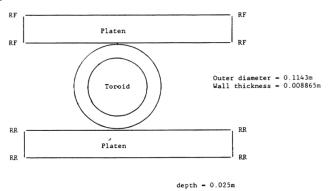
### **Keyword**

Nonlinear, Contact, Implicit, Static, Slidelines

### **Description**

The crushing of a long, straight pipe between two flat, frictionless anvils is presented in this example. The objectives are to establish the load deflection response of the pipe and to describe the overall deformation of the section. The example is a simple test for modelling contact problems involving deformable bodies and rigid, impenetrable surfaces.

### Geometry



### **Discretisation**

The cylinder is modelled using 8x4x2 eight noded solid elements (HX8), whilst the platen utilises an 8x1x2 element mesh (Figure 1). The boundary between the outer cylinder surface and the platen face is represented using the frictionless slideline option (type 1). The boundary conditions are depicted in the preceeding diagram and loading is applied over 36 increments to the upper platen via a prescribed displacement to give a total relative displacement of -0.0315m. Note that there are two symmetry planes in this problem.

Young's modulus, E	30 x10 <sup>6</sup> kN/m <sup>2</sup>
Poisson's ratio, v	0.3
Density, ρ	$7800 \text{ Kg/m}^3$
Yield stress, $\sigma_o$	$241 \text{ N/m}^2$

where  $\sigma_0$  is the von Mises initial yield stress. The hardening curve is as follows:-

Hardness, C <sub>i</sub> (N/m <sup>2</sup> )	Plastic strain, ε
$9.850 \times 10^9$	0.0035
$5.386 \times 10^9$	0.0083
$3.447x10^9$	0.0133
$1.741 \times 10^9$	0.0281
$1.144x10^9$	1.0000

## **Slideline Properties**

Default values are specified for all slideline properties, except for the slideline extension where a value of 0.1 is used.

### **Lusas results**

The undeformed mesh is shown in Figure 1, together with the deformed mesh at load increment 36. Figure 2 shows the deformation history. The force displacement variation of total normal interface force corresponds well with the experimental curve given in [1]. Note that the relative displacement is plotted, i.e. twice the applied displacement.

### Reference

 Taylor, L.M., "A finite Element Analysis for Large Deformation Metal Forming Problems Involving Contact and Friction" TICOM report number 81-15, 1981

## **Input Data**

X10D14A.DAT

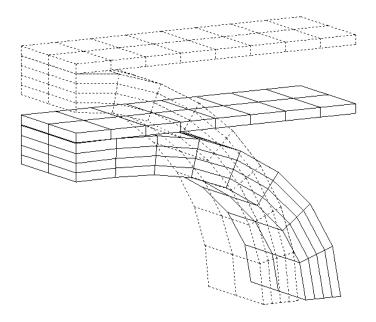


Figure 1: Deformation of cylinder at increment 36

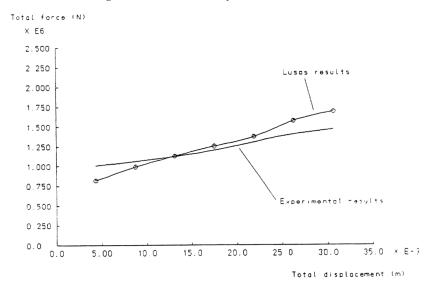


Figure 2: Variation of normal interface force with total platen displacement (3D)

# **Example 10.2.1**

# Propped Cantilever With Joint Elements And Contact/Gap Model

### **Keywords**

Cantilever, Contact, Gap, Joint

### **Description**

Determine the end deflection and reactions of point loaded cantilever beam fully restrained at one end and supported at its mid length by a sunken support.

### **Discretisation**

The cantilever is modelled using two BS4 element. The prop is modelled using one JSH4 element and the nonlinear contact/gap model.

Three support conditions are considered:

#### 1. Free cantilever

Since the contact stiffness is only invoked by a downward (negative) deflection of the cantilever, a positive end load results in a free cantilever condition.

#### 2. Propped cantilever

The propped cantilever condition is modelled using an initial gap of 0 mm and a large contact stiffness (10E16). The end load is applied in the negative Z direction.

#### 3. Cantilever with sunken prop

The sunken propped cantilever is modelled using an initial gap of 1 mm and a large contact stiffness (10E16). The end load is applied in the negative Z direction.

## Geometry

The dimensions and geometry of the cantilever are shown in figure 1.

Length, L	40 mm
Distance to prop, 1	20 mm
Second moment of area, I	833.3 mm <sup>4</sup>

	Young's modulus, E	$2.0x10^6$
--	--------------------	------------

## Loading

The cantilever is subjected to a concentrated end load (P= 390609.3 N) designed to produce a free cantilever end deflection of 5 mm. The total load is applied in five equal increments:

Increment	Case 1	Case 2	Case 3
1	0.2P	-0.2P	-0.2P
2	0.4P	-0.4P	-0.4P
3	0.6P	-0.6P	-0.6P
4	0.8P	-0.8P	-0.8P
5	1.0P	-1.0P	-1.0P

### **Theory**

### Case 1. Free cantilever condition (Figure 2)

The end deflection of a free cantilever subject to a transverse end load is given by the expression [1]

$$d_{\text{max}} = \frac{Pa^3}{3EI} \left( 1 + \frac{3b}{2a} \right) \tag{0.0.144}$$

Hence, since (a=L) and (b=0)

$$d_{\text{max}} = \frac{PL^3}{3EI} \tag{0.0.145}$$

### Case 2. Propped cantilever condition (Figure 3)

The end deflection of a propped cantilever is given by the expression

$$d_D = \frac{P1^3}{12EI} \left( 4p^2 + 6pq + 3p + 3q \right) \tag{0.0.146}$$

and the reactions by the expressions

$$R_{A} = \frac{3Pp}{2}$$
 and  $R_{B} = -P\left(1 + \frac{3p}{2}\right)$  (0.0.147)

where

$$p = \frac{a}{1}$$
 and  $q = \frac{b}{1}$  (0.0.148)

Hence, since (a=1) and (b=0), therefore (p=1) and (q=0), and

$$D = \frac{Pl^3}{12EI} (4+3) = \frac{7Pl^3}{12EI}$$
 (0.0.149)

and

$$R_{A} = \frac{3P}{2}$$
 and  $R_{B} = -P\left(\frac{5}{2}\right)$  (0.0.150)

### Case 3. Propped cantilever with sunken support

The propped cantilever with the sunken mid support behaves as the free cantilever until the initial gap of 1 mm is closed. Following this deflection the stiff contact spring is invoked, the end deflection restricted, and the reaction at the prop progressively increased.

$$f P = K_1 R_A + K_2 R_B (0.0.151)$$

where

f is the load factor (Total Load level = fP)

 $K_1$ ,  $K_2$  are the support reaction scaling factors

### Comparison

The LUSAS results for the end deflection and the support reactions are compared to the theoretical solutions.

Case 1. Free Cantilever.

		End Displacement			
Increment	Load Level	Theory	LUSAS		
1	0.2P	1.000	0.9999		
2	0.4P	2.000	1.9999		
3	0.6P	3.000	2.9988		
4	0.8P	4.000	3.9988		
5	1.0P	5.000	4.9988		

**Case 2. Propped Cantilever** 

		End Displacement		End Reaction R <sub>A</sub>		Prop Reaction R <sub>B</sub>	
Inc.	Load	Theory	Lusas	Theory	Lusas	Theory	Lusas
1	-0.2P	-0.2187	-0.2187	-117182	-117183	195304	195305
2	-0.4P	-0.4374	-0.4375	-234365	-234366	390609	390609
3	-0.6P	-0.6562	-0.6562	-351548	-351548	585914	585914
4	-0.8P	-0.8749	-0.8749	-468730	-468731	781217	781219
5	-1.0P	-1.0937	-1.0937	-585913	-585913	976523	976523

Case 3. Cantilever with sunken prop.

			nd cement	End Rea	ction R <sub>A</sub>	Prop Rea	action R <sub>B</sub>		
Inc.	Load	Theory	Lusas	Theory	Lusas	Theory	Lusas	K <sub>1</sub>	$K_2$
1	-0.2P	-0.2187	-0.2187	-117182	-117183	195304	195305	1.0	0.0
2	-0.4P	-0.4374	-0.4375	-234365	-234366	390609	390609	1.0	0.0
3	-0.6P	-0.6562	-0.6562	-351548	-351548	585914	585914	1.0	0.0
4	-0.8P	-0.8749	-0.8749	-468730	-468731	781217	781219	0.5	0.5
5	-1.0P	-1.0937	-1.0937	-585913	-585913	976523	976523	0.1	0.9

### **References**

1. C.S.Gray et al. 'Steel Designers Manual : Second Edition', Crosby-Lockwood and Sons Publishers Ltd. (1960)

## **Input Data**

X10D21A.DAT

X10D21B.DAT

X10D21C.DAT

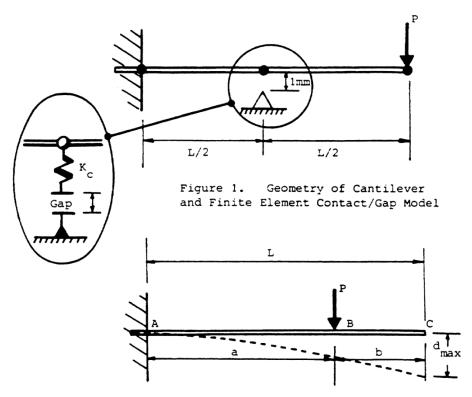


Figure 2. Theory of Free Cantilever

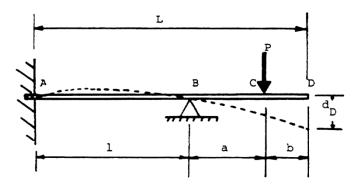


Figure 3. Theory of Propped Cantilever.

# **Example 11.1.1**

# **Eigenvalue Analysis Of Pinned Double Cross**

### **Keywords**

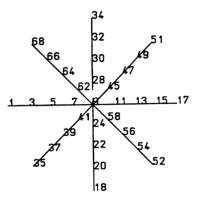
Superelements, Eigenvalue Analysis, Generalised Coordinates

### **Description**

The eigenvalues of a pinned double cross are calculated using a full mesh. The structure is then partitioned into three superelements and the same analysis is repeated. Generalised coordinates are used to improve the solution.

#### **Finite Element Model**

Superelement 1 is defined by arms with node numbers 68-9, 1-9, and 35-9, superelement 2 by 34-9 and 18-9 and superelement 3 by 51-9, 17-9, and 52-9. The master freedoms are those of node 9.



### Geometry

Length of arms	5 m
Arm cross-sectional area	$0.125 \; m^2$

Young's modulus	200.0 x10 <sup>9</sup>
Density	$8x10^3\ kg/m^3$

### **Boundary Conditions**

Pinned at each support.

## **Solution Comparison**

The lowest 16 eigenvalues of the full mesh compared to [1] are:

	Eigenvalue (Hz)		
Modes	Reference	Full mesh	
1	11.336	11.336	
2-3	17.687	17.687	
4-8	17.715	17.715	
9	45.477	45.477	
10-11	57.364	57.364	
12-16	57.683	57.683	

The problem has multiple and close eigenvalues for which the corresponding eigenvectors are detailed in Figure 1. Surprisingly, there is no symmetry associated with the multiple vectors at the higher modes. The structure is split into superelements which are connected at node 9. There are only three eigenvalues associated with the reduced structure.

	Eigenvalue (Hz)		
Modes	Full mesh	Reduced mesh (master freedoms only)	
1	11.336	14.411	
2-3	17.687	175.866	

The eigenvalues of the reduced structure do not adequately describe the behaviour of the complete structure since only three degrees of freedom are used to simulate the complete structure. To improve the estimate, additional freedoms called generalised coordinates are introduced. Generalised coordinates are fictitious freedoms that are used to enhance the reduced stiffness and mass of the individual superelements by using the modes of vibration of the individual superelements.

Figure 2 details the three lowest modes of each of the superelements. In their calculation, node 9 was fully restrained; therefore they correspond to the first mode of a beam which is simply supported at one end and encastre at the other. Superelement 2 has only two arms, thus the third mode corresponds to the next highest mode for the beam. Including generalised coordinates into the computations the lowest 9 modes are:

	Eigenvalue (Hz)		
Modes	Full mesh	Reduced mesh (master freedoms + 3 generalised coordinates)	
1	11.336	11.343	
2-3	17.687	17.687	
3		17.687	
4-8	17.715	17.715	
9	45.477	53.404	

There is a large improvement in the accuracy of the analysis. The best results are obtained for modes 4-8. In this instance node 9 is stationary and the generalised coordinates exactly represent the full eigenvectors.

## **Superelement Analysis Sequence**

- a) Solve with Master freedoms only
- (i) Generate superelements 1, 2, 3
- (ii) Combine superelements and solve eigen problem
- b) Solve with Master freedoms + Generalised coordinates
- (i) Generate superelements with modal freedoms defined
- (ii) Combine superelements and solve eigen problem

#### References

Test No FV2. The Standard NAFEMS benchmarks, NAFEMS.

## **Input Data**

X11D11A.DAT

X11D11B.DAT

X11D11C.DAT

X11D11D.DAT

X11D11E.DAT

X11D11F.DAT

X11D11G.DAT

X11D11H.DAT

X11D11I.DAT

- a) Solve complete structure
- b) Use superelements with master freedoms only
- (i) Create superelement 1
- (ii) Create superelement 2
- (iii) Create superelement 3
- (iv) Solve eigenvalue analysis
- c) Use superelements with master freedoms and generalised coordinates
- (i) Create superelement 1

Note for other superelements 2 and 3 include GENERALISED COORDINATES command

(ii) solve for eigenvalues of combined structure

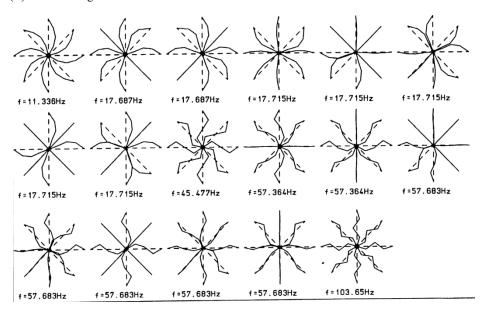


Figure 1 Eigenvectors of Pin-Ended Double Cross.

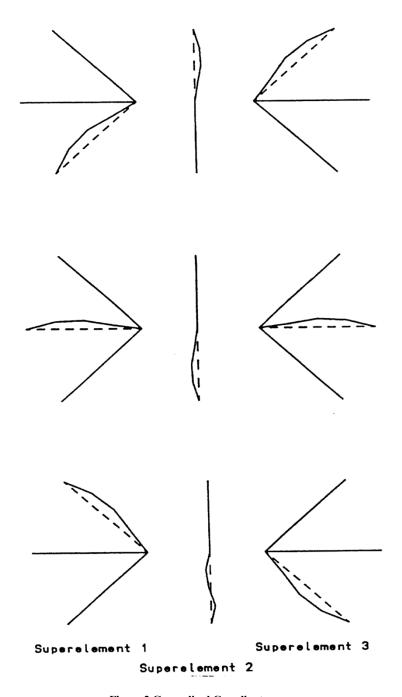


Figure 2 Generalised Coordinates.

# **Example 11.5.1**

# **Tip Loaded Cantilever Analysed Using Superelements**

## **Keywords**

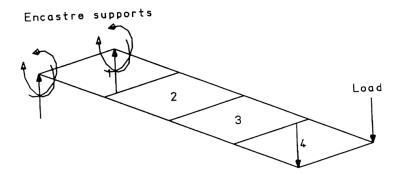
Superelements, Static Analysis

### **Description**

A cantilever is divided into two superelements, one modelling the root, the other the free end. The superelements are assembled and load is applied. Displacement and stresses of the superelements are then recovered using the restart file for the root and by re-defining the superelement data for the tip.

### **Finite Element Model**

Elements 1 and 2 form the root superelement, whilst 3 and 4 constitute the tip superelement.



## **Geometry**

Length	4
Width	1
Thickness	0.05

Young's modulus	200.0x10 <sup>11</sup>
Poisson's ratio	0.3

#### Load

Concentrated load of -1 applied at cantilever tip (total load=-2)

## **Boundary Conditions**

Encastre at fixed end

## **Solution Comparison**

Moment at restrained nodes = 4x-2/2 = -4

Shear force = 2/2 = 1

## **Analysis Sequence**

- (i) Generate root superelement use RESTART WRITE to save superelement data
- (ii) Generate tip superelement RESTART WRITE is not used so a full solution is required to recover superelement data
- (iii) Solve superelement with applied concentrated loads
- (iv) Recover root data using RESTART READ
- (v) Recover tip data using full re-definition of problem

Both superelements are stored in the same external file

#### CANTLVR.SDA

If a restart file is not saved, then data defining the elements comprising the superelement must be input in the data file.

## **Input Data**

X11D51A.DAT

- (i) Create root superelement
- (ii) Create tip superelement
- (iii) Assemble root and tip superelement into cantilever and apply load
- (iv) Recover root stresses and displacements using restart file to define superelement
- (v) Recover tip stresses by re-defining superelement data input

# **Example 12.1.1**

### 1-D Soil Consolidation

## **Keywords**

Consolidation, Linear, Nonlinear, Transient

### **Description**

A soil column fixed along its base and subjected to uniform vertical load applied along its upper face is analysed by plane strain and 3D two phase elements. In order to establish 1D soil consolidation, proper horizontal constraints are applied to the column.

## **Geometry, Boundary and Loading Conditions**

distributed load q = 100, undrained conditions (p free) applied at time 0; drainage (p = 0) is provided in the subsequent time steps

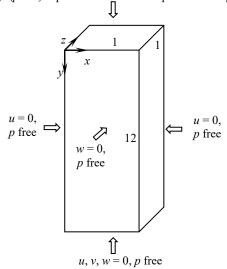


Figure 1

The 3D model of the column is shown in Figure 1, together with boundary and loading conditions. In the plane strain analysis, only one *xy* section (projection of the 3D model in *z*-direction) is considered.

#### **Discretisation**

In the 3D analysis, the column is discretised by 96 hexahedral (HX20P) elements (four columns of 24 elements as shown in Figure 2); in the plane strain analysis, it is discretised by 48 quadrilateral (QPN8P) elements (two columns of 24 elements) (without discretisation in the Z-direction).

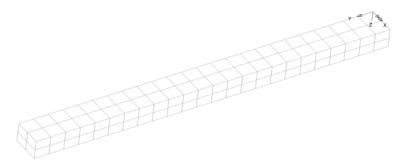


Figure 2

## **Material properties**

Linear material properties:

Effective Young's modulus of the soil  $E'=6.0\times10^3$ 

Poisson's ratio v = 0.0 (1D deformation)

Consolidation data:

Bulk modulus of pore fluid  $K_w = 2.2 \times 10^{15}$ 

Soil permeability factor  $\kappa_x = \kappa_y = \kappa_z = 0.02$ 

Porosity  $\eta = 1$ 

Unit weight of water  $y_w = \rho g = 10$ 

Mohr-Coulomb material properties:

Cohesion of the soil skeleton c = 20

Angle of friction  $\phi = \pi/6$ 

## **Theory**

Firstly, a linear analysis is carried out, then, the same linear problem is solved using non-linear solution facilities. The target results for these two cases are the Terzaghi 1D consolidation analytical solution. The results for the displacement at the top and the pore pressure at the bottom at specific times are (Craig, 1974):

Time ( <i>t</i> )	0.0	0.1	0.5	1.0	1.999
v(0, t)	0.0	0.02061	0.04607	0.06515	0.09208
p(12, t)	100.0	100.0	99.8936	97.1388	83.3619

Finally, the problem is solved using the non-linear Mohr-Coulomb material model; the final displacements, stresses and plastic strains should be the same as those obtained by their counterpart continuum elements using drained properties.

All analyses are run for a total time of 2 units using a constant time step of length  $\Delta t = 0.1$  units after the initial conditions have been established at time t = 0.001 units.

#### Reference

- 1. Craig, R.F. Soil Mechanics. Van Nostrand Reinhold Ltd., U.K., 1974.
- **2.** Norris, V.A. The elasto-plastic analysis of soil consolidation with special reference to kinematic hardening. Ph.D. Thesis, Department of Civil Engineering, University College of Swansea, 1980.

#### **LUSAS** results

Displacements and pore pressures from QPN8P and HX20P elements for the linear problem using linear (L) and nonlinear (NL) facilities (QPN8P gives identical results for the two cases)

Time (t)		0.001	0.101	0.501	1.001	2.000
v(0, t)	QPN8P	0.0	0.01842	0.04497	0.06437	0.09149
	HX20P(L)	0.0	0.01841	0.04493	0.06422	0.09097
	HX20P(NL)	0.0	0.01842	0.04497	0.06437	0.09149
p(12, t)	QPN8P	100.0	99.9968	99.5893	96.5290	83.3734
	HX20P(L)	99.9984	99.8649	99.0530	95.6493	82.1417
	HX20P(NL)	100.000	99.9968	99.5893	96.5290	83.3734

The above linear consolidation solutions by elements QPN8P and HX20P are satisfactory considering the coarse mesh adopted. A finer mesh at the permeable boundary and a smaller time step can further improve the accuracy of the solution.

In 3D linear consolidation analysis, slight out-of-balance may occur, i.e. the reaction force becomes slightly different from the applied load. This effect becomes more serious with softer soils and when increasing the number of elements. This is because of the ill conditioning of the stiffness matrix due to the big difference in the amplitude of displacements and pressures, as also noticed by Norris (1980). This problem disappears when we solve the linear transient problem using nonlinear control.

The results from QPN8P and KHX20P elements using a Mohr-Coulomb plasticity model are:

Time (t)		0.001	0.101	0.501	1.001	2.000
v(0, t)	QPN8P	0.0	0.01862	0.04556	0.06523	0.09273
	HX20P	0.0	0.01862	0.04556	0.06523	0.09273
p(12, t)	QPN8P	100.0	99.9968	99.5902	96.5392	83.4284
	HX20P	100.0	99.9968	99.5902	96.5392	83.4284

They are close to the above linear solution; when using a non-linear material model, the final displacements are somewhat larger than the linear material model due to the onset of yielding. The final displacements, stresses and plastic strains agree well when compared with their counterpart continuum elements.

## **Input data**

**X12d11a.dat** (Plane strain, linear analysis)

**X12d11b.dat** (Plane strain, linear problem with nonlinear control)

**X12d11c.dat** (Plane strain, nonlinear material)

x12d11d.dat (3D, linear analysis)

**X12d11e.dat** (3D, linear problem with nonlinear control)

**X12d11f.dat** (3D, nonlinear material)

## **Example 12.1.2**

## **Isothermal Drainage of Water from a Sand Column**

## **Keywords**

Consolidation, Geostatic, Nonlinear, Transient, Unsaturated

## **Description**

This example simulates the experiment performed by Liakopoulos [1] on desaturation of a column of very fine Del Monte sand due to gravitational effects. Before the start of the experiment, water was continuously added from the top and allowed to drain freely at the bottom through a filter, until uniform flow (i.e. fully saturated) conditions were established. At the start of the experiment, the water supply ceased and water flows out from the bottom due to the desaturation process, and also to the squeezing effect of soil deformation.

## **Geometry, Boundary and Loading Conditions**

In accordance with [2, 3], a column of soil with height 1m and width 0.1m is analysed. The boundary conditions are

lateral surface	$q^w = 0$ , $u = 0$ (impervious and constrained)
top surface	$q^w = 0, t > 0$
bottom surface	$p^c = 0$ (free water outflow) for $t > 0$ , $u = v = 0$

The initial conditions are: at t = 0,  $p^c = 0$  at all nodes (i.e. full saturation with water), which corresponds to a steady flow of water through the sand column. Furthermore, a state of mechanical equilibrium is assumed for t = 0. All the displacements are related to these initial displacements which correspond to the equilibrium state.

Only gravity of soil and water is considered.

#### **Discretisation**

The column is discretised into 10 uniform 2D QPN8P or 3D HX20P (with thickness of 0.1m) elements along the height.

## **Material properties**

Deformable linear elastic soil and isotropic permeability are assumed. The porosity and hydraulic properties of Del Monte sand were measured by Liakopoulos [1] in an independent set of experiments.

Young's modulus*	E = 1.3  MPa
Poisson's ratio*	ν= 0.4
Solid grain density	$\rho^{\rm s} = 2000 \text{ kg m}^{-3}$
Liquid density	$\rho^{W} = 1000 \text{ kg m}^{-3}$
Porosity	n = 0.2975
Intrinsic permeability	$\kappa = 4.5 \times 10^{-13} \text{ m}^2$
Water viscosity	$\mu^{w} = 1 \times 10^{-3} \text{ Pa s}$
Air viscosity	$\mu^g = 1.8 \times 10^{-5} \text{ Pa s}$
Gravitational acceleration	$g = 9.80665 \text{ m s}^{-2}$
Atmospheric pressure ( T = 293.15K)	$p_{atm} = 101 \ 325 \ Pa$
Biot's constant*	$\alpha_b = 1$
Bulk modulus of solid grains*	$K_s = 1.0 \times 10^6 \text{MPa}$
Bulk modulus of water*	$K_w = 2.0 \times 10^3 \text{MPa}$

<sup>(\*</sup> These mechanical parameters adopted by Lewis and Schrefler [2] were not measured by Liakopoulos [1])

Liakopoulos [1] used the following saturation-capillary pressure and relative permeability-saturation relationships in isothermal conditions, valid for water saturation  $S_w \ge 0.91$ 

$$S_w = 1 - 1.9722 \times 10^{-11} p_c^{2.4279}$$

$$\kappa^{rw} = 1 - 2.207 (1 - S_w)^{1.0121}$$

which can be fitted by the relationships of Valiantzas [4] with n = 3.207, w = 0.522 and  $h_c = -1.654$ ; the maximum errors in  $S_w$  and  $\kappa^{rw}$  are 5.4% and -3.2%, respectively.

#### Reference

- Liakopoulos, A.C. Transient flow through unsaturated porous media. PhD thesis, University of California, Berkeley (CA), USA, 1965.
- Lewis, R.W. and Schrefler, B.A. The finite element method in the static and dynamic deformation and consolidation of porous media. John Wiley & Sons Ltd, Chichester, 1998.
- Kolditz, O. and Shao, H. (Eds) OpenGeoSys developer-benchmark-book. OGS-DBB 5.04, UFZ Publisher, 2010.

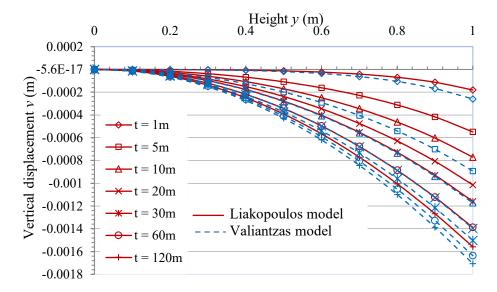
 Valiantzas, J.D. Combined Brooks-Corey/Burdine and van Genuchten/Mualem closed-form model for improving prediction of unsaturated conductivity. J. Irrig. Drain Eng. 137: 223-233, 2011.

#### **LUSAS** results

We first carry out an initial (t = 0) nonlinear static (or Geostatic) analysis to the column under gravity with the given displacement condition and  $p^w = 0$  at all nodes to ensure an equilibrium geostatic stress field. The obtained effective stresses are transformed into initial stresses for the subsequent transient coupled pore diffusion/stress analysis from t > 0.

The timestep size is controlled by the target change in water pressure and the target change in saturation. Up until 8 seconds the change in water pressure is critical afterwards the timestep is controlled by the change in saturation. The target change of pressure is set to 1Pa and the target change in saturation to 0.001. 3×3 full Gauss integration was adopted.

Figure 1 shows the vertical displacement and water pore pressure versus height at different times obtained by QPN8P. The results from HX20P are identical to those from QPN8P. When the  $S_w - p^c$  and  $\kappa^{TW} - S_w$  relationships of Liakopoulos are adopted (input in tabular form), the obtained results (solid lines in Figure 1) are the same as in the literature [2, 3]. If the deformability of the solid grains is considered, i.e.  $K_s = 10^6$  MPa is used instead of an infinite value, the change in the results is negligible. However, when the fitted Valiantzas relationships [4] with n = 3.207, w = 0.522 and  $h_c = -1.654$  are used, evident differences are noticed in the results (dash lines in Figure 1), especially at intermediate stages, e.g. t = 5, 10 and 20m.



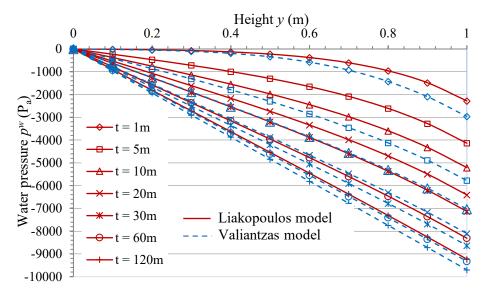


Figure 1

## **Input data**

**X12d12a.dat** (Plane strain QPN8P element)

**X12d12b.dat** (3D HX20P element)

# **Example 12.1.3**

### Volumetric deformation of unsaturated soils

## **Keywords**

Nonlinear, Unsaturated, Barcelona Basic Model

## **Description**

Selected triaxial stress paths involving wetting, drying and increasing confining stress are modelled to demonstrate the capibilities of the Barcelona Basic Model (BBM) and agreement with the results originally presented in [1].

## **Geometry, Boundary and Loading Conditions**

A unit cube is restrained on three faces and loaded with face loads in each of the other faces to impose a uniform pressure. Pore water pressure is prescribed at all of the nodes with a pore water pressure degree of freedom. The face load and pore water pressure are then changed through the use of load curves to give different combinations of changes in suction and pressure.

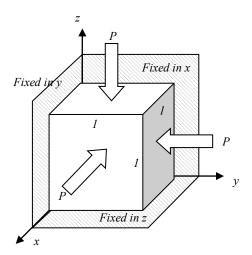


Figure 1

The 3D hexahedral element (HX20P) is shown in Figure 1, together with boundary and loading conditions. In the plane strain analysis, only one xz section (projection of the 3D model in y-direction) is considered using a quadrilateral (QPN8P) element.

The pressure applied through face loads (P) and the suction (equal to the negative pore water pressure) (u) are varied using load curves as follows:

Path	u	P	u	P	u	P	u	P
	(MPa)							
Wetting 1	0.2	0.15	0.2	0.6	0.1	0.6	0.0	0.6
Wetting 2	0.2	0.15	0.1	0.15	0.1	0.6	0.0	0.6
Wetting 3	0.2	0.15	0.1	0.15	0.0	0.15	0.0	0.6
Drying 1	0.0	0.15	0.0	0.6	0.1	0.6	0.2	0.6
Drying 2	0.0	0.15	0.1	0.15	0.1	0.6	0.2	0.6
Drying 3	0.0	0.15	0.1	0.15	0.2	0.15	0.2	0.6

The wetting tests here begin at a net pressure (input as a residual stress) of 0.15MPa and a suction of 0.2MPa and are subsequently loaded and wetted through different paths until a final net pressure of 0.6MPa and a suction of 0MPa are reached. The drying tests also begin at 0.15MPa but in a fully saturated and are dried and loaded until a final net pressure of 0.6MPa and a suction of 0.2MPa are reached.

## **Material properties**

The following parameters are given in [1]

М	G	$\lambda_0$	κ	r	$K_S$	$p_{c0}$	$p_{ m ref}$	β	
1	10MPa	0.2	0.02	0.75	0.008	0.1MPa	0.1MPa	12.5MPa <sup>-1</sup>	

These have been adjusted to allow for use of Poisson's ratio instead of shear modulus (although due to the applied boundary conditions this does not have any effect here). Additionally, to account for the bi-logarithmic definition of  $\lambda$  and  $\kappa$  they have been modified to the following. To exactly replicate the existing behaviour the semi-log parameters would be divided by the specific volume to get the bi-log parameters. As the void ratio changes, an average value which fits the data has been chosen. It can be seen in the following examples that this approach gives a good overall fit to the response.

М	ν	$\lambda_0$	к	r	$\kappa_{_S}$	$p_{c0}$	$p_{ m ref}$	β	
1	0.3	0.111	0.0111	0.75	0.00421	0.1MPa	0.1MPa	12.5MPa <sup>-1</sup>	

The initial yield surface position at zero suction is set at 0.2MPa and the movement of the LC curve can be seen as plastic yielding causes this to increase to 0.6MPa. The specific volume is set initially to 1.9.

The tests are undertaken without any two phase material definition so that an effective stress is not calculated in order to replicate the behaviour seen in [1] which uses net stress.

#### Reference

**1.** Alonso, E. E., Gens, A. & Josa, A. A constitutive model for partially saturated soils Géotechnique 40, No. 3, 405-430 (1990).

## **LUSAS** results

The stress paths described above are plotted along with the development of the LC curve. The volumetric response is plotted on top of the behaviour in [1], it can be seen that excellent agreement is observed for all stress paths.

Wetting and loading:

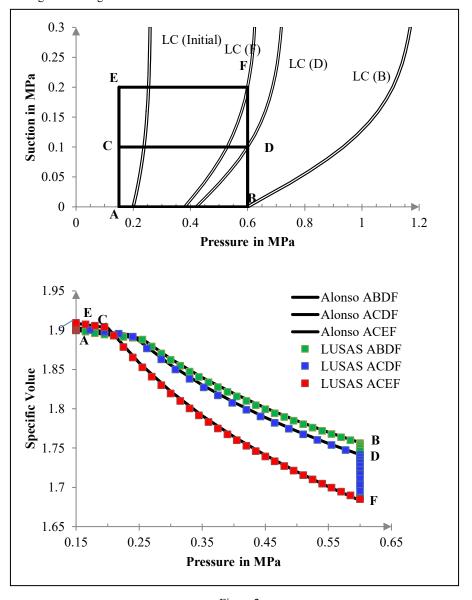


Figure 2

### Drying and loading:

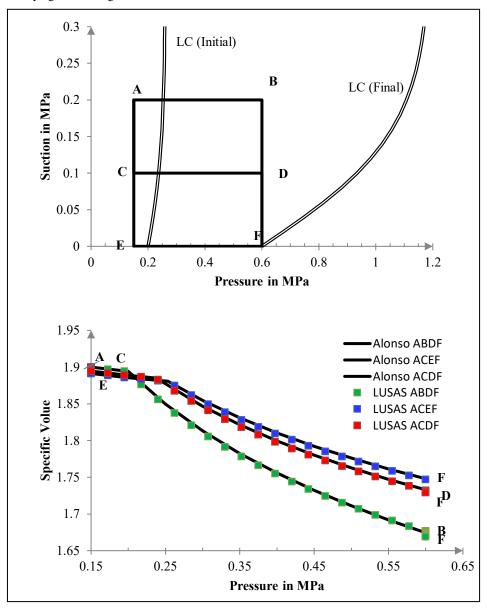


Figure 3

The 2D and 3D elements yielded identical results so only one set of graphs are shown.

## Input data

E02N96.dat	(2D wetting then loading of soil at u=0MPa)
E02N102.dat	(2D wetting, loading of soil at u=0.1 MPa and further wetting)
E02N98.dat	(2D loading of soil at u=0.2MPa and then wetting)
E02N104.dat	(2D drying and then loading of soil at u=0.2MPa)
E02N105.dat	(2D drying, loading of soil at u=0.1MPa and further drying)
E02N106.dat	(2D loading of soil at u=0MPa and then drying)
E18N46.dat	(3D wetting then loading of soil at u=0MPa)
E18N52.dat	(3D wetting, loading of soil at u=0.1 MPa and further wetting)
E18N48.dat	(3D loading of soil at u=0.2MPa and then wetting)
E18N54.dat	(3D drying and then loading of soil at u=0.2MPa)
E18N55.dat	(3D drying, loading of soil at u=0.1MPa and further drying)
E18N56.dat	(3D loading of soil at u=0MPa and then drying)

## **Example 13.1.1**

## **Coupled Hygro-Thermal-Mechanical Analysis**

## **Keywords**

Coupled Hygro-Thermal-Mechanical Analysis

## **Description**

This example is a concrete reinforced beam under a constant sustained service load for a period in excess of 400 days, experimentally tested by Gilbert, [1, 2].

The model is shown in Fig 1. Dimensions are in [mm].

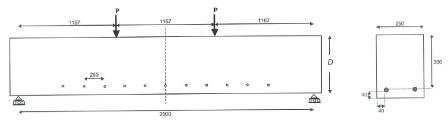


Fig. 1

Load and dimension of beam tested by Gilbert [1, 2]. D=348mm; bar diameter  $\phi$ =16 mm;

Case B1-a: P=18.6 kN, (50% of ultimate load) Case B1-b: P=11.8 kN, (30% of ultimate load)

The properties of concrete at various ages after casting were measured from companion cylinders and prisms using standard test procedures and are give in Table 1.

Material Property (in MPa)	Age (days)						
	7	14	21	28			
Compressive strength	12.9	18.3	23.1	24.8			
Flexural tensile strength (mod. of rupture)	3.0	3.7	4.3	5.6			
Indirect tensile strength (split cylinder)		2.0	2.6	2.8			
Modulus of elasticity ()	21090	22820	23990	24950			

Table 1. Concrete properties

The creep coefficient  $\varphi_{cc}$  for concrete first loaded at age 14 days and the shrinkage strain  $\varepsilon_{sh}$  were measured on 300mm diameter cylinders and slab specimens 160 mm thick, respectively, and the values at various concrete ages are given in Table 2. At present, this data is not used in the analysis.

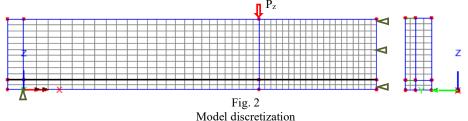
Age (days)	14	16	21	27	53	96	136	200	242	332	394
Фсс	0	0.14	0.36	0.48	0.92	1.15	1.29	1.40	1.50	1.64	1.71
Pak.	0	-14	-109	-179	-403	-591	-731	-772	-784	-816	-825

Table 2. Creep coefficient and shrinkage strain (×10<sup>-6</sup>)

The specimens were undisturbed in their moulds for 3 days. After the 3<sup>rd</sup> day the specimens were removed from the moulds and kept continuously moist by a thick and complete covering of wet hessian to minimise the loss of moisture from the specimens. After 14 days the covers and wet Hessian were removed, the demec strain targets were glued to the concrete surface and initial measurements were recorded. Beam specimens were subjected to two concentrated loads located at the third of the span and applied via steel channel beams.

#### **Discretisation**

A quarter of the beam is modelled by 2208 8-node solid elements HX8/HHT8 and 46 2-node bar elements BRS2/BFS2, Fig. 2.



## **Material properties**

#### Concrete hygro-thermal - main

$k_{T0}$	Thermal conductivity [W/m.K]	= 2.0	(*)
$C_{s0}$	Specific heat capacity of dry concrete [J/kg.K]	= 900	(*)
$ ho_{s0}$	Density of dry concrete [kg/m³]	= 2400	(*)
$K_{I0}$	Permeability (intrinsic) [m <sup>2</sup> ]	= 0.1E-1	8 (*)
$V_{ea}$	Entrapped air volume ratio [-]	= 0.023	(*)

#### Concrete hygro-thermal - advanced

$f_{s}$	Tortuosity coefficient [ - ]	= 0.01	(*)
$H_{v}$	Latent heat of evaporation [J/kg]	= 0.0	
$k_{T1}$	Change of thermal conductivity with temperature [ - ]	=-0.01	(*)
$C_{s1}$	Change of specific heat capacity with temperature [ - ]	= 0.0	(*)
$\rho_{s1}$	Change of density with temperature [ - ]	= 0.0	(*)
$f_{\iota}$	Check box 'Thermal conductivity depends on water satura	tion' = $0$	or 1

Properties that change with temperature are computed from:

$$\begin{split} & \rho_s\left(T\right) = \rho_{s0} + \rho_{s1}\left(T - T_{\text{ref}}\right) \\ & C_s\left(T\right) = C_{s0} + C_{s1}\left(T - T_{\text{ref}}\right) \\ & k_T(T, S_w) = \left[k_{T0} + k_{T1}\left(T - T_{\text{ref}}\right)\right] \left[1 + f_k \Phi\left(S_w\right)\right] \end{split}$$

where:

 $T_{\text{ref}}$  Built-in reference temperature = 20 °C or 293.14 K, based on model units.

 $\Phi(S_w)$  Built-in function of the water saturation  $S_w$ 

#### Concrete hygro-thermal - Heat of Hydration

Cement type		Type I	(*)
$W_{cem}$	Mass of cement per unit volume [kg/m³]	= 350	
$W_{c}$	Water/Cementitious mass ratio [ - ]	=0.45	
$W_{slg}$	Mass of slag per unit volume [kg/m <sup>3</sup> ]	= 0.0	(*)
$W_{pfa}$	Mass of fly ash per unit volume [kg/m <sup>3</sup> ]	= 0.0	(*)
$CaO_{pfa}$	CaO content of fly ash [%]	= 0.0	(*)
delav	Delay of heat of hydration	= 0.0	(*)

#### Steel bars thermal

$k_{\scriptscriptstyle T}$	Thermal conductivity [W/m.K]	= 45.0
$\rho_{\scriptscriptstyle s}$	Density [kg/m³]	= 7800
C	Specific heat capacity [J/kg.K]	= 450

*Note*: Since the concrete and steel bars share the same volume it would be more accurate for the bars to have equivalent properties that are computed by subtracting the value of the concrete property from the corresponding steel property. Thus:

$k_T^*$	Equivalent bar thermal conductivity $[W/m.K]$	= 45 - 2 = 43	
$ ho_s^*$	Equivalent bar density [kg/m³]	=7800 - 2400 = 5400	
$C_s^*$	Equivalent bar specific heat capacity [J/kg.K]	$=\frac{7800\times450-2400\times900}{5400}=250$	0

#### Concrete mechanical - main

E	Young's modulus [N/m <sup>2</sup> ]	= 25.0E9
$\nu$	Poisson's ratio	= 0.2
$\rho$	Mass density [kg/m³]	= 2400
$\alpha$	Expansion coefficient	= 1.0E-5
$f_c$	Compressive strength [N/m <sup>2</sup> ]	= 25.0E6
$f_t$	Tensile strength [N/m <sup>2</sup> ]	= 2.24E6
$\mathcal{E}_{\mathcal{C}}$	Strain at peak uniaxial compression	= 2.2E-3 (*)
$G_f$	Fracture energy per unit area [N/m]	= 130 (*)
$eta_{ m ds}$	Coefficient of dry shrinkage	= 0.9E-3 (*)
$eta_{ m cs}$	Coefficient of chemical shrinkage	= 0.1E-3 (*)

#### Concrete mechanical - advanced

$\beta_r$	Biaxial to uniaxial stress ratio	= 1.15	(*)
$Z_{o}$	Initial relative position of yield surface	= 0.6	(*)
Ψ	Dilatancy factor	=-0.1	(*)
$m_{\rm g}$	Constant in interlock state function	= 0.3	(*)
$m_{\text{hi}}$	Contact multiplier on $\varepsilon_o$	= 0.5	(*)
$m_{\text{ful}}$	Final contact multiplier on $\varepsilon_o$	= 5.0	(*)

$r_{\sigma}$	Shear intercept to tensile strength	=1.25	(*)	
μ	Slope of friction asymptote for damage	=0.8	(*)	
$m_{cf}$	Multiplier on ε <sub>0</sub> to define crack fixity strain	=0.5	(*)	
$f_{pz} \\$	Fracture process zone width [m]	= 0.06	(*)	
$it_{fx} \\$	No. of iterations after which cracks are fixed	= 2	(*)	
$it_{mx} \\$	Max. iterations for fracture strain update	= 2	(*)	
$R_{\rm s}$	Residual softening factor	= 0.001	(*)	
$A_p$	Material constant A <sub>p</sub>	= 0.6	(*)	
$A_{n} \\$	Material constant A <sub>n</sub>	= 0.75	(*)	
$Z_1$	Contact sharpness $Z_1$	= 8.0	(*)	
$Z_2$	Contact sharpness Z <sub>2</sub>	= 8.0	(*)	
$C_{LM}$	Limit of contact tension	= 3.0	(*)	
$\mathrm{fm}_{\mathrm{r}}$	Multiplier on relaxation times	= 1.0	(*)	
$fm_c$	Creep factor multiplier	= 1.0	(*)	
$fm_p$	Pickett factor multiplier	= 1.0	(*)	
$\Gamma_{\rm r,mi}$	Degree of hydration percolation		= 0.05	(*)
$\Gamma_{\rm r,1}$	Degree of hydration at 1st cure	= 0.5	(*)	
$\Gamma_{r,l} \\ m_p$	Degree of hydration at 1st cure  Moisture rate Pickett factor effect [sec-1]	= 0.5 =10.0E-		
-,-				
m <sub>p</sub>	Moisture rate Pickett factor effect [sec <sup>-1</sup> ]	=10.0E-	12 (*)	
$m_p$ $S_{wd}$	Moisture rate Pickett factor effect [sec <sup>-1</sup> ] Saturation start Pickett effect	=10.0E- = 0.85	12 (*) (*)	
$m_{ m p} \ { m S}_{ m wd} \ { m  au}_{ m sh}$	Moisture rate Pickett factor effect [sec <sup>-1</sup> ] Saturation start Pickett effect Short relaxation time [days or its equivalent]	=10.0E- = 0.85 = 10	12 (*) (*) (*)	
$m_{ m p}$ $S_{ m wd}$ $ au_{ m sh}$ $ au_{ m Ll}$	Moisture rate Pickett factor effect [sec <sup>-1</sup> ] Saturation start Pickett effect Short relaxation time [days or its equivalent] 1 <sup>st</sup> long relaxation time [days or its equivalent]	=10.0E- = 0.85 = 10 = 150	12 (*) (*) (*) (*)	
$m_{ m p}$ $S_{ m wd}$ $ au_{ m sh}$ $ au_{ m L1}$ $ au_{ m L2}$	Moisture rate Pickett factor effect [sec <sup>-1</sup> ] Saturation start Pickett effect Short relaxation time [days or its equivalent] 1 <sup>st</sup> long relaxation time [days or its equivalent] 2 <sup>nd</sup> long relaxation time [days or its equivalent]	=10.0E- = 0.85 = 10 = 150 = 1100	12 (*) (*) (*) (*) (*) (*)	
$m_{ m p}$ $S_{ m wd}$ $ au_{ m sh}$ $ au_{ m L1}$ $ au_{ m L2}$ $c_{ m \it E}$	Moisture rate Pickett factor effect [sec <sup>-1</sup> ] Saturation start Pickett effect Short relaxation time [days or its equivalent] 1 <sup>st</sup> long relaxation time [days or its equivalent] 2 <sup>nd</sup> long relaxation time [days or its equivalent] Logistic model coefficient for E	=10.0E- = 0.85 = 10 = 150 = 1100 = 12.5	(*) (*) (*) (*) (*) (*) (*) (*)	
$m_{ m p}$ $S_{ m wd}$ $ au_{ m sh}$ $ au_{ m L1}$ $ au_{ m L2}$ $c_{\it E}$	Moisture rate Pickett factor effect [sec <sup>-1</sup> ] Saturation start Pickett effect Short relaxation time [days or its equivalent] $1^{st}$ long relaxation time [days or its equivalent] $2^{nd}$ long relaxation time [days or its equivalent] Logistic model coefficient for E Power model coefficient for $f_c$	=10.0E- = 0.85 = 10 = 150 = 1100 = 12.5 = 3.0	(*) (*) (*) (*) (*) (*) (*) (*)	
$egin{aligned} & \mathbf{m_p} & \\ & \mathbf{S_{wd}} & \\ & oldsymbol{ au_{sh}} & \\ & oldsymbol{ au_{L1}} & \\ & oldsymbol{ au_{L2}} & \\ & oldsymbol{c_fc} & \\ & oldsymbol{c_{fc}} & \\ & \oldsymbol{c_{fc}} $	Moisture rate Pickett factor effect [sec <sup>-1</sup> ] Saturation start Pickett effect Short relaxation time [days or its equivalent] $1^{st}$ long relaxation time [days or its equivalent] $2^{nd}$ long relaxation time [days or its equivalent] Logistic model coefficient for E Power model coefficient for $f_c$ Power model coefficient for $f_t$	=10.0E- = 0.85 = 10 = 150 = 1100 = 12.5 = 3.0 = 3.0	(*) (*) (*) (*) (*) (*) (*) (*) (*) (*)	
$egin{aligned} & \mathbf{m_p} & \mathbf{S_{wd}} & \\ & \mathbf{\mathcal{T}_{sh}} & \\ & \mathbf{\mathcal{T}_{L1}} & \\ & \mathbf{\mathcal{T}_{L2}} & \\ & c_{fc} & \\ & c_{fi} & \\ &$	Moisture rate Pickett factor effect [sec <sup>-1</sup> ] Saturation start Pickett effect Short relaxation time [days or its equivalent] $1^{st}$ long relaxation time [days or its equivalent] $2^{nd}$ long relaxation time [days or its equivalent] Logistic model coefficient for E Power model coefficient for $f_c$ Power model coefficient for $f_t$ Temperature when degradation starts [°C]	=10.0E- = 0.85 = 10 = 150 = 1100 = 12.5 = 3.0 = 3.0 = 40	(*) (*) (*) (*) (*) (*) (*) (*) (*) (*)	
$egin{array}{l} m_{ m p} & & & & \\ S_{ m wd} & & & & \\ oldsymbol{ au}_{ m sh} & & & & \\ oldsymbol{ au}_{ m L1} & & & & \\ oldsymbol{ au}_{ m L2} & & & & \\ oldsymbol{ au}_{ m fe} & & \\$	Moisture rate Pickett factor effect [sec <sup>-1</sup> ] Saturation start Pickett effect Short relaxation time [days or its equivalent] $1^{st}$ long relaxation time [days or its equivalent] $2^{nd}$ long relaxation time [days or its equivalent] Logistic model coefficient for E Power model coefficient for $f_c$ Power model coefficient for $f_t$ Temperature when degradation starts [°C] Temperature when degradation stops [°C]	=10.0E- = 0.85 = 10 = 150 = 1100 = 12.5 = 3.0 = 3.0 = 40 = 80	(*) (*) (*) (*) (*) (*) (*) (*) (*) (*)	

### (\*) Default property value

#### Note:

The true tensile strength  $f_t$  is about 80% of the cylinder splitting strength [3]. Thus  $f_t$  (t=25days) = 0.8\*2.8 = 2.24 MPa  $f_t$  (t=14days) = 0.8\*2.0 = 1.60 MPa To achieve  $f_t$ (t=14days)=1.60MPa,  $C_{ft}$  (default = 1) is set to 3.0

#### Steel bars mechanical

E	Young's modulus [N/m <sup>2</sup> ]	= 200.0E9
$\nu$	Poisson's ratio [-]	= 0.3
$\rho$	Mass density [kg/m³]	= 7800
$\alpha$	Expansion coefficient [-]	= 11.0E-6

*Note*: Since the concrete and steel bars share the same volume it would be more accurate the bars to have an equivalent properties that are computed by subtracting the value of concrete property from the corresponding steel property. Thus:

E Equivalent bar Young's modulus  $[N/m^2]$  = 175.0E9  $\rho$  Equivalent bar mass density  $[kg/m^3]$  = 5400

## **Boundary conditions**

#### (i) Hygro-thermal

Time 0.0 to time 14.0 days (wet Hessian cover) on all external surfaces:

ENVT load: Tenv = 20 °C;  $\alpha_T = 5.0$  W/m<sup>2</sup>K TPDSP load: RH=1, contact with water.

Time 40.0 to time 400 days. All external surfaces exposed to air.

ENVT load:  $Tenv = 20 \text{ °C}; \ \alpha_T = 5.0 \text{ W/m}^2\text{K}; \ \text{RH} = 0.5; \ \alpha_T = -1$ 

where:

Tenv Environmental temperature

 $\alpha_{\scriptscriptstyle T}$  Convective heat transfer coefficient

RH Relative humidity

 $\alpha_{\nu}$  Vapour mass transfer coefficient (negative - envokes Chiltern-Colburn analogy)

#### (ii) Structural

Support Z=0 along Y axis; Support X=0 on plane of symmetry.

Global distributed load along the half-width:  $P_z = -P/(\frac{1}{2}0.125)$ 

Case B1-a (P=18.6/kN):  $P_z=18.6/(\frac{1}{2}0.125) = -74.4 \text{ N/m}$ 

Case B1-b (P=11.8 kN)  $P_z = 11.8/(\frac{1}{2}0.125) = -47.2 \text{ N/m}$ 

## **Time steps**

#### Hygro-thermal analysis

A reduction of the time step at time 14 days is required when Hygro TPDSP is removed and replaced by ENVT. Time step of 1 day is used until the end of the simulation. Fig 3a

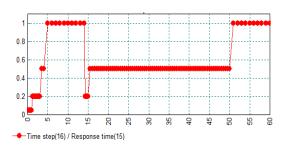


Fig 3a
Hygro-thermal analysis
Variation of Time step [days] vs Response time [days]

#### Structural analysis

The beam tests start at age 14 days. In the LUSAS analysis the self weight is applied at this age. After that, the sustained load is applied in 200 short time steps. After that, the load remains constant for 400 days. The time steps used up to day 60 are plotted in Fig. 3b. A constant time step of 3 days is used until the end of the simulation (Time=400 days).

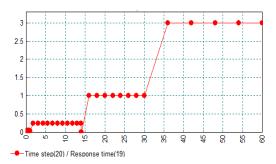


Fig 3b Structural analysis Variation of Time step [days] vs Response time [days]

## **Theory**

LUSAS results are compared to experimental results [1, 2], wherever available.

#### Reference

- 1. Gilbert, R. I., Nejadi, S., (2004). An Experimental Study of Flexural Cracking in R.C. Members under Sustained Loads, UNICIV Report R-435, School of Civil & Env. Eng., University of New South Wales, Sydney, Australia, (http://www.civeng.unsw.edu.au/staff/ian\_gilbert/).
- 2. Gilbert, R. I. (2013). Time-Dependent Stiffness of Cracked Reinforced and Composite Concrete Slabs, Procedia Engineering 57, 19 34.
- 3. Neville, A.M. (2013) Properties of Concrete, 5th ed., Pearson Education.

## **Input Data**

X13D11a.DAT X13D11b.DAT

### **LUSAS Results**

The computed maximum net deflection (i.e. the deflection due to self weight is subtracted) is compared to the experimentally measured one in Fig 4. When the load level is 30% of the ultimate load, test B1-b, LUSAS results agree very well with the experimental results. When the load level is 50% of the ultimate load, test B1-a, there is an 11% underestimation of the final deflection.

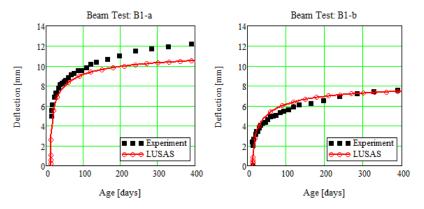


Fig. 4 Creep deflection.

Temperature, degree of hydration and relative humidity distributions are shown in Figs. 5a, 5b and 5c, resp.

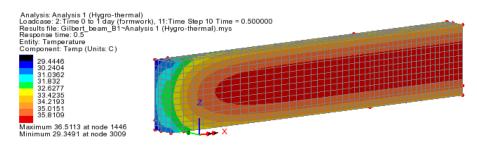


Figure 5a
Temperature distribution at time 0.5 days

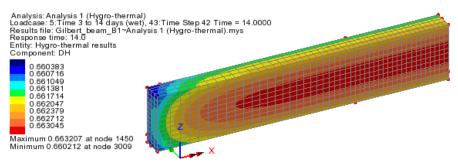


Figure 5.9b Degree of hydration at time 14.0 days

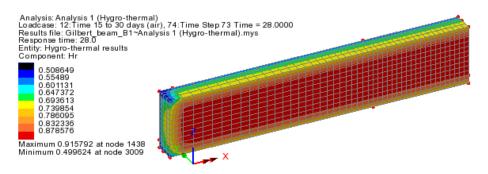
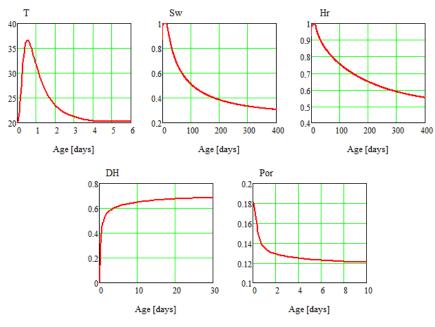


Figure 5.9b Relative humidity at time 28.0 days



LUSAS time history hygro-thermal results at the beam centre are given in Fig. 6.

Fig. 6 Hygro-thermal results at beam centre

T - temperature
Sw - water saturation
Hr - relative humidity
DH - degree of hydration
Por - porosity

There are no experimental hygro-thermal results to compare with.

The experimental data for the crack widths in beams B1-a and B1-b is given in Gilbert's Table 7, shown below:

Beam or	00101	Crack width (mm) At first loading		Crack width (mm) After 400 days			
slab		Avge w <sub>avge</sub>	Min <i>w</i> <sub>min</sub>	Max W <sub>max</sub>	Avge Wavge	Min w <sub>min</sub>	Max w <sub>max</sub>
В1-а	40	0.11	0.06	0.13	0.28	0.10	0.38
B1-b	40	0.04	0.02	0.05	0.13	0.10	0.18

Table 7 Initial and final crack widths for beams containing 2 N16 bars.

Response time: 400.0 Entity: Concrete Results - Solids

LUSAS crack distribution and crack width, CWmax, contour plot for test B1-a at Time=14 days (i.e. just after loading) and Time=400 days, are presented in Figs 7a and 7b, resp.

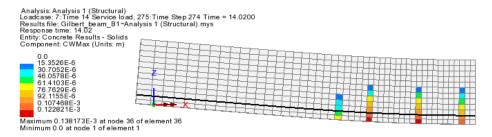
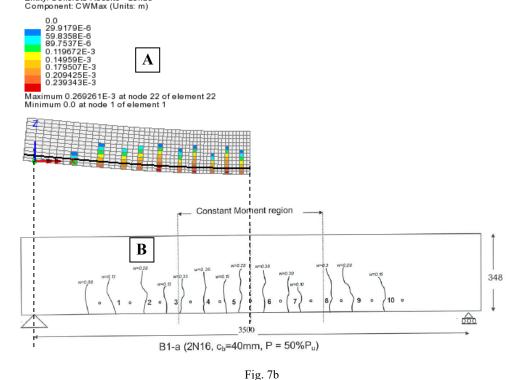


Fig. 7a

Test B1-a: Crack distribution and crack widths at time 14 days, i.e. at first loading A-LUSAS: CWmax=0.14 mm B-Experiment; CWmax=0.13mm



Test B1-a: Crack distribution and crack widths at time 400 days:

A – LUSAS: CWmax=0.27 mm. Number of cracks along half length – 10.

B – Experiment; CWmax=0.38mm. Number of cracks along full length – 14

LUSAS crack distribution and crack width, CWmax, contour plot for test B1-b at Time=14 days (i.e. just after loading) and Time=400 days, are presented in Figs 8a and 8b, resp.

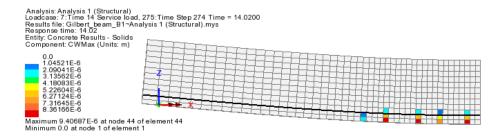
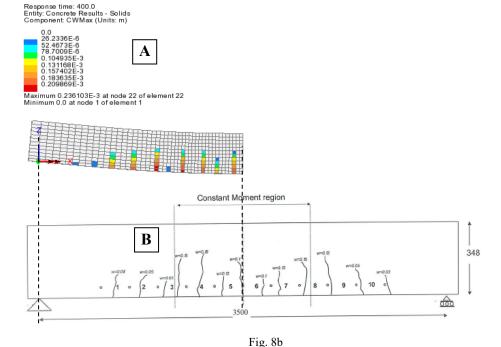


Fig. 7b

Test B1-b: Crack distribution and crack widths at time 14 days, i.e. at first loading A – LUSAS: CWmax=0.01 mm

B-Experiment; CWmax=0.05mm



Test B1-b: Crack distribution and crack widths at time 400 days:

A – LUSAS: CWmax=0.236 mm; No of cracks along half length -7.

B – Experiment; CWmax=0.18mm (at crack 4). No of cracks along full length – 13

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