Finite Element Analysis and Genetic Algorithm Optimization Design for the Actuator Placement on a Large Adaptive Structure

By

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A DISSERTATION SUBMITTED TO THE FACULTY OF VIRGINIA POLYTECHNIC INSTITUTE AND STATE UNIVERSITY IN PARTIAL FULFILLMENT OF THE REQUIREMENTS FOR THE DEGREE OF DOCTOR OF PHILOSOPHY

IN

AEROSPACE ENGINEERING

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December 9, 2004

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Keywords: Smart Structures, Finite Element Analysis, Genetic Algorithms, Multidisciplinary Design & Optimization, Shape Control, Piezoelectric Actuator Locations, Parallel Computing.

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(Abstract)

The dissertation focuses on one of the major research needs in the area of adaptive /intelligent/smart structures, the development and application of finite element analysis and genetic algorithms for optimal design of large-scale adaptive structures. We first review some basic concepts in finite element method and genetic algorithms, along with the research on smart structures. Then we propose a solution methodology for solving a critical problem in the design of a next generation of large-scale adaptive structures -- optimal placements of a large number of actuators to control thermal deformations. After briefly reviewing the three most frequently used general approaches to derive a finite element formulation, the dissertation presents techniques associated with general shell finite element analysis using flat triangular laminated composite elements. The element used here has three nodes and eighteen degrees of freedom and is obtained by combining a triangular membrane element and a triangular plate bending element. The element includes the coupling effect between membrane deformation and bending deformation. The membrane element is derived from the linear strain triangular element using Cook's transformation. The discrete Kirchhoff triangular (DKT) element is used as the plate bending element. For completeness, a complete derivation of the DKT is presented. Geometrically nonlinear finite element formulation is derived for the analysis of adaptive structures under the combined thermal and electrical loads. Next, we solve the optimization problems of placing a large number of piezoelectric

actuators to control thermal distortions in a large mirror in the presence of four different thermal loads. We then extend this to a multi-objective optimization problem of determining only one set of piezoelectric actuator locations that can be used to control the deformation in the same mirror under the action of any one of the four thermal loads. A series of genetic algorithms, GA Version 1, 2 and 3, were developed to find the optimal locations of piezoelectric actuators from the order of $10^{21} \sim 10^{56}$ candidate placements. Introducing a variable population approach, we improve the flexibility of selection operation in genetic algorithms. Incorporating mutation and hill climbing into micro-genetic algorithms, we are able to develop a more efficient genetic algorithm. Through extensive numerical experiments, we find that the design search space for the optimal placements of a large number of actuators is highly multi-modal and that the most distinct nature of genetic algorithms is their robustness. They give results that are random but with only a slight variability. The genetic algorithms can be used to get adequate solution using a limited number of evaluations. To get the highest quality solution, multiple runs including different random seed generators are necessary. The investigation time can be significantly reduced using a very coarse grain parallel computing. Overall, the methodology of using finite element analysis and genetic algorithm optimization provides a robust solution approach for the challenging problem of optimal placements of a large number of actuators in the design of next generation of adaptive structures.

Acknowledgments

I would like to acknowledge the technical guidance of Professor Rakesh K. Kapania. I am grateful to my former department head Professor Bernard Grossman, current department head Professor Robert W. Walters, and my advisor Professor Rakesh K. Kapania for their encouragement and support. I would also like to thank Professor Romesh C. Batra, Professor Raymond H. Plaut, Professor Alan Brown, and Professor Eric Johnson for serving on my advisory committee and their patience. The helpful comments from Professor Raphael T. Haftka at the University of Florida and other AIAA Journal reviewers are appreciated. The support through the National Science Foundation (NSF Award Number: 9714128, Principal Investigator: Professor Shiv P. Joshi at the University of Texas at Arlington) and the Carilion Biomedical Institute is greatly appreciated.

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Chapter 1. Introduction

Over the last two decades, the area of multidisciplinary adaptive/intelligent/smart structures and systems has been studied extensively under numerous programs sponsored by US Army, Air Force, Navy, Defense Agency, NASA, NSF, etc. Much of the work has been focused on shape control, vibration control, buckling control, acoustic reduction, aeroelastic control, aerodynamic and hydrodynamic flow control, optimization of lifting surfaces, structural health monitoring, active damping, new smart materials, sensors, actuators development, smart systems modeling, robust control techniques, and multidisciplinary optimization techniques. The systems range from next generation telescopes to antenna, fixed wing aircraft, rotary-wing helicopters, spacecraft, missiles, submarines, automobiles, civil structures, and medical systems.

The structure for our study is a thin hexagonal spherical primary mirror of the next generation astronomical telescope. The mirror has adaptability to its changing thermal environments through the control of distributed piezoelectric actuators. It is well-known that the effectiveness of the control system strongly depends on the actuator locations. However, the problem of the optimal placement of actuators is hard to solve, especially for a large number of actuators.

This dissertation is focused on solving this challenging problem. We propose a solution methodology, which combines the finite element analysis and genetic algorithms. In this chapter, we first review some basic concepts in finite element method and genetic algorithms, along with the research on smart structures. Then we present the objective and scope of present research.

1.1 Finite Element Method

The finite element method has become one of the universally adopted methods for the analysis of general structural and continuum problems. The finite element analysis in many applications has become both routine and essential in engineering design. This is an approximate method in which the structure or the continuum is divided into small pieces of various shapes, sizes and types which are then assembled together using compatibility of displacements and equilibrium of forces to form an approximate mathematical model. The finite element method is used extensively in Aerospace, Automobile, Civil, Naval, and other applications.

An approach similar to the finite element method, involving the use of piecewise continuous trial functions defined over triangular regions, was first suggested by Courant in 1943. The original formulation of the finite element method was presented by Turner, et al. in 1956. The name finite element was coined by Clough in 1960. The more information about the history of finite element method can be found in the references [Zienkiewicz and Taylor, 2000, Cook, et al., 2002, Yang, 1986, etc.].

1.1.1 Some basic Concepts of Finite Element Method.

Convergence

Requirements for the convergence are described in the following.

Continuity/compatibility requirement: If the nth-order derivative of a function occurs in the integral form then the function has to be such that its n-1st derivative is continuous (C_{n-1} continuity) within the elements and across the element boundaries. Physically, this requirement ensures that no gaps or overlaps occur between elements when the assemblage is performed. From a variational viewpoint, this requirement means that there is no strain energy gained or lost at the interface.

Completeness requirement: If the *n*th-order derivative of a function occurs in the integral form then the function has to be such that a constant value of the *n*th-order derivative is attainable in the element domain when the size of the element tends to zero. If the function is approximated by a polynomial this requirement means that the function has to be complete up to the *n*th order. In the displacement-based finite element method this requirement means that the displacement functions of the element must be able to represent the rigid body displacements and the constant strain states, including the condition of zero strain under rigid-body motion.

If each element in a model satisfies strain completeness (that is, assumed displacement functions contain zero strain under rigid body translations and rotations, and constant strain modes), and the model is a compatible one (internal compatibility, nodal compatibility, and inter-element boundary compatibility are all satisfied), the results will converge monotonically and will be a lower bound. On the other hand, if each element in a model satisfies stress completeness (that is, assumed stress functions contain zero stress and constant stress modes), and the model is in equilibrium (internal equilibrium, nodal equilibrium, and inter-element boundary equilibrium all satisfied), the results will converge monotonically and will be an upper bound. If the elements are complete but the model is a non-compatible or non-equilibrated one, the results may still converge but, in general, they will converge non-monotonically.

The completeness requirement for an individual element (element completeness test) is said to be satisfied if the constant strain/stress states and the rigid body modes are represented.

In the context of Kirchhoff/thin plate bending, the presence of the second derivatives of

the transverse deflection w in the internal work indicates that w should be C_1 continuous at element interface. This means that along element interfaces we must have continuity of w as well as that of $\frac{\partial w}{\partial t}$, the derivative of w normal to the interface. However, it is difficult to implement the C_1 compatibility. In fact, it is impossible to specify simple polynomial expressions for shape functions ensuring full C_1 compatibility when only w and its first derivatives are prescribed at corner nodes. In the context of Kirchhoff/thin plate bending, the completeness means that w has to be complete up to 2^{nd} order polynomial. Therefore, the terms 1, x, y, x^2 , xy, and y^2 must be included in the expression for w, which is capable of representing rigid body, constant slope, and constant curvature states of deformation. If the element satisfies the compatibility and completeness requirements we can be sure that successive solutions obtained from element mesh refinement will converge monotonically to the correct displacements and stresses. Note that C_1 continuity is not always a necessary condition for convergence. If such non-conforming elements satisfy the patch test then convergence will still occur. Experience indicates that convergence is more dependent on the completeness than on the compatibility of the element [Huebner, et al. 2001].

Similarly, it is easy to show that only C_0 continuity is required for the Reissner-Mindlin/thick plate. Though it is easy to implement C_0 continuity, there exists the so-called locking problem that is hard to solve for lower-order C_0 elements.

Patch test and benchmark/test problems

For the results for non-compatible or non-equilibrated models to converge, it is necessary that an assemblage of elements can also represent the constant strain or stress states. It must be realized that although an individual element may contain the constant strain or stress states an assemblage of the same elements may not do so for non-compatible or non-

equilibrium models. To check whether or not an assemblage of elements can represent the constant states the patch test was proposed by Irons [Zienkiewicz and Taylor, 2000]. The test consists of the following:

- 1. Assemble a group (patch) of elements with, at least, one internal node
- 2. Apply a system of forces or displacements and constraints to the external boundary nodes only which are consistent with a state of constant stress or strain
- 3. If the same constant state of stress or strain occurs in each element of the patch then it is assumed that any patch of the same elements will behave well in a general structure in a constant stress/strain field

The patch test is a completeness condition for an assemblage of elements as against an individual element. It has been found that the patch test may be passed by one kind of geometries or meshes but failed by the others. Therefore, it is important for the patch test to be performed on the different geometries or meshes before the element is claimed to have passed the patch test.

Since its first introduction by Irons [Bazeley, et al., 1966], the patch test has been analyzed, discussed, and reinterpreted at length [Zienkiewicz and Taylor, 1997]. The basic idea was that in the limit of decreasing mesh size the nonconforming finite element solution to the given problem posed on an arbitrary patch of elements should be able to represent the appropriate constant strain states (constant curvature for plate bending). A patch includes one or more finite elements which share some common interfaces. Note that in this definition even a single element constitutes a patch. Patch test is used to determine if a given nonconforming element (inter-element continuity was violated) is permissible. The

be wrong to leave the impression that the only interesting elements are those that pass the test. Patch test contains a single element patch and multi-element patch test in which at least one node is an internal node. Patch test tests both the consistency and stability of finite element equations [Zienkiewicz and Taylor, 2000].

The simple patch account immediately indicates which elements fail and which have a chance of success. The count condition $[n_{\theta} + n_{w} \ge n_{s}]$ and $n_{s} \ge n_{w}$, where n_{w} , n_{θ} , and n_{s} are the number of deflection, rotation and shear force variables, respectively. Zienkiewicz and Taylor, 2000] is necessary but not sufficient to define successful elements and numerical testing is always needed. The failure of the patch account test means that under some circumstances the element will fail. However, in many problems a reasonable performance can still be obtained and a non-singularity will be observed in the performance, provided consistency is also satisfied.

The accuracy of the finite elements is of primary concern. A standard set of problems to test finite element accuracy of shell elements was proposed [MacNeal and Harder, 1985].

Spurious modes/rank deficiencies and Locking problems

The problems occur for the low order pure displacement finite elements if additional constraints need to be satisfied. In the extreme case these constraints may restrict the solution to such a degree that only a trivial solution will be obtained. In the other case locking effects may be present, which means that the finite element solution will ultimately converge, but the rate of convergence is drastically reduced so that the computational costs become unacceptable. Locking results from the relative inability of the finite element space to approximate the exact solution which is constrained in the analysis of beams, plates and shells or incompressible media where the poisson's ratio approaches 0.5.

Mindlin elements can exhibit spurious transverse shear stiffness (or locking) as the plate become thin. The spurious transverse shear stiffness arises from an interpolation inconsistency that the Kirchhoff condition from being satisfied as the plate becomes thin. The spurious shear stiffness can be alleviated by using a reduced/selective integration to compute certain terms in the element stiffness matrix, or by using higher-order elements. Although the reduced integration solution is the most economical alternative, such inexact integration can result in an element stiffness matrix rank deficiencies that have an excess number of zero eigenvalues and the corresponding spurious displacement modes (i.e. deformation modes that result in zero strain at the Gauss integration points) [Hughes, 1977]. These spurious deformation modes can be superimposed on the true displacement solution and make the solution meaningless. Thus, the elements with spurious modes must be used with care.

Locking can also result from the element distortions. Both membrane locking and shear locking can occur in the shell finite element analysis. The locking phenomenon can be alleviated by using higher-order elements(Lagrangian, not serendipity), extremely fine mesh, reduced and selective integration—whether the rank of the element stiffness matrix is less than (number of nodal degrees of freedom) minus (feasible rigid-body motions), which tests whether a single element can have spurious zero energy modes, assumed strain method, enhanced assumed strain methods, mixed interpolation, linked interpolation, patch account test, drilling degrees of freedom (vertex rotations), robust element, hybrid/mixed finite element formulation and higher order shear deformation theories.

Singularities

A singularity may occur with the mesh refinement at some points such as the obtuse corners of the Morley skew plate [Huebner, et al., 2001]. The finite element solution

converges to the solution of corresponding mathematical model. Theoretically, what is not allowed [Hartmann and Katz, 2004]:

Kirchhoff plate: single moments;

Reissner-Mindlin plate: point loads, single moments;

Plates (2-D elasticity): point loads;

Elastic solids (3-D elasticity): point loads, line loads.

At these points the finite element solution for displacements and stresses will increase without limit with the mesh refinement. Of course, a concentrated force is a convenient mathematical tool. All real forces are distributed over an area greater than zero. The easiest way to apply a concentrated force at a prescribed location is to arrange the FE mesh so that a node appears at this location. A concentrated moment load cannot be applied at a node unless the node includes the appropriate rotational d.o.f. If nodes have only translational d.o.f., a moment must be applied as couple-forces on a pair of nodes [Cook, 2002].

1.1.2 Advances of Plate and Shell Finite Elements

Plates and Shells are a particular form of a three dimensional solid. The thickness of such structures is very small when compared with other dimensions, and complete three-dimensional treatment is not only costly but often leads to serious numerical ill-conditioning problems. Although the finite element analysis for plate and shell structures now spans over forty years, the establishment of reliable and efficient finite element models continues to be the subject of research effort at the present. An enormous amount of effort has been devoted to the development of finite elements for the plate and shell analysis. The literature is extensive and we will not make an attempt to review it here. Review of the many contributions on the subject is given in a number of survey papers and books [Yang, et al.,

1990, 2000, MacNeal, 1998, Zienkiewicz and Taylor, 2000, etc.].

In 1965, Irons and Draper noted that an expression for deflection w which assures uniqueness of the bending curvatures over the surface can not assure slope $w_{,x}$ and $w_{,y}$ continuity along the common edges of adjacent elements when only w, $w_{,x}$ and $w_{,y}$ are prescribed at nodes. Therefore, it is impossible to specify simple polynomial expressions for shape functions ensuring full C^1 continuity when only deflection and its slopes are prescribed at corner nodes [Zienkiewicz and Taylor, 2000]. However, a full conforming rectangular element with w, $w_{,x}$, $w_{,y}$ and $w_{,xy}$ prescribed at the corner nodes has been successfully developed by Bogner, et al. in 1966.

The hybrid and mixed finite elements were first studied by T. H. H. Pian in 1964. Although the motivation for the early developments in hybrid and mixed finite elements was to avoid the difficulty in C^1 continuity problem, such elements are found to be effective in plate and shell bending problems, incompressible or nearly incompressible materials, crack, elastic-plastic and creep problems.

The Kirchhoff thin plate and shell theory is very widely used in practice and proves adequate for a large number of structural problems. The theory is based on the following assumptions:

- (i) Cross-sections normal to the middle plane remain plane during the deformation;
- (ii) The direct strain in the normal direction is zero;
- (iii) Plates and Shells are assumed to be in plane stress state;
- (iv) The normal to the middle plane remain normal to it during deformation.

 It is obvious that the assumptions (ii) and (iii) are contradictory from the perspective of theory of elasticity. However, this inconsistency in approximation is acceptable for most

engineering applications.

By removing the fourth assumption, we get the Reissner-Mindlin thick plates and shells. By replacing the assumptions (i) and (ii) with the assumption that the in-plane displacements be expanded as polynomial of z in the normal direction, we get the higher order shear deformation plate and shell theories.

By removing all of the assumptions and assuming that the in-plane and lateral displacements be expanded as polynomial of z in the normal direction, we get the so-called higher order shear deformation plate and shell theories that consider transverse shear deformation and transverse normal strain. These are actually three dimensional models.

The membrane elements with drilling d.o.f. have been considerably improved since Allman published his original work in 1984 [Allman, 1988, Fellipa, 2003].

The concept of discrete Kirchhoff constraints was first introduced by Wempner et al., Stricklin et al., and Dhatt in 1968-69. An interesting study of three-node plate bending finite elements with nine degrees of freedom was presented by Batoz et al. in 1980 where it was concluded that the discrete Kirchhoff triangular element (DKT) was the most reliable and efficient. The displacements of this element are cubic polynomials defined only along the sides and the independent rotations are quadratic over the triangle. The Kirchhoff condition of zero transverse shear strain is employed to link the displacements and rotations at certain points of the element only. A new explanation of the DKT element was presented by Zienkiewicz, et al. in the context of mixed formulations of Reissner-Mindlin thick plate elements using collocation constraints [Zienkiewicz and Taylor, 2000].

The potential of the conventional six-node shell finite element for practical application is limited, however, because the element does not match with other important elements such as

standard edge beams; modeling stiffened shell structures like airframes therefore presents serious problems. The practical requirement for general thin shells is a three-node triangular finite element with the complete six degrees of freedom at each vertex node, namely: three displacement components and three rotational components. This matches with other standard elements.

The equivalence of certain mixed finite element methods with displacement methods employing the reduced and selective integration was presented by Malkus and Hughes in 1978.

Various elements, presented by Bogner, et al., 1966, Stricklin, et. al, 1969, Dhatt, 1969, Specht, 1988, etc., for comparison of performance for the thin plate analysis can be found in Table 4.3 in the reference [Zienkiewicz and Taylor, 2000].

An effective shell finite element must be applicable to membrane- and bending-dominated shell problems. Ideally, the element shows optimal convergence characteristics in both problem areas. Elements formulated using pure displacement interpolations are effective in membrane-dominated situations but lock when bending is encountered. To arrive at a general shell finite element discretization, a mixed formulation is usually used. For the analysis of general shell problems, the MITC (mixed interpolation of tensorial components) shell elements are shown to be effective [Bathe, 1996].

The derivation of finite elements has evolved from direct stiffness method to variational methods (Ritz method, Lagrange mutiplier, general variational principles) to weak forms of equilibrium equations (Galerkin weighted residual methods -- generalized finite element method). However, the fact that a finite element can be derived from a variational method or Galerkin weighted residual methods does not necessarily mean that the element is reliable.

The element shape functions should be capable of reproducing basic conditions of rigid body movement and constant strain states, as dictated by the patch test.

To get reliable FEA results, it is important to choose an appropriate mathematical model, appropriate elements, and appropriate meshes. It is helpful to note that no single element is likely to be best for all applications.

1.2 Genetic Algorithms

Genetic algorithms belong to evolutionary computation. The evolutionary computation includes evolution strategies (Rechenburg, I., 1965), genetic algorithms (Holland, J., 1992, and Goldberg, D., 1989), evolutionary programming (Fogel, L., et al., 1966) and genetic programming (Koza, J., et al., 1999). Other adaptive search algorithms include simulated annealing (Kirkpatrick, S., et al., 1983), tabu search (Glover, F., 1997). Genetic algorithms (GAs) are defined as search procedures based on the mechanics of natural selection and genetics. Genetic algorithms as an optimization method have been applied to a number of fields. The general procedure of genetic algorithms is as follows.

GA procedure

The first step we must perform to apply a genetic algorithm to a specific problem is to encode it as an artificial chromosome or chromosomes. These artificial chromosomes can be strings of 1s and 0s, parameter lists, permutation codes, real number, etc.

The second step that we must perform in solving a problem is to have some means or a procedure for discriminating good solutions from a bad solution. This is usually done by choosing a proper fitness function in the GA context, which is called an objective function in the optimization context. The function will be used by the genetic algorithm to guide the evolution of future generations.

After encoding the problem and choosing fitness function, we create an initial population of encoded solutions. The population can be created randomly or by using prior knowledge of possibly good solutions, but either way a key idea is that the GA will search from a population, not a single point.

After creating an initial population, we apply genetic operators to process the population iteratively to create a sequence of populations that hopefully will contain more and more good solutions as time goes on. There is a wide variety of the types of operators that are used in GAs, but quite often selection, crossover, and mutation are used.

Selection is based on the survival of the fittest principle of natural evolution. It can be accomplished in various ways, including weighted roulette wheels, local tournaments, various ranking schemes, etc. The key idea is to prefer better solutions to worse ones.

Crossover is a genetic operator that exchange bits and pieces of parental solutions to form new, possibly better offspring. Again, there are many ways of accomplishing this, including one point, two points, multiple points, uniform crossover, etc. The key idea is that the offspring will not be identical to any particular parent and will instead combine parental traits in a novel manner.

Mutation acts by simply modifying a single individual. There are many variations of mutation, but the key idea is that the offspring be identical to the parental individual except that one or more changes are made to an individual's trait.

Selection is a stochastic process where better solutions have a greater probability of survival and duplication. However, there generally remains a probability that some low quality of solutions may also pass through to the next generation. The reason is that such solutions may also contain information that could be of benefit at a later stage. A random

mutation is also introduced to perform small numbers of change to individuals. This supports diversity and exploration whilst preventing premature convergence of the system.

1.3 Adaptive/Intelligent/Smart Structures

Although the classification of adaptive structures was suggested in a paper 1990 [Wada, et al., 1990] the following use of these concepts in the literature doesn't seem to strictly adhere to it. Instead, the three concepts of adaptive/intelligent/smart structures are often used interchangeably. Similarly do the terms of adaptive/intelligent/smart materials. In the dissertation we accepted the conventional use. The smart material generally designates a material that changes one or more of its properties in response to an external stimulus. The most popular smart materials are piezoelectric materials, electrostrictive materials, magnetostrictive materials, shape memory alloys, electrorheological fluids, and optical fibers. An adaptive/intelligent/smart structures can be viewed as a structure or system that is capable of a corresponding desired response to the changing external or internal environment such as external loads change or internal damage, etc. Thus, an adaptive structure is an integration of structure, sensors, actuators, and control systems. It is an emerging multidisciplinary area. The development of such structures depends on the advances of structures and materials, sensors and actuators, power electronics, digital signal processors, microprocessors, communication, and high performance computing technologies.

Numerous applications of smart-structures technology exist obviously such as shape control, vibration control, noise control, aeroelastic control, and health monitoring.

Applications range from space systems to fixed-wing and rotary wing aircraft, automotive, civil structures, machine tools, and medical systems. Much of the early development of smart-structures methodology was driven by space application such as vibration and shape

control of large flexible space structures, but now wider applications are envisaged for aeronautical and other systems. Embedded or surface-bonded smart actuators on an airplane wing or a helicopter blade, for example, can induce airfoil twist/camber change that in turn can cause a variation of lift distribution and can help to control static and dynamic aeroelastic problems.

1.3.1 Piezoelectric Actuators and Sensors

Piezoelectric materials are the most popular smart materials. The typical piezoelectric materials include certain crystals such as quartz crystal and Rochelle salt, piezoelectric ceramics such as lead zirconate titanate (PZT), and piezoelectric polymer such as polyvinylidene fluoride (PVDF). Piezoelectricity, literally pressure electricity, is a property of these materials that develop a voltage when pressure is applied, and vice versa, produce deformation when a voltage is applied. The former is called the direct piezoelectric effect; the later is called the converse piezoelectric effect. The brothers Pierre and Jacques Curie discovered the direct piezoelectric effect in 1880 while studying crystals compressed in particular directions. The converse piezoelectric effect was predicted by the Gabriel Lippmann in 1881 and confirmed by the Curies that same year [Cady, 1964]. The direct and converse piezoelectric effects form a basis for the use of a piezoelectric material as a sensor and actuator, respectively. The common uses of piezoelectricity include in the devices such as electromechanical transducers, accelerometers, phonograph pickups, loudspeakers, sonar transmitters and receivers, and quartz crystal oscillators in clocks, watches, microprocessors, etc.

Piezoelectricity occurs naturally in some crystalline materials, such as quartz crystal and Rochelle salt. Piezoceramics are polycrystalline in nature and do not have piezoelectric

characteristics in their original state. Piezoelectric effects are induced in these materials through simple poling (application of high dc electric field at elevated temperatures results in polarization). The most commonly used piezoceramics is lead zirconate titanate (PZT). These are solid solutions of lead zirconate and lead titanate, often doped with other elements to obtain specific properties. These ceramics are manufactured by mixing together proportional amounts of lead, zirconium, and titanium oxide powers and heating the mixture to around 800–1000°C. They then react to form PZT perovskite aggregates, which are then milled into the PZT perovskite powder. This powder is mixed with a binder and sintered into the desired shape. During the cooling process, the material undergoes a paraelectric to ferroelectric phase transition at the Curie temperature and the cubic unit cell becomes tetragonal. As a result, the unit cell becomes elongated in one direction and yields an electric dipole. Due to randomly oriented ferroelectric domains, the unpoled ceramic has no net polarization [Chopra, 2002, Jordan and Ounaies, 2002].

Application of high electric field at elevated temperatures aligns most of the tetragonal domains in such a way that the polar axes of unit cells are oriented mostly parallel to the applied field. This process is called poling, and it imparts a permanent net polarization and mechanical deformation to the ceramic. Poling is performed in a silicon oil bath at an elevated temperature under a DC electric field of 1 to 3kV/mm.

Piezoceramic materials are linear at low electric fields and low mechanical stress levels; they show considerable nonlinearity at high values of electric field and mechanical stress.

The piezoceramic material itself is mechanically isotropic, and by virtue of the poling process, is assumed transversely isotropic in the plane normal to the poling direction.

The most commonly used piezoelectric polymer is polyvinylidene fluoride (PVDF). The

PVDF consists of long chains of the repeating monomer ($-CH_2 - CF_2 -$). The hydrogen atoms are positively charged and the fluorine atoms are negatively charged with respect to the carbon atoms and this leaves each monomer unit with an inherent dipole. PVDF film is manufactured by solidification of the film from a molten phase, which is then stretched in a particular direction and finally poled through the application of a strong electric field. In the liquid form the polymer has no net polarization because of randomly oriented polymer chains. After solidification, and stretching the film in one direction, the polymer chains are mostly aligned along the direction of stretching. This, combined with poling, imparts a net polarization to the film [Harrison and Ounaies, 2002].

The process of stretching the film, which orients the polymer chains in a specific direction, renders the material piezoelectrically orthotropic. The stretching direction is taken as the 1-direction.

The actuation strain can be modeled like an equivalent thermal strain. The constitutive relations are based on the assumption that the total strain in the actuator is the sum of the mechanical strain induced by the stress, the thermal strain caused by temperature, and the controllable actuation strain caused by the electric voltage. Assume that the axis 3 is in the direction of the initial polarization. For linear analysis, the piezoelectric actuator equations (the converse effect) are as follows,

$$\begin{cases} \mathcal{E}_{1} \\ \mathcal{E}_{2} \\ \mathcal{E}_{3} \\ \gamma_{23} \\ \gamma_{12} \end{cases} = \begin{bmatrix} S_{11} & S_{12} & S_{13} & 0 & 0 & 0 \\ S_{12} & S_{11} & S_{13} & 0 & 0 & 0 \\ S_{13} & S_{13} & S_{33} & 0 & 0 & 0 \\ 0 & 0 & 0 & S_{44} & 0 & 0 \\ 0 & 0 & 0 & 0 & S_{55} & 0 \\ 0 & 0 & 0 & 0 & S_{66} \end{bmatrix} \begin{bmatrix} \sigma_{1} \\ \sigma_{2} \\ \sigma_{3} \\ \tau_{23} \\ \tau_{12} \end{cases} + \begin{bmatrix} 0 & 0 & d_{31} \\ 0 & 0 & d_{32} \\ 0 & 0 & d_{33} \\ 0 & d_{24} & 0 \\ d_{15} & 0 & 0 \\ 0 & 0 & 0 \end{bmatrix} \begin{bmatrix} E_{1} \\ E_{2} \\ E_{3} \end{cases} + \begin{bmatrix} \alpha_{1} \\ \alpha_{2} \\ \alpha_{3} \\ 0 \\ 0 \\ 0 \end{bmatrix} \Delta T$$

or

$$\begin{cases} \sigma_1 \\ \sigma_2 \\ \sigma_3 \\ \tau_{23} \\ \tau_{31} \\ \tau_{12} \end{cases} = \begin{vmatrix} C_{11} & C_{12} & C_{13} & 0 & 0 & 0 \\ C_{12} & C_{22} & C_{23} & 0 & 0 & 0 \\ C_{13} & C_{23} & C_{33} & 0 & 0 & 0 \\ 0 & 0 & 0 & C_{44} & 0 & 0 \\ 0 & 0 & 0 & 0 & C_{55} & 0 \\ 0 & 0 & 0 & 0 & 0 & C_{66} \end{vmatrix} \begin{pmatrix} \varepsilon_1 \\ \varepsilon_2 \\ \varepsilon_3 \\ \gamma_{23} \\ \gamma_{31} \\ \gamma_{12} \end{pmatrix} - \begin{pmatrix} 0 & 0 & e_{31} \\ 0 & 0 & e_{32} \\ 0 & 0 & e_{33} \\ 0 & e_{24} & 0 \\ e_{15} & 0 & 0 \\ 0 & 0 & 0 & 0 \end{pmatrix} \begin{pmatrix} E_1 \\ \lambda_2 \\ E_3 \end{pmatrix} - \begin{pmatrix} \lambda_1 \\ \lambda_2 \\ \lambda_3 \\ 0 \\ 0 \\ 0 \end{pmatrix} \Delta T$$

The piezoelectric sensor equations (the direct effect) are as follows,

$$\begin{cases} D_1 \\ D_2 \\ D_3 \end{cases} = \begin{bmatrix} 0 & 0 & 0 & 0 & d_{15} & 0 \\ 0 & 0 & 0 & d_{24} & 0 & 0 \\ d_{31} & d_{32} & d_{33} & 0 & 0 & 0 \end{bmatrix} \begin{bmatrix} \sigma_1 \\ \sigma_2 \\ \sigma_3 \\ \tau_{23} \\ \tau_{31} \\ \tau_{12} \end{bmatrix} + \begin{bmatrix} e_{11}^{\sigma} & 0 & 0 \\ 0 & e_{22}^{\sigma} & 0 \\ 0 & 0 & e_{33}^{\sigma} \end{bmatrix} \begin{bmatrix} E_1 \\ E_2 \\ E_3 \end{bmatrix} + \begin{bmatrix} a_1 \\ a_2 \\ a_3 \end{bmatrix} \Delta T$$

where D_i is the electric displacement (charge per unit area)in coulombs/square meter, d_{ij} is the piezoelectric coefficients in meters/volt or coulombs/Newton, (e_{ij} is also the piezoelectric coefficients), e_{ij}^{σ} is the dielectric permittivity in newtons/square volt or farads/meter, S_{ij} is the elastic compliance matrix, C_{ij} is the stiffness matrix, α_i is the thermal coefficient in 1/degrees Kelvin, (λ_i is also the thermal coefficients), a_i is the pyroelectric constants in newtons/volt-meter-degrees Kelvin, E_i is the applied electric field (V/t) in volts/meter, ΔT is the temperature change. For piezoelectric ceramics such as PZT, $d_{31} = d_{32}$ and $d_{15} = d_{24}$; for piezoelectric polymer such as PVDF, $d_{31} \neq d_{32}$ and $d_{15} \neq d_{24}$. Normally, the converse effect is used to determine piezoelectric coefficients.

The maximum actuation strain of the PZT and PVDF is about 1000 microstrain and 700 microstrain, respectively. Most recently, strain levels more than 1.2% have been reported for the lead zirconate niobate and lead titanate (PZN-PT).

PVDF is flexible as opposed to PZT which is usually brittle. Most commonly used

sensors are piezofilm (PVDF) because of PVDF's low stiffness. Sometimes, piezoceramic (PZT) sensors are used for specific applications. It has been found that the performance of piezoelectric sensors surpasses that of conventional foil strain gages, with much less signal conditioning required, especially in applications involving low strain levels, and high noise levels.

It is not advisable to use piezoelectric sensors to measure strain above 150 microstrains because nonlinearities and change in material properties, especially d_{31} , with stress will affect the accuracy of the calibration. PVDF sensors are relatively sensitive to temperature compared with PZT sensor, and suitable temperature compensation must be included in measurements.

1.3.2 Advances in Smart Structures

The area of adaptive/intelligent/smart structures has received significant attention since the late in 1980s. Numerous conferences and journals are dedicated to this high potential area for many technological revolutions. The literature in this multidisciplinary area has experienced tremendous growth. The comprehensive review of this field can be find in several excellent papers [Crawley, 1994, Sater, et. al., 2000, Chopra, 2002, etc.].

Crawley and de Luis in 1987 compared a variety of piezoelectric materials for possible use as actuators. A number of factors were used in the comparison, including embedability in composites, ratio of strain to applied voltage, and a performance criteria derived by maximizing the actuator effectiveness. The superiority of piezoelectric ceramics over other piezoelectric materials such as polymer film was conclusively demonstrated by this comparison.

A coupled thermal-piezoelectric-mechanical model was developed by Chattopadhyay, et

al. in 1999. A higher-order (third-order) shear deformation was used to capture the transverse shear effects in the moderately thick composite laminates. Numerical results show that the coupling could have significant effects on the static and dynamic response of some piezoelectric composite plates.

A completely coupled thermal-piezoelectric-mechanical theory, based on the improved layer wise displacement field and higher order electrical and temperature fields, was developed to study dynamic responses and control effects of smart composite shells [Chattopadhyay, et al., 2002]. The theory was implemented using a finite element technique that ensures the application to practical geometry and boundary conditions. Numerical analysis was conducted for simply-supported cylindrical shells with distributed self-sensing piezoelectric actuators. Comparisons were made with three dimensional NASTRAN solutions to verify the accuracy of the developed theory. Control authority was investigated using Linear Quadratic Gaussian (LQG) controller. Parametric studies were conducted to investigate the effects of the two-way coupling, actuator location and bending-torsion coupling and flexibility of the host structure.

Inflatable structures have special properties such as lightweight, low deflated volume, and high strength to-mass ratio. These remarkable properties make them suitable for cost-effective large space applications such as large solar antennas and optical mirrors. Finding optimal actuators and sensors is particularly important for these types of structures, as their vibration control and sensing require a large number of actuators and sensors, and their mode shapes are relatively complicated. Based on actuator and sensor performance indices, optimal locations and sizes of 5 actuators and 5 sensors for the vibration control of an inflated torus were determined by genetic algorithms so that the actuators and sensors provide good control

and sensing authorities in the considered modes. The simulation study indicated that actuators and sensors made of PVDF material can be used in the vibration control of an inflated torus [Jha and Inman, 2003].

Numerous analytical and computational models for linear static and dynamic response of smart structures using various theories was reviewed by Saravanos and Heyliger in 1999.

Nonlinear finite element formulation of piezoelectric laminated plates undergoing large displacements and rotations was presented by Varelis and Saravanos in 2004. The coupled model uses both displacements and electric potential as field variables. The nonlinear static response of piezoelectric laminate under mechanical and electric loads was investigated by an eight-node element. The significance of the nonlinear effects in piezoelectric adaptive structures caused by large deformation was illustrated by the computational results.

The Defense Advanced Research Projects Agency (DARPA) has supported several programs with a focus on smart materials and structures such as Smart Rotor Program, Smart Wing program, and Smart Aircraft and Marine Propulsion System demonstration (SAMPSON) program. The primary focus of these programs has been to apply existing smart materials in an appropriate device form to reduce noise and vibration and to achieve aerodynamic and hydrodynamic flow control in a variety of structures. Achievement of the program objectives will potentially create paradigm shifts for the design of undersea vehicles, helicopter rotor blades, aircraft wings, and engine inlets [Sanders, et al., 2004].

The Smart Rotor Program, managed by the Boeing Company in Philadelphia,

Pennsylvania, aimed to achieve active, real-time blade tracking adjustments through

individual blade control by using smart materials and structures technology. The

DARPA/AFRL/NASA Smart Wing program, performed by Northrop Grumman Corporation,

investigated techniques for wing twisting and camber control using smart structures. The overall objective of the smart wing program was to develop and demonstrate smart materials based control surface to improve the aerodynamic and aero-elastic performance of military aircraft. The demonstration of high actuation rate, large deformation, hinge-less, smoothly contoured control surfaces with chord-wise and span-wise shape variability represent an important milestone in the development and application of smart materials and structures technologies.

The SAMPSON program, conducted by the Boeing Company, Lockheed Martin Astronautics, General Dynamics, etc., is focused on both aircraft and undersea vehicle applications, with a specific interest in inlet shape control by applying smart materials based solutions.

More recently, the DARPA initiated another program -- Morphing air structures program. The program aims to create and advance enabling technologies and incorporate them into the design, build, and demonstration of a seamless, aero-efficient, radically shape changing vehicle during operation. A morphing wing is designed to drastically change wing geometric parameters such as aspect ratio, area, and twist to achieve truly multi-mission.

The National Aeronautics and Space Agency (NASA) Morphing project, led by the Langley Research Center, aims to develop and assess advanced technologies and integrated component concepts to enable efficient, multi-point adaptability in air and space vehicles. The three focus areas of the project are: adaptive structural morphing, micro-aero adaptive control, and biologically-inspired flight systems. These areas are supported by the core enabling areas of smart, nano and biologically-inspired materials, multi-disciplinary optimization, controls and electronics. An overview of NASA's Morphing Project as well as

highlights of recent results can be found in the reference [McGowan, et al., 2002].

1.4 Objective and Scope of Present Research

The dissertation is focused on one of the major research needs in the area of adaptive /intelligent/smart structures, that is, the development of finite element analysis and genetic algorithms optimization design for optimal placement of a large number of piezoelectric actuators on a thin hexagonal spherical primary mirror of the next generation astronomical telescope. First, we systematically derive and document the general flat triangular composite shell element under thermal and electric loads. Then we develop the genetic algorithms and combine the finite element analysis and the genetic algorithms for the optimization of piezoelectric actuator locations. Finally, we perform extensive experiments on genetic algorithms for the optimization of piezoelectric actuator locations through coarse-grain parallel computation to best understand genetic algorithms and to obtain the best design of smart structures. The following is the scope of remaining chapters.

In Chapter 2, after briefly reviewing the three most frequently used general approaches to derive the finite element formulation, we systematically present the key techniques associated with general shell finite element analysis using the flat shell elements by a combination of membrane element and plate bending element. The membrane element is obtained using the Cook's transformation from linear strain triangular element. It is clearly demonstrated that such an element is indeed the same as the Allman triangular element. The discrete Kirchhoff triangular element is chosen as plate bending element. Because of the high efficiency and accuracy of DKT element for thin plate bending analysis but lack of document for its derivation we present the detail derivation for DKT element. We present the geometrically nonlinear finite element formulation for general shell structures under the

combination of thermal and electrical loads.

In Chapter 3, we solve the two kinds of optimization problems of piezoelectric actuator locations in case of selecting 30 actuators from 193 candidate locations using our proposed methodology – combination of finite element analysis as an analyzer and genetic algorithms as an optimizer. We first find the optimal locations and optimal voltages for correcting thermal distortions caused by only one type of thermal loads. Then, we extend the optimization problems to the multi-objective optimization problems, that is, to determine just one set of actuator location for correcting thermal distortions caused by four different types of thermal loads. The advantages and disadvantage of genetic algorithms are presented. The genetic algorithms used in this chapter is GA version 2, which is developed from GA version 1 (see Appendix A) by mainly changing the mechanism of preserving the best solution and introducing the variable population in the genetic algorithms.

In Chapter 4, the more effective genetic algorithms, GA version 3, is developed from GA version 2 by mainly incorporating mutation and hill-climbing into micro-genetic algorithms. Using GA version 3, we resolve the two kinds of optimization problems studied in Chapter 3 and even larger problems of selecting 121 piezoelectric actuators from 193 candidate locations.

In Chapter 5, extensive numerical experiments are conducted on the multi-objective problems of the optimization of piezoelectric actuator locations using course grain parallel computing. The genetic algorithms used in this chapter is basically the same as the GA version 3 in Chapter 4 except we get results faster by reducing file operations. The time of investigation are significantly reduced by running multiple jobs at the same time on a Sun machine with 17 processors.

The major contributions, perspective and future directions are briefly summarized in Chapter 6. The feasibility investigation of using finite element analysis and genetic algorithms GA version 1 to optimize the piezoelectric actuator location is presented in Appendix A.

Chapter 2. Finite Element Formulation

Basically, there are three most frequently used general approaches to derive the finite element formulation. The first approach is called the Galerkin weighted residual method, in which the original shape functions of the finite element are used as weights in the integral form or weak form of the differential equations (Not necessarily. The test functions may or may not be same as trial functions). The weak form is obtained by performing the integration by part of the integral form of the differential equations, obtained by multiplying the differential equations by so-called test functions, and integrating the resulting equations over the domain of the problem. The test functions generally satisfy the homogeneous part of the essential boundary condition. The integral form or the weak form forms the basis of finite element discretization.

The second approach is using the principle of virtual work i.e. the principle of virtual displacements, or alternatively the variational principle of total potential energy, either of which can be regarded as the basis of most frequently used displacement-based finite element method. Using the Galerkin procedure, it is easy to show that the principle of virtual work is precisely the weak form of the equilibrium equations. Because the variation of displacements can be regarded as virtual displacements, it is also easy to show that for conservative systems using the variational principle of minimum total potential energy is equivalent to using the principle of virtual work. It is important to note that the principle of virtual work is not same as the principle of total potential energy and that the principle of virtual work is more universal than the principle of total potential energy.

The last approach of the derivation of the finite element formulation is using general variational principles such as the Hu-Washizu variational principle, which can be derived

from the variational principle of total potential energy using Lagrange multiplier method. The Hu-Washizu variational principle can be regard as a generalization of the principle of virtual displacements, in which variations are performed on all unknown displacements, strains, stresses, and unknown surface tractions. The Hu-Washizu variational principle and its derivatives such as Hellinger-Reissner variational principles form the basis of mixed finite element methods, in which the strains and stresses can also be used as the unknown variables as displacements.

The mixed finite element formulations derived using either Galerkin weighted residual method from reducible differential equations or general variational principles are found to be effective in the analysis of plates and shells and incompressible media. The first approach -Galerkin weighted residual method has found wide applications in heat transfer, fluid mechanics, various field problems and also in the analysis of plates and shells and incompressible media. The last two approaches have been extensively used in solid mechanics. Using the natural variational principle will always produce a symmetric stiffness matrix for linear problems and in these cases Galerkin and variational procedures will produce the same results. It is well known that the natural variational principles do not exist for all the physical problems in which well-defined differential equations may be formulated. In this sense it seems that the Galerkin process has a greater range of applicability. However, there is another category of variational principles which can always be constructed for any set of differential equations by the Lagrange multiplier method. Therefore, we can derive the finite element formulation using either Galerkin procedure or variational procedure, and which procedure to choose depends on which one is more convenient for the specific problem or user's preference.

In this chapter, we will use the second approach, that is, the principle of virtual work, to derive the finite element formulation. It is important to note that for all of the approaches the key step is the assumption of the solution in the form of trial functions.

2.1 Galerkin weighted residual method, principle of virtual work and general variational principles

Galerkin weighted residual method

Given a set of differential equations of a problem A(u) = 0 in V, the integral form of the differential equations is

$$\int_{V} V^{T} A(u) dV = 0 \tag{2.1a}$$

where v is a set of arbitrary functions (called test functions) satisfying the homogeneous essential boundary conditions and u is chosen to satisfy the essential boundary conditions. Integrating the left side of the equations by parts (see Appendix B) and applying the boundary conditions we can get the weak form of the differential equations

$$\int_{V} C(v)^{T} D(u) dV + \int_{S_{n}} E(v)^{T} F(u) dS = 0$$
(2.1b)

where S_p is that part of the boundary on which natural or force boundary conditions are specified. In eqn. (2.1b), a lower order of continuity is required in the choice of the function used to express u. This is achieved at a price of requiring higher continuity for the v function (the test function).

The finite element approximation u is expressed as

$$u = \sum_{i=1}^{n} N_i a_i = Na \tag{2.2}$$

where N_i are shape functions that assume value 1 at the node corresponding to the degree i

and 0 at the other nodes, and a_i are the nodal parameters. In Galerkin weighted residual method, we substitute v in the form of $v = \sum_{j=1}^{n} N_j \delta a_j$ into the integral form or weak form

and because δa_j are arbitrary (except $\delta a_j = 0$ on the essential boundary S_u), we have

$$\int_{V} N_{j}^{T} A(Na) dV = 0 \qquad j = 1 \text{ to } n$$
(2.3a)

or

$$\int_{V} C(N_{j})^{T} D(Na) dV + \int_{S_{p}} E(N_{j})^{T} F(Na) dS = 0 j = 1 \text{ to } n$$
 (2.3b)

Principle of Virtual Work (Principle of Virtual displacement)

In the following presentation, the symbol \hat{o} represents the partial differential operator on displacements u to obtain the strains ε as,

$$\partial = \begin{bmatrix} \frac{\partial}{\partial x} & 0 & 0 \\ 0 & \frac{\partial}{\partial y} & 0 \\ 0 & 0 & \frac{\partial}{\partial z} \\ 0 & \frac{\partial}{\partial z} & \frac{\partial}{\partial y} \\ \frac{\partial}{\partial z} & 0 & \frac{\partial}{\partial x} \\ \frac{\partial}{\partial y} & \frac{\partial}{\partial x} & 0 \end{bmatrix}$$

For any kinematically compatible deformation, that is, $\delta u = 0$ on S_u and $\delta \varepsilon = \partial \delta u$ in V, where δu is a variation of displacement u and may represent a virtual displacement field from a given equilibrium configuration, the necessary and sufficient condition for equilibrium is

$$\delta W_i = \delta W_e \tag{2.4}$$

where δW_i and δW_e are internal virtual work and external virtual work, respectively,

$$\delta W_{i} = \int_{V} \delta \varepsilon^{T} \sigma dV$$

$$\delta W_e = \int_V \delta u^T f dV + \int_{S_p} \delta u^T p dS$$

where f is body force in V and p is surface force on the natural boundary S_p . Most significant is the fact that the principle of virtual work is independent of the material constitutive relation and applies to any materials, linear or nonlinear. Similarly, it is applicable to structures with follower forces.

General Variational Principles

The Hu-Washizu variational principle is one of the most general unconstrained variational principles in which independent variations are performed on all unknown quantities namely displacements, strains, stresses, and the surface traction. The significance of Lagrange multiplier method is that it converts a constrained variational principle to an unconstrained variational principle. The principle of virtual work is a constrained variational principle because the virtual displacements and virtual strains must be compatible, that is, $\delta u = 0$ on S_u , the boundary where essential boundary conditions $u = \overline{u}$ are specified, and $\delta \varepsilon = \partial \delta u$ in V. Using Lagrange multiplier method to introduce the constraints in the principle of virtual work, the Hu-Washizu variational principle can be derived as follows,

$$\delta \prod^* = \delta W_i - \delta W_e - \delta \int_V \lambda_1^T (\varepsilon - \partial u) dV - \delta \int_S \lambda_2^T (u - \overline{u}) dS = 0$$

where δW_i and δW_e are internal virtual work and external virtual work, respectively,

$$\delta W_{i} = \int_{V} \delta \varepsilon^{T} \sigma dV$$

$$\delta W_e = \int_V \delta u^T f dV + \int_{S_p} \delta u^T p dS$$

where f is body force in V and p is surface force on the natural boundary S_p . From the equation above, by integrating the term $\int_V \lambda_1^T \delta \partial u dV$ by parts, the Lagrange multipliers λ_1 and λ_2 are identified, respectively, as the stresses σ in V and reactions p_u on S_u . Therefore, we get the following Hu-Washizu variational principle,

$$\delta \prod_{HW} = \delta W_i - \delta W_e - \delta \int_V \sigma^T (\varepsilon - \partial u) dV - \delta \int_S p_u^T (u - \overline{u}) dS = 0$$
 (2.5a)

in which there are no constraints on the unknown displacements u, strains ε , stresses σ , and reactions p_u on S_u and they can be assumed independently. However, it doesn't mean that these quantities can be assumed to be completely arbitrary in seeking a finite element solution. In general, to get a good solution, the assumption should reflect the true response as much as possible. For example, the assumption of these quantities needs to satisfy the completeness and keep an appropriate "balance" among these quantities. For the completeness, the assumption should be such that the rigid-body displacements including rigid-body translations and rotations, constant strains, constant stresses, and constant tractions can be represented in the limit as the element size approaches zero. Therefore, if these quantities are assumed to be a polynomial, the minimum requirements are that the displacements be linear, strains, stresses, and tractions be constant. It should be noted that the above derivation hasn't used the constitutive equation. Therefore, the formulation applies to any materials, linear or nonlinear. For example, substituting linear constitutive equation $\sigma = C\varepsilon$ in the first item δW_i , we get the following unconstrained Hu-Washizu variational principle in linear elasticity,

$$\delta \prod_{HW} = \int_{V} \delta \varepsilon^{T} (C\varepsilon - \sigma) dV - \int_{V} \delta \sigma^{T} (\varepsilon - \partial u) dV + \int_{V} \delta \partial u^{T} \sigma dV - \int_{V} \delta u^{T} f dV - \int_{S_{p}} \delta u^{T} p dS - \int_{S_{u}} \delta p_{u}^{T} (u - \overline{u}) dS - \int_{S_{u}} \delta u^{T} p_{u} = 0$$
(2.5b)

in which there are no constraints on unknown displacements u, strains ε , stresses σ , and tractions p on S_u and they can be assumed independently. In most finite element applications, the displacement boundary condition $u = \overline{u}$ on S_u is enforced. In these cases, the Hu-Washizu variational principle becomes,

$$\delta \Pi_{HW} = \int_{V} \delta \varepsilon^{T} (C \varepsilon - \sigma) dV - \int_{V} \delta \sigma^{T} (\varepsilon - \partial u) dV + \int_{V} \delta \partial u^{T} \sigma dV - \int_{V} \delta u^{T} f dV - \int_{S_{p}} \delta u^{T} p dS = 0$$
(2.5c)

where unknown displacements u, strains ε and stresses σ are the independent variables and they can be assumed independently. By enforcing the constitutive equation $\sigma = C\varepsilon$, we get the following two forms of Hellinger-Reissner variational principles,

$$\delta \prod_{HR} (u, \sigma) = -\int_{V} \delta \sigma^{T} (S\sigma - \partial u) dV + \int_{V} \delta \partial u^{T} \sigma dV - \int_{V} \delta u^{T} f dV - \int_{S_{p}} \delta u^{T} p dS = 0$$
 (2.6d)

$$\delta \prod_{HR} (u, \varepsilon) = -\int_{V} \delta \varepsilon^{T} C(\varepsilon - \partial u) dV + \int_{V} \delta \partial u^{T} C \varepsilon dV - \int_{V} \delta u^{T} f dV - \int_{S_{n}} \delta u^{T} p dS = 0$$
 (2.6b)

where S is the compliance matrix. In the first equation unknown displacements u and stresses σ can be assumed independently, whereas in the second equation unknown displacements u and strains ε are the independent variables and they can be assumed independently.

It is obvious that with various displacement, stress, and strain assumptions many different mixed finite element can be designed. However, by the limitation principle [B. Fraeijs de Veubeke, 1965], if the mixed formulation is capable of producing the same approximation as that produced by direct displacement form then it will in fact reproduce that form exactly and give identical and not any improved results. Any one considering for instance the addition of many higher order polynomials representing stress distribution in a linear triangle will soon find out that although the formulation is correct nothing is gained by its use as all the higher-order terms disappear [Zienkiewicz and Taylor, 2000].

2.2 Coordinate Transformation

Transformation of coordinates is vital in many contexts and must be fully understood. A general argument on the coordinate transformation is as follows.

We wish to replace a set of parameters $\{a\}$ in which the system equations have been written by another one related to it by a transformation matrix [T] as

$${a} = [T]{b}$$
 (2.7)

In the linear case the system equations are of the form

$$[K]{a} = {r}$$

$$(2.8)$$

and on the substitution we have

$$[K][T]{b} = {r}$$

The new system can be pre-multiplied simply by $[T]^T$, yielding

$$[T]^{T}[K][T]\{b\} = [T]^{T}\{r\}$$
 (2.9)

which will preserve the symmetry of equations if the matrix [K] is symmetric. It is important to note that the argument does not demand that [T] be either orthogonal or square.

Transformation matrices exist that have neither of these properties.

Transformation of stress, strain, stiffness and compliance matrix

Let the two sets of coordinate systems denoted by xyz and x'y'z', respectively, refer to Figure 2.1.

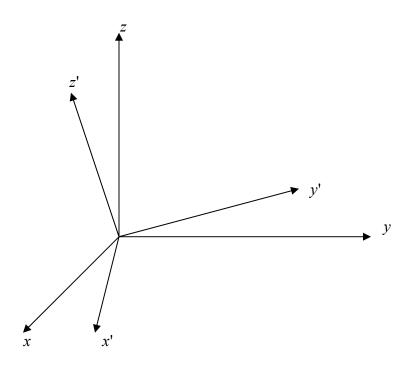


Figure 2.1 Cartesian coordinate systems xyz and x'y'z'

Table 2.1 Direction cosines between the two sets of axes

	x	y	Z
x'	$\cos(x',x)$	$\cos(x',y)$	$\cos(x',z)$
<i>y</i> '	$\cos(y',x)$	$\cos(y',y)$	$\cos(y',z)$
z'	$\cos(z',x)$	$\cos(z',y)$	$\cos(z',z)$

where cos(x',x) is the cosine of the angle between the x' axis and the x axis, and so on.

Transformation of stress in tensor form is

$$\sigma'_{ij} = \lambda_{im} \lambda_{jn} \sigma_{mn} \tag{2.10a}$$

From which we can easily get the matrix form as

$$[\sigma'] = [\lambda][\sigma][\lambda]^T \tag{2.10b}$$

or

$$\{\sigma'\} = [T_{\sigma}]\{\sigma\} \tag{2.10c}$$

where $[\lambda]$ is a 3×3 matrix of direction cosines between the two sets of axes, that is,

$$[\lambda] = \begin{bmatrix} \cos(x',x) & \cos(x',y) & \cos(x',z) \\ \cos(y',x) & \cos(y',y) & \cos(y',z) \\ \cos(z',x) & \cos(z',y) & \cos(z',z) \end{bmatrix}$$
(2.10d)

$$[T_{\sigma}] = \begin{bmatrix} \lambda_{11}^{2} & \lambda_{12}^{2} & \lambda_{13}^{2} & 2\lambda_{12}\lambda_{13} & 2\lambda_{13}\lambda_{11} & 2\lambda_{11}\lambda_{12} \\ \lambda_{21}^{2} & \lambda_{22}^{2} & \lambda_{23}^{2} & 2\lambda_{22}\lambda_{23} & 2\lambda_{23}\lambda_{21} & 2\lambda_{21}\lambda_{22} \\ \lambda_{31}^{2} & \lambda_{32}^{2} & \lambda_{33}^{2} & 2\lambda_{33}\lambda_{33} & 2\lambda_{33}\lambda_{31} & 2\lambda_{31}\lambda_{32} \\ \lambda_{21}\lambda_{31} & \lambda_{22}\lambda_{32} & \lambda_{23}\lambda_{33} & \lambda_{22}\lambda_{33} + \lambda_{23}\lambda_{32} & \lambda_{23}\lambda_{31} + \lambda_{21}\lambda_{33} & \lambda_{21}\lambda_{32} + \lambda_{22}\lambda_{31} \\ \lambda_{31}\lambda_{11} & \lambda_{32}\lambda_{12} & \lambda_{33}\lambda_{13} & \lambda_{32}\lambda_{13} + \lambda_{33}\lambda_{12} & \lambda_{33}\lambda_{11} + \lambda_{31}\lambda_{13} & \lambda_{31}\lambda_{12} + \lambda_{32}\lambda_{11} \\ \lambda_{11}\lambda_{21} & \lambda_{12}\lambda_{22} & \lambda_{13}\lambda_{23} & \lambda_{12}\lambda_{23} + \lambda_{13}\lambda_{22} & \lambda_{13}\lambda_{21} + \lambda_{11}\lambda_{23} & \lambda_{11}\lambda_{22} + \lambda_{12}\lambda_{21} \end{bmatrix}$$
 (2.10e)

Similarly, and noting that the shear components of strain tensor are half of the corresponding engineering shear strains, we can get the transformation of strain in matrix form as

$$\{\varepsilon'\} = [T_{\varepsilon}]\{\varepsilon\} \tag{2.11a}$$

where

$$[T_{\varepsilon}] = \begin{bmatrix} \lambda_{11}^{2} & \lambda_{12}^{2} & \lambda_{13}^{2} & \lambda_{12}\lambda_{13} & \lambda_{13}\lambda_{11} & \lambda_{11}\lambda_{12} \\ \lambda_{21}^{2} & \lambda_{22}^{2} & \lambda_{23}^{2} & \lambda_{22}\lambda_{23} & \lambda_{23}\lambda_{21} & \lambda_{21}\lambda_{22} \\ \lambda_{31}^{2} & \lambda_{32}^{2} & \lambda_{33}^{2} & \lambda_{32}\lambda_{33} & \lambda_{32}\lambda_{33} & \lambda_{33}\lambda_{31} & \lambda_{31}\lambda_{32} \\ 2\lambda_{21}\lambda_{31} & 2\lambda_{22}\lambda_{32} & 2\lambda_{23}\lambda_{33} & \lambda_{22}\lambda_{33} + \lambda_{23}\lambda_{32} & \lambda_{23}\lambda_{31} + \lambda_{21}\lambda_{33} & \lambda_{21}\lambda_{32} + \lambda_{22}\lambda_{31} \\ 2\lambda_{31}\lambda_{11} & 2\lambda_{32}\lambda_{12} & 2\lambda_{33}\lambda_{13} & \lambda_{32}\lambda_{13} + \lambda_{33}\lambda_{12} & \lambda_{33}\lambda_{11} + \lambda_{31}\lambda_{13} & \lambda_{31}\lambda_{12} + \lambda_{32}\lambda_{11} \\ 2\lambda_{11}\lambda_{21} & 2\lambda_{12}\lambda_{22} & 2\lambda_{13}\lambda_{23} & \lambda_{12}\lambda_{23} + \lambda_{13}\lambda_{22} & \lambda_{13}\lambda_{21} + \lambda_{11}\lambda_{23} & \lambda_{11}\lambda_{22} + \lambda_{12}\lambda_{21} \end{bmatrix}$$
 (2.11b)

Using the properties of direction cosines, we can easily verify

$$[T_{\sigma}]^{-1} = [T_{\varepsilon}]^{T}, \qquad [T_{\varepsilon}]^{-1} = [T_{\sigma}]^{T}$$

$$(2.12)$$

We can also get the relation from $\{\varepsilon'\}^T \{\sigma'\} = \{\varepsilon\}^T [T_{\varepsilon}]^T [T_{\sigma}] \{\sigma\} = \{\varepsilon\}^T \{\sigma\}$

From $\{\mathcal{E}'\} = [T_{\mathcal{E}}]\{\mathcal{E}\}$ and the relation (2.12), we get the transformation of stiffness matrix as

$$[C] = [T_{\varepsilon}]^{T} [C'][T_{\varepsilon}], \qquad [C'] = [T_{\sigma}][C][T_{\sigma}]^{T}$$

$$(2.13)$$

From $\{\sigma'\} = [T_{\sigma}]\{\sigma\}$ and the relation (2.12), we get the transformation of compliance matrix

as

$$[S] = [T_{\sigma}]^{T} [S'][T_{\sigma}], \qquad [S'] = [T_{\varepsilon}][S][T_{\varepsilon}]^{T}$$
(2.14)

Coordinate Transformation for Composite Laminates

In the analysis of composite laminated plates and shells, we frequently assume the lamina of orthotropic material under plane stress. Refer to Figure 2.2, in which the 1-2 coordinate system is the principal material coordinate system and θ is the angle from the x axis to the 1 axis.

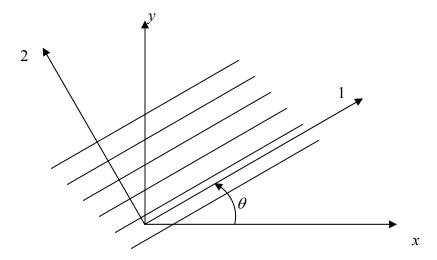


Figure 2.2 Positive rotation of principal material axes from xy axes.

We need to find the transformations from the principal material coordinate system to *xy* coordinate system. The matrix of direction cosines is,

$$[\lambda] = \begin{vmatrix} \cos(x,1) & \cos(x,2) & \cos(x,3) \\ \cos(y,1) & \cos(y,2) & \cos(y,3) \\ \cos(z,1) & \cos(z,2) & \cos(z,3) \end{vmatrix} = \begin{vmatrix} \cos\theta & -\sin\theta & 0 \\ \sin\theta & \cos\theta & 0 \\ 0 & 0 & 1 \end{vmatrix}$$
 (2.15a)

In these problems we are concerned with $\{\varepsilon\} = \{\varepsilon_x, \varepsilon_y, \gamma_{xy}\}^T$ and with $\{\sigma\} = \{\sigma_x, \sigma_y, \sigma_{xy}\}^T$. Accordingly, discarding rows and columns 3, 4 and 5 from Eqs. (2.10e) and (2.11b) and using the matrix of direction cosines, we have

$$[T_{\sigma}] = \begin{bmatrix} c^2 & s^2 & -2cs \\ s^2 & c^2 & 2cs \\ cs & -cs & c^2 - s^2 \end{bmatrix}, \qquad [T_{\varepsilon}] = \begin{bmatrix} c^2 & s^2 & -cs \\ s^2 & c^2 & cs \\ 2cs & -2cs & c^2 - s^2 \end{bmatrix}$$
 (2.15b)

Therefore, the stiffness transformation is,

$$\begin{bmatrix} \overline{Q}_{11} & \overline{Q}_{12} & \overline{Q}_{16} \\ \overline{Q}_{12} & \overline{Q}_{22} & \overline{Q}_{26} \\ \overline{Q}_{16} & \overline{Q}_{26} & \overline{Q}_{66} \end{bmatrix} = \begin{bmatrix} c^2 & s^2 & -2cs \\ s^2 & c^2 & 2cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} \begin{bmatrix} Q_{11} & Q_{12} & 0 \\ Q_{12} & Q_{22} & 0 \\ 0 & 0 & Q_{66} \end{bmatrix} \begin{bmatrix} c^2 & s^2 & -2cs \\ s^2 & c^2 & 2cs \\ cs & -cs & c^2 - s^2 \end{bmatrix}^T$$
(2.16)

$$\begin{split} & \overline{Q}_{11} = Q_{11}c^4 + 2(Q_{12} + 2Q_{66})s^2c^2 + Q_{22}s^4 \\ & \overline{Q}_{12} = (Q_{11} + Q_{22} - 4Q_{66})s^2c^2 + Q_{12}(s^4 + c^4) \\ & \overline{Q}_{22} = Q_{11}s^4 + 2(Q_{12} + 2Q_{66})s^2c^2 + Q_{22}c^4 \\ & \overline{Q}_{16} = (Q_{11} - Q_{12} - 2Q_{66})sc^3 + (Q_{12} - Q_{22} + 2Q_{66})s^3c \\ & \overline{Q}_{26} = (Q_{11} - Q_{12} - 2Q_{66})s^3c + (Q_{12} - Q_{22} + 2Q_{66})sc^3 \\ & \overline{Q}_{66} = (Q_{11} + Q_{22} - 2Q_{12} - 2Q_{66})s^2c^2 + Q_{66}(s^4 + c^4) \end{split}$$

where $c = \cos \theta$, and $s = \sin \theta$. The \overline{Q}_{ij} is the transformed reduced stiffness coefficients in the xy coordinate system and the Q_{ij} is the reduced stiffness coefficients in the principal material coordinate system given by

$$Q_{11} = \frac{E_1}{1 - v_{12}v_{21}}$$

$$Q_{12} = \frac{v_{12}E_2}{1 - v_{12}v_{21}}$$

$$Q_{22} = \frac{E_2}{1 - v_{12}v_{21}}$$

$$Q_{66} = G_{12}$$

The relation between the three-dimensional stiffness coefficient and reduced stiffness coefficient is given by [Chia, 1980]

$$Q_{ij} = C_{ij} - \frac{C_{i3}C_{j3}}{C_{33}} \qquad (i, j = 1, 2, 6)$$
(2.17)

Transformation of element stiffness matrix and force vector

It is often convenient to establish the characteristics of an individual element in a coordinate system which is different from that in which the external forces and displacements of the assembled structure or system will be measured. A different coordinate system may, in fact, be used for every element, to ease the computation. Clearly, it is necessary to do the transformation of coordinates to a common global system before the assembly of elements. Let the common coordinate system denoted by xyz and the local coordinate system denoted by x'y'z', refer to Figure 2.3. Define the nodal displacement vector of a node i as

$$\{a_i\} = \{u_i, v_i, w_i, \theta_{xi}, \theta_{yi}, \theta_{zi}\}^T$$

and the corresponding nodal force vector as

$${f_i} = {F_{xi}, F_{yi}, F_{zi}, M_{xi}, M_{yi}, M_{zi}}^T$$

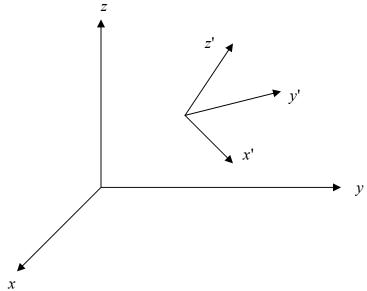


Figure 2.3 Cartesian coordinate systems xyz and x'y'z'

The displacements and forces of a node i transform from the global, to the local system (shown by prime) by a matrix $[T_i]$ given by

$$\{a_i'\} = [T_i]\{a_i\} \tag{2.18a}$$

where

$$\begin{bmatrix} T_i \end{bmatrix} = \begin{bmatrix} \begin{bmatrix} \lambda \end{bmatrix} & 0 \\ 0 & \begin{bmatrix} \lambda \end{bmatrix} \end{bmatrix}$$

with $[\lambda]$ being a 3×3 matrix of direction cosines between the two sets of axes, that is,

$$[\lambda] = \begin{bmatrix} \cos(x',x) & \cos(x',y) & \cos(x',z) \\ \cos(y',x) & \cos(y',y) & \cos(y',z) \\ \cos(z',x) & \cos(z',y) & \cos(z',z) \end{bmatrix}$$

Therefore, the nodal displacements and forces of a three-node triangular plate bending element transform from the global to the local system by a matrix [T] given by

$$\{a'\} = [T]\{a\}, \qquad \{f'\} = [T]\{f\}$$
 (2.18b)

where

$$[T] = \begin{bmatrix} T_1 \\ 0 \\ 0 \\ 0 \end{bmatrix} \begin{bmatrix} T_2 \\ 0 \end{bmatrix} \begin{bmatrix} 0 \\ 0 \\ 0 \end{bmatrix}$$

Using the rules of orthogonal transformation, the inverse of [T] is given by its transpose, thus we have

$${a} = [T]^T {a'}, {f} = [T]^T {f'}$$

which permits the stiffness matrix of an element in the global coordinates to be computed as

$$[K] = [T]^{T} [K'][T]$$
 (2.20)

2.3 Area/Natural Coordinates and Shape Functions for Triangular Elements

As compared to other types of elements, a triangular element has one great advantage it can approximate to any boundary shape.

For studying the triangular element, area coordinates are more convenient than Cartesian

coordinates. Let us first define area coordinates. In Figure 2.4, an arbitrarily located point P divides triangle 1-2-3 into three sub-triangles. Area coordinates of the point P are defined as ratios of the area of these sub-triangles to that of the total triangle (For formulation of the area of the triangle, see appendix C):

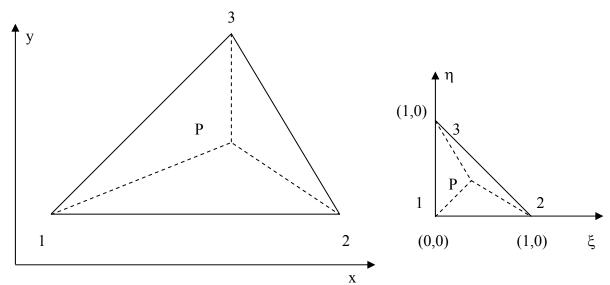


Figure 2.4 a) Area coordinates

b) Natural coordinates

$$\xi_{1} = \frac{S_{\Delta P23}}{S_{\Delta 123}} = \frac{\frac{1}{2} \begin{vmatrix} 1 & x & y \\ 1 & x_{2} & y_{2} \\ 1 & x_{3} & y_{3} \end{vmatrix}}{\frac{1}{2} \begin{vmatrix} 1 & x_{1} & y_{1} \\ 1 & x_{2} & y_{2} \\ 1 & x_{3} & y_{3} \end{vmatrix}} = \frac{a_{1} + b_{1}x + c_{1}y}{2A}$$
(2.21a)

$$\xi_{2} = \frac{S_{\Delta P31}}{S_{\Delta 123}} = \frac{\frac{1}{2} \begin{vmatrix} 1 & x & y \\ 1 & x_{3} & y_{3} \\ 1 & x_{1} & y_{1} \end{vmatrix}}{A} = \frac{a_{2} + b_{2}x + c_{2}y}{2A}$$
(2.21b)

$$\xi_{3} = \frac{S_{\Delta P12}}{S_{\Delta 123}} = \frac{\frac{1}{2} \begin{vmatrix} 1 & x & y \\ 1 & x_{1} & y_{1} \\ 1 & x_{2} & y_{2} \end{vmatrix}}{A} = \frac{a_{3} + b_{3}x + c_{3}y}{2A}$$
(2.21c)

where

$$a_1 = x_2y_3 - x_3y_2$$
, $b_1 = y_2 - y_3$, $c_1 = x_3 - x_2$
 $a_2 = x_3y_1 - x_1y_3$, $b_2 = y_3 - y_1$, $c_2 = x_1 - x_3$
 $a_3 = x_1y_2 - x_2y_1$, $b_3 = y_1 - y_2$, $c_3 = x_2 - x_1$

$$A = S_{\Delta 123} = \frac{1}{2} \begin{vmatrix} 1 & x_1 & y_1 \\ 1 & x_2 & y_2 \\ 1 & x_3 & y_3 \end{vmatrix} = \frac{1}{2} (b_2 c_3 - c_2 b_3)$$

In the matrix form, we have,

$$\begin{cases}
\xi_1 \\ \xi_2 \\ \xi_3
\end{cases} = \frac{1}{2A} \begin{bmatrix} a_1 & b_1 & c_1 \\ a_2 & b_2 & c_2 \\ a_3 & b_3 & c_3 \end{bmatrix} \begin{bmatrix} 1 \\ x \\ y \end{bmatrix} \tag{2.21d}$$

Because the total area of the triangle equals the sum of the areas of its three sub-triangles, we have

$$\xi_1 + \xi_2 + \xi_3 = 1$$
 (2.22a)

From the definition of area coordinates we can see that the Cartesian coordinates, x and y, of the point P have linear relations with the area coordinates. Here we have

$$x = \xi_1 x_1 + \xi_2 x_2 + \xi_3 x_3 \tag{2.22b}$$

$$y = \xi_1 y_1 + \xi_2 y_2 + \xi_3 y_3 \tag{2.22c}$$

because these relations holds at points 1, 2, and 3. The above relations can also be obtained from the geometry of a triangle (see Appendix D).

In the matrix form, we have,

$$\begin{cases} 1 \\ x \\ y \end{cases} = \begin{bmatrix} 1 & 1 & 1 \\ x_1 & x_2 & x_3 \\ y_1 & y_2 & y_3 \end{bmatrix} \begin{bmatrix} \xi_1 \\ \xi_2 \\ \xi_3 \end{cases}$$
 (2.22d)

The integral of a polynomial term in terms of area coordinates over the triangular region can

be exactly computed (see Appendix E).

Along a side of the triangle, the partial derivatives of area coordinates with respect to s are,

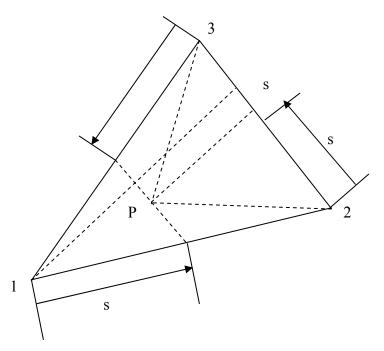


Figure 2.5 Area coordinates along the side of the triangle

Along the side 12,
$$\xi_1 = \frac{S_{\Delta P23}}{S_{\Delta 123}} = 1 - \frac{s}{l_{12}}, \qquad \frac{\partial \xi_1}{\partial s} \Big|_{side12} = -\frac{1}{l_{12}}$$
Along the side 31,
$$\xi_1 = \frac{S_{\Delta P23}}{S_{\Delta 123}} = \frac{s}{l_{31}}, \qquad \frac{\partial \xi_1}{\partial s} \Big|_{side31} = \frac{1}{l_{31}}$$
Similarly,

Along the side 12,
$$\xi_2 = \frac{S_{\Delta P31}}{S_{\Delta 123}} = \frac{s}{l_{12}}, \qquad \frac{\partial \xi_2}{\partial s} \Big|_{side12} = \frac{1}{l_{12}}$$
Along the side 23,
$$\xi_2 = \frac{S_{\Delta P31}}{S_{\Delta 123}} = 1 - \frac{s}{l_{23}}, \qquad \frac{\partial \xi_2}{\partial s} \Big|_{side23} = -\frac{1}{l_{23}}$$
Along the side 23,
$$\xi_3 = \frac{S_{\Delta P12}}{S_{\Delta 123}} = \frac{s}{l_{23}}, \qquad \frac{\partial \xi_3}{\partial s} \Big|_{side23} = \frac{1}{l_{23}}$$
Along the side 31,
$$\xi_3 = \frac{S_{\Delta P12}}{S_{\Delta 123}} = 1 - \frac{s}{l_{31}}, \qquad \frac{\partial \xi_3}{\partial s} \Big|_{side23} = -\frac{1}{l_{31}}$$

Shape Functions for Triangular Elements

The shape functions for triangular elements can be generated directly in terms of area coordinates by using following characteristics: the shape function N_i for node i gives a unit value at node i and zero at all other nodes and sides of the element that do not pass through node i. Polynomials in only one coordinate having this property are known as Lagrange polynomials and can be written as

$$l_i^n(x) = \frac{(x - x_0)(x - x_1)\cdots(x - x_{i-1})(x - x_{i+1})\cdots(x - x_n)}{(x_i - x_0)(x_i - x_1)\cdots(x_i - x_{i-1})(x_i - x_{i+1})\cdots(x_i - x_n)}$$
(2.23)

In general, we shall seek element expansions which possess the highest order of a complete polynomial for a minimum number of degrees of freedom. In this context, it is useful to recall the Pascal triangle given below.

Figure 2.6 The Pascal triangle

The Pascal triangle not only gives the terms to be included in a complete polynomial of a given degree, but it also gives the location for the nodes. For example, for quadratic triangular elements, the shape functions are complete polynomials of degree two and the nodes located at the vertices and the midpoints of each side.

For a general triangular element, see Figure 2.7, denoting a typical node i by three integer numbers I, J, and K corresponding to the position of coordinates ξ_{1i} , ξ_{2i} , and ξ_{3i} we can generate the shape function by a direct product of three Lagrange polynomials.

$$N_{i} = l_{I}^{I}(\xi_{1})l_{J}^{J}(\xi_{2})l_{K}^{K}(\xi_{3})$$
(2.24)

where $l_I^I(\xi_1)$, etc., are given by expression (2.23), with ξ_1 taking the place of x, etc.

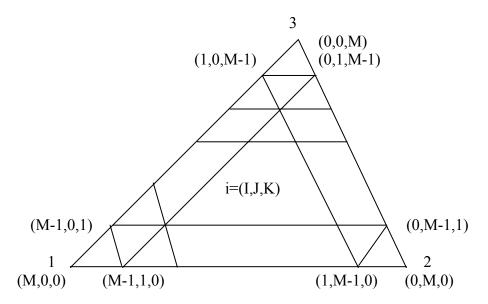


Figure 2.7 A general triangular element

Therefore, the shape functions for quadratic and cubic triangular elements can be generated as follows.

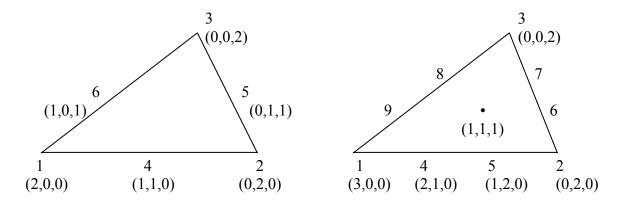


Figure 2.8 a) Quadratic triangular element

b) Cubic triangular element

Quadratic triangular element (LST, linear strain triangular element)

Corner nodes:

$$N_1 = \frac{(\xi_1 - 0)(\xi_1 - \frac{1}{2})}{(1 - 0)(1 - \frac{1}{2})} = \xi_1(2\xi_1 - 1)$$

$$N_2 = \frac{(\xi_2 - 0)(\xi_2 - \frac{1}{2})}{(1 - 0)(1 - \frac{1}{2})} = \xi_2(2\xi_2 - 1)$$

$$N_3 = \frac{(\xi_3 - 0)(\xi_3 - \frac{1}{2})}{(1 - 0)(1 - \frac{1}{2})} = \xi_3(2\xi_3 - 1)$$

Mid-side nodes:

$$N_4 = \frac{(\xi_1 - 0)(\xi_2 - 0)}{(\frac{1}{2} - 0)(\frac{1}{2} - 0)} = 4\xi_1 \xi_2$$

$$N_5 = \frac{(\xi_2 - 0)(\xi_3 - 0)}{(\frac{1}{2} - 0)(\frac{1}{2} - 0)} = 4\xi_2\xi_3$$

$$N_6 = \frac{(\xi_1 - 0)(\xi_3 - 0)}{(\frac{1}{2} - 0)(\frac{1}{2} - 0)} = 4\xi_1 \xi_3$$

Cubic triangular element

Corner nodes:

$$N_{1} = \frac{(\xi_{1} - 0)(\xi_{1} - \frac{1}{3})(\xi_{1} - \frac{2}{3})}{(1 - 0)(1 - \frac{1}{3})(1 - \frac{2}{3})} = \frac{1}{2}\xi_{1}(3\xi_{1} - 1)(3\xi_{1} - 2)$$

similarly,

$$N_2 = \frac{1}{2}\xi_2(3\xi_2 - 1)(3\xi_2 - 2)$$

$$N_3 = \frac{1}{2}\xi_3(3\xi_3 - 1)(3\xi_3 - 2)$$

Mid-side nodes:

$$N_4 = \frac{(\xi_1 - 0)(\xi_1 - \frac{1}{3})(\xi_2 - 0)}{(\frac{2}{3} - 0)(\frac{2}{3} - \frac{1}{3})(\frac{1}{3} - 0)} = \frac{9}{2}\xi_1\xi_2(3\xi_1 - 1)$$

similarly

$$N_6 = \frac{9}{2} \xi_2 \xi_3 (3\xi_2 - 1)$$

$$N_8 = \frac{9}{2} \xi_1 \xi_3 (3\xi_3 - 1)$$

$$N_5 = \frac{(\xi_1 - 0)(\xi_2 - 0)(\xi_2 - \frac{1}{3})}{(\frac{1}{3} - 0)(\frac{2}{3} - 0)(\frac{2}{3} - \frac{1}{3})} = \frac{9}{2}\xi_1\xi_2(3\xi_2 - 1)$$

$$N_7 = \frac{9}{2}\xi_2\xi_3(3\xi_3 - 1)$$

$$N_9 = \frac{9}{2} \xi_1 \xi_3 (3\xi_1 - 1)$$

In practice, it is frequently expedient to use a natural coordinate system. It contains a right triangle of unit base and unit height, called a 'parent' element in the context of mapped elements. Using the natural coordinate system, we have

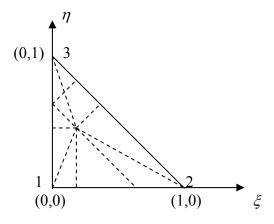


Figure 2.9 Relation between the area coordinates and natural coordinates

$$\xi_1 = 1 - \xi - \eta, \qquad \xi_2 = \xi, \qquad \xi_3 = \eta$$
 (2.25)

It is easily seen that the 'parent' element will be mapped into an actual element by the following transformation.

$$x = (1 - \xi - \eta)x_1 + \xi x_2 + \eta x_3 \tag{2.26a}$$

$$y = (1 - \xi - \eta)y_1 + \xi y_2 + \eta y_3 \tag{2.26b}$$

where (x_1, y_1) , (x_2, y_2) , and (x_3, y_3) are the coordinates of nodes 1, 2 and 3 of the actual element.

$$dxdy = |J|d\xi d\eta \tag{2.27}$$

where |J| is the determinant of the Jacobian matrix,

$$|J| = \begin{vmatrix} \frac{\partial x}{\partial \xi} & \frac{\partial y}{\partial \xi} \\ \frac{\partial x}{\partial \eta} & \frac{\partial y}{\partial \eta} \end{vmatrix} = \begin{vmatrix} x_2 - x_1 & y_2 - y_1 \\ x_3 - x_1 & y_3 - y_1 \end{vmatrix} = \begin{vmatrix} 1 & x_1 & y_1 \\ 1 & x_2 & y_2 \\ 1 & x_3 & y_3 \end{vmatrix} = 2A$$

where A is the area of triangle. Furthermore, all integrations are carried out over the natural coordinate, i.e., the ξ integration goes from 0 to 1 and the η integration goes from 0 to $(1-\xi)$.

Using the chain rule, we have

$$\frac{\partial}{\partial x} = \frac{\partial}{\partial \xi} \frac{\partial \xi}{\partial x} + \frac{\partial}{\partial \eta} \frac{\partial \eta}{\partial x}$$
 (2.28a)

$$\frac{\partial}{\partial y} = \frac{\partial}{\partial \xi} \frac{\partial \xi}{\partial y} + \frac{\partial}{\partial \eta} \frac{\partial \eta}{\partial y}$$
 (2.28b)

To evaluate $\frac{\partial \xi}{\partial x}$, $\frac{\partial \xi}{\partial y}$, etc., we note that

$$\begin{cases}
d\xi \\
d\eta
\end{cases} = \begin{bmatrix}
\frac{\partial \xi}{\partial x} & \frac{\partial \xi}{\partial y} \\
\frac{\partial \eta}{\partial x} & \frac{\partial \eta}{\partial y}
\end{bmatrix} dx \\
dy
\end{cases} \tag{2.29}$$

and

or

$$\begin{cases}
 d\xi \\
 d\eta
\end{cases} = \begin{bmatrix}
 \frac{\partial x}{\partial \xi} & \frac{\partial x}{\partial \eta} \\
 \frac{\partial y}{\partial \xi} & \frac{\partial y}{\partial \eta}
\end{bmatrix}^{-1} \begin{cases}
 dx \\
 dy
\end{cases} = \frac{1}{|J|} \begin{bmatrix}
 \frac{\partial y}{\partial \eta} & -\frac{\partial x}{\partial \eta} \\
 \frac{\partial y}{\partial \eta} & \frac{\partial x}{\partial \xi}
\end{bmatrix} \begin{bmatrix}
 dx \\
 dy
\end{cases} = (2.30)$$

A comparison of (2.29) and (2.30) yields

$$\frac{\partial \xi}{\partial x} = \frac{1}{|J|} \frac{\partial y}{\partial \eta}, \qquad \frac{\partial \xi}{\partial y} = -\frac{1}{|J|} \frac{\partial x}{\partial \eta}, \qquad \frac{\partial \eta}{\partial x} = -\frac{1}{|J|} \frac{\partial y}{\partial \xi}, \qquad \frac{\partial \eta}{\partial y} = \frac{1}{|J|} \frac{\partial x}{\partial \xi}$$

$$\frac{\partial}{\partial x} = \frac{\partial}{\partial \xi} \frac{\partial \xi}{\partial x} + \frac{\partial}{\partial \eta} \frac{\partial \eta}{\partial x} = \frac{1}{|J|} \left(\frac{\partial}{\partial \xi} \frac{\partial y}{\partial \eta} - \frac{\partial}{\partial \eta} \frac{\partial y}{\partial \xi} \right)$$

$$\frac{\partial}{\partial y} = \frac{\partial}{\partial \xi} \frac{\partial \xi}{\partial y} + \frac{\partial}{\partial \eta} \frac{\partial \eta}{\partial y} = \frac{1}{|J|} \left(-\frac{\partial}{\partial \xi} \frac{\partial x}{\partial \eta} + \frac{\partial}{\partial \eta} \frac{\partial x}{\partial \xi} \right)$$

in matrix form,

where
$$b_2 = y_3 - y_1$$
, $b_3 = y_1 - y_2$, $c_2 = x_1 - x_3$, $c_3 = x_2 - x_1$

The shape functions and their derivatives of linear strain triangular element (quadratic triangular element) in the natural coordinate system, see Appendix F.

2.4 Numerical Integration

In finite element analysis a large number of integrations are required, including the evaluation of the element stiffness matrix, mass matrix and force vector. To develop general finite element software, numerical integration is essential. Because the Gauss numerical

integration requires the least number of functional evaluations as compared with other methods, it is used almost exclusively in the finite element formulations.

With n sampling points Gauss integration can exactly integrate any polynomial of degree 2n-1. That is, if $f(\xi)$ is any polynomial of degree less than 2n, we have

$$\int_{-1}^{1} f(\xi) d\xi = \sum_{i=1}^{n} W_{i} f(\xi_{i})$$
 (2.32a)

where ξ_i , $i = 1, 2, \dots, n$, are the roots of the Legender polynomial of degree n, given by

$$P_n(\xi) = \frac{1}{2^n n!} \frac{d^n (\xi^2 - 1)^n}{d\xi^n}$$
 (2.32b)

 W_i $i = 1, 2, \dots, n$, are the corresponding weights given by

$$W_i = \int_{-1}^{1} N_i(\xi) d\xi$$
 (2.32c)

where $N_i(\xi)$, $i = 1, 2, \dots, n$, are the corresponding Lagrange shape function given by

$$N_{i}(\xi) = \frac{(\xi - \xi_{1})\cdots(\xi - \xi_{i-1})(\xi - \xi_{i+1})\cdots(\xi - \xi_{n})}{(\xi_{i} - \xi_{1})\cdots(\xi_{i} - \xi_{i-1})(\xi_{i} - \xi_{i+1})\cdots(\xi_{i} - \xi_{n})}$$
(2.32d)

The weights and sampling points for integrating an even order polynomial on [-1,1] are the same as that for the next odd order polynomial. The sampling points and the corresponding weights for n=1 to 4 are given in Table 2.2[Yang, 1986]

Table 2.2 Sampling points and weights in Gauss integration $\int_{-1}^{1} f(\xi) d\xi = \sum_{i=1}^{n} W_{i} f(\xi_{i})$

<u>n</u>	ξ_i	W_{i}
1	0	2
2	$\pm 1/\sqrt{3}$	1
3	0	8/9
	$\pm\sqrt{0.6}$	5/9
4	±0.339981043584856	0.652145154862546
	±0.861136311594053	0.347854845137454

Note: The third digit from the end in the number 0.861136311594053 in the Table was

mistyped to 9 in the Zienkiewicz's book.

The above Gauss integration can be easily extended to the rectangular region as follows

$$\int_{-1}^{1} \int_{-1}^{1} f(\xi, \eta) d\xi d\eta = \sum_{i=1}^{m} \sum_{j=1}^{n} W_{i} W_{j} f(\xi_{i}, \eta_{i})$$
(2.33)

For the triangular region, the Gauss integration is

$$\int_{0}^{1} \int_{0}^{1-\xi} f(\xi, \eta) d\xi d\eta = \frac{1}{2} \sum_{i=1}^{n} W_{i} f(\xi_{i}, \eta_{i})$$
(2.34)

where the coefficient $\frac{1}{2}$ represents the area of triangular integration region. The sum of

weights equals 1 because the formulation must be true when $f(\xi,\eta)=1$. ξ_i and η_i are corresponding to the area coordinates ξ_2 and ξ_3 , respectively. The sampling points are symmetric with respect to the three area coordinates. That is, the sampling points occur only in groups of one, three or six points with equal weights in each group, in the form of

$$\left(\frac{1}{3}, \frac{1}{3}, \frac{1}{3}\right)$$
, (a,b,b) , (b,a,b) , (b,b,a) or (a,b,c) , (b,c,a) , (c,a,b) , (b,a,c) , (c,b,a) and (a,c,b) ,

respectively. Note that the sum of three area coordinates equals one.

The choice of the order of numerical integration is important in practice. In theory, if a high enough order is used, all matrices will be evaluated very accurately. However, the cost of analysis increases when a high-order integration is employed. On the other hand, if the order of numerical integration is too low, the matrices may be evaluated very inaccurately and singularities may arise. The integration order required to evaluate a specific element matrix can be determined by studying the order of the function to be integrated. If the integrand is not a polynomial function, one should experiment with an increasing number of Gauss points till the difference between two successive values obtained is less than the acceptable tolerance. In general, the order of integration for the evaluation of the consistent mass matrix

is higher than that for the evaluation of the stiffness matrix because the mass matrix is obtained from the displacement interpolation functions, whereas the stiffness matrix is computed from the derivatives of the displacement functions. If the element geometric distortions are very large, and in nonlinear analysis, a higher order of the integration may be appropriate. For a reliable finite element analysis, full (exact) numerical integration should be used.

The sampling points and the corresponding weights for n=3 to 4 are given in Table 2.3 [Yang, 1986]. For more comprehensive list, see the book [Yang, 1986] and papers [Hammer et al., 1956; Cowper, 1973; and Dunavant, 1985].

Table 2.3 Gauss Integration for the triangular region

Number and Locations of	ξ_i	$\eta_{_i}$	Weights	Degree of
Sampling Points			W_{i}	Accuracy
n=3	$\frac{1}{2}$	0	$\frac{1}{3}$	2
\wedge	2		3	
	$\frac{1}{2}$	$\frac{1}{2}$	$\frac{1}{3}$	
	0	$\frac{1}{2}$	$\frac{1}{3}$	
n=4	1	1	27	3
	$\frac{1}{3}$	$\frac{1}{3}$	$-\frac{27}{48}$	
^		3	40	
/ _i \	1	1	25	
	$\frac{1}{5}$	$\frac{1}{5}$	$\frac{25}{48}$	
	$\frac{3}{5}$	$\frac{1}{5}$	$\frac{25}{48}$	
	5	5	48	
	1	3	25	
	$\frac{1}{5}$	$\frac{3}{5}$	$\frac{25}{48}$	
			-10	

In the finite element history, the reduced/selective integration that mitigates the shear locking in the displacement-based plate/shell element should be viewed as merely an effective way to accurately compute the finite element matrices of the mixed formulation. The equivalence between the reduced/selective integration and the mixed formulation was first shown by Malkus and Hughes in 1978 and later in a general context by Zienkiewicz and Nakazawa in 1984.

2.5 Membrane Element

Allman Triangular Element

One way to improve the performance of elements is to add drilling d.o.f. A drilling d.o.f in a plane element is a rotational d.o.f. whose vector is normal to the analysis plane. A benefit of the drilling d.o.f. is that they can enable elements having only corner nodes to provide acceptable performance while using fewer d.o.f. than elements having both corner and side nodes. For example, a triangular element having drilling d.o.f. and only vertex nodes performs much better than the six d.o.f. CST element, although not as well as the 12 d.o.f. LST element. The number of d.o.f. in a large 2D mesh is reduced by a factor of 5/8 comparing with using LST element. Another benefit of using drilling degrees of freedom lies in shell analysis. A shell element can easily be formed as the combination of a plane element and an element for plate bending. A shell element that uses all six d.o.f at each node is perhaps best suited to analysis of folded plates, where many elements are coplanar. In modeling a continuously curved shell, drilling d.o.f. may interact unfavorably with bending deformation [MacNeal, 1994].

The LST element is known to give much improved estimates of displacements and stresses, for a given finite element mesh size, compared to the CST element. However, this

improvement in accuracy is achieved at a price of a big increase in the total degrees of freedom in the finite element model used to represent a structure. The Allman triangle, which has three nodes and three degrees of freedom at each node, combines the best features of the two elements, namely: compatible quadratic displacements with the degrees of freedom located at the vertices only.

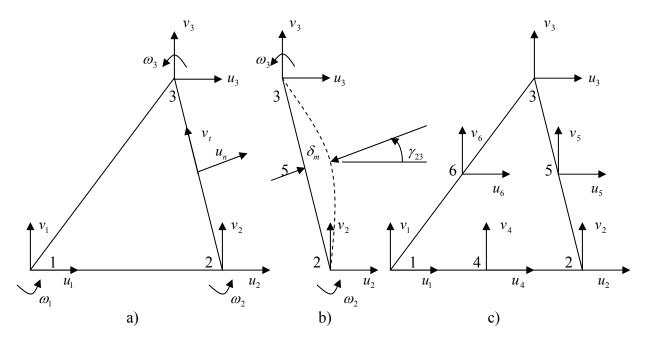


Figure 2.10 a) Nodal d.o.f. of the Allman triangle b) Parabolic edge displacement created by d.o.f. ω_2 and ω_3 c) D.o.f. of the linear strain triangle (LST).

In the Allman triangle, the normal and tangential displacements u_n and u_t along the side of triangle are assumed to be quadratic and linear, respectively. Consider a typical side of a plane element, Figure 2.10 (b), in which δ_m is the mid-side normal displacement relative to the straight-side condition. Therefore, we have

$$u_{n} = \xi_{2}u_{n2} + \xi_{3}u_{n3} + 4\xi_{2}\xi_{3}\delta_{m} = (1 - \frac{s}{l_{23}})u_{n2} + \frac{s}{l_{23}}u_{n3} + 4(1 - \frac{s}{l_{23}})\frac{s}{l_{23}}\delta_{m}$$

$$u_{t} = \xi_{2}u_{t2} + \xi_{3}u_{t3} = (1 - \frac{s}{l_{23}})u_{t2} + (\frac{s}{l_{23}})u_{t3}$$

$$(2.35)$$

We can also write $u_n = N_2 u_{n2} + N_3 u_{n3} + N_5 u_{n5} = \xi_2 (2\xi_2 - 1)u_{n2} + \xi_3 (2\xi_3 - 1)u_{n3} + 4\xi_2 \xi_3 u_{n5}$, but as long as we note $u_{n5} = \frac{u_{n2} + u_{n3}}{2} + \delta_m$, it reduces to eqn. (2.35).

Our objective is to express δ_m by the vertex degrees of freedom. Using the following condition

$$\frac{\partial u_n}{\partial s}\Big|_{s=l_{23}} - \frac{\partial u_n}{\partial s}\Big|_{s=0} = -\omega_3 + \omega_2 \tag{2.36}$$

we get δ_m

$$\delta_{m} = \frac{l_{23}}{8}(\omega_{3} - \omega_{2}) \tag{2.37}$$

Clearly, ω_2 and ω_3 are not true rotations as defined in the theory of elasticity and the quadratic term in the normal displacement u_n is caused by these drilling degrees of freedom. We also note here that the use of eqn. (2.36) neither prevents the compatibility of the displacements nor constrains the change in the vertex angles of a triangular element. Therefore, all constant states of strain can be represented exactly by the present formulation and convergence to an exact solution with consistent mesh refinement is thus assured. Substituting eqn. (2.37) into (2.35) gives the boundary displacements along side 23 as

$$u_n = \xi_2 u_{n2} + \xi_3 u_{n3} + \frac{1}{2} l_{23} (\omega_3 - \omega_2) \xi_2 \xi_3 = (1 - \frac{s}{l_{23}}) u_{n2} + \frac{s}{l_{23}} u_{n3} + \frac{1}{2} s (1 - \frac{s}{l_{23}}) (\omega_3 - \omega_2)$$

$$u_{t} = \xi_{2}u_{t2} + \xi_{3}u_{t3} = (1 - \frac{s}{l_{23}})u_{t2} + (\frac{s}{l_{23}})u_{t3}$$
(2.38)

and the boundary displacements along sides 31 and 12 follow from a cyclic permutation of the indices in eqn (2.38).

Based on the eqns (2.38), Allman introduced the element displacement field as follows

$$u = u_{1}\xi_{1} + u_{2}\xi_{2} + u_{3}\xi_{3} + \frac{1}{2}l_{12}\cos\gamma_{12}(\omega_{2} - \omega_{1})\xi_{1}\xi_{2}$$

$$+ \frac{1}{2}l_{23}\cos\gamma_{23}(\omega_{3} - \omega_{2})\xi_{2}\xi_{3} + \frac{1}{2}l_{31}\cos\gamma_{31}(\omega_{1} - \omega_{3})\xi_{3}\xi_{1}$$

$$v = v_{1}\xi_{1} + v_{2}\xi_{2} + v_{3}\xi_{3} + \frac{1}{2}l_{12}\sin\gamma_{12}(\omega_{2} - \omega_{1})\xi_{1}\xi_{2}$$

$$+ \frac{1}{2}l_{23}\sin\gamma_{23}(\omega_{3} - \omega_{2})\xi_{2}\xi_{3} + \frac{1}{2}l_{31}\sin\gamma_{31}(\omega_{1} - \omega_{3})\xi_{3}\xi_{1}$$

$$\text{where } \sin\gamma_{12} = \frac{x_{1} - x_{2}}{l_{12}}, \cos\gamma_{12} = -\frac{y_{1} - y_{2}}{l_{12}}, \text{ etc.}$$

$$(2.39)$$

To verify that the displacement field eqn. (2.39) is corresponding to the boundary displacements eqn. (2.38), we note that along the side 23, for example,

$$\begin{split} \xi_1 &= 0, \qquad \xi_2 = 1 - \frac{s}{l_{23}}, \qquad \xi_3 = \frac{s}{l_{23}} \\ u_n &= u \cos \gamma_{23} + v \sin \gamma_{23} \\ &= (u_2 \xi_2 + u_3 \xi_3 + \frac{1}{2} l_{23} \cos \gamma_{23} (\omega_3 - \omega_2) \xi_2 \xi_3) \cos \gamma_{23} \\ &+ (v_2 \xi_2 + v_3 \xi_3 + \frac{1}{2} l_{23} \sin \gamma_{23} (\omega_3 - \omega_2) \xi_2 \xi_3) \sin \gamma_{23} \\ &= \xi_2 (u_2 \cos \gamma_{23} + v_2 \sin \gamma_{23}) + \xi_3 (u_3 \cos \gamma_{23} + v_3 \sin \gamma_{23}) + \frac{1}{2} l_{23} (\omega_3 - \omega_2) \xi_2 \xi_3 \\ &= \xi_2 u_{n2} + \xi_3 u_{n3} + \frac{1}{2} l_{23} (\omega_3 - \omega_2) \xi_2 \xi_3 \\ &= (1 - \frac{s}{l_{23}}) u_{n2} + (\frac{s}{l_{23}}) u_{n3} + \frac{1}{2} s (1 - \frac{s}{l_{23}}) (\omega_3 - \omega_2) \\ u_t &= -u \sin \gamma_{23} + v \cos \gamma_{23} \\ &= -(u_2 \xi_2 + u_3 \xi_3 + \frac{1}{2} l_{23} \cos \gamma_{23} (\omega_3 - \omega_2) \xi_2 \xi_3) \sin \gamma_{23} \\ &+ (v_2 \xi_2 + v_3 \xi_3 + \frac{1}{2} l_{23} \sin \gamma_{23} (\omega_3 - \omega_2) \xi_2 \xi_3) \cos \gamma_{23} \\ &= \xi_2 (-u_2 \sin \gamma_{23} + v_2 \cos \gamma_{23}) + \xi_3 (-u_3 \sin \gamma_{23} + v_3 \cos \gamma_{23}) \\ &= \xi_2 u_{t2} + \xi_3 u_{t3} \\ &= (1 - \frac{s}{l_{23}}) u_{t2} + (\frac{s}{l_{23}}) u_{t3} \end{split}$$

From the displacement field eqn. (2.39), Allman constructs the 9 by 9 element stiffness matrix. Tests show that the element performs much better than the CST element, although not as well as the LST element. However, in a large mesh the LST may produce more than twice as many d.o.f. One nodal ω of the entire mesh must be set to zero to prevent an unusual type of zero energy mode in which all u_i and v_i are zero and all ω_i are equal. Cook noted that the Allman triangle can be regarded as the result of applying a coordinate transformation to an element that has mid-side nodes. From eqn. (2.39), we have the displacements at the mid-side nodes

$$u_{4} = \frac{1}{2}u_{1} + \frac{1}{2}u_{2} + \frac{l_{12}}{8}(\omega_{2} - \omega_{1})\cos\gamma_{12} = \frac{1}{2}u_{1} + \frac{1}{2}u_{2} - \frac{y_{1} - y_{2}}{8}(\omega_{2} - \omega_{1}) = \frac{1}{2}u_{1} + \frac{1}{2}u_{2} - \frac{b_{3}}{8}(\omega_{2} - \omega_{1})$$

$$v_{4} = \frac{1}{2}v_{1} + \frac{1}{2}v_{2} + \frac{l_{12}}{8}(\omega_{2} - \omega_{1})\sin\gamma_{12} = \frac{1}{2}v_{1} + \frac{1}{2}v_{2} - \frac{x_{2} - x_{1}}{8}(\omega_{2} - \omega_{1}) = \frac{1}{2}v_{1} + \frac{1}{2}v_{2} - \frac{c_{3}}{8}(\omega_{2} - \omega_{1})$$

$$u_{5} = \frac{1}{2}u_{2} + \frac{1}{2}u_{3} + \frac{l_{23}}{8}(\omega_{3} - \omega_{2})\cos\gamma_{23} = \frac{1}{2}u_{2} + \frac{1}{2}u_{3} - \frac{y_{2} - y_{3}}{8}(\omega_{3} - \omega_{2}) = \frac{1}{2}u_{2} + \frac{1}{2}u_{3} - \frac{b_{1}}{8}(\omega_{3} - \omega_{2})$$

$$v_{5} = \frac{1}{2}v_{2} + \frac{1}{2}v_{3} + \frac{l_{23}}{8}(\omega_{3} - \omega_{2})\sin\gamma_{23} = \frac{1}{2}v_{2} + \frac{1}{2}v_{3} - \frac{x_{3} - x_{2}}{8}(\omega_{3} - \omega_{2}) = \frac{1}{2}v_{2} + \frac{1}{2}v_{3} - \frac{c_{1}}{8}(\omega_{3} - \omega_{2})$$

$$u_{6} = \frac{1}{2}u_{3} + \frac{1}{2}u_{1} + \frac{l_{31}}{8}(\omega_{1} - \omega_{3})\cos\gamma_{31} = \frac{1}{2}u_{3} + \frac{1}{2}u_{1} - \frac{y_{3} - y_{1}}{8}(\omega_{1} - \omega_{3}) = \frac{1}{2}u_{3} + \frac{1}{2}u_{1} - \frac{b_{2}}{8}(\omega_{1} - \omega_{3})$$

$$v_{6} = \frac{1}{2}v_{3} + \frac{1}{2}v_{1} + \frac{l_{31}}{8}(\omega_{1} - \omega_{3})\sin\gamma_{31} = \frac{1}{2}v_{3} + \frac{1}{2}v_{1} - \frac{x_{1} - x_{3}}{8}(\omega_{1} - \omega_{3}) = \frac{1}{2}v_{3} + \frac{1}{2}v_{1} - \frac{c_{2}}{8}(\omega_{1} - \omega_{3})$$

Therefore, the coordinate transformation between the degrees of freedom of the LST and the Allman triangle is

$$\{a_{lst}\} = [T]\{a_{allman}\} \tag{2.40}$$

where

$$\{a_{lst}\} = \{u_1 \quad v_1 \quad u_2 \quad v_2 \quad u_3 \quad v_3 \quad u_4 \quad v_4 \quad u_5 \quad v_5 \quad u_6 \quad v_6\}^T$$

$$\{a_{allman}\} = \{u_1 \quad v_1 \quad \omega_1 \quad u_2 \quad v_2 \quad \omega_2 \quad u_3 \quad v_3 \quad \omega_3\}^T$$

where

$$b_1 = y_2 - y_3,$$
 $c_1 = x_3 - x_2$
 $b_2 = y_3 - y_1,$ $c_2 = x_1 - x_3$
 $b_3 = y_1 - y_2,$ $c_3 = x_2 - x_1$

So the 9 by 9 stiffness matrix $[k_A]$ of the Allman triangle is

$$\begin{bmatrix} k_A \end{bmatrix} = \begin{bmatrix} T \end{bmatrix}^T \begin{bmatrix} k_{LST} \end{bmatrix} \begin{bmatrix} T \end{bmatrix} \tag{2.41}$$

where $[k_{LST}]$ is the 12 by 12 stiffness matrix of the corresponding LST element. Clearly, the transformation symbolized by eqn (2.40) can be applied to elements with more than three sides. Accordingly, various elements with nodal rotations can be produced from similar elements with midside nodes and straight edges merely by employing a coordinate transformation subroutine. Thus, Allman type elements can easily be added as an option to existing programs.

Next, we verify that the element obtained by the above transformation from the LST element is indeed the Allman triangle. The element displacement field is

$$\begin{split} u &= \sum_{i=1}^6 N_i u_i \\ &= \xi_1 (2\xi_1 - 1) u_1 + \xi_2 (2\xi_2 - 1) u_2 + \xi_3 (2\xi_3 - 1) u_3 + 4\xi_1 \xi_2 (\frac{1}{2} u_1 + \frac{1}{2} u_2 + \frac{l_{12}}{8} (\omega_2 - \omega_1) \cos \gamma_{12}) \\ &+ 4\xi_2 \xi_3 (\frac{1}{2} u_2 + \frac{1}{2} u_3 + \frac{l_{23}}{8} (\omega_3 - \omega_2) \cos \gamma_{23}) + 4\xi_1 \xi_3 (\frac{1}{2} u_3 + \frac{1}{2} u_1 + \frac{l_{31}}{8} (\omega_1 - \omega_3) \cos \gamma_{31}) \\ &= (2\xi_1 (\xi_1 + \xi_2 + \xi_3) - \xi_1) u_1 + (2\xi_2 (\xi_1 + \xi_2 + \xi_3) - \xi_2) u_2 + (2\xi_3 (\xi_1 + \xi_2 + \xi_3) - \xi_3) u_3 \\ &+ \frac{l_{12}}{2} (\omega_2 - \omega_1) \cos \gamma_{12} \xi_1 \xi_2 + \frac{l_{23}}{2} (\omega_3 - \omega_2) \cos \gamma_{23} \xi_2 \xi_3 + \frac{l_{31}}{2} (\omega_1 - \omega_3) \cos \gamma_{31} \xi_3 \xi_1 \\ &= \xi_1 u_1 + \xi_2 u_2 + \xi_3 u_3 + \frac{l_{12}}{2} (\omega_2 - \omega_1) \cos \gamma_{12} \xi_1 \xi_2 + \frac{l_{23}}{2} (\omega_3 - \omega_2) \cos \gamma_{23} \xi_2 \xi_3 + \frac{l_{31}}{2} (\omega_1 - \omega_3) \cos \gamma_{31} \xi_3 \xi_1 \\ v &= \sum_{i=1}^6 N_i v_i \\ &= \xi_1 (2\xi_1 - 1) v_1 + \xi_2 (2\xi_2 - 1) v_2 + \xi_3 (2\xi_3 - 1) v_3 + 4\xi_1 \xi_2 (\frac{1}{2} u_1 + \frac{1}{2} u_2 + \frac{l_{12}}{8} (\omega_2 - \omega_1) \sin \gamma_{12}) \\ &+ 4\xi_2 \xi_3 (\frac{1}{2} u_2 + \frac{1}{2} u_3 + \frac{l_{23}}{8} (\omega_3 - \omega_2) \sin \gamma_{23}) + 4\xi_1 \xi_3 (\frac{1}{2} u_3 + \frac{1}{2} u_1 + \frac{l_{31}}{8} (\omega_1 - \omega_3) \sin \gamma_{31}) \\ &= (2\xi_1 (\xi_1 + \xi_2 + \xi_3) - \xi_1) v_1 + (2\xi_2 (\xi_1 + \xi_2 + \xi_3) - \xi_2) v_2 + (2\xi_3 (\xi_1 + \xi_2 + \xi_3) - \xi_3) v_3 \\ &+ \frac{l_{12}}{2} (\omega_2 - \omega_1) \sin \gamma_{12} \xi_1 \xi_2 + \frac{l_{23}}{2} (\omega_3 - \omega_2) \sin \gamma_{23} \xi_2 \xi_3 + \frac{l_{31}}{2} (\omega_1 - \omega_3) \sin \gamma_{31} \xi_3 \xi_1 \\ &= \xi_1 v_1 + \xi_2 v_2 + \xi_3 v_3 + \frac{l_{12}}{2} (\omega_2 - \omega_1) \sin \gamma_{12} \xi_1 \xi_2 + \frac{l_{23}}{2} (\omega_3 - \omega_2) \sin \gamma_{23} \xi_2 \xi_3 + \frac{l_{31}}{2} (\omega_1 - \omega_3) \sin \gamma_{31} \xi_3 \xi_1 \\ &= \xi_1 v_1 + \xi_2 v_2 + \xi_3 v_3 + \frac{l_{12}}{2} (\omega_2 - \omega_1) \sin \gamma_{12} \xi_1 \xi_2 + \frac{l_{23}}{2} (\omega_3 - \omega_2) \sin \gamma_{23} \xi_2 \xi_3 + \frac{l_{31}}{2} (\omega_1 - \omega_3) \sin \gamma_{31} \xi_3 \xi_1 \\ &= \xi_1 v_1 + \xi_2 v_2 + \xi_3 v_3 + \frac{l_{12}}{2} (\omega_2 - \omega_1) \sin \gamma_{12} \xi_1 \xi_2 + \frac{l_{23}}{2} (\omega_3 - \omega_2) \sin \gamma_{23} \xi_2 \xi_3 + \frac{l_{31}}{2} (\omega_1 - \omega_3) \sin \gamma_{31} \xi_3 \xi_1 \\ &= \xi_1 v_1 + \xi_2 v_2 + \xi_3 v_3 + \frac{l_{12}}{2} (\omega_2 - \omega_1) \sin \gamma_{12} \xi_1 \xi_2 + \frac{l_{23}}{2} (\omega_3 - \omega_2) \sin \gamma_{23} \xi_2 \xi_3 + \frac{l_{31}}{2} (\omega_1 - \omega_3) \sin \gamma_{31} \xi_3 \xi_1 \\ &= \xi_1 v_1 + \xi_2 v_2 + \xi_3 v_3 + \frac{l_{12}}{2} (\omega_2 - \omega_1) \sin \gamma_{12} \xi_1 \xi_2 + \frac{l_$$

Comparing eqn. (2.42) with eqn. (2.39), we can see that the element really has the same element displacement field as the Allman triangular element. Therefore, the element obtained by the above transformation from the LST element is indeed the Allman triangular element. In the displacement field of the Allman triangle, eqn. (2.39), the true rotation Ω in the theory of elasticity is (The derivation, see Appendix G)

(2.42)

$$\Omega = \frac{1}{2} \left(\frac{\partial v}{\partial x} - \frac{\partial u}{\partial y} \right) = \Omega_0 + \frac{1}{4} \sum_{i=1}^{3} (3\xi_i - 1)\omega_i$$
(2.43)

where Ω_0 , the rotation at the centroid $\xi_1 = \xi_2 = \xi_3 = \frac{1}{3}$, is

$$\Omega_0 = \frac{1}{4A} [(x_2 - x_3)u_1 + (x_3 - x_1)u_2 + (x_1 - x_2)u_3 + (y_2 - y_3)v_1 + (y_3 - y_1)v_2 + (y_1 - y_2)v_3]$$

From the eqn. (2.43), we have at the vertices

$$\Omega_1 = \Omega_0 + \frac{1}{4}(2\omega_1 - \omega_2 - \omega_3)$$

$$\Omega_2 = \Omega_0 + \frac{1}{4}(2\omega_2 - \omega_1 - \omega_3)$$

$$\Omega_3 = \Omega_0 + \frac{1}{4}(2\omega_3 - \omega_1 - \omega_2)$$

therefore,

$$\Omega_0 = \frac{1}{3}(\Omega_1 + \Omega_2 + \Omega_3)$$

Moreover, denoting ω_0 as the average value of the ω_i connectors, thus

$$\omega_0 = \frac{1}{3}(\omega_1 + \omega_2 + \omega_3)$$

$$\Omega_i - \Omega_0 = \frac{3}{4}(\omega_i - \omega_0) \tag{2.44}$$

Eqs. (2.43) and (2.44) show that continuity of the true rotation field is not enforced in the Allman element either across element sides or at the nodes. It is to be noted that this level of continuity is not required for a correct application of the principle of minimum potential energy.

Notes:

(1) ω true meaning. Though we get $\delta_m = \frac{l_{23}}{8}(\omega_3 - \omega_2)$ from the condition

$$\frac{\partial u_n}{\partial s}\Big|_{s=l_{23}} - \frac{\partial u_n}{\partial s}\Big|_{s=0} = -\omega_3 + \omega_2$$
, we can not assume $\omega = -\frac{\partial u_n}{\partial s}$, concept wrong. Applying the

boundary conditions should include the drilling degrees when we choose the Allman element. Whether the drilling degrees $\omega_i = 0$ or not should be based on whether $\delta_m = 0$ or not.

- (2) ω is not true rotation, which is defined as $(v_{,x}-u_{,y})/2$ in the theory of elasticity, but it is closely related to the true rotation as in eqn. (2.43) and (2.44).
- (3) In addition to the rigid body motion, the Allman element also exhibits an unusual type of zero energy mode $u_i = v_i = 0$ and $\omega_i = c$, a constant. The result is that a mesh of elements formulated in this way displays no strain energy if all drilling d.o.f. in the mesh are equal. Therefore, the structure stiffness matrix is singular even if the rigid body movements are constrained. One nodal ω of the entire mesh must be set to zero to prevent this singularity. (4) Because of the zero energy mode, the element has only 8 d.o.f. available to model deformation, despite having a total of 9 nodal d.o.f. Therefore, the element uses incomplete quadratic fields. The complete quadratic field in a triangular element needs 12 d.o.f to describe it, with each component having 6 d.o.fs as in the LST element.

2.6 Bending Element

Because of the high efficiency and accuracy of discrete Kirchhoff triangular (DKT) element for thin plate bending analysis but lack of document for its derivation we present the detail derivation for DKT element in this section.

Discrete Kirchhoff Triangular (DKT) Element

The formulation of elements based on the discrete Kirchhoff theory for the bending of thin plates is obtained by first considering a theory of plates including transverse shear deformations. The independent variables are the deflection w, and the rotations β_x and β_y and only C^0 continuity requirements need to be satisfied. The transverse shear energy is

neglected altogether and the Kirchhoff hypothesis is introduced in a discrete way along the edges of the element to relate the rotations to the transverse displacements.

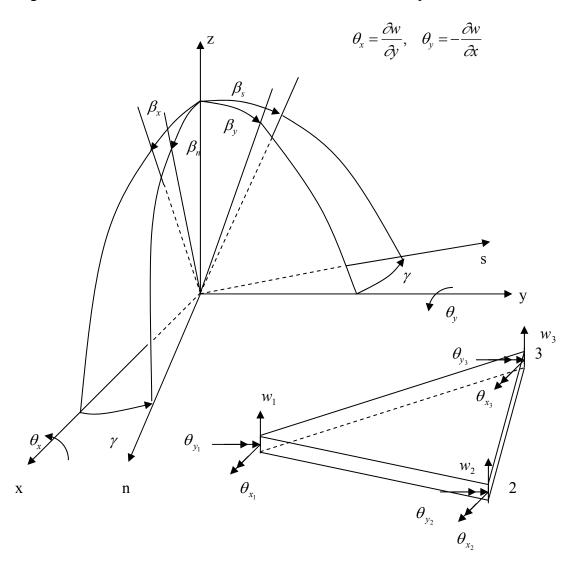


Figure 2.11 Definitions of variables in the formulation of the DKT element

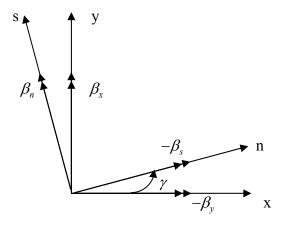


Figure 2.12 Relation between β_x , β_y and β_n , β_s

Table 2.4 Direction cosines between the two sets of axes

	n	S
X	$\cos \gamma$	$-\sin\gamma$
У	$\sin \gamma$	$\cos \gamma$

$$x = n\cos\gamma - s\sin\gamma$$
$$y = n\sin\gamma + s\cos\gamma$$

Using tensor transformation,

$$\begin{cases}
-\beta_{y} \\
\beta_{x}
\end{cases} =
\begin{bmatrix}
\cos \gamma & -\sin \gamma \\
\sin \gamma & \cos \gamma
\end{bmatrix}
\begin{bmatrix}
-\beta_{s} \\
\beta_{n}
\end{bmatrix}$$

that is,

$$\begin{cases}
\beta_x \\
\beta_y
\end{cases} = \begin{bmatrix}
\cos \gamma & -\sin \gamma \\
\sin \gamma & \cos \gamma
\end{bmatrix} \begin{bmatrix}
\beta_n \\
\beta_s
\end{cases}$$
(2.45a)

or,

$$\begin{cases} \beta_n \\ \beta_s \end{cases} = \begin{bmatrix} \cos \gamma & \sin \gamma \\ -\sin \gamma & \cos \gamma \end{bmatrix} \begin{bmatrix} \beta_x \\ \beta_y \end{bmatrix} \tag{2.45b}$$

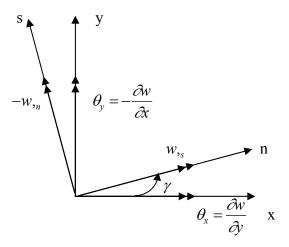


Figure 2.13 Relation between $w_{,n}$, $w_{,s}$ and θ_x , θ_y

Table 2.5 Direction cosines between the two sets of axes

	X	y
n	$\cos \gamma$	$\sin \gamma$
S	$-\sin\gamma$	$\cos \gamma$

$$\begin{cases} w_{,s} \\ -w_{,n} \end{cases} = \begin{bmatrix} \cos \gamma & \sin \gamma \\ -\sin \gamma & \cos \gamma \end{bmatrix} \begin{pmatrix} w_{,y} \\ -w_{,x} \end{pmatrix} = \begin{bmatrix} \cos \gamma & \sin \gamma \\ -\sin \gamma & \cos \gamma \end{bmatrix} \begin{pmatrix} \theta_{x} \\ \theta_{y} \end{pmatrix}$$

that is,

$$\begin{cases}
 w_{x_s} \\
 w_{y_n}
 \end{cases} = \begin{bmatrix}
 \cos \gamma & \sin \gamma \\
 \sin \gamma & -\cos \gamma
 \end{bmatrix} \begin{bmatrix}
 \theta_x \\
 \theta_y
 \end{bmatrix}$$
(2.46)

Another way,

$$w(x,y) = w(n,s)$$

so

$$w_{,n} = \frac{\partial w}{\partial n} = \frac{\partial w}{\partial x} \frac{\partial x}{\partial n} + \frac{\partial w}{\partial y} \frac{\partial y}{\partial n} = \cos \gamma \frac{\partial w}{\partial x} + \sin \gamma \frac{\partial w}{\partial y} = \sin \gamma \theta_x - \cos \gamma \theta_y$$

$$w_{,s} = \frac{\partial w}{\partial s} = \frac{\partial w}{\partial s} \frac{\partial x}{\partial s} + \frac{\partial w}{\partial s} \frac{\partial y}{\partial s} = -\sin \gamma \frac{\partial w}{\partial s} + \cos \gamma \frac{\partial w}{\partial y} = \cos \gamma \theta_x + \sin \gamma \theta_y$$

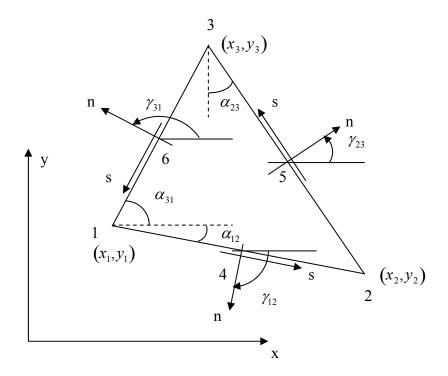


Figure 2.14 Geometry of the triangular element

$$\sin \gamma_{12} = \sin(-(90 + \alpha_{12})) = -\cos \alpha_{12} = \frac{x_1 - x_2}{l_{12}} = \frac{x_{12}}{l_{12}}$$

$$\cos \gamma_{12} = \cos(-(90 + \alpha_{12})) = -\sin \alpha_{12} = -\frac{y_1 - y_2}{l_{12}} = -\frac{y_{12}}{l_{12}}$$

$$\sin \gamma_{23} = \sin \alpha_{23} = \frac{x_2 - x_3}{l_{23}} = \frac{x_{23}}{l_{23}}$$

$$\cos \gamma_{23} = \cos \alpha_{23} = -\frac{y_2 - y_3}{l_{23}} = -\frac{y_{23}}{l_{23}}$$

$$\sin \gamma_{31} = \sin(90 + \alpha_{31}) = \cos \alpha_{31} = \frac{x_3 - x_1}{l_{31}} = \frac{x_{31}}{l_{31}}$$

$$\cos \gamma_{31} = \cos(90 + \alpha_{31}) = -\sin \alpha_{31} = -\frac{y_3 - y_1}{l_{31}} = -\frac{y_{31}}{l_{31}}$$

$$l_{ij} = \sqrt{x_{ij}^2 + y_{ij}^2}, \quad x_{ij} = x_i - x_j, \quad y_{ij} = y_i - y_j$$

$$\sin \gamma_{ij} = \frac{x_{ij}}{l_{ij}}, \quad \cos \gamma_{ij} = -\frac{y_{ij}}{l_{ij}}$$

$$x_k = \frac{1}{2}(x_i + x_j), \quad y_k = \frac{1}{2}(y_i + y_j)$$

where k = 4, 5, 6 for the sides ij = 12, 23, 31, respectively.

The formulation of the DKT element is based on the following assumptions

(1) β_x and β_y are given by complete quadratic polynomials, i.e.

$$\beta_{x} = \sum_{i=1}^{6} N_{i} \beta_{x_{i}}, \quad \beta_{y} = \sum_{i=1}^{6} N_{i} \beta_{y_{i}}$$
(2.47)

Shape functions N_i are given in Appendix F.

- (2) The Kirchhoff hypothesis is imposed at,
 - (a) the corner nodes

$$\gamma = \begin{cases} \beta_x + w_{,x} \\ \beta_y + w_{,y} \end{cases} = 0$$
 at the nodes 1, 2, and 3 (2.48)

(b) the mid-nodes (defined anticlockwise around the element boundary)
$$\beta_{s_k} + w_{,s_k} = 0 \qquad \qquad k=4, 5, 6$$
 (2.49)

(3) The variation of w along the sides is cubic. The slope w_{s} at the mid-nodes can be obtained from beam shape functions by differentiating once and evaluating at the mid-span(see Appendix H), i.e.

$$w_{s_{s_k}} = -\frac{3}{2l_{ii}} w_i - \frac{1}{4} w_{s_{s_i}} + \frac{3}{2l_{ii}} w_j - \frac{1}{4} w_{s_{s_j}}$$
 $k = 4, 5, 6$ (2.50)

(4) The normal slope β_n along each side vary linearly. Therefore,

$$\beta_{n_k} = \frac{1}{2} (\beta_{n_i} + \beta_{n_j}) \qquad k = 4, 5, 6$$
 (2.51)

where k = 4, 5, 6 denotes the mid-node of the sides 12, 23 and 31, respectively.

$$\beta_{x} = \begin{bmatrix} H_{x_{1}} & H_{x_{2}} & H_{x_{3}} & H_{x_{4}} & H_{x_{5}} & H_{x_{6}} & H_{x_{7}} & H_{x_{8}} & H_{x_{9}} \end{bmatrix} \begin{pmatrix} w_{1} \\ \theta_{x_{1}} \\ \theta_{y_{1}} \\ w_{2} \\ \theta_{y_{2}} \\ \theta_{y_{2}} \\ w \\ \theta_{x_{3}} \\ \theta_{y_{3}} \end{pmatrix} = H_{x}^{T} \begin{pmatrix} w_{1} \\ \theta_{x_{1}} \\ \theta_{y_{1}} \\ w_{2} \\ \theta_{y_{2}} \\ w \\ \theta_{x_{3}} \\ \theta_{y_{3}} \end{pmatrix}$$

$$\beta_{x} = N_{1}\beta_{x_{1}} + N_{2}\beta_{x_{2}} + N_{3}\beta_{x_{3}} + N_{4}\beta_{x_{4}} + N_{5}\beta_{x_{5}} + N_{6}\beta_{x_{6}}$$

using (2.48) and (2.45a)

$$\beta_{x} = -N_{1}w_{,x_{1}} - N_{2}w_{,x_{2}} - N_{3}w_{,x_{3}} + N_{4}(\cos\gamma_{12}\beta_{n_{4}} - \sin\gamma_{12}\beta_{s_{4}})$$

$$+ N_{5}(\cos\gamma_{23}\beta_{n_{5}} - \sin\gamma_{23}\beta_{s_{5}}) + N_{6}(\cos\gamma_{31}\beta_{n_{6}} - \sin\gamma_{31}\beta_{s_{6}})$$

using (2.49) and (2.51)

$$\beta_{x} = N_{1}\theta_{y_{1}} + N_{2}\theta_{y_{2}} + N_{3}\theta_{y_{3}} + N_{4}\cos\gamma_{12}\frac{1}{2}(\beta_{n_{1}} + \beta_{n_{2}}) + N_{4}\sin\gamma_{12}w_{,s_{4}}$$

$$+ N_{5}\cos\gamma_{23}\frac{1}{2}(\beta_{n_{2}} + \beta_{n_{3}}) + N_{5}\sin\gamma_{23}w_{,s_{5}} + N_{6}\cos\gamma_{31}\frac{1}{2}(\beta_{n_{3}} + \beta_{n_{1}}) + N_{6}\sin\gamma_{31}w_{,s_{6}}$$

using (2.45b)

$$\begin{split} \boldsymbol{\beta}_{x} &= N_{1}\boldsymbol{\theta}_{y_{1}} + N_{2}\boldsymbol{\theta}_{y_{2}} + N_{3}\boldsymbol{\theta}_{y_{3}} \\ &+ N_{4}\cos\gamma_{12}\frac{1}{2}(\cos\gamma_{12}\boldsymbol{\beta}_{x_{1}} + \sin\gamma_{12}\boldsymbol{\beta}_{y_{1}} + \cos\gamma_{12}\boldsymbol{\beta}_{x_{2}} + \sin\gamma_{12}\boldsymbol{\beta}_{y_{2}}) + N_{4}\sin\gamma_{12}\boldsymbol{w},_{s_{4}} \\ &+ N_{5}\cos\gamma_{23}\frac{1}{2}(\cos\gamma_{23}\boldsymbol{\beta}_{x_{2}} + \sin\gamma_{23}\boldsymbol{\beta}_{y_{2}} + \cos\gamma_{23}\boldsymbol{\beta}_{x_{3}} + \sin\gamma_{23}\boldsymbol{\beta}_{y_{3}}) + N_{5}\sin\gamma_{23}\boldsymbol{w},_{s_{5}} \\ &+ N_{6}\cos\gamma_{31}\frac{1}{2}(\cos\gamma_{31}\boldsymbol{\beta}_{x_{3}} + \sin\gamma_{31}\boldsymbol{\beta}_{y_{3}} + \cos\gamma_{31}\boldsymbol{\beta}_{x_{1}} + \sin\gamma_{31}\boldsymbol{\beta}_{y_{1}}) + N_{6}\sin\gamma_{31}\boldsymbol{w},_{s_{6}} \end{split}$$

using (2.48)

$$\begin{split} \beta_{x} &= N_{1}\theta_{y_{1}} + N_{2}\theta_{y_{2}} + N_{3}\theta_{y_{3}} \\ &+ N_{4}\cos\gamma_{12}\frac{1}{2}(\cos\gamma_{12}(-w_{,x_{1}}) + \sin\gamma_{12}(-w_{,y_{1}}) + \cos\gamma_{12}(-w_{,x_{2}}) + \sin\gamma_{12}(-w_{,y_{2}})) + N_{4}\sin\gamma_{12}w_{,s_{4}} \\ &+ N_{5}\cos\gamma_{23}\frac{1}{2}(\cos\gamma_{23}(-w_{,x_{2}}) + \sin\gamma_{23}(-w_{,y_{2}}) + \cos\gamma_{23}(-w_{,x_{3}}) + \sin\gamma_{23}(-w_{,y_{3}})) + N_{5}\sin\gamma_{23}w_{,s_{5}} \\ &+ N_{6}\cos\gamma_{31}\frac{1}{2}(\cos\gamma_{31}(-w_{,x_{3}}) + \sin\gamma_{31}(-w_{,y_{3}}) + \cos\gamma_{31}(-w_{,x_{1}}) + \sin\gamma_{31}(-w_{,y_{1}})) + N_{6}\sin\gamma_{31}w_{,s_{6}} \\ &= N_{1}\theta_{y_{1}} + N_{2}\theta_{y_{2}} + N_{3}\theta_{y_{3}} \\ &+ N_{4}\cos\gamma_{12}\frac{1}{2}(\cos\gamma_{12}\theta_{y_{1}} - \sin\gamma_{12}\theta_{x_{1}} + \cos\gamma_{12}\theta_{y_{2}} - \sin\gamma_{12}\theta_{x_{2}}) + N_{4}\sin\gamma_{12}w_{,s_{4}} \\ &+ N_{5}\cos\gamma_{23}\frac{1}{2}(\cos\gamma_{23}\theta_{y_{2}} - \sin\gamma_{23}\theta_{x_{2}} + \cos\gamma_{23}\theta_{y_{3}} - \sin\gamma_{23}\theta_{x_{3}}) + N_{5}\sin\gamma_{23}w_{,s_{5}} \\ &+ N_{6}\cos\gamma_{31}\frac{1}{2}(\cos\gamma_{31}\theta_{y_{3}} - \sin\gamma_{31}\theta_{x_{3}} + \cos\gamma_{31}\theta_{y_{1}} - \sin\gamma_{31}\theta_{x_{1}}) + N_{6}\sin\gamma_{31}w_{,s_{6}} \end{split}$$

using (2.50)

$$\begin{split} \beta_{x} &= N_{1}\theta_{y_{1}} + N_{2}\theta_{y_{2}} + N_{3}\theta_{y_{3}} \\ &+ N_{4}\cos\gamma_{12}\frac{1}{2}(\cos\gamma_{12}\theta_{y_{1}} - \sin\gamma_{12}\theta_{x_{1}} + \cos\gamma_{12}\theta_{y_{2}} - \sin\gamma_{12}\theta_{x_{2}}) \\ &+ N_{4}\sin\gamma_{12}(-\frac{3}{2l_{12}}w_{1} - \frac{1}{4}w_{,s_{1}} + \frac{3}{2l_{12}}w_{2} - \frac{1}{4}w_{,s_{2}}) \\ &+ N_{5}\cos\gamma_{23}\frac{1}{2}(\cos\gamma_{23}\theta_{y_{2}} - \sin\gamma_{23}\theta_{x_{2}} + \cos\gamma_{23}\theta_{y_{3}} - \sin\gamma_{23}\theta_{x_{3}}) \\ &+ N_{5}\sin\gamma_{23}(-\frac{3}{2l_{23}}w_{2} - \frac{1}{4}w_{,s_{2}} + \frac{3}{2l_{23}}w_{3} - \frac{1}{4}w_{,s_{3}}) \\ &+ N_{6}\cos\gamma_{31}\frac{1}{2}(\cos\gamma_{31}\theta_{y_{3}} - \sin\gamma_{31}\theta_{x_{3}} + \cos\gamma_{31}\theta_{y_{1}} - \sin\gamma_{31}\theta_{x_{1}}) \\ &+ N_{6}\sin\gamma_{31}(-\frac{3}{2l_{31}}w_{3} - \frac{1}{4}w_{,s_{3}} + \frac{3}{2l_{31}}w_{1} - \frac{1}{4}w_{,s_{1}}) \end{split}$$

using (2.46)

$$\begin{split} \beta_{x} &= N_{1}\theta_{y_{1}} + N_{2}\theta_{y_{2}} + N_{3}\theta_{y_{3}} \\ &+ N_{4}\cos\gamma_{12}\frac{1}{2}(\cos\gamma_{12}\theta_{y_{1}} - \sin\gamma_{12}\theta_{x_{1}} + \cos\gamma_{12}\theta_{y_{2}} - \sin\gamma_{12}\theta_{x_{2}}) \\ &+ N_{4}\sin\gamma_{12}(-\frac{3}{2l_{12}}w_{1} - \frac{1}{4}(\cos\gamma_{12}\theta_{x_{1}} + \sin\gamma_{12}\theta_{y_{1}}) + \frac{3}{2l_{12}}w_{2} - \frac{1}{4}(\cos\gamma_{12}\theta_{x_{2}} + \sin\gamma_{12}\theta_{y_{2}}) \\ &+ N_{5}\cos\gamma_{23}\frac{1}{2}(\cos\gamma_{23}\theta_{y_{2}} - \sin\gamma_{23}\theta_{x_{2}} + \cos\gamma_{23}\theta_{y_{3}} - \sin\gamma_{23}\theta_{x_{3}}) \\ &+ N_{5}\sin\gamma_{23}(-\frac{3}{2l_{23}}w_{2} - \frac{1}{4}(\cos\gamma_{23}\theta_{x_{2}} + \sin\gamma_{23}\theta_{y_{2}}) + \frac{3}{2l_{23}}w_{3} - \frac{1}{4}(\cos\gamma_{23}\theta_{x_{3}} + \sin\gamma_{23}\theta_{y_{3}})) \\ &+ N_{6}\cos\gamma_{31}\frac{1}{2}(\cos\gamma_{31}\theta_{y_{3}} - \sin\gamma_{31}\theta_{x_{3}} + \cos\gamma_{31}\theta_{y_{1}} - \sin\gamma_{31}\theta_{x_{1}}) \\ &+ N_{6}\sin\gamma_{31}(-\frac{3}{2l_{31}}w_{3} - \frac{1}{4}(\cos\gamma_{31}\theta_{x_{3}} + \sin\gamma_{31}\theta_{y_{3}}) + \frac{3}{2l_{31}}w_{1} - \frac{1}{4}(\cos\gamma_{31}\theta_{x_{1}} + \sin\gamma_{31}\theta_{y_{1}})) \end{split}$$

Collecting the coefficient of w_1 , θ_{x_1} , θ_{y_1} , etc., we get the shape function

$$\begin{split} H_{x_1} &= -N_4 \sin \gamma_{12} \frac{3}{2l_{12}} + N_6 \sin \gamma_{31} \frac{3}{2l_{31}} = -N_4 \frac{x_{12}}{l_{12}} \frac{3}{2l_{12}} + N_6 \frac{x_{31}}{l_{31}} \frac{3}{2l_{31}} = 1.5(a_4 N_4 - a_6 N_6) \\ H_{x_2} &= -\frac{3}{4} N_4 \sin \gamma_{12} \cos \gamma_{12} - \frac{3}{4} N_6 \sin \gamma_{31} \cos \gamma_{31} = -\frac{3}{4} N_4 \frac{x_{12}}{l_{12}} (-\frac{y_{12}}{l_{12}}) - \frac{3}{4} N_6 \frac{x_{31}}{l_{31}} (-\frac{y_{31}}{l_{31}}) = b_4 N_4 + b_6 N_6 \\ H_{x_3} &= N_1 - (\frac{1}{4} \sin^2 \gamma_{12} - \frac{1}{2} \cos^2 \gamma_{12}) N_4 - (\frac{1}{4} \sin^2 \gamma_{31} - \frac{1}{2} \cos^2 \gamma_{31}) N_6 \\ &= N_1 - (\frac{1}{4} (\frac{x_{12}}{l_{12}})^2 - \frac{1}{2} (-\frac{y_{12}}{l_{12}})^2) N_4 - (\frac{1}{4} (\frac{x_{31}}{l_{31}})^2 - \frac{1}{2} (-\frac{y_{31}}{l_{31}})^2) N_6 \\ &= N_1 - c_4 N_4 - c_6 N_6 \\ H_{x_4} &= N_4 \sin \gamma_{12} \frac{3}{2l_{12}} - N_5 \sin \gamma_{23} \frac{3}{2l_{23}} = N_4 \frac{x_{12}}{l_{12}} \frac{3}{2l_{12}} - N_5 \frac{x_{23}}{l_{23}} \frac{3}{2l_{23}} = 1.5(a_5 N_5 - a_4 N_4) \\ H_{x_5} &= -\frac{3}{4} N_4 \sin \gamma_{12} \cos \gamma_{12} - \frac{3}{4} N_5 \sin \gamma_{23} \cos \gamma_{23} = -\frac{3}{4} N_4 \frac{x_{12}}{l_{12}} (-\frac{y_{12}}{l_{12}}) - \frac{3}{4} N_5 \frac{x_{23}}{l_{23}} (-\frac{y_{23}}{l_{23}}) = b_4 N_4 + b_5 N_5 \\ H_{x_6} &= N_2 - (\frac{1}{4} \sin^2 \gamma_{12} - \frac{1}{2} \cos^2 \gamma_{12}) N_4 - (\frac{1}{4} \sin^2 \gamma_{23} - \frac{1}{2} \cos^2 \gamma_{23}) N_5 \\ &= N_2 - (\frac{1}{4} (\frac{x_{12}}{l_{12}})^2 - \frac{1}{2} (-\frac{y_{12}}{l_{12}})^2) N_4 - (\frac{1}{4} (\frac{x_{23}}{l_{23}})^2 - \frac{1}{2} (-\frac{y_{23}}{l_{23}})^2) N_5 \\ &= N_2 - c_4 N_4 - c_5 N_5 \end{split}$$

$$\begin{split} H_{x_7} &= -N_6 \sin \gamma_{31} \frac{3}{2l_{31}} + N_5 \sin \gamma_{23} \frac{3}{2l_{23}} = -N_6 \frac{x_{31}}{l_{31}} \frac{3}{2l_{31}} + N_5 \frac{x_{23}}{l_{23}} \frac{3}{2l_{23}} = 1.5(a_6 N_6 - a_5 N_5) \\ H_{x_8} &= -\frac{3}{4} N_5 \sin \gamma_{23} \cos \gamma_{23} - \frac{3}{4} N_6 \sin \gamma_{31} \cos \gamma_{31} = -\frac{3}{4} N_5 \frac{x_{23}}{l_{23}} (-\frac{y_{23}}{l_{23}}) - \frac{3}{4} N_6 \frac{x_{31}}{l_{31}} (-\frac{y_{31}}{l_{31}}) = b_5 N_5 + b_6 N_6 \\ H_{x_9} &= N_3 - (\frac{1}{4} \sin^2 \gamma_{23} - \frac{1}{2} \cos^2 \gamma_{23}) N_5 - (\frac{1}{4} \sin^2 \gamma_{31} - \frac{1}{2} \cos^2 \gamma_{31}) N_6 \\ &= N_3 - (\frac{1}{4} (\frac{x_{23}}{l_{23}})^2 - \frac{1}{2} (-\frac{y_{23}}{l_{23}})^2) N_5 - (\frac{1}{4} (\frac{x_{31}}{l_{31}})^2 - \frac{1}{2} (-\frac{y_{31}}{l_{31}})^2) N_6 \\ &= N_3 - c_5 N_5 - c_6 N_6 \end{split}$$

where

$$a_4 = -x_{12}/l_{12}^2, \quad a_5 = -x_{23}/l_{23}^2, \quad a_6 = -x_{31}/l_{31}^2$$

$$b_4 = \frac{3}{4} \frac{x_{12}y_{12}}{l_{12}^2}, \quad b_5 = \frac{3}{4} \frac{x_{23}y_{23}}{l_{23}^2}, \quad b_6 = \frac{3}{4} \frac{x_{31}y_{31}}{l_{31}^2}$$

$$c_4 = (\frac{1}{4}x_{12}^2 - \frac{1}{2}y_{12}^2)/l_{12}^2, \quad c_5 = (\frac{1}{4}x_{23}^2 - \frac{1}{2}y_{23}^2)/l_{23}^2, \quad c_6 = (\frac{1}{4}x_{31}^2 - \frac{1}{2}y_{31}^2)/l_{31}^2$$

Similarly

$$\beta_{y} = \begin{bmatrix} H_{y_{1}} & H_{y_{2}} & H_{y_{3}} & H_{y_{4}} & H_{y_{5}} & H_{y_{6}} & H_{y_{7}} & H_{y_{8}} & H_{y_{9}} \end{bmatrix} \begin{pmatrix} w_{1} \\ \theta_{x_{1}} \\ \theta_{y_{1}} \\ w_{2} \\ \theta_{x_{2}} \\ \theta_{y_{2}} \\ w \\ \theta_{x_{3}} \\ \theta_{y_{3}} \end{pmatrix} = H_{y}^{T} \begin{pmatrix} w_{1} \\ \theta_{x_{1}} \\ \theta_{y_{1}} \\ w_{2} \\ \theta_{x_{2}} \\ \theta_{y_{2}} \\ w \\ \theta_{x_{3}} \\ \theta_{y_{3}} \end{pmatrix}$$

$$\beta_y = N_1 \beta_{y_1} + N_2 \beta_{y_2} + N_3 \beta_{y_3} + N_4 \beta_{y_4} + N_5 \beta_{y_5} + N_6 \beta_{y_6}$$

using (2.48) and (2.45a)

$$\beta_{y} = -N_{1}w_{,y_{1}} - N_{2}w_{,y_{2}} - N_{3}w_{,y_{3}} + N_{4}(\sin\gamma_{12}\beta_{n_{4}} + \cos\gamma_{12}\beta_{s_{4}})$$

$$+ N_{5}(\sin\gamma_{23}\beta_{n_{5}} + \cos\gamma_{23}\beta_{s_{5}}) + N_{6}(\sin\gamma_{31}\beta_{n_{6}} + \cos\gamma_{31}\beta_{s_{6}})$$

using (2.49) and (2.51)

$$\beta_{y} = -N_{1}\theta_{x_{1}} - N_{2}\theta_{x_{2}} - N_{3}\theta_{x_{3}} + N_{4}\sin\gamma_{12}\frac{1}{2}(\beta_{n_{1}} + \beta_{n_{2}}) - N_{4}\cos\gamma_{12}w_{,s_{4}}$$

$$+ N_{5}\sin\gamma_{23}\frac{1}{2}(\beta_{n_{2}} + \beta_{n_{3}}) - N_{5}\cos\gamma_{23}w_{,s_{5}} + N_{6}\sin\gamma_{31}\frac{1}{2}(\beta_{n_{3}} + \beta_{n_{1}}) - N_{6}\cos\gamma_{31}w_{,s_{6}}$$

using (2.45b)

$$\begin{split} \boldsymbol{\beta}_{y} &= -N_{1}\boldsymbol{\theta}_{x_{1}} - N_{2}\boldsymbol{\theta}_{x_{2}} - N_{3}\boldsymbol{\theta}_{x_{3}} \\ &+ N_{4}\sin\gamma_{12}\frac{1}{2}(\cos\gamma_{12}\boldsymbol{\beta}_{x_{1}} + \sin\gamma_{12}\boldsymbol{\beta}_{y_{1}} + \cos\gamma_{12}\boldsymbol{\beta}_{x_{2}} + \sin\gamma_{12}\boldsymbol{\beta}_{y_{2}}) - N_{4}\cos\gamma_{12}\boldsymbol{w},_{s_{4}} \\ &+ N_{5}\sin\gamma_{23}\frac{1}{2}(\cos\gamma_{23}\boldsymbol{\beta}_{x_{2}} + \sin\gamma_{23}\boldsymbol{\beta}_{y_{2}} + \cos\gamma_{23}\boldsymbol{\beta}_{x_{3}} + \sin\gamma_{23}\boldsymbol{\beta}_{y_{3}}) - N_{5}\cos\gamma_{23}\boldsymbol{w},_{s_{5}} \\ &+ N_{6}\sin\gamma_{31}\frac{1}{2}(\cos\gamma_{31}\boldsymbol{\beta}_{x_{3}} + \sin\gamma_{31}\boldsymbol{\beta}_{y_{3}} + \cos\gamma_{31}\boldsymbol{\beta}_{x_{1}} + \sin\gamma_{31}\boldsymbol{\beta}_{y_{1}}) - N_{6}\cos\gamma_{31}\boldsymbol{w},_{s_{6}} \end{split}$$

using (2.48)

$$\begin{split} \beta_y &= -N_1 \theta_{x_1} - N_2 \theta_{x_2} - N_3 \theta_{x_3} \\ &+ N_4 \sin \gamma_{12} \frac{1}{2} (\cos \gamma_{12} (-w_{,x_1}) + \sin \gamma_{12} (-w_{,y_1}) + \cos \gamma_{12} (-w_{,x_2}) + \sin \gamma_{12} (-w_{,y_2})) - N_4 \cos \gamma_{12} w_{,s_4} \\ &+ N_5 \sin \gamma_{23} \frac{1}{2} (\cos \gamma_{23} (-w_{,x_2}) + \sin \gamma_{23} (-w_{,y_2}) + \cos \gamma_{23} (-w_{,x_3}) + \sin \gamma_{23} (-w_{,y_3})) - N_5 \cos \gamma_{23} w_{,s_5} \\ &+ N_6 \sin \gamma_{31} \frac{1}{2} (\cos \gamma_{31} (-w_{,x_3}) + \sin \gamma_{31} (-w_{,y_3}) + \cos \gamma_{31} (-w_{,x_1}) + \sin \gamma_{31} (-w_{,y_1})) - N_6 \cos \gamma_{31} w_{,s_6} \\ &= -N_1 \theta_{x_1} - N_2 \theta_{x_2} - N_3 \theta_{x_3} \\ &+ N_4 \sin \gamma_{12} \frac{1}{2} (\cos \gamma_{12} \theta_{y_1} - \sin \gamma_{12} \theta_{x_1} + \cos \gamma_{12} \theta_{y_2} - \sin \gamma_{12} \theta_{x_2}) - N_4 \cos \gamma_{12} w_{,s_4} \\ &+ N_5 \sin \gamma_{23} \frac{1}{2} (\cos \gamma_{23} \theta_{y_2} - \sin \gamma_{23} \theta_{x_2} + \cos \gamma_{23} \theta_{y_3} - \sin \gamma_{23} \theta_{x_3}) - N_5 \cos \gamma_{23} w_{,s_5} \\ &+ N_6 \sin \gamma_{31} \frac{1}{2} (\cos \gamma_{31} \theta_{y_3} - \sin \gamma_{31} \theta_{x_3} + \cos \gamma_{31} \theta_{y_1} - \sin \gamma_{31} \theta_{x_1}) - N_6 \cos \gamma_{31} w_{,s_6} \end{split}$$

using (2.50)

$$\begin{split} \beta_{y} &= -N_{1}\theta_{x_{1}} - N_{2}\theta_{x_{2}} - N_{3}\theta_{x_{3}} \\ &+ N_{4}\sin\gamma_{12}\frac{1}{2}(\cos\gamma_{12}\theta_{y_{1}} - \sin\gamma_{12}\theta_{x_{1}} + \cos\gamma_{12}\theta_{y_{2}} - \sin\gamma_{12}\theta_{x_{2}}) \\ &- N_{4}\cos\gamma_{12}(-\frac{3}{2l_{12}}w_{1} - \frac{1}{4}w_{,s_{1}} + \frac{3}{2l_{12}}w_{2} - \frac{1}{4}w_{,s_{2}}) \\ &+ N_{5}\sin\gamma_{23}\frac{1}{2}(\cos\gamma_{23}\theta_{y_{2}} - \sin\gamma_{23}\theta_{x_{2}} + \cos\gamma_{23}\theta_{y_{3}} - \sin\gamma_{23}\theta_{x_{3}}) \\ &- N_{5}\cos\gamma_{23}(-\frac{3}{2l_{23}}w_{2} - \frac{1}{4}w_{,s_{2}} + \frac{3}{2l_{23}}w_{3} - \frac{1}{4}w_{,s_{3}}) \\ &+ N_{6}\sin\gamma_{31}\frac{1}{2}(\cos\gamma_{31}\theta_{y_{3}} - \sin\gamma_{31}\theta_{x_{3}} + \cos\gamma_{31}\theta_{y_{1}} - \sin\gamma_{31}\theta_{x_{1}}) \\ &- N_{6}\cos\gamma_{31}(-\frac{3}{2l_{31}}w_{3} - \frac{1}{4}w_{,s_{3}} + \frac{3}{2l_{31}}w_{1} - \frac{1}{4}w_{,s_{1}}) \end{split}$$

using (2.46)

$$\begin{split} \beta_{y} &= -N_{1}\theta_{x_{1}} - N_{2}\theta_{x_{2}} - N_{3}\theta_{x_{3}} \\ &+ N_{4}\sin\gamma_{12}\frac{1}{2}(\cos\gamma_{12}\theta_{y_{1}} - \sin\gamma_{12}\theta_{x_{1}} + \cos\gamma_{12}\theta_{y_{2}} - \sin\gamma_{12}\theta_{x_{2}}) \\ &- N_{4}\cos\gamma_{12}(-\frac{3}{2l_{12}}w_{1} - \frac{1}{4}(\cos\gamma_{12}\theta_{x_{1}} + \sin\gamma_{12}\theta_{y_{1}}) + \frac{3}{2l_{12}}w_{2} - \frac{1}{4}(\cos\gamma_{12}\theta_{x_{2}} + \sin\gamma_{12}\theta_{y_{2}})) \\ &+ N_{5}\sin\gamma_{23}\frac{1}{2}(\cos\gamma_{23}\theta_{y_{2}} - \sin\gamma_{23}\theta_{x_{2}} + \cos\gamma_{23}\theta_{y_{3}} - \sin\gamma_{23}\theta_{x_{3}}) \\ &- N_{5}\cos\gamma_{23}(-\frac{3}{2l_{23}}w_{2} - \frac{1}{4}(\cos\gamma_{23}\theta_{x_{2}} + \sin\gamma_{23}\theta_{y_{2}}) + \frac{3}{2l_{23}}w_{3} - \frac{1}{4}(\cos\gamma_{23}\theta_{x_{3}} + \sin\gamma_{23}\theta_{y_{3}})) \\ &+ N_{6}\sin\gamma_{31}\frac{1}{2}(\cos\gamma_{31}\theta_{y_{3}} - \sin\gamma_{31}\theta_{x_{3}} + \cos\gamma_{31}\theta_{y_{1}} - \sin\gamma_{31}\theta_{x_{1}}) \\ &- N_{6}\cos\gamma_{31}(-\frac{3}{2l_{31}}w_{3} - \frac{1}{4}(\cos\gamma_{31}\theta_{x_{3}} + \sin\gamma_{31}\theta_{y_{3}}) + \frac{3}{2l_{31}}w_{1} - \frac{1}{4}(\cos\gamma_{31}\theta_{x_{1}} + \sin\gamma_{31}\theta_{y_{1}})) \end{split}$$

collecting the coefficient of w_1 , θ_{x_1} , θ_{y_1} , etc., we get the shape function

$$H_{y_1} = N_4 \cos \gamma_{12} \frac{3}{2l_{12}} - N_6 \cos \gamma_{31} \frac{3}{2l_{31}} = N_4 \left(-\frac{y_{12}}{l_{12}}\right) \frac{3}{2l_{12}} - N_6 \left(-\frac{y_{31}}{l_{31}}\right) \frac{3}{2l_{31}} = 1.5(d_4 N_4 - d_6 N_6)$$

$$H_{y_2} = -N_1 + \left(\frac{1}{4}\cos^2 \gamma_{12} - \frac{1}{2}\sin^2 \gamma_{12}\right) N_4 + \left(\frac{1}{4}\cos^2 \gamma_{31} - \frac{1}{2}\sin^2 \gamma_{31}\right) N_6$$

$$= -N_1 + \left(\frac{1}{4}\left(-\frac{y_{12}}{l_{12}}\right)^2 - \frac{1}{2}\left(\frac{x_{12}}{l_{12}}\right)^2\right) N_4 + \left(\frac{1}{4}\left(-\frac{y_{31}}{l_{31}}\right)^2 - \frac{1}{2}\left(\frac{x_{31}}{l_{31}}\right)^2\right) N_6$$

$$= -N_1 + e_4 N_4 + e_6 N_6$$

$$H_{y_3} = \frac{3}{4} N_4 \sin \gamma_{12} \cos \gamma_{12} + \frac{3}{4} N_6 \sin \gamma_{31} \cos \gamma_{31} = \frac{3}{4} N_4 \frac{x_{12}}{l_{12}} \left(-\frac{y_{12}}{l_{12}} \right) + \frac{3}{4} N_6 \frac{x_{31}}{l_{31}} \left(-\frac{y_{31}}{l_{31}} \right) = -b_4 N_4 - b_6 N_6 = -H_{x_2}$$

$$H_{y_4} = -N_4 \cos \gamma_{12} \frac{3}{2l_{12}} + N_5 \cos \gamma_{23} \frac{3}{2l_{23}} = -N_4 \left(-\frac{y_{12}}{l_{12}} \right) \frac{3}{2l_{12}} + N_5 \left(-\frac{y_{23}}{l_{23}} \right) \frac{3}{2l_{23}} = 1.5 (d_5 N_5 - d_4 N_4)$$

$$H_{y_5} = -N_2 + \left(\frac{1}{4} \cos^2 \gamma_{12} - \frac{1}{2} \sin^2 \gamma_{12} \right) N_4 + \left(\frac{1}{4} \cos^2 \gamma_{23} - \frac{1}{2} \sin^2 \gamma_{23} \right) N_5$$

$$= -N_2 + \left(\frac{1}{4} \left(-\frac{y_{12}}{l_{12}} \right)^2 - \frac{1}{2} \left(\frac{x_{12}}{l_{12}} \right)^2 \right) N_4 + \left(\frac{1}{4} \left(-\frac{y_{23}}{l_{23}} \right)^2 - \frac{1}{2} \left(\frac{x_{23}}{l_{23}} \right)^2 \right) N_5$$

$$= -N_2 + e_4 N_4 + e_5 N_5$$

$$H_{y_6} = \frac{3}{4} N_4 \sin \gamma_{12} \cos \gamma_{12} + \frac{3}{4} N_5 \sin \gamma_{23} \cos \gamma_{23} = \frac{3}{4} N_4 \frac{x_{12}}{l_{12}} \left(-\frac{y_{12}}{l_{12}} \right) + \frac{3}{4} N_5 \frac{x_{23}}{l_{23}} \left(-\frac{y_{23}}{l_{23}} \right) = -b_4 N_4 - b_5 N_5 = -H_{x_5} \left(-\frac{y_{12}}{l_{12}} \right) + \frac{3}{4} N_5 \frac{x_{23}}{l_{23}} \left(-\frac{y_{23}}{l_{23}} \right) = -b_4 N_4 - b_5 N_5 = -H_{x_5} \left(-\frac{y_{12}}{l_{12}} \right) + \frac{3}{4} N_5 \frac{x_{23}}{l_{23}} \left(-\frac{y_{23}}{l_{23}} \right) = -b_4 N_4 - b_5 N_5 = -H_{x_5} \left(-\frac{y_{12}}{l_{12}} \right) + \frac{3}{4} N_5 \frac{x_{23}}{l_{23}} \left(-\frac{y_{23}}{l_{23}} \right) = -b_4 N_4 - b_5 N_5 = -H_{x_5} \left(-\frac{y_{12}}{l_{12}} \right) + \frac{3}{4} N_5 \frac{x_{23}}{l_{23}} \left(-\frac{y_{23}}{l_{23}} \right) = -b_4 N_4 - b_5 N_5 = -H_{x_5} \left(-\frac{y_{12}}{l_{12}} \right) + \frac{3}{4} N_5 \frac{x_{23}}{l_{23}} \left(-\frac{y_{23}}{l_{23}} \right) = -b_4 N_4 - b_5 N_5 = -H_{x_5} \left(-\frac{y_{12}}{l_{12}} \right) + \frac{3}{4} N_5 \frac{x_{23}}{l_{23}} \left(-\frac{y_{23}}{l_{23}} \right) = -b_4 N_4 - b_5 N_5 = -H_{x_5} \left(-\frac{y_{12}}{l_{12}} \right) + \frac{3}{4} N_5 \frac{x_{23}}{l_{23}} \left(-\frac{y_{23}}{l_{23}} \right) = -b_4 N_4 - b_5 N_5 = -H_{x_5} \left(-\frac{y_{12}}{l_{12}} \right) + \frac{3}{4} N_5 \frac{x_{23}}{l_{23}} \left(-\frac{y_{23}}{l_{23}} \right) = -b_4 N_4 - b_5 N_5 = -H_{x_5} \left(-\frac{y_{12}}{l_{23}} \right) = -b_4 N_5 + b_5 N_5 = -H_{x_5} \left(-\frac{y_{12}}{l_{12}} \right) = -b_5 N_5 + b_5 N_5 = -h_5 N$$

$$H_{y_7} = -N_5 \cos \gamma_{23} \frac{3}{2l_{23}} + N_6 \cos \gamma_{31} \frac{3}{2l_{31}} = -N_5 \left(-\frac{y_{23}}{l_{23}}\right) \frac{3}{2l_{23}} + N_6 \left(-\frac{y_{31}}{l_{31}}\right) \frac{3}{2l_{31}} = 1.5(d_6 N_6 - d_5 N_5)$$

$$\begin{split} H_{y_8} &= -N_3 + (\frac{1}{4}\cos^2\gamma_{23} - \frac{1}{2}\sin^2\gamma_{23})N_5 + (\frac{1}{4}\cos^2\gamma_{31} - \frac{1}{2}\sin^2\gamma_{31})N_6 \\ &= -N_3 + (\frac{1}{4}(-\frac{y_{23}}{l_{23}})^2 - \frac{1}{2}(\frac{x_{23}}{l_{23}})^2)N_5 + (\frac{1}{4}(-\frac{y_{31}}{l_{31}})^2 - \frac{1}{2}(\frac{x_{31}}{l_{31}})^2)N_6 \\ &= -N_3 + e_5N_5 + e_6N_6 \end{split}$$

$$H_{y_9} = \frac{3}{4} N_5 \sin \gamma_{23} \cos \gamma_{23} + \frac{3}{4} N_6 \sin \gamma_{31} \cos \gamma_{31} = \frac{3}{4} N_5 \frac{x_{23}}{l_{23}} \left(-\frac{y_{23}}{l_{23}} \right) + \frac{3}{4} N_6 \frac{x_{31}}{l_{31}} \left(-\frac{y_{31}}{l_{31}} \right) = -b_5 N_5 - b_6 N_6 = -H_{x_8}$$

where

$$b_4 = \frac{3}{4} \frac{x_{12} y_{12}}{l_{12}^2}, \quad b_5 = \frac{3}{4} \frac{x_{23} y_{23}}{l_{23}^2}, \quad b_6 = \frac{3}{4} \frac{x_{31} y_{31}}{l_{31}^2}$$

$$d_4 = -y_{12}/l_{12}^2, \quad d_5 = -y_{23}/l_{23}^2, \quad d_6 = -y_{31}/l_{31}^2$$

$$e_4 = (\frac{1}{4} y_{12}^2 - \frac{1}{2} x_{12}^2)/l_{12}^2, \quad e_5 = (\frac{1}{4} y_{23}^2 - \frac{1}{2} x_{23}^2)/l_{23}^2, \quad e_6 = (\frac{1}{4} y_{31}^2 - \frac{1}{2} x_{31}^2)/l_{31}^2$$

For a summary of the shape functions and their derivatives of the Discrete Kirchhoff Triangular (DKT) Element, see Appendix I.

$$\kappa = \left\{ \begin{array}{c} \boldsymbol{\beta}_{x,x} \\ \boldsymbol{\beta}_{y,y} \\ \boldsymbol{\beta}_{x,y} + \boldsymbol{\beta}_{y,x} \end{array} \right\} = \begin{bmatrix} \boldsymbol{H}_{x,x}^T \\ \boldsymbol{H}_{y,y}^T \\ \boldsymbol{H}_{x,y}^T + \boldsymbol{H}_{y,x}^T \end{bmatrix} \left\{ \begin{array}{c} \boldsymbol{W}_1 \\ \boldsymbol{\theta}_{x1} \\ \boldsymbol{W}_2 \\ \boldsymbol{\theta}_{x2} \\ \boldsymbol{\theta}_{y2} \\ \boldsymbol{W}_3 \\ \boldsymbol{\theta}_{x3} \\ \boldsymbol{\theta}_{x3} \end{array} \right\} = \begin{bmatrix} \frac{\partial \boldsymbol{H}_x^T}{\partial \xi} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_x^T}{\partial \eta} \frac{\partial \eta}{\partial x} \\ \frac{\partial \boldsymbol{H}_y^T}{\partial \xi} \frac{\partial \xi}{\partial y} + \frac{\partial \boldsymbol{H}_y^T}{\partial \eta} \frac{\partial \eta}{\partial y} \\ \frac{\partial \boldsymbol{H}_y^T}{\partial \xi} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_y^T}{\partial \eta} \frac{\partial \eta}{\partial y} \\ \frac{\partial \boldsymbol{H}_y^T}{\partial \xi} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_x^T}{\partial \eta} \frac{\partial \eta}{\partial y} \\ \frac{\partial \boldsymbol{H}_y^T}{\partial \xi} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_y^T}{\partial \eta} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_y^T}{\partial \eta} \frac{\partial \eta}{\partial x} \\ \frac{\partial \boldsymbol{H}_y^T}{\partial \xi} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_x^T}{\partial \eta} \frac{\partial \xi}{\partial y} + \frac{\partial \boldsymbol{H}_y^T}{\partial \eta} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_y^T}{\partial \eta} \frac{\partial \eta}{\partial x} \\ \frac{\partial \boldsymbol{H}_y^T}{\partial \xi} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_x^T}{\partial \eta} \frac{\partial \xi}{\partial y} + \frac{\partial \boldsymbol{H}_y^T}{\partial \eta} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_y^T}{\partial \eta} \frac{\partial \eta}{\partial x} \\ \frac{\partial \boldsymbol{H}_y^T}{\partial \xi} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_x^T}{\partial \eta} \frac{\partial \xi}{\partial y} + \frac{\partial \boldsymbol{H}_y^T}{\partial \eta} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_y^T}{\partial \eta} \frac{\partial \eta}{\partial x} \\ \frac{\partial \boldsymbol{H}_y^T}{\partial \xi} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_x^T}{\partial \eta} \frac{\partial \xi}{\partial y} + \frac{\partial \boldsymbol{H}_x^T}{\partial \eta} \frac{\partial \xi}{\partial x} + \frac{\partial \boldsymbol{H}_y^T}{\partial \eta} \frac{\partial \eta}{\partial x} \\ \frac{\partial \boldsymbol{H}_y^T}{\partial \xi} \frac{\partial \boldsymbol{H}_y^T}{\partial \eta} \frac{\partial \boldsymbol{H}$$

$$=\frac{1}{2A}\begin{bmatrix} y_{31}H_{x,\xi}^{T}+y_{12}H_{x,\eta}^{T} \\ -x_{31}H_{y,\xi}^{T}-x_{12}H_{y,\eta}^{T} \\ -x_{31}H_{x,\xi}^{T}-x_{12}H_{x,\eta}^{T}+y_{31}H_{y,\xi}^{T}+y_{12}H_{y,\eta}^{T} \end{bmatrix} \begin{pmatrix} w_{1} \\ \theta_{x1} \\ \theta_{y1} \\ w_{2} \\ \theta_{x2} \\ \theta_{y2} \\ w_{3} \\ \theta_{y3} \end{pmatrix}$$

(2.52)

where

$$B = \frac{1}{2A} \begin{bmatrix} y_{31}H_{x,\xi}^T + y_{12}H_{x,\eta}^T \\ -x_{31}H_{y,\xi}^T - x_{12}H_{y,\eta}^T \\ -x_{31}H_{x,\eta}^T + y_{31}H_{y,\xi}^T + y_{12}H_{y,\eta}^T \end{bmatrix}_{3\times 9} = \frac{1}{2A} \begin{bmatrix} b_2H_{x,\xi}^T + b_3H_{x,\eta}^T \\ c_2H_{y,\xi}^T + c_3H_{y,\eta}^T \\ c_2H_{x,\xi}^T + c_3H_{x,\eta}^T + b_2H_{y,\xi}^T + b_3H_{y,\eta}^T \end{bmatrix}_{3\times 9} \\ b_2 = y_{31} = y_3 - y_1, \quad b_3 = y_{12} = y_1 - y_2, \quad c_2 = -x_{31} = x_1 - x_3, \quad c_3 = -x_{12} = x_2 - x_1 \end{bmatrix}$$

$$\begin{split} U_b &= \frac{1}{2} \iiint_V \left(\mathcal{E}_x \quad \mathcal{E}_y \quad \gamma_{xy} \right) \begin{cases} \sigma_x \\ \sigma_y \\ \tau_{xy} \end{cases} dV = \frac{1}{2} \iint_A \left(\beta_{x,x} \quad \beta_{y,y} \quad \beta_{x,y} + \beta_{y,x} \right) D_b \begin{cases} \beta_{x,x} \\ \beta_{y,y} \\ \beta_{x,y} + \beta_{y,x} \end{cases} dx dy \\ &= \frac{1}{2} U^T \iint_A B^T D_b B dx dy U \\ &= \frac{1}{2} U^T k_{DKT} U \end{split}$$

$$k_{DKT} = \iint_{A} B^{T} D_{b} B dx dy = \int_{0}^{1} \int_{0}^{1-\xi} B^{T} D_{b} B |J| d\xi d\eta$$
 (2.53)

$$|J| = \begin{vmatrix} \frac{\partial x}{\partial \xi} & \frac{\partial y}{\partial \xi} \\ \frac{\partial x}{\partial \eta} & \frac{\partial y}{\partial \eta} \end{vmatrix} = \begin{vmatrix} x_2 - x_1 & y_2 - y_1 \\ x_3 - x_1 & y_3 - y_1 \end{vmatrix} = \begin{vmatrix} 1 & x_1 & y_1 \\ 1 & x_2 & y_2 \\ 1 & x_3 & y_3 \end{vmatrix} = 2A$$

$$k_{DKT} = \int_0^1 \int_0^{1-\xi} B^T D_b B |J| d\xi d\eta = 2A \int_0^1 \int_0^{1-\xi} B^T D_b B d\xi d\eta$$

$$D_{b} = \frac{1}{3} \sum_{i=1}^{n} \begin{bmatrix} \overline{Q}_{11} & \overline{Q}_{12} & \overline{Q}_{16} \\ \overline{Q}_{21} & \overline{Q}_{22} & \overline{Q}_{26} \\ \overline{Q}_{16} & \overline{Q}_{26} & \overline{Q}_{66} \end{bmatrix} (h_{i+1}^{3} - h_{i}^{3})$$

It is helpful to note the following points (Batoz, et al., 1980),

- (1) From the relations (2.47)-(2.51) it is seen that the relation between the rotations and the transverse displacement w is given by equation (2.49) by assuming a cubic variation of w along the sides (or a quadratic variation of w_{s_s}).
- (2) There is no need to define an interpolation function for w on the element (but the assumption of a cubic variation of w along the sides is the property of a cubic polynomial on the element).
- (3) Since w varies cubically along the sides, w_{s} varies quadratically and so does β_{s} . Since w_{s} matches β_{s} at the three points along each side, the Kirchhoff hypothesis $(\gamma_{s} = \beta_{s} + w_{s} = 0)$ is satisfied along the entire boundary.
- (4) Convergence towards the classical thin plate solution is obtained because the transverse shear strain energy is neglected, equation (2.53), and because the Kirchhoff hypothesis is satisfied along the element boundary.
- (5) It follows from equations (2.47-2.51) that w, $w_{,s}$, β_s and β_n are compatible (interelement continuous) along the sides (note that $w_{,n}$ has not been introduced since w is not defined in the interior).

- (6) With the restriction imposed on β_n in equation (2.51), β_x and β_y are given by complete polynomials of degree one in the element. The global discretization error in the energy is $o(l^2)$, where 1 is the measure of the size of the element.
- (7) When the formulation is applied to a one-dimensional beam, the exact stiffness matrix of a thin beam (with a cubic polynomial w) is obtained.

It is also noted that lateral deflection w is defined only on element sides, using d.o.f. of vertex nodes. On each side w is a cubic in the side-tangent coordinate s. Because w is defined only along element sides, consistent formulations for an element load vector and an elastic foundation stiffness matrix are not available. It is helpful that there is an alternative formulation of the DKT element that defines w within the element[Crisfield, 1986]. The nine term cubic polynomial in area coordinates given by Specht (1988) was also used to represent the transverse displacement by Kapania and Mohan (1996) to compute the mass matrix for free vibration analysis.

2.7 Shell Element

Flat Triangular Shell Element

Local coordinates and direction cosines for a triangular element arbitrarily orientated in space

When we use triangular elements to model an arbitrary shell, we will face the problem of how to define local coordinates and their direction cosines. Because the direction of one of the local axes is arbitrary, the local coordinates are not unique to choose. In our study, we will define them as followed from vector algebra.

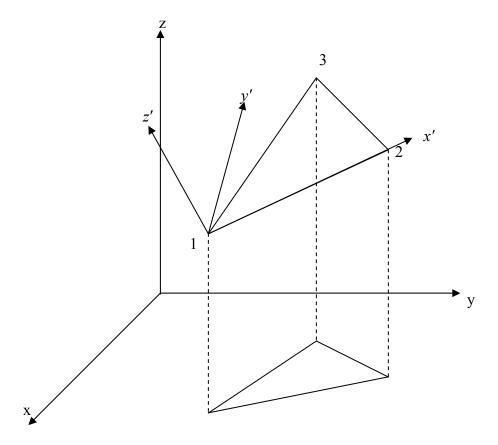


Figure 2.15 Local and global coordinate systems for a triangular element

The local x' axis is chosen to be directed along the side 12 of the triangle. The vector \vec{V}_{21} defines this side and in terms of global coordinates we have

$$\vec{V}_{x'} = \vec{V}_{21} = \begin{cases} x_2 - x_1 \\ y_2 - y_1 \\ z_2 - z_1 \end{cases} = \begin{cases} x_{21} \\ y_{21} \\ z_{21} \end{cases}$$

The z' direction is normal to the plane of the triangle. We choose this direction by means of cross-product of the two vectors \vec{V}_{21} and \vec{V}_{31} formed by the two sides of the triangle. The vector \vec{V}_{31} is

$$\vec{V}_{31} = \begin{cases} x_3 - x_1 \\ y_3 - y_1 \\ z_3 - z_1 \end{cases} = \begin{cases} x_{31} \\ y_{31} \\ z_{31} \end{cases}$$

Thus, we have

$$\vec{V}_{z'} = \vec{V}_{21} \times \vec{V}_{31} = \begin{vmatrix} \vec{i} & \vec{j} & \vec{k} \\ x_{21} & y_{21} & z_{21} \\ x_{31} & y_{31} & z_{31} \end{vmatrix} = \begin{cases} y_{21}z_{31} - z_{21}y_{31} \\ z_{21}x_{31} - x_{21}z_{31} \\ x_{21}y_{31} - y_{21}x_{31} \end{cases} = \begin{cases} V_{z'1} \\ V_{z'2} \\ V_{z'3} \end{cases}$$

The y' direction can be obtained from the cross-product of two vectors $\vec{V}_{z'}$ and $\vec{V}_{x'}$,

$$\vec{V}_{y'} = \vec{V}_{z'} \times \vec{V}_{x'} = \begin{vmatrix} \vec{i} & \vec{j} & \vec{k} \\ V_{z'1} & V_{z'2} & V_{z'3} \\ x_{21} & y_{21} & z_{21} \end{vmatrix} = \begin{cases} V_{z'2} z_{21} - V_{z'3} y_{21} \\ V_{z'3} x_{21} - V_{z'1} z_{21} \\ V_{z'1} y_{21} - V_{z'2} x_{21} \end{cases} = \begin{cases} V_{y'1} \\ V_{y'2} \\ V_{y'3} \end{cases}$$

The direction cosines are given by dividing the components of each vector by its length, that is, defining a vector of unit length

$$\vec{v}_{x'} = \begin{cases} \lambda_{x'x} \\ \lambda_{y'y} \\ \lambda_{z'z} \end{cases} = \frac{1}{l_{12}} \begin{cases} x_{21} \\ y_{21} \\ z_{21} \end{cases}$$
 (2.54a)

$$\vec{v}_{y'} = \begin{cases} \lambda_{y'x} \\ \lambda_{y'y} \\ \lambda_{y'z} \end{cases} = \frac{1}{l_{y'}} \begin{cases} V_{z'2} z_{21} - V_{z'3} y_{21} \\ V_{z'3} x_{21} - V_{z'1} z_{21} \\ V_{z'1} y_{21} - V_{z'2} x_{21} \end{cases}$$
(2.54b)

$$\vec{v}_{z'} = \begin{cases} \lambda_{z'x} \\ \lambda_{z'y} \\ \lambda_{z'z} \end{cases} = \frac{1}{l_{z'}} \begin{cases} y_{21}z_{31} - z_{21}y_{31} \\ z_{21}x_{31} - x_{21}z_{31} \\ x_{21}y_{31} - y_{21}x_{31} \end{cases}$$
(2.54c)

where

$$l_{12} = \sqrt{x_{21}^2 + y_{21}^2 + z_{21}^2}$$

$$l_{y'} = \sqrt{V_{y'1}^2 + V_{y'2}^2 + V_{y'3}^2}$$

 $l_{z'}=2\Delta=\sqrt{V_{z'1}^2+V_{z'2}^2+V_{z'3}^2}$, with Δ representing the area of the triangle.

Derivation of a consistent plate/shell model

The term "consistent plate model" comes from the assumption of consistent deformations in the actuators and substructures. The strain distribution is assumed to result from a linear combination of in-plane extensional (constant strain through the thickness) and bending (linearly varying through the thickness) displacements. This assumed strain distribution, in contrast to several other possibilities, has been found to most accurately represent the actual behavior, even in the case of discrete surface bonded actuators [Crawley and Lazarus, 1991]. In this section we will use the principle of virtual work to derive the equations of finite element. It is assumed that the solution for the configuration C_0 at time t = 0 is available and that a new configuration C_{τ} at time $t = \tau$ has to be determined. The principle of virtual work can be expressed for a single element as follows:

$$\delta W_i = \delta W_e \tag{2.55}$$

where δW_i is the virtual work done by internal forces and δW_e is the virtual work done by external forces. The internal virtual work is given by

$$\delta W_i = \int_{C_0} \delta \{\varepsilon\}^T \{s\} dV \tag{2.56}$$

where V is the volume of the element in C_0 , $\{\varepsilon\}$ is the vector of incremental Green-Lagrange strains and $\{s\}$ is the vector of second Piola-Kirchhoff (PK2) stresses. Under infinitesimal strain condition the stress and strain reduce to the engineering stress and strain [Shames and Dym, 1985]. The components of the Green-Lagrange strains and second Piola-Kirchhoff stresses do not change under rigid body motions including rigid body translation or rigid body rotation. This fundamental fact implies that any material description which has been developed for infinitesimal displacement analysis using engineering stress and strain measures can directly be employed for large displacement, large rotation but small strain analysis by simply substituting Green-Lagrange strains and second Piola-Kirchhoff stresses

for the engineering strains and stresses. In updated Lagrangian formulation, the PK2 stresses at any time $0 < t < \tau$ during the solution process can be decomposed as

$$\{s\} = \{\sigma\} + \{\Delta s\} \tag{2.57}$$

where $\{\sigma\}$ is the vector of Cauchy stresses in C_0 at time t=0 and $\{\Delta s\}$ is the vector of incremental PK2 stresses. The basic difference between the Cauchy stresses and the PK2 stresses is that the Cauchy stresses are measured in the current configuration, whereas the PK2 stresses are measured in the reference configuration. The term "current configuration" is used to denote any configuration between C_0 and C_τ obtained during the solution process. The incremental Green-Lagrange strains and PK2 stresses are related by the constitutive equations

$$\{\Delta s\} = \overline{Q} \{\varepsilon\} \tag{2.58}$$

where [Q] is the transformed reduced stiffness matrix of the plate or one of its plies in the local element coordinate system and is given by

where

$$Q_{11} = \frac{E_1}{1 - v_{12}v_{21}}$$

$$Q_{12} = \frac{v_{12}E_2}{1 - v_{12}v_{21}}$$

$$Q_{22} = \frac{E_2}{1 - v_{12}v_{21}}$$

$$Q_{66} = G_{12}$$

 $s = \sin \theta$, $c = \cos \theta$

 E_1 , E_2 , v_{12} , G_{12} are the properties in the principal material directions and θ is the angle which the E_1 direction makes with the local x-axis, measured counter clock wise from the local x-axis.

 E_i = Young's moduli in the i^{th} principal material direction,

 v_{ij} = Poisson's ration for transverse strain in the j-direction when stressed in the i-direction, that is,

$$V_{ij} = -\frac{\mathcal{E}_j}{\mathcal{E}_i}$$

Because of the symmetric property of the compliance matrix

$$\frac{V_{ij}}{E_i} = \frac{V_{ji}}{E_j}$$

 G_{12} = shear modulus in the 1-2 plane

At the end of the current step ($t = \tau$), the PK2 stresses in C_0 , computed using Eqs. 2.57 and 2.58, are converted to Cauchy stresses in C_{τ} by the following transformation,

$$\sigma_{mn} = \frac{\rho_t}{\rho_0} {}^t x_{m,i} {}^t x_{n,j} s_{ij}$$
 (2.60a)

or in matrix form,

$$[\sigma] = \frac{\rho_t}{\rho_0} \begin{bmatrix} {}^t_0 X \end{bmatrix} [s] \begin{bmatrix} {}^t_0 X \end{bmatrix}^T \tag{2.60b}$$

where ρ_0 and ρ_t are the mass density of the structure at time 0 and t, respectively. They are related by the deformation gradient $\begin{bmatrix} t \\ 0 \end{bmatrix}$ as follows,

$$\rho_t = \frac{\rho_0}{\left| {t \atop 0} X \right|}$$

The deformation gradient $\begin{bmatrix} {}^{t}_{0}X \end{bmatrix}$ is defined as

$$\begin{bmatrix} {}^{t}X \end{bmatrix} = \begin{bmatrix} \frac{\partial x_{1}}{\partial^{0}x_{1}} & \frac{\partial x_{1}}{\partial^{0}x_{2}} & \frac{\partial x_{1}}{\partial^{0}x_{3}} \\ \frac{\partial x_{2}}{\partial^{0}x_{1}} & \frac{\partial x_{2}}{\partial^{0}x_{2}} & \frac{\partial x_{2}}{\partial^{0}x_{3}} \\ \frac{\partial x_{3}}{\partial^{0}x_{1}} & \frac{\partial x_{3}}{\partial^{0}x_{2}} & \frac{\partial x_{3}}{\partial^{0}x_{3}} \end{bmatrix} = \begin{bmatrix} {}^{t}_{0}x_{1,1} & {}^{t}_{0}x_{1,1} & {}^{t}_{0}x_{1,1} \\ {}^{t}_{0}x_{2,1} & {}^{t}_{0}x_{2,1} & {}^{t}_{0}x_{2,1} \\ {}^{t}_{0}x_{3,1} & {}^{t}_{0}x_{3,1} & {}^{t}_{0}x_{3,1} \end{bmatrix}$$
(2.61)

where ${}^{0}x_{1}$, ${}^{0}x_{2}$, and ${}^{0}x_{3}$ are the coordinates of a typical point P in the structure at time 0 and ${}^{t}x_{1}$, ${}^{t}x_{2}$, and ${}^{t}x_{3}$ are the coordinates of the new position of P at time t.

Under mechanical, thermal and electric loads, the variation of strain through the thickness can be expressed as

$$\{\varepsilon\} = \{\varepsilon^0\} + z\{\kappa\} - \{\varepsilon\}_E - \{\varepsilon\}_T \tag{2.62}$$

where $\{\varepsilon^0\}$ and $\{\kappa\}$ are the vectors of incremental membrane strains and bending strains, respectively.

$$\left\{ \mathcal{E}^{0} \right\} = \left\{ \begin{array}{l} \mathcal{E}^{0}_{x} \\ \mathcal{E}^{0}_{y} \\ \mathcal{Y}^{0}_{xy} \end{array} \right\} = \left\{ \begin{array}{l} u_{,x} + \frac{1}{2} (u_{,x}^{2} + v_{,x}^{2} + w_{,x}^{2}) \\ v_{,y} + \frac{1}{2} (u_{,y}^{2} + v_{,y}^{2} + w_{,y}^{2}) \\ u_{,y} + v_{,x} + u_{,x} u_{,y} + v_{,x} v_{,y} + w_{,x} w_{,y} \end{array} \right\}$$

$$(2.63)$$

$$\{\kappa\} = \begin{cases} \kappa_{x} \\ \kappa_{y} \\ \kappa_{xy} \end{cases} = \begin{cases} \frac{\partial \beta_{x}}{\partial x} \\ \frac{\partial \beta_{y}}{\partial y} \\ \frac{\partial \beta_{x}}{\partial y} + \frac{\partial \beta_{y}}{\partial x} \end{cases}$$

$$(2.64)$$

where u, v, and w are the incremental displacements of the mid-plane of the element, β_x and β_y are the incremental rotations of the normal to the undeformed mid-plane in the local x-z and y-z planes, respectively.

 $\{\varepsilon\}_{\scriptscriptstyle E}$ and $\{\varepsilon\}_{\scriptscriptstyle T}$ are actuation strains and thermal strains, respectively, and given by

$$\{\varepsilon\}_{E} = \begin{cases} \varepsilon_{x} \\ \varepsilon_{y} \\ \gamma_{xy} \end{cases}_{E} = \begin{bmatrix} 0 & 0 & \overline{d}_{31} \\ 0 & 0 & \overline{d}_{32} \\ 0 & 0 & \overline{d}_{36} \end{bmatrix} \begin{bmatrix} E_{x} \\ E_{y} \\ E_{z} \end{bmatrix} = \begin{bmatrix} \overline{d}_{31} \\ \overline{d}_{32} \\ \overline{d}_{36} \end{bmatrix} E_{z} = \begin{bmatrix} c^{2} & s^{2} & -cs \\ s^{2} & c^{2} & cs \\ 2cs & -2cs & c^{2} - s^{2} \end{bmatrix} \begin{bmatrix} d_{31} \\ d_{32} \\ d_{36} \end{bmatrix} E_{3}$$
(2.65)

$$\left\{\varepsilon\right\}_{T} = \begin{cases} \varepsilon_{x} \\ \varepsilon_{y} \\ \gamma_{xy} \end{cases}_{T} = \begin{cases} \alpha_{x} \\ \alpha_{y} \\ \alpha_{xy} \end{cases} \Delta T = \begin{bmatrix} c^{2} & s^{2} & -cs \\ s^{2} & c^{2} & cs \\ 2cs & -2cs & c^{2} - s^{2} \end{bmatrix} \begin{bmatrix} \alpha_{1} \\ \alpha_{2} \\ \alpha_{12} \end{bmatrix} \Delta T$$

$$(2.66)$$

where E_i is the applied electric field in volts/meter, and d_{ij} is piezoelectric coefficients in meters/volt. α_x , α_y and α_{xy} are the coefficients of thermal expansion with respect to the element local coordinate system. α_1 , α_2 and α_{12} are the coefficients of thermal expansion with respect to the principal material coordinate system in 1/degrees Kelvin, ΔT is the prescribed temperature increment.

Under mechanical, thermal and electric loads, the incremental Green-Lagrange strains and PK2 stresses are related by

$$\{\Delta s\} = \overline{Q} \{\varepsilon\} = \overline{Q} \{\varepsilon\} + z\{\kappa\} - \{\varepsilon\}_{E} - \{\varepsilon\}_{T} = \overline{Q} \{\varepsilon\} + z\{\kappa\} - \overline{Q} \{\varepsilon\}_{E} - \overline{Q} \{\varepsilon\}_{T} (2.67) = \overline{Q} \{\varepsilon\}_{T} + z\{\kappa\} - \overline{Q} \{\varepsilon\}_{T} (2.67) = \overline{Q} \{\varepsilon\}_{T} (2.67$$

where the second term $[Q]\{\varepsilon\}_E$ in Eq. (2.67) represents the equivalent stress created as a result of the actuation strains. The actuation strain vector $\{\varepsilon\}_E$ contains in-plane normal and shear strain components, and enters into the elasticity equations in the same manner as does thermal strain. Actuation strain is the strain that physically causes induced strains to be produced, and can be due to thermal expansion, piezoelectricity, electrostriction, etc. The last term $[Q]\{\varepsilon\}_T$ in Eq. (2.67) represents the equivalent stress created as a result of the thermal strains.

The internal force and moment resultants are defined by

$$\{N\} = \int_{I} \{s\} dz, \qquad \{M\} = \int_{I} \{s\} z dz$$
 (2.68)

where h is the thickness of the element. Substituting Eqs. (2.57) and (2.67) into Eqs. (2.68) and integrating through the thickness t of the plate, we can get the plate load deformation relations as follows

$$\begin{cases}
N \\
M
\end{cases} = \begin{cases}
N_I \\
M_I
\end{cases} + \begin{bmatrix}
A & B \\
B & D
\end{bmatrix} \begin{bmatrix}
\varepsilon^0 \\
K
\end{bmatrix} - \begin{Bmatrix}
N_E \\
M_E
\end{bmatrix} - \begin{Bmatrix}
N_T \\
M_T
\end{Bmatrix}$$
(2.69)

where the initial force and moment resultants are

$$\{N_I\} = \begin{cases} N_{xI} \\ N_{yI} \\ N_{xyI} \end{cases} = \int_t \sigma dz, \qquad \{M_I\} = \begin{cases} M_{xI} \\ M_{yI} \\ M_{xyI} \end{cases} = \int_t \sigma z dz$$

$$(2.70)$$

The matrices A, B, and D are the usual extensional, coupling and bending stiffnesses of the plate.

$$A = \int_{t} \overline{Q} dz = \begin{bmatrix} A_{11} & A_{12} & A_{16} \\ A_{12} & A_{22} & A_{26} \\ A_{16} & A_{26} & A_{66} \end{bmatrix}, \qquad A_{ij} = \sum_{k=1}^{N} (\overline{Q}_{ij})_{k} (h_{k} - h_{k-1})$$

$$B = \int_{t} \overline{Q}zdz = \begin{bmatrix} B_{11} & B_{12} & B_{16} \\ B_{12} & B_{22} & B_{26} \\ B_{16} & B_{26} & B_{66} \end{bmatrix}, \qquad B_{ij} = \sum_{k=1}^{N} (\overline{Q}_{ij})_{k} (\frac{h_{k}^{2} - h_{k-1}^{2}}{2})$$

$$D = \int_{t} \overline{Q} z^{2} dz = \begin{vmatrix} D_{11} & D_{12} & D_{16} \\ D_{12} & D_{22} & D_{26} \\ D_{16} & D_{26} & D_{66} \end{vmatrix}, \qquad D_{ij} = \sum_{k=1}^{N} (\overline{Q}_{ij})_{k} (\frac{h_{k}^{3} - h_{k-1}^{3}}{3})$$

The equivalent actuator forces N_E and moments M_E per unit length are

$$\{N_E\} = \int_t \overline{Q} \{\varepsilon_E\} dz, \qquad \{M_E\} = \int_t \overline{Q} \{\varepsilon_E\} z dz$$
 (2.71)

The equivalent thermal forces N_T and moments M_T per unit length are

$$\{N_T\} = \int \overline{Q} \{\varepsilon_T\} dz, \qquad \{M_T\} = \int \overline{Q} \{\varepsilon_T\} z dz$$
 (2.72)

Care must be taken in performing the necessary integrations to obtain the correct stiffness, actuator and thermal forcing terms. Both actuator and substrate plies contribute to the stiffness matrices A, B, and D and the thermal forcing vectors N_T and M_T , whereas only actuator plies contribute to the actuator forcing vectors N_E and M_E , which are dependent on the mode of actuation (prescribed actuation strains). Extensional actuation is produced by prescribing actuation strains that are symmetric about the neutral axis, while bending actuation results when the actuation strains in the actuators above the neutral axis are prescribed to be in a direction opposite (180 deg out of phase) to those below the neutral axis. Equation (2.69) relates the resultant total strains and curvatures found in the actuator/substrate system to the actuation strains, thermal strains, external loads, and stiffness properties of the system in a general and compact form. The presence of numerous coupling terms shows that it is possible to create a variety of deformations (e.g., bending or twisting) using several different actuation modes (e.g., extension or bending). Thus, by careful selection of the laminate ply orientations, an actuator/substrate system can be designed to effect control for a variety of applications. In addition, various couplings may be introduced by the boundary conditions, as would be the case for a swept cantilever wing.

The internal virtual work can now be expressed as

$$\delta W_{i} = \int_{C_{0}} \delta \{\varepsilon\}^{T} \{s\} dV = \int_{C_{0}} \{\delta \{\varepsilon^{0}\}^{T} + z\delta \{\kappa\}^{T}\} \{s\} dV$$

$$= \int_{C_{0}} \{\delta \{\varepsilon^{0}\}^{T} \int_{t} \{s\} dz + \delta \{\kappa\}^{T} \int_{t} \{s\} z dz\} dA$$

$$= \int_{C_{0}} \{\delta \{\varepsilon^{0}\}^{T} \{N\} + \delta \{\kappa\}^{T} \{M\}\} dA$$

$$(2.73)$$

The variation of the membrane strains can be expressed as

$$\left\{\delta\varepsilon^{0}\right\} = \left\{ \begin{array}{c} \delta u_{x} + u_{x}\delta u_{x} + v_{x}\delta v_{x} + w_{x}\delta w_{x} \\ \delta v_{y} + u_{y}\delta u_{y} + v_{y}\delta v_{y} + w_{y}\delta w_{y} \\ \delta u_{y} + \delta v_{x} + u_{x}\delta u_{y} + u_{y}\delta u_{x} + v_{x}\delta v_{y} + v_{y}\delta v_{x} + w_{x}\delta w_{y} + w_{y}\delta w_{x} \end{array} \right\}$$

$$= \left[G_{1}\right] \left\{\delta u_{x} \quad \delta u_{y} \quad \delta v_{x} \quad \delta v_{y} \quad \delta w_{x} \quad \delta w_{y}\right\}^{T}$$

$$(2.74a)$$

where

$$[G_1] = \begin{bmatrix} 1 + u_{,x} & 0 & v_{,x} & 0 & w_{,x} & 0 \\ 0 & u_{,y} & 0 & 1 + v_{,y} & 0 & w_{,y} \\ u_{,y} & 1 + u_{,x} & 1 + v_{,y} & v_{,x} & w_{,y} & w_{,x} \end{bmatrix}$$
(2.74b)

Noting that the coordinate transformation between the degrees of freedom of the LST element and Allman triangle element, the in-plane Allman displacement field u and v can be expressed in terms of the shape function of the LST element and the nodal degrees of freedom of the shell element $\{a\}$ as follows,

$$\begin{cases} u \\ v \end{cases} = \begin{bmatrix} \{N\}^T & 0 \\ 0 & \{N\}^T \end{bmatrix} \{u_1, u_2, u_3, u_4, u_5, u_6, v_1, v_2, v_3, v_4, v_5, v_6 \}^T \\
= \begin{bmatrix} \{N\}^T & 0 \\ 0 & \{N\}^T \end{bmatrix} [T_{lst}] \{a\}$$
(2.75)

where $\{N\}$ is the vector of the shape functions of the LST element in the natural coordinates (Appendix F), $\{a\}$ is the nodal degrees of freedom of the shell element,

$$\{a\} = \left\{u_1, v_1, w_1, \theta_{x1}, \theta_{y1}, \theta_{z1}, u_2, v_2, w_2, \theta_{x2}, \theta_{y2}, \theta_{z2}, u_3, v_3, w_3, \theta_{x3}, \theta_{y3}, \theta_{z3}\right\}^T$$

 $[T_{lst}]$ is transformation matrix between the degrees of freedom of the LST element and Allman triangle element

The derivatives of the in-plane displacements are given by

$$\begin{cases} u_{,x} \\ u_{,y} \\ v_{,x} \\ v_{,y} \end{cases} = \begin{bmatrix} \frac{\partial \{N\}^T}{\partial x} & 0 \\ \frac{\partial \{N\}^T}{\partial y} & 0 \\ 0 & \frac{\partial \{N\}^T}{\partial x} \\ 0 & \frac{\partial \{N\}^T}{\partial y} \end{bmatrix} [T_{lst}] \{a\} = \frac{1}{2A} \begin{bmatrix} b_2 \{N_{,\xi}\}^T + b_3 \{N_{,\eta}\}^T & 0 \\ c_2 \{N_{,\xi}\}^T + c_3 \{N_{,\eta}\}^T & 0 \\ 0 & b_2 \{N_{,\xi}\}^T + b_3 \{N_{,\eta}\}^T \\ 0 & c_2 \{N_{,\xi}\}^T + c_3 \{N_{,\eta}\}^T \end{bmatrix} [T_{lst}] \{a\}$$

$$= [B_{lst}] \{a\}$$

(2.76a)

where $[B_{lst}]$ is strain-displacement matrix

$$[B_{lst}] = \frac{1}{2A} \begin{bmatrix} b_2 \{N_{,\xi}\}^T + b_3 \{N_{,\eta}\}^T & 0 \\ c_2 \{N_{,\xi}\}^T + c_3 \{N_{,\eta}\}^T & 0 \\ 0 & b_2 \{N_{,\xi}\}^T + b_3 \{N_{,\eta}\}^T \end{bmatrix} [T_{lst}]$$

$$0 & c_2 \{N_{,\xi}\}^T + c_3 \{N_{,\eta}\}^T \end{bmatrix}$$
(2.76b)

 $\{N_{,\xi}\}$ and $\{N_{,\eta}\}$ are the vectors of derivatives of the LST shape functions with respect to the natural coordinates ξ and η

$$\{N_{,\xi}\} = \frac{\partial\{N\}}{\partial\xi} = \begin{cases} \frac{\partial N_1}{\partial\xi} \\ \frac{\partial N_2}{\partial\xi} \\ \frac{\partial N_3}{\partial\xi} \\ \frac{\partial N_4}{\partial\xi} \\ \frac{\partial N_5}{\partial\xi} \\ \frac{\partial N_6}{\partial\xi} \end{cases} = \begin{cases} 4\xi + 4\eta - 3 \\ 4\xi - 1 \\ 0 \\ 4 - 8\xi - 4\eta \\ 4\eta \\ -4\eta \end{cases}$$

$$\{N_{,\eta}\} = \frac{\partial\{N\}}{\partial\eta} = \begin{cases} \frac{\partial N_1}{\partial\eta} \\ \frac{\partial N_2}{\partial\lambda_3} \\ \frac{\partial N_3}{\partial\eta} \\ \frac{\partial N_4}{\partial\eta} \\ \frac{\partial N_5}{\partial\eta} \\ \frac{\partial N_5}{\partial\eta} \end{cases} = \begin{cases} 4\xi + 4\eta - 3 \\ 0 \\ 4\eta - 1 \\ -4\xi \\ 4\xi \\ 4 - 4\xi - 8\eta \end{cases}$$

(2.76c)

The transverse displacement w is not defined explicitly in the DKT formulation. Farfard et al. in 1989 have shown that evaluating the derivatives $w_{,x}$ and $w_{,y}$ at the integration points from the linear interpolation for w gives better results than from the linear interpolation for the derivatives in terms of the nodal values. If the linear interpolation for w is used, that is, $w(\xi,\eta) = (1-\xi-\eta)w_1 + \xi w_2 + \eta w_3$, the derivatives $w_{,x}$ and $w_{,y}$ can be obtained as

$$\begin{cases}
w_{,x} \\ w_{,y}
\end{cases} = \frac{1}{2A} \begin{bmatrix} -b_2 - b_3 & b_2 & b_3 \\ -c_2 - c_3 & c_2 & c_3 \end{bmatrix} \begin{bmatrix} w_1 \\ w_2 \\ w_3 \end{bmatrix} = \frac{1}{2A} \begin{bmatrix} -b_2 - b_3 & b_2 & b_3 \\ -c_2 - c_3 & c_2 & c_3 \end{bmatrix} [T_w] \{a\} = [B_w] \{a\}$$
(2.77a)

where the transformation matrix $[T_w]$ is used to express $\{w_1, w_2, w_3\}^T$ in terms of the degrees of freedom of the shell element and given by

 $\left[B_{\scriptscriptstyle W}\right]$ is strain-displacement matrix and given by

$$[B_{w}] = \frac{1}{2A} \begin{vmatrix} -b_{2} - b_{3} & b_{2} & b_{3} \\ -c_{3} - c_{3} & c_{3} & c_{3} \end{vmatrix} [T_{w}]$$
(2.77c)

Combining the variations of Eqs. (2.40) and (2.41), we have

$$\left\{ \delta u_{,x} \quad \delta u_{,y} \quad \delta v_{,x} \quad \delta v_{,y} \quad \delta w_{,x} \quad \delta w_{,y} \right\}^{T} = \begin{bmatrix} B_{lst} \\ B_{w} \end{bmatrix} \left\{ \delta a \right\} = \begin{bmatrix} G_{2} \end{bmatrix} \left\{ \delta a \right\}$$
(2.78a)

where $[G_2]$ is a matrix of size 6 by 18 with the first 4 rows made of $[B_{lst}]$ and the last 2 rows made of $[B_w]$, that is,

$$[G_2] = \begin{vmatrix} [B_{lst}] \\ [B_w] \end{vmatrix}$$
 (2.78b)

Substituting (2.42) into (2.38), the variation of the membrane strains can be expressed as

$$\{\delta \varepsilon^0\} = [G_1][G_2]\{\delta a\} \tag{2.79}$$

The normal rotations β_x and β_y can be expressed in terms of the nodal degrees of freedom of the DKT element as

$$\beta_{x} = \{H_{x}\}^{T} \{a_{dkt}\}, \qquad \beta_{v} = \{H_{v}\}^{T} \{a_{dkt}\}$$
 (2.80)

where $\{H_x\}$ and $\{H_y\}$ are the vectors of the shape functions of the DKT element (Appendix I), $\{a_{dkt}\}$ is the vector of nodal degrees of freedom of the DKT element given by

$$\{a_{dkt}\}^{T} = \{w_{1}, \theta_{x1}, \theta_{y1}, w_{2}, \theta_{x2}, \theta_{y2}, w_{3}, \theta_{x3}, \theta_{y3}\}$$

The first variation of the bending strains can be expressed as

$$\{\delta\kappa\} = \frac{1}{2A} \begin{vmatrix} y_{31}\{H_{x,\xi}\}^{T} + y_{12}\{H_{x,\eta}\}^{T} \\ -x_{31}\{H_{y,\xi}\}^{T} - x_{12}\{H_{y,\eta}\}^{T} \\ -x_{31}\{H_{x,\xi}\}^{T} - x_{12}\{H_{x,\eta}\}^{T} + y_{31}\{H_{y,\xi}\}^{T} + y_{12}\{H_{y,\eta}\}^{T} \end{vmatrix} [T_{dkt}]\{\delta a\} = [B_{dkt}]\{\delta a\} (2.81)$$

where $\{H_{x,\xi}\}$, $\{H_{x,\eta}\}$, $\{H_{y,\xi}\}$, $\{H_{y,\eta}\}$ are the derivatives of the shape functions with respect to the natural coordinates (Appendix I).

The transformation matrix $[T_{dkt}]$ is used to express the nodal degrees of freedom of the DKT element $\{a_{dkt}\}$ in terms of the nodal degrees of freedom of the shell element $\{a\}$ as

 $\{a_{dkt}\} = [T_{dkt}]\{a\}$ and given by

 $\left[B_{\mathit{dkt}}\right]$ is strain-displacement matrix and given by

$$[B_{dkt}] = \frac{1}{2A} \begin{vmatrix} y_{31} \{H_{x,\xi}\}^T + y_{12} \{H_{x,\eta}\}^T \\ -x_{31} \{H_{y,\xi}\}^T - x_{12} \{H_{y,\eta}\}^T \\ -x_{31} \{H_{x,\xi}\}^T - x_{12} \{H_{x,\eta}\}^T + y_{31} \{H_{y,\xi}\}^T + y_{12} \{H_{y,\eta}\}^T \end{vmatrix} [T_{dkt}]$$

Substituting the Eqs. (2.79) and (2.81) into Eqs. (2.73), the internal virtual work can be written as

$$\delta W_{i} = \int_{C_{0}} \{ [G_{1}] [G_{2}] \{\delta a\} \}^{T} \{N\} + \{ [B_{dkt}] \{\delta a\} \}^{T} \{M\} \} dA$$

$$= \{\delta a\}^{T} \int_{C_{0}} \{ [G_{2}]^{T} [G_{1}]^{T} \{N\} + [B_{dkt}]^{T} \{M\} \} dA$$

$$= \{\delta a\}^{T} \{q\}$$

where $\{q\}$ is the element internal force vector given by

$$\{q\} = \int_{C_0} \{ [G_2]^T [G_1]^T \{N\} + [B_{dkt}]^T \{M\} \} dA$$
 (2.82)

The external virtual work can be expressed as

$$\delta W_e = \left\{ \delta a \right\}^T \left\{ f \right\} \tag{2.83}$$

where $\{f\}$ is the element external force vector due to mechanical loads. The equations of equilibrium at the element level can be expressed as

$$\{g\} = \{q\} - \{f\} = 0$$
 (2.84)

The element tangent stiffness matrix $[k] = \frac{\partial \{q\}}{\partial \{a\}}$ can be obtained by taking the first variation

of the element internal force vector $\{q\}$ in Eq. (2.82) and using the relation $\delta\{q\} = \frac{\partial \{q\}}{\partial \{a\}} \delta\{a\}$.

The first variation of the element internal force vector $\{q\}$ is

$$\delta\{q\} = \int_{C_0} \left\{ \mathcal{S}([G_2]^T [G_1]^T \{N\}) + \mathcal{S}([B_{dkt}]^T \{M\}) \right\} dA$$

$$= \int_{C_0} \left\{ [G_2]^T (\mathcal{S}[G_1]^T) \{N\} + [G_2]^T [G_1]^T \mathcal{S}\{N\} + [B_{dkt}]^T \mathcal{S}\{M\} \right\} dA$$
(2.85)

$$\left(\delta[G_{1}]^{T}\right)(N) = \begin{bmatrix} \delta u_{,x} & 0 & \delta u_{,y} \\ 0 & \delta u_{,y} & \delta u_{,x} \\ \delta v_{,x} & 0 & \delta v_{,y} \\ 0 & \delta v_{,y} & \delta v_{,x} \\ \delta w_{,x} & 0 & \delta w_{,y} \\ 0 & \delta w_{,y} & \delta w_{,x} \end{bmatrix} \begin{bmatrix} N_{x} \\ N_{y} \\ N_{xy} \end{bmatrix} = \begin{bmatrix} N_{x} & N_{xy} & 0 & 0 & 0 & 0 & 0 \\ N_{xy} & N_{y} & 0 & 0 & 0 & 0 & 0 \\ 0 & 0 & N_{x} & N_{xy} & 0 & 0 & 0 \\ 0 & 0 & N_{xy} & N_{y} & 0 & 0 & 0 \\ 0 & 0 & 0 & 0 & N_{x} & N_{xy} \\ 0 & 0 & 0 & 0 & N_{xy} & N_{y} \end{bmatrix} \begin{bmatrix} \delta u_{,x} \\ \delta u_{,y} \\ \delta v_{,x} \\ \delta v_{,y} \\ \delta w_{,x} \end{bmatrix}$$

$$= \begin{bmatrix} \hat{N} & 0 & 0 \\ 0 & \hat{N} & 0 \\ 0 & 0 & \hat{N} \end{bmatrix} [G_{2}]\{\delta a\}$$

$$\delta\{N\} = [A]\{\delta\varepsilon^0\} + [B]\{\delta\kappa\} = [A][G_1][G_2]\{\delta a\} + [B][B_{dkt}]\{\delta a\}$$

$$(2.86)$$

$$(2.87)$$

$$\delta\{M\} = [B]\{\delta\varepsilon^0\} + [D]\{\delta\kappa\} = [B][G_1][G_2]\{\delta a\} + [D][B_{dkt}]\{\delta a\}$$

$$(2.88)$$

Substituting (2.86), (2.87) and (2.88) into (2.85), we get the first variation of the element internal force vector $\{q\}$

$$\delta\{q\} = \lceil k \rceil \{\delta a\} \tag{2.89}$$

where [k] is the element tangent stiffness matrix given by

$$[k] = \int_{C_1} (G_2)^T [G_1]^T [A] [G_1] [G_2] + [B_{dkt}]^T [D] [B_{dkt}] + [G_2]^T [G_1]^T [B] [B_{dkt}]$$

$$+ [B_{dkt}]^T [B] [G_1] [G_2] + [G_2]^T \begin{bmatrix} \hat{N} & 0 & 0 \\ 0 & \hat{N} & 0 \\ 0 & 0 & \hat{N} \end{bmatrix} [G_2] dA \qquad (2.90a)$$

$$= [k_a] + [k_{dkt}] + [k_{mb}] + [k_{bm}] + [k_n]$$

where the first term $[k_a]$ is the Allman membrane stiffness,

$$[k_a] = \int_{C_1} [G_2]^T [G_1]^T [A] [G_1] [G_2] dA$$
 (2.90b)

The second $[k_{dkt}]$ is the DKT bending stiffness,

$$[k_{dkt}] = \int_{C_1} [B_{dkt}]^T [D] [B_{dkt}] dA$$
 (2.90c)

The third $[k_{mb}]$ is the membrane-bending coupling stiffness, associated with in-plane forces resulting from bending deformation,

$$[k_{mb}] = \int_{C_1} [G_2]^T [G_1]^T [B] [B_{dkt}] dA$$
 (2.90d)

The fourth $[k_{bm}]$ is the bending-membrane coupling stiffness, associated with bending moment resulting from in-plane deformation,

$$[k_{bm}] = \int_{C} [B_{dkl}]^{T} [B] [G_{1}] [G_{2}] dA$$
 (2.90e)

Obviously, this matrix is simply the transpose of the membrane-bending coupling stiffness.

That is,

$$\begin{bmatrix} k_{bm} \end{bmatrix} = \begin{bmatrix} k_{mb} \end{bmatrix}^T \tag{2.90f}$$

The last is the contribution from the geometrically nonlinear terms in the membrane strains due to the large deformation,

$$[k_n] = \int_{C_1} [G_2]^T \begin{bmatrix} \hat{N} & 0 & 0 \\ 0 & \hat{N} & 0 \\ 0 & 0 & \hat{N} \end{bmatrix} [G_2] dA$$
 (2.90g)

From the derivation this term is zero matrix in the case of linear analysis.

In the case of linear analysis, we can easily get the corresponding linear formulation from the above. For example, the in-plane strain-displacement matrix $[G_1]$, $[G_2]$ and element stiffness [k] become

$$\begin{bmatrix} G_1 \end{bmatrix} = \begin{vmatrix} 1 & 0 & 0 & 0 & 0 & 0 \\ 0 & 0 & 0 & 1 & 0 & 0 \\ 0 & 1 & 1 & 0 & 0 & 0 \end{vmatrix}$$

$$[G_2] = \begin{bmatrix} [B_{lst}]_{4\times 18} \\ [0]_{2\times 18} \end{bmatrix}$$

$$[k] = \int_{C_1} ([G_2]^T [G_1]^T [A] [G_1] [G_2] + [B_{dkt}]^T [D] [B_{dkt}] + [G_2]^T [G_1]^T [B] [B_{dkt}]$$
$$+ [B_{dkt}]^T [B] [G_1] [G_2] dA$$

Of course, we can get the above linear element stiffness using the total potential energy of the system as follows

$$\Pi = \frac{1}{2} \int_{A} \left\{ \boldsymbol{\varepsilon}^{0T} \quad \boldsymbol{\kappa}^{T} \left\{ \begin{bmatrix} A & B \\ B & D \end{bmatrix} \right\} \boldsymbol{\varepsilon}^{0} dA - \int_{A} \left\{ N_{E} \quad M_{E} \right\} \left\{ \begin{bmatrix} \boldsymbol{\varepsilon}^{0} \\ \boldsymbol{\kappa} \end{bmatrix} dA - \int_{A} \left\{ N_{T} \quad M_{T} \right\} \left\{ \begin{bmatrix} \boldsymbol{\varepsilon}^{0} \\ \boldsymbol{\kappa} \end{bmatrix} dA \right\} dA$$

Derivation of the thermal load vector

The virtual work done by thermal load is given by

$$\delta W_{T} = \int_{C_{0}} \left\{ \delta \left\{ \varepsilon^{0} \right\}^{T} \left\{ N_{T} \right\} + \delta \left\{ \kappa \right\}^{T} \left\{ M_{T} \right\} \right\} dA$$

$$= \left\{ \delta a \right\}^{T} \int_{C_{0}} \left\{ \left[G_{2} \right]^{T} \left[G_{1} \right]^{T} \left\{ N_{T} \right\} + \left[B_{dkt} \right]^{T} \left\{ M_{T} \right\} \right\} dA$$

$$(2.91)$$

From which the thermal load can be identified as

$$\{f_T\} = \int_{C_0} \{ [G_2]^T [G_1]^T \{N_T\} + [B_{dkt}]^T \{M_T\} \} dA$$
 (2.92)

Substituting Eqs. (2.59), (2.66) and (2.72) into Eq. (2.92),

$$\{f_T\} = \int_{C_0} \left(\left[G_2 \right]^T \left[G_1 \right]^T + z \left[B_{dkt} \right]^T \right) \left[\overline{Q} \right] \{ \varepsilon_T \} dV$$
(2.93)

Using Eqs. (2.59) and (2.66),

$$\begin{aligned} & \boxed{Q} \big[\mathcal{E}_T \big\} = \begin{bmatrix} c^2 & s^2 & -2cs \\ s^2 & c^2 & 2cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} & \boxed{Q_{12}} & \boxed{Q_{12}} & \boxed{Q_{22}} & \boxed{0} \\ 0 & 0 & \boxed{Q_{66}} & \boxed{c^2 & s^2 & -2cs} \\ cs & -cs & \boxed{c^2 - s^2} & \boxed{c^2 & 2cs} \\ cs & -cs & \boxed{c^2 - s^2} & \boxed{d_{12}} & \boxed{Q_{12}} & \boxed{0} \\ cs & -cs & \boxed{c^2 - s^2} & \boxed{d_{12}} & \boxed{Q_{12}} & \boxed{0} \\ cs & -cs & \boxed{c^2 - s^2} & \boxed{d_{12}} & \boxed{Q_{12}} & \boxed{0} \\ cs & -cs & \boxed{c^2 - s^2} & \boxed{d_{12}} & \boxed{Q_{12}} & \boxed{0} \\ 0 & 0 & \boxed{Q_{66}} & \boxed{d_{12}} & \boxed{\Delta}T = \begin{bmatrix} \overline{\lambda}_x \\ \overline{\lambda}_y \\ \overline{\lambda}_{xy} \end{bmatrix} \Delta T \end{aligned}$$

where

$$\begin{cases}
\overline{\lambda}_{x} \\
\overline{\lambda}_{y}
\end{cases} = \begin{bmatrix}
c^{2} & s^{2} & -2cs \\
s^{2} & c^{2} & 2cs \\
cs & -cs & c^{2} - s^{2}
\end{bmatrix} \begin{bmatrix}
Q_{11} & Q_{12} & 0 \\
Q_{12} & Q_{22} & 0 \\
0 & 0 & Q_{66}
\end{bmatrix} \begin{bmatrix}
\alpha_{1} \\
\alpha_{2} \\
\alpha_{12}
\end{bmatrix}$$
(2.94)

Therefore, the thermal load vector can be written as

$$\{f_T\} = \int_{C_0} \left(\left[G_2 \right]^T \left[G_1 \right]^T + z \left[B_{dkt} \right]^T \right) \left(\frac{\overline{\lambda}_x}{\overline{\lambda}_{xy}} \right) \Delta T dV$$
(2.95)

The temperature distribution at any point (ξ, η, z) within the element is given by

$$\Delta T = \frac{1}{2} \left(T_u(\xi, \eta) + T_l(\xi, \eta) \right) + \frac{z}{h} \left(T_u(\xi, \eta) - T_l(\xi, \eta) \right)$$

$$= T_m(\xi, \eta) + \frac{z}{h} T_d(\xi, \eta)$$
(2.96a)

where

$$T_{m}(\xi,\eta) = \frac{1}{2} (T_{u}(\xi,\eta) + T_{l}(\xi,\eta)), \qquad T_{d}(\xi,\eta) = T_{u}(\xi,\eta) - T_{l}(\xi,\eta)$$
 (2.96b)

 $T_u(\xi,\eta)$ and $T_l(\xi,\eta)$ are the temperatures at the top and bottom surfaces of the laminate, respectively. They are represented in terms of the nodal quantities T_{u1} , T_{u2} , T_{u3} , and T_{l1} , T_{l2} , T_{l3} using linear interpolation functions in the natural coordinates as follows

$$T_{u}(\xi,\eta) = T_{u1} + T_{u21}\xi + T_{u31}\eta, \qquad T_{l}(\xi,\eta) = T_{l1} + T_{l21}\xi + T_{l31}\eta$$
(2.96c)

where
$$T_{u21} = T_{u2} - T_{u1}$$
, $T_{u31} = T_{u3} - T_{u1}$, $T_{l21} = T_{l2} - T_{l1}$, $T_{l31} = T_{l3} - T_{l1}$

Substituting Eq. (2.96) into (2.95) and integrating over the thickness, the thermal load vector can be expressed as

$$\begin{aligned}
\{f_{T}\} &= \iint_{\Delta} [G_{2}]^{T} [G_{1}]^{T} \begin{cases} n_{t11} \\ n_{t12} \\ n_{t13} \end{cases} T_{m}(\xi, \eta) dA + \iint_{\Delta} [G_{2}]^{T} [G_{1}]^{T} \begin{cases} n_{t21} \\ n_{t22} \\ n_{t23} \end{cases} T_{d}(\xi, \eta) dA \\
&+ \iint_{\Delta} [B_{dkt}]^{T} \begin{cases} m_{t11} \\ m_{t12} \\ m_{t13} \end{cases} T_{m}(\xi, \eta) dA + \iint_{\Delta} [B_{dkt}]^{T} \begin{cases} m_{t21} \\ m_{t22} \\ m_{t23} \end{cases} T_{d}(\xi, \eta) dA
\end{aligned} (2.97)$$

where

$$\begin{cases}
n_{t11} \\
n_{t12} \\
n_{t13}
\end{cases} = \int_{t}^{t} \left\{ \frac{\overline{\lambda}_{x}}{\overline{\lambda}_{y}} \right\}_{k} dz = \sum_{k}^{t} \left\{ \frac{\overline{\lambda}_{x}}{\overline{\lambda}_{y}} \right\}_{k} (z_{k+1} - z_{k}), \qquad \begin{cases}
n_{t21} \\
n_{t22} \\
n_{t23}
\end{cases} = \int_{t}^{t} \left\{ \frac{\overline{\lambda}_{x}}{\overline{\lambda}_{y}} \right\}_{k} \frac{z}{h} dz = \sum_{k}^{t} \left\{ \frac{\overline{\lambda}_{x}}{\overline{\lambda}_{y}} \right\}_{k} \frac{1}{2h} (z_{k+1}^{2} - z_{k}^{2})$$

$$\begin{cases}
 m_{t11} \\
 m_{t12} \\
 m_{t13}
\end{cases} = \int_{t}^{t} \left\{ \frac{\overline{\lambda}_{x}}{\overline{\lambda}_{y}} \right\}_{k} z dz = \sum_{k}^{t} \left\{ \frac{\overline{\lambda}_{x}}{\overline{\lambda}_{y}} \right\}_{k}^{t} \frac{1}{2} (z_{k+1}^{2} - z_{k}^{2}), \qquad \begin{cases}
 m_{t21} \\
 m_{t22} \\
 m_{t23}
\end{cases} = \int_{t}^{t} \left\{ \frac{\overline{\lambda}_{x}}{\overline{\lambda}_{y}} \right\}_{k}^{t} \frac{z^{2}}{h} dz = \sum_{k}^{t} \left\{ \frac{\overline{\lambda}_{x}}{\overline{\lambda}_{y}} \right\}_{k}^{t} \frac{1}{3h} (z_{k+1}^{3} - z_{k}^{3})$$

Derivation of the electric load vector

The virtual work done by electric load is given by

$$\delta W_{E} = \int_{C_{0}} \left\{ \delta \left\{ \varepsilon^{0} \right\}^{T} \left\{ N_{E} \right\} + \delta \left\{ \kappa \right\}^{T} \left\{ M_{E} \right\} \right\} dA$$

$$= \left\{ \delta a \right\}^{T} \int_{C_{0}} \left\{ \left[G_{2} \right]^{T} \left[G_{1} \right]^{T} \left\{ N_{E} \right\} + \left[B_{dkt} \right]^{T} \left\{ M_{E} \right\} \right\} dA$$

$$(2.98)$$

From which the electric load can be identified as

$$\{f_E\} = \int_{C_0} \{ [G_2]^T [G_1]^T \{N_E\} + [B_{dkt}]^T \{M_E\} \} dA$$
(2.99)

Substituting Eqs. (2.59), (2.65) and (2.71) into Eq. (2.99),

$$\{f_E\} = \int_{C_0} \left(\left[G_2 \right]^T \left[G_1 \right]^T + z \left[B_{dkt} \right]^T \right) \overline{Q} \left[\mathcal{E}_E \right\} dV$$
(2.100)

Using Eqs. (2.59) and (2.65),

$$\begin{aligned} & [\overline{Q}] \{ \varepsilon_E \} = \begin{bmatrix} c^2 & s^2 & -2cs \\ s^2 & c^2 & 2cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} \underbrace{ \begin{bmatrix} Q_{11} & Q_{12} & 0 \\ Q_{12} & Q_{22} & 0 \\ 0 & 0 & Q_{66} \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} c^2 & s^2 & -2cs \\ s^2 & c^2 & 2cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -2cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} Q_{11} & Q_{12} & 0 \\ Q_{12} & Q_{22} & 0 \\ 0 & 0 & Q_{66} \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -2cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 - s^2 \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 & -cs \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 & -cs \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 & -cs \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 & -cs \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 & -cs \end{bmatrix} }_{cs} \underbrace{ \begin{bmatrix} C^2 & s^2 & -cs \\ cs & -cs & c^2 & -cs$$

where

$$\begin{cases}
\overline{e}_{31} \\
\overline{e}_{32} \\
\overline{e}_{36}
\end{cases} = \begin{bmatrix}
c^2 & s^2 & -2cs \\
s^2 & c^2 & 2cs \\
cs & -cs & c^2 - s^2
\end{bmatrix} \begin{bmatrix}
Q_{11} & Q_{12} & 0 \\
Q_{12} & Q_{22} & 0 \\
0 & 0 & Q_{66}
\end{bmatrix} \begin{bmatrix}
d_{31} \\
d_{32} \\
d_{36}
\end{bmatrix} \tag{2.101}$$

Therefore, the electric load vector can be written as

$$\{f_E\} = \int_{C_0} \left(\left[G_2 \right]^T \left[G_1 \right]^T + z \left[B_{dkt} \right]^T \right) \left[\overline{e}_{31} \right] E_{3d} V$$

$$(2.102)$$

Integrating over the thickness, the electric load vector can be expressed as

$$\{f_E\} = \iint_{\Delta} [G_2]^T [G_1]^T \begin{Bmatrix} n_{e1} \\ n_{e2} \\ n_{e3} \end{Bmatrix} dA + \iint_{\Delta} [B_{dkt}]^T \begin{Bmatrix} m_{e1} \\ m_{e2} \\ m_{e3} \end{Bmatrix} dA$$
(2.103)

where

$$\begin{cases}
 n_{e1} \\
 n_{e2} \\
 n_{e3}
 \end{cases} = \int_{t}^{t} \begin{cases}
 \overline{e}_{31} \\
 \overline{e}_{32} \\
 \overline{e}_{36}
 \end{cases}_{k} E_{3} dz = \sum_{k}^{t} \begin{cases}
 \overline{e}_{31} \\
 \overline{e}_{32} \\
 \overline{e}_{36}
 \end{cases}_{k} (E_{3})_{k} (z_{k+1} - z_{k})$$

Integrating the stiffness matrix, thermal load and electric load vector by Gauss numerical integration, we have, respectively

The Allman membrane stiffness $[k_a]$,

$$[k_a] = \int_0^1 \int_0^{1-\xi} [G_2]^T [G_1]^T [A] [G_1] [G_2] J | d\xi d\eta$$

$$= \frac{1}{2} \sum_{i=1}^n W_i [G_2]^T [G_1]^T [A] [G_1] [G_2] J | d\xi d\eta$$
(2.104a)

The DKT bending stiffness $[k_{dkt}]$,

$$[k_{dkt}] = \int_{0}^{1} \int_{0}^{1-\xi} [B_{dkt}]^{T} [D] [B_{dkt}] J | d\xi d\eta$$

$$= \frac{1}{2} \sum_{i=1}^{n} W_{i} [B_{dkt}]^{T} [D] [B_{dkt}] J |$$
(2.104b)

The membrane-bending coupling stiffness $[k_{mb}]$,

$$[k_{mb}] = \int_{0}^{1} \int_{0}^{1-\xi} [G_{2}]^{T} [G_{1}]^{T} [B] [B_{dkt}] J | d\xi d\eta$$

$$= \frac{1}{2} \sum_{i=1}^{n} W_{i} [G_{2}]^{T} [G_{1}]^{T} [B] [B_{dkt}] J |$$
(2.104c)

The contribution from the geometrically nonlinear terms in the membrane strains $[k_n]$,

$$\begin{split} \begin{bmatrix} k_n \end{bmatrix} &= \int_0^1 \int_0^{1-\xi} \begin{bmatrix} G_2 \end{bmatrix}^T \begin{bmatrix} \hat{N} & 0 & 0 \\ 0 & \hat{N} & 0 \\ 0 & 0 & \hat{N} \end{bmatrix} \begin{bmatrix} G_2 \end{bmatrix} J | d\xi d\eta \\ & & & & \\ \end{bmatrix} \\ &= \frac{1}{2} \sum_{i=1}^n W_i \begin{bmatrix} G_2 \end{bmatrix}^T \begin{bmatrix} \hat{N} & 0 & 0 \\ 0 & \hat{N} & 0 \\ 0 & 0 & \hat{N} \end{bmatrix} \begin{bmatrix} G_2 \end{bmatrix} J | & & \\ \end{bmatrix} \end{split}$$

The thermal load vector can be computed as

$$\begin{aligned}
\{f_{T}\} &= \int_{0}^{1} \int_{0}^{1-\xi} \left[G_{2}\right]^{T} \left[G_{1}\right]^{T} \begin{cases} n_{t11} \\ n_{t12} \\ n_{t13} \end{cases} T_{m}(\xi,\eta) |J| d\xi d\eta + \int_{0}^{1} \int_{0}^{1-\xi} \left[G_{2}\right]^{T} \left[G_{1}\right]^{T} \begin{cases} n_{t21} \\ n_{t23} \end{cases} T_{d}(\xi,\eta) |J| d\xi d\eta \\
&+ \int_{0}^{1} \int_{0}^{1-\xi} \left[B_{dkt}\right]^{T} \begin{cases} m_{t11} \\ m_{t12} \\ m_{t13} \end{cases} T_{m}(\xi,\eta) |J| d\xi d\eta + \int_{0}^{1} \int_{0}^{1-\xi} \left[B_{dkt}\right]^{T} \begin{cases} m_{t21} \\ m_{t22} \\ m_{t23} \end{cases} T_{d}(\xi,\eta) |J| d\xi d\eta \\
&= \frac{1}{2} \sum_{i=1}^{n} W_{i} \left[G_{2}\right]^{T} \left[G_{1}\right]^{T} \begin{cases} n_{t11} \\ n_{t12} \\ n_{t13} \end{cases} T_{m}(\xi,\eta) |J| + \frac{1}{2} \sum_{i=1}^{n} W_{i} \left[G_{2}\right]^{T} \left[G_{1}\right]^{T} \begin{cases} n_{t21} \\ n_{t22} \\ n_{t23} \end{cases} T_{d}(\xi,\eta) |J| \\
&+ \frac{1}{2} \sum_{i=1}^{n} W_{i} \left[B_{dkt}\right]^{T} \begin{cases} m_{t11} \\ m_{t12} \\ m_{t13} \end{cases} T_{m}(\xi,\eta) |J| + \frac{1}{2} \sum_{i=1}^{n} W_{i} \left[B_{dkt}\right]^{T} \begin{cases} m_{t21} \\ m_{t22} \\ m_{t23} \end{cases} T_{d}(\xi,\eta) |J| \end{aligned} \tag{2.105}$$

The electric load vector can be computed as

$$\begin{aligned}
\{f_{E}\} &= \int_{0}^{1} \int_{0}^{1-\xi} \left[G_{2}\right]^{T} \left[G_{1}\right]^{T} \begin{cases} n_{e1} \\ n_{e2} \\ n_{e3} \end{cases} |J| d\xi d\eta + \int_{0}^{1} \int_{0}^{1-\xi} \left[B_{dkt}\right]^{T} \begin{cases} m_{e1} \\ m_{e2} \\ m_{e3} \end{cases} |J| d\xi d\eta \\
&= \frac{1}{2} \sum_{i=1}^{n} W_{i} \left[G_{2}\right]^{T} \left[G_{1}\right]^{T} \begin{cases} n_{e1} \\ n_{e2} \\ n_{e3} \end{cases} |J| + \frac{1}{2} \sum_{i=1}^{n} W_{i} \left[B_{dkt}\right]^{T} \begin{cases} m_{e1} \\ m_{e2} \\ m_{e3} \end{cases} |J| \\
&M_{e2} \end{cases} |J| \\
&M_{e3} \end{cases} |J| \\
\end{aligned} (2.106)$$

where |J| is the determinant of the Jacobian matrix given by

$$|J| = 2A$$

 W_i is the weight corresponding to the sampling point. n is the number of the sampling points. Because the integrand is quadratic polynomial in ξ and η , the 3-point Gauss integration is used.

Chapter 3. Genetic Algorithms for the Optimization of Piezoelectric Actuator Locations

3.1 Abstract

Genetic algorithms (GAs) are becoming increasingly popular due to their ability to solve large complex optimization problems which can't be solved by any other method. GAs are robust stochastic global search techniques based on the mechanism of natural selection and natural genetics. In this paper, the advantages and disadvantages of GAs are presented. An improved GA (termed Version 2), developed by changing the mechanism that preserves the best solution and making the population size variable, from an existing version (termed Version 1) is employed to study an important issue in the design of smart structures: The selection of piezoelectric actuator location for shape control. The problem is to find both an optimal placement and optimal voltages for 30 piezoelectric actuators, selected from a maximum of 193 candidate locations, to obtain the best correction to the surface thermal distortions of a thin hexagonal spherical primary mirror. The mirror considered is a representative of the mirrors to be used in the next generation astronomical telescopes. Four different types of thermal distortions are considered. Two optimization problems are considered: (i) to find the optimal locations and optimal voltages suitable for only one type of thermal loads at a time, and (ii) to determine just one set of actuator locations which will reduce the distortion caused by all the four types of thermal loads. Although both these problems are large, difficult and computationally intensive, combinatorial and continuous optimization problems, the latter is a more challenging, multi-criterion optimization problem. A recently developed laminated triangular shell element is used to model the mirror. The

This chapter contains the materials presented at AIAA/ASME/ASCE/AHS/AHC 41st SDM Conference (in MDO session) and published by the conference proceedings and AIAA Journal, Vol. 39, No. 9, 2001, pp. 1818-1822.

results reveal an important phenomenon in the application of GAs to practical problems, namely that the convergence may occur to a non-optimal solution. The improved GAs are found to be more flexible and can get modestly better results than the DeLorenzo algorithm and the computational cost of the GA is found to be less than DeLorenzo's algorithm for the case of 30 actuators. Moreover, using GAs, it is possible to determine one set of actuator locations that reduces distortions caused by all the four different thermal loads.

3.2 Introduction

Placement of sensors/actuators in an optimal fashion has drawn significant attention recently due to its importance in many applications such as shape control, vibration control, acoustic control, buckling control, aeroelastic control, and health monitoring of structures. Various methods have been used to address this issue, including the method of placing piezoelectric actuators in the region of high average strain and away from areas of zero strain, heuristic integer programming, tabu search, simulated annealing, and genetic algorithms, tet.

The problem in this study is that given a set of 193 candidate piezoelectric actuator locations, find a subset of 30 locations and corresponding voltages that provide the best performance i.e. obtain the best correction to the surface thermal distortions of a thin hexagonal spherical primary mirror in the next generation of astronomical telescopes under four different types of thermal loads (Figure 3.1 a)). This is a very large, difficult and computationally intensive optimization problem with more than 1.28*10³⁵ candidate schemes for the placement of piezoelectric actuators. Kapania, Mohan, and Jakubowski studied the above problem using DeLorenzo algorithm.² The solution obtained by such an approach may be a local minimum.

Genetic algorithms (GAs) have attracted considerable attention due to their ability to solve large complex optimization problems. Genetic algorithms are also very general optimization methods which can be applied to virtually any optimization problem. However, implementing GAs for practical problems in an effective way is not easy. When one wants to apply GAs to a particular problem, especially a large complex practical problem, one faces a huge number of choices about how to proceed, with little theoretical guidance on how to make the most appropriate choice. There are a number of questions which must be answered. For example, which encoding scheme should be used? Which is the best selection scheme? Which is the best crossover scheme? How to set the values for the various parameters such as population size, crossover rate, and mutation rate? There are no conclusive answers to these questions. Besides these, one must choose what number of generations or evaluations GAs should run because GAs are not generally allowed to run until the convergence is reached but instead are halted after a certain time. If these issues are not addressed properly, solutions which are of both low quality and are costly may result.

As a first step, the present authors successfully applied genetic algorithms to the large-scale complex optimization problem, the selection of piezoelectric actuator location from more than 1.28*10³⁵ candidate schemes in the case of 30 actuators and more than 1.38*10⁵⁴ candidate schemes in the case of 121 actuators to obtain the best correction to the surface distortion of the primary mirror caused by one type of thermal loads. However, the GAs (termed Version 1) used in that study got results that were only slightly better than DeLorenzo algorithm. Also the cost of the GA was found to be less than that for the DeLorenzo's algorithm for the case of 30 actuators but was more than that for the

distributions are considered in the present study. In general, the optimal locations determined for one type of thermal loads are not optimal for the others. Here, we use the improved GAs not only to find optimal actuator location for each of the four types of thermal loads considered but also to find a single set of actuator location, but with different voltage in a given actuator for each thermal load, that will reduce the thermal distortions caused by all the four different types of thermal loads.

3.3 Genetic Algorithms

Genetic algorithms (GAs) are robust stochastic global optimization techniques based on the mechanism of natural selection and natural genetics. Genetic algorithms were invented by Holland in the 1960s. ¹⁰ They are population-based search algorithms with selection, crossover, mutation, and inversion operations. They work with a population of strings or chromosomes which represent potential solutions from one generation to another until some criteria are satisfied. These population-based search techniques distinguish GAs from traditional point-by-point engineering optimization techniques. Point-by-point search moves from a single point in the search space to the next, using some decision rule such as gradient information in the steepest descent algorithm to decide how to reach the next point in the design space. This point-by-point method may at times be highly inaccurate because it often locates a false peak in a search space with multiple peaks, almost the case in practical problems. But genetic algorithms work with a population of points simultaneously climbing many peaks in parallel, thus are less likely to get trapped at a false peak.

Selection is a bias operation much like Darwinian survival of the fittest, through which the better solutions will be retained and the worse ones will be discarded. Crossover exchanges subparts of two strings much like swapping genes of two chromosomes. Mutation randomly

changes the genes of some locations in the chromosome. Inversion reverses the order of a contiguous section of the chromosome, thus rearranging the order in which genes are arrayed.

Genetic algorithms have been developed by a large number of researchers. There are as many different GAs as there are GA projects. But, the basic idea is the same and is based on the mechanism of natural selection and natural genetics. There are two kinds of operations: one is evolution operation i.e. selection operation; the other is genetic operations such as crossover, mutation and the like. GAs have the following advantages and disadvantages: Advantages:

- --- very general optimization techniques which can be applied to virtually any optimization problem.
 - --- applicable to either continuous or discrete or both types of optimization problems.
 - --- applicable to either linear or nonlinear or both types of optimization problems.
 - --- robust stochastic global optimization techniques.
 - --- population-based search techniques.
 - --- require only objective function evaluations, not the gradient or sensitivity information.
 - --- can find many optimal or near-optimal solutions.
- --- can solve large, complex optimization problems which can't be solved by any other method.
 - --- can get better results than or at least the same results as any other method.

Disadvantages:

- --- not fully understood even now
- --- computation cost would be prohibitive if the effective GAs could not be developed.

To get an insight into the power of genetic algorithms we need to understand the concept of schemas (or schemata) introduced by Holland. A schema is a set of bit strings that can be described by a template made up of ones, zeros, and asterisks, the asterisks representing wild cards or don't care symbols. For example, the schema H=1101**01 represents the set of four strings {11011001, 11010001, 11011101, 11010101}. It is easy to see that a schema with k asterisks represents the set of 2^k strings. On the other hand, a string of length L is matched by 2^L schemata. For example, the string 1101 is matched by sixteen schemata {1101, *101, 1*01, 11*1, 110*, **01, *1*1, *10*, 1**1, 1*0*, 11**, ***1, **0*, *1**, 1***, ****}. The largest region which contains many wild cards will typically be sampled by a large fraction of all the strings in a population. Thus, genetic algorithms that manipulate a population of strings actually sample a vastly large number of regions. This implicit parallelism gives genetic algorithms one of their central advantages over other optimization schemes.

In the field of shape control, noise and vibration control, buckling control, and aeroelastic control of smart structures, the effectiveness of the control system strongly depends on the actuator locations. In this paper, the improved GA (termed Version 2), developed by changing the mechanism that preserves the best solution and making the population size variable (as opposed to being fixed), from our preliminary study (termed Version 1), a microgenetic algorithm, is employed to solve an important problem in the design of smart structures, namely, the selection of actuator locations. Genetic algorithms usually run with a large population. The low convergence rate of GAs is partially due to evaluating the large population, especially in large complex practical problems. Goldberg noted that a small population could be used successfully with GAs (hence called micro-genetic algorithms) if the population is restarted a sufficient number of times. ⁷ This is mainly due to the fact that

genetic algorithms using smaller populations converge in fewer generations than the ones with large populations. In this study, encoding scheme is the same as that used in Version 1, which is, the string would contain a 1 or 0 in the bits corresponding to the presence or absence of actuators. The selection scheme is the same as that used in Version 1, which is, the tournament selection. Crossover scheme is also the same as that used in Version 1, and is the uniform crossover with a rate of 0.5. The main differences between the previous and the present versions are as follows:

--- First, the mechanism that preserves the best solution is different. In Version 1, a randomly selected individual in the population is replaced by the best solution, but in Version 2, the worst individual in the population is replaced by the best solution.

--- Second, there are two important issues in the design of GAs: population diversity and selective pressure. Both of these are influenced by the size of the population. In Version 1, the size of the population is fixed, but in Version 2, the population size can vary during evolution.

3.4 Problem Definition

In the design of next generation astronomical telescopes, one of the most stringent requirements is the maintenance of high surface accuracy of the primary mirror during their operation. A promising method is to use a certain number of piezoelectric actuators bonded onto the rear surface of the primary mirror to correct its distortions without imposing a significant weight penalty. The problem is how to find the optimal location of piezoelectric actuators to maximize their effectiveness. Our problem is as follows:

With 30 piezoelectric actuators, determine from a total of 193 candidate locations an optimal placement, and corresponding optimal voltage for each actuator, to obtain the best

correction to the surface thermal distortions of a thin hexagonal spherical primary mirror subjected to four different types of thermal loads (Figure 3.1 a)). There are two kinds of optimization problems: one is to find a set of locations and corresponding voltages that get the best correction to the surface thermal distortions under each of the four types of thermal loads; the other is to find one set of locations and corresponding voltages that provide the best possible correction to the four surface thermal distortions caused by the four different thermal loads. Note that, for the second case, while the actuator locations are same for all the four cases, the corresponding voltages may not be. The second problem is a multi-criterion problem and obviously is a more challenging problem.

These are very large, difficult and computationally intensive combinatorial optimization problems. Total number of different candidate sets are:

$$_{193}C_{30} = {193 \choose 30} = \frac{193!}{30!(193 - 30)!} = 1.28866 * 10^{35}$$

The geometry and material properties of the mirror and piezoelectric actuators are given in Table 3.1. The temperature distribution at the lower surface of the mirror is assumed to be in the form of linear combination of the first few terms of the Zernike series expressed in terms of Cartesian coordinates x and y with the origin at the center of the mirror. The Cartesian coordinates used to express the temperature distributions are normalized such that they are in the range [-1,1]. The temperature distributions that are considered in this study are given in Table 3.2, where the constant C is used to scale the temperature distributions such that the upper, light-reflecting surface is at a lower temperature than the lower one, with a constant temperature difference $\Delta T_z^{\circ}C$, taken as $0.2^{\circ}C$. The maximum temperature difference between any two grid points across the lower surface of the mirror, $\Delta T_{xy}^{\circ}C$, is taken as $0.5^{\circ}C$.

Table 3.1: Properties and geometry of the mirror and piezoelectric actuators

	Mirror	Piezoelectric strips
	(beryllium)	-
Young's modulus, GPa	293	63
Poisson's ratio	0.1	0.3
Coefficient of thermal	11.5E-6	0.9E-6
expansion, /°C		
d_{31} , d_{32} , m/V		254E-12
Radius, m	10	
Side of the hexagon, m	0.5	0.04166
Thickness, m	0.012	0.25E-3

Table 3.2: Temperature distributions at the lower surface of the mirror

	Temperature distribution
T1	$C[2(x^2+y^2)-1]$
T2	$C[(x+y)(3x^2+3y^2-2)]$
T3	$C(\sum_{i=1}^{i=9} K_i Z_i)$
_T4	C[x+y+2xy]

3.5 Finite Element Modeling

A recently developed laminated triangular shell element in Chapter 2 is used to model the mirror. The element is a combination of the DKT plate bending element and a membrane element derived from the linear strain triangular element with a total of 18 degrees-of-freedom (3 translations and 3 rotations per node). The piezoelectric strips are assumed to be perfectly bonded on the lower surface of the mirror and are modeled as a separate layer. The finite element model consists of 864 flat shell elements, 469 grid points (Figure 3.1 b)). The mirror segment is assumed to be simple-supported at the six vertices 1, 13, 223, 247, 457, and 469.

3.6 Control Algorithms

The surface thermal distortions or the transverse displacements w of the mirror segment are corrected by applying the voltage across the thickness of the strip, which induces a distributed strain in the strip and hence in the mirror. In this study, the thermal deformation w

due to any one type of thermal loads is computed by the finite element analysis. The finite element formulation suggested in Chapter 2 is capable of analyzing panels under thermal loads.

The deformations considered are so small (of order of a few micrometers) that the correction u_i at any point can be assumed to be:

$$u_i = \sum_{j=1}^n \alpha_{ij} V_j$$

where the control input V_j is the voltage applied across the *j*th piezoelectric strip and the influence coefficient α_{ij} is defined as the deformation caused at node *i* due to a unit voltage applied across the *j*th piezoelectric strip alone.

A matrix of influence coefficients of size $m \times n$ (where m represents the total number of grid points in the FEM model, n represents the given number of piezoelectric actuators) is obtained from the finite element model by applying a unit voltage across each of the piezoelectric strips, one at a time. A measure of the overall deviation or the RMS error is given by

$$E = \sqrt{\frac{1}{m} \sum_{i=1}^{m} (w_i + u_i)^2} = \sqrt{\frac{1}{m} \sum_{i=1}^{m} (w_i + \sum_{j=1}^{n} \alpha_{ij} V_j)^2}$$

To obtain the best correction, setting $\partial E/\partial V_k = 0$ gives

$$\sum_{i=1}^{m} (w_i + \sum_{j=1}^{n} \alpha_{ij} V_j) \alpha_{ik} = 0$$

i.e.
$$[A]{V} = {b}$$
, where $A_{kj} = \sum_{i=1}^{m} \alpha_{ij} \alpha_{ik}$, $b_k = -\sum_{i=1}^{m} w_i \alpha_{ik}$

For each set of locations we can get the optimal voltages to minimize RMS error. Different settings of actuator location have different optimal voltages and corresponding minimum RMS error. Thus, **the first optimization problem** is to find a set of locations and

corresponding voltages that minimizes the minimum RMS error for a given type of distortions i.e. of the form,

$$E = \mathop{Min}_{L} \mathop{Min}_{V} E(T, L, V)$$

the second optimization problem is to find a set of locations and corresponding voltages that minimizes the maximum of the minimum RMS error for all the four different distortions i.e. of the form,

$$E = \underset{L}{Min} \underset{T}{Max} \underset{V}{Min} E(T, L, V)$$

3.7 Results and Discussion

In this section, the results obtained by using the improved GAs to solve the above two kinds of optimization problems are presented. To show the effectiveness of Version 2, the corresponding results obtained by DeLorenzo algorithm and Version 1 are also presented.

Following parameters were used:

Version 1: Population size 5, restart control parameter diffrac=0.06.

Version 2: run1 -- Initial population size 10, population size 5, scale=0.5, random=0, restart control parameter diffrac=0.06.

Version 2: run2 -- Initial population size 10, population size 5, scale=0.5, random=0, restart control parameter diffrac=0.0.

Note: The new parameters Initial population size, scale, and random were introduced in Version 2. The parameter scale is used to adjust the selective pressure. The parameter random is used to control whether the initial population size and population size are randomly generated or not. When the parameter random equals 0, the initial population size and population size equal the preset values respectively; otherwise, they equal the numbers generated randomly. The parameter diffrac is used to check the convergence of population.

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When this value is less than the preset value, the new population are randomly generated.

The number of evaluations for the case of 30 strips using DeLorenzo algorithm is 18256. Note that DA in Figure 3.3 to Figure 3.6 represents the results by using DeLorenzo algorithm. The performance of the GAs on different types of thermal loads versus the number of evaluations is shown in Figure 3.2 a) to d) and Figure 3.7 a). Version 2 outperforms Version 1. Results for optimal actuator locations for various cases of thermal loads considered are presented in Figure 3.3 to Figure 3.9. For example, Figure 3.3 a) indicates the actuator locations as obtained using DeLorenzo algorithm to minimize the distortions caused by the type 1 (T1) of thermal loads. It is of interest to see how the optimal locations, obtained to minimize a given type of thermal loads, perform for other types of thermal distortions. In general, it was seen that this performance of the actuators deteriorates (substantially, in some cases) when used for any other type of thermal loads. For example, the actuator location obtained to minimize distortions for the type 1 thermal load, Figure 3.3 a), is not as effective for reducing the other types of thermal distortions. Using this set of actuator location, the RMS error for the type 1 distortions reduces to 0.204 (see top left corner of Figure 3.3 a), whereas the corresponding number is 0.272 for type 2 (top right corner in Figure 3.3 a); 0.359 for type 3 (bottom left corner in Figure 3.3 a); and 0.323 for type 4 (bottom right corner in Figure 3.3 a)). Indeed, it was this deterioration in the performance of a given set of actuator locations for other types of thermal loads that led to the present authors to seek one set of actuator locations that will give acceptable distortion reduction for all the four types of thermal distortions (the second optimization problem). Figure 3.7 shows the one set of actuator locations that can be used for all the four types of thermal distortions. The number of evaluations using genetic algorithms is limited to 15000

except for results shown in Figure 3.8 and Figure 3.9. Figure 3.8 presents the results for the first optimization problem obtained using Version 2 with 30000 evaluations. Figure 3.9 presents the results for the second optimization problem obtained using Version 2 with 20000 evaluations.

The optimal voltages corresponding to this location under each type of thermal loads are shown in Table 3.3 to Table 3.6. The negative sign indicates that the voltage is applied in the direction opposite to the direction of polarization of the piezoelectric material. The voltages shown on some actuators may be too high to generate in space. Several promising methods can be used to lower the control voltages: one is using the piezoelectric materials with higher strain constant as actuators; another is using more piezoelectric actuators; the third is to optimize the design of composite mirror so that result in the smallest thermal deformations and hence reduce the control voltages, and so on. One of the most practical methods is to optimize the actuator location and corresponding voltages by applying constraints to electric voltages such as given the maximum voltages that can be provided. This is another different large and difficult optimization problem.

Table 3.3: Optimal voltages corresponding to optimal location (see Figure 3.9) for thermal load type T1.

Strip	Voltage	Strip	Voltage	Strip	Voltage
Loc.	V	Loc.	V	Loc.	V
6	-1017	65	350	126	-201
7	723	74	-474	138	-659
11	-357	79	-1075	141	-368
23	-497	80	567	145	-191
28	-1011	88	-456	156	-471
29	502	89	188	159	-641
40	-699	93	-587	161	-34
47	-293	98	-222	164	-87
59	-502	119	-351	176	-483
64	-420	121	-313	179	-175

Table 3.4: Optimal voltages corresponding to optimal location (see Figure 3.9) for thermal load type T2.

Strip	Voltage	Strip	Voltage	Strip	Voltage
Loc.	V	Loc.	V	Loc.	V
6	-1656	65	388	126	30
7	961	74	-647	138	-276
11	-749	79	-1417	141	-218
23	-688	80	836	145	-37
28	-1322	88	-774	156	-199
29	595	89	511	159	-380
40	-921	93	-685	161	-81
47	-469	98	-349	164	-11
59	-779	119	-44	176	-303
64	-586	121	-121	179	-181

Table 3.5: Optimal voltages corresponding to optimal location (see Figure 3.9) for thermal load type T3.

Strip	Voltage	Strip	Voltage	Strip	Voltage
Loc.	V	Loc.	V	Loc.	V
6	-1744	65	983	126	-109
7	732	74	-192	138	-66
11	-1111	79	-1306	141	190
23	-967	80	1156	145	-143
28	-1594	88	-797	156	-139
29	826	89	772	159	20
40	-270	93	-539	161	222
47	-102	98	-266	164	-222
59	-527	119	53	176	-50
64	-843	121	213	179	197

Table 3.6: Optimal voltages corresponding to optimal location (see Figure 3.9) for thermal load type T4.

Strip	Voltage	Strip	Voltage	Strip	Voltage
Loc.	V	Loc.	V	Loc.	V
6	-1791	65	338	126	-208
7	817	74	-930	138	-322
11	-1103	79	-1626	141	-146
23	-799	80	695	145	-129
28	-1608	88	-771	156	-245
29	964	89	154	159	-258
40	-49	93	-988	161	120
47	73	98	-571	164	-133
59	-387	119	-268	176	-113
64	-389	121	-166	179	129

3.8 Conclusions

In this study, a version of GAs (Version 2) obtained by changing the mechanism that preserves the best solution and making the population size variable from an existing version (Version 1) were used to solve two kinds of combinatorial and continuous optimization problems in the design of a thin hexagonal spherical primary mirror to be used in the next

generation of astronomical telescopes. One type of optimization problem is to find a set of locations and the corresponding voltages that gives us the best correction to the surface thermal distortions of the primary mirror under a given type of thermal loads; the other is to find one set of locations and corresponding voltages which provide the best correction to all the surface thermal distortions caused by each of the four different thermal loads. The second type is a difficult and computationally intensive multi-criterion optimization problem. The results show that both types of problems are multi-modal optimization problems, that is, there are more than one optimal or near optimal solution. This feature provides a great flexibility in the placement of piezoelectric actuators. The results reveal an important phenomenon in the application of GAs to a practical problem, namely that the convergence to a solution may occur without reaching an optimal or near-optimal solution. This study shows that the improved GAs are more flexible and can get modestly better results than DeLorenzo algorithm and that the cost of the GAs is still less than DeLorenzo's algorithm for the case of 30 actuators. Moreover, a very good solution to the optimization of one set of actuator locations applicable for the correction to all the thermal distortions under each of four different thermal loads has been found. The research shows that the improved GAs are effective in solving optimization problems of determining actuator locations for thermal distortion control.

Acknowledgments

The work was performed under a subcontract from University of Texas at Arlington; themselves working under a grant from the National Science Foundation. We would like to thank Dr. Bernard Grossman, Department Head, Aerospace and Ocean Engineering, for providing considerable computational resources.

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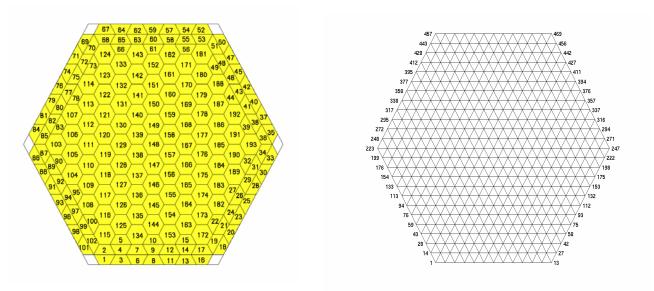


Figure 3.1 a) Piezoelectric actuator candidate locations

b) Finite element mesh

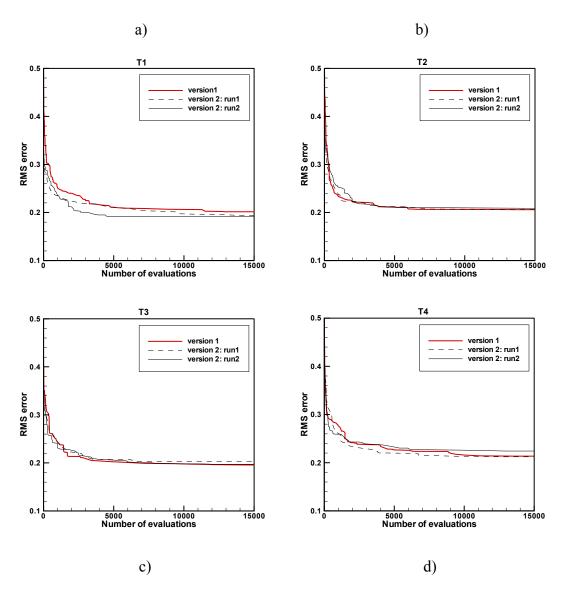


Figure 3.2 Performance of the GAs versus the number of evaluations on a) thermal load T1; b) thermal load T2; c) thermal load T3; and d) thermal load T4.

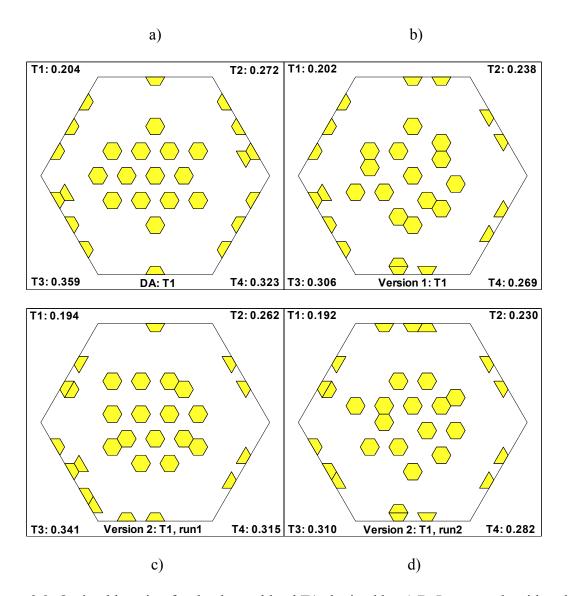


Figure 3.3 Optimal location for the thermal load T1 obtained by a) DeLorenzo algorithm; b) GA Version 1; c) GA Version 2, run1; and d) GA Version 2, run2.

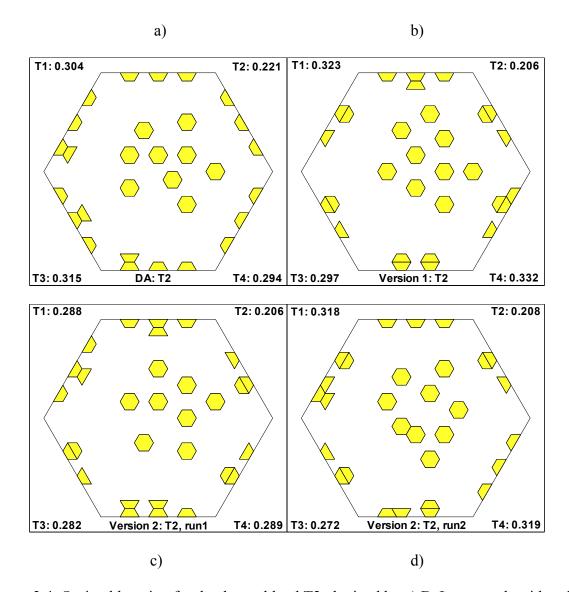


Figure 3.4 Optimal location for the thermal load T2 obtained by a) DeLorenzo algorithm; b) GA Version 1; c) GA Version 2, run1; and d) GA Version 2, run2.

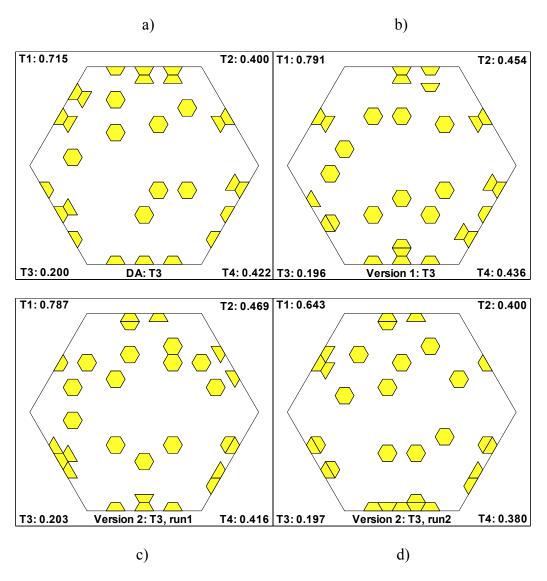


Figure 3.5 Optimal location for the thermal load T3 obtained by a) DeLorenzo algorithm; b) GA Version 1; c) GA Version 2, run1; and d) GA Version 2, run2.

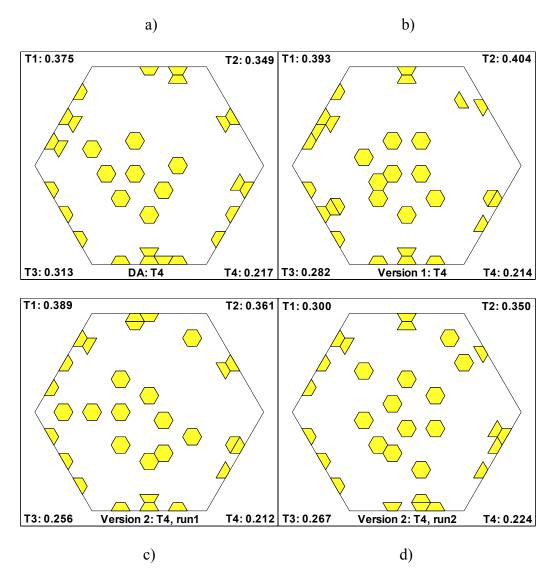


Figure 3.6 Optimal location for the thermal load T4 obtained by a) DeLorenzo algorithm; b) GA Version 1; c) GA Version 2, run1; and d) GA Version 2, run2.

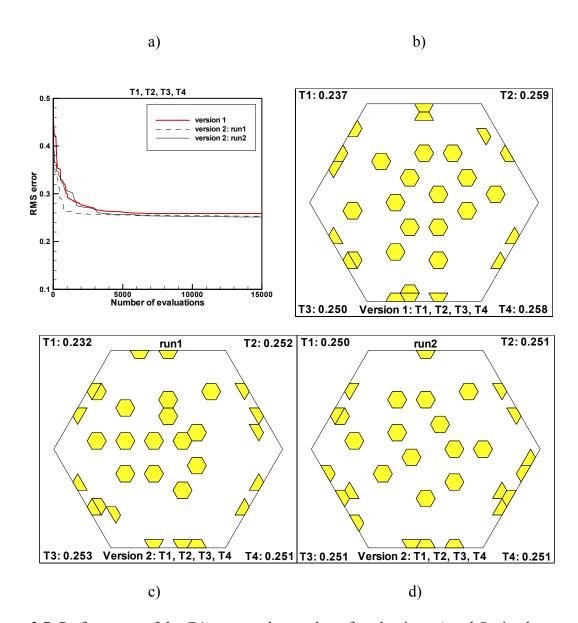


Figure 3.7 Performance of the GAs versus the number of evaluations a) and Optimal location for the thermal loads T1, T2, T3, and T4 obtained by b) GA Version 1; c) GA Version 2, run1; and d) GA Version 2, run2.

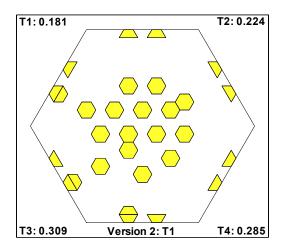


Figure 3.8 Optimal location for the thermal load T1 obtained using Version 2 with 30000 evaluations. Note that for the thermal load T1, the RMS error has reduced by 11.3%.

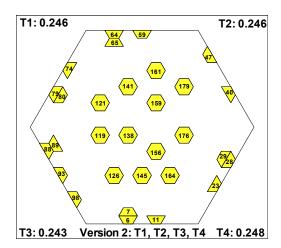


Fig 3.9 Optimal location for the thermal loads T1, T2, T3, and T4 obtained using Version 2 with 20000 evaluations

Chapter 4. Toward More Effective Genetic Algorithms for the **Optimization of Piezoelectric Actuator Locations**

4.1 Abstract

Genetic algorithms (GAs), inspired by natural evolution, have drawn considerable attention during the past two decades due to their ability to solve large complex optimization problems that may be difficult to solve using conventional gradient-based optimization techniques. In our previous paper, some advantages and disadvantages of GAs were presented, and two versions of GA's (termed GA Version 1 and GA Version 2) were used to solve two kinds of difficult, computationally intensive, combinatorial and continuous large-scale optimization problems. The problems consisted of finding both an optimal placement and optimal voltages of 30 piezoelectric actuators, from 193 candidate locations, with more than 1.28*10³⁵ possible solutions, to obtain the best correction to the surface thermal distortions of a thin hexagonal spherical primary mirror. The thermal distortions were caused by four different types of spatial temperature distributions. The first problem was the one in which a set of actuator locations corresponding to each of the four types of thermal loads was obtained and in the other one set of actuator location was obtained for controlling thermal distortions caused by all the four types of thermal loads. The latter is a more challenging, multi-criterion optimization problem. A laminated triangular shell element was used to model the mirror. In this paper, an improved GA, termed GA Version 3, is developed from the GA Version 2 and is employed to resolve the problems studied in our previous paper and even larger problems. Two key differences between the GA Version 3 and the earlier GA Version 2 are: (i) to apply random-mutation hill climbing to elitists, and

This chapter contains the materials presented at AIAA/ASME/ASCE/AHS/AHC 42nd SDM Conference (in MDO session) and published by the conference proceedings and AIAA Journal, Vol. 40, No. 6, 2002, pp. 1246-1250.

(ii) to apply mutation to micro-genetic algorithms. For the problem of interest in this study, it is seen that the above two modifications significantly improve the performance of GAs. The results of using same parameter settings of the GA Version 3 to even a larger problem, that is, choosing a set of 121 piezoelectric actuator locations from 193 candidate locations with more than 1.38*10⁵⁴ possible solutions, are also reported. The results show that the RMS error, for the first type of problem, for the 121- piezoelectric actuator case obtained using the GA Version 3, is basically the same as that obtained using the DeLorenzo's algorithm, but a different setting of piezoelectric actuator locations is obtained. Moreover, a very good setting of 121 actuators for the second optimization problem is found.

4.2 Introduction

Due to the well-known advantages of adaptive structures over traditional structures, significant research has been conducted in this field in the past two decades. The performance of such systems strongly depends on piezoelectric actuator locations. In our previous paper, various methods addressing this issue were mentioned and some advantages and disadvantages of GAs were presented.

In the previous studies, the present authors successfully applied two versions of GA's (termed GA Version 1 and GA Version 2), developed by the present authors from Carroll's FORTRAN Genetic Algorithm Driver ², to solve two kinds of large-scale optimization problems. These problems entailed choosing a set of 30 piezoelectric actuator locations, from 193 possible candidate locations, and with more than 1.28*10³⁵ possible solutions, to correct the thermal distortions of a thin hexagonal spherical primary mirror (Figure 4.1 a)). The thermal distortions were generated from four different thermal load distributions. The two types of optimization problems considered in the earlier study were: (i) to individually find

the optimal locations and optimal voltages suitable for each type of thermal loads; (ii) to determine just one set of actuator locations which will reduce the distortion caused by all the four types of thermal loads. Both these problems are difficult and computationally intensive. The latter type is a more challenging, multi-criterion optimization problem. A laminated triangular shell element developed in Chapter 2 was used to model the mirror. The first problem was studied by Kapania, Mohan, and Jakubowski ⁴ using DeLorenzo algorithm. The solution obtained by such an approach may be a local minimum. The main conclusions from our previous studies are as follows:

- --- The design search space is highly multi-modal.
- --- GA Version 1 got only slightly better results than the DeLorenzo algorithm and the computation cost was less than that of the DeLorenzo's algorithm for the case of 30 piezoelectric actuators but higher for the case of 121 piezoelectric actuators.
- --- Both GA Version 1 and GA Version 2 are robust for the optimization of piezoelectric actuator locations.
 - --- The GA Version 2 has more flexibility than GA Version 1.
- --- The GA Version 2 can get modestly better results than DeLorenzo algorithm for both optimization problems for the case of 30 piezoelectric actuators.
- --- The convergence to a solution may occur without reaching an optimal or near-optimal solution.
 - --- More than one sub-optimal solution to each problem were found.
- --- Optimal location obtained for one type of thermal loads may perform poorly for other types of thermal loads.
 - --- GAs can determine one set of actuator locations which is good for all the four types of

thermal loads considered for these studies. The needed voltages will be different for different thermal loads.

In the present paper, an improved GA, termed GA Version 3, developed by modifying the GA Version 2, is employed to solve the two problems studied in Ref. 1 and also an even larger problem. Two key differences between GA Version 3 and GA Version 2 are:

- (i) Application of random-mutation hill climbing to elitists, and
- (ii) Application of mutation to micro-genetic algorithms.

From the results of running GA Version 3, for the problem of interest in this study, it can be seen that the above two modifications significantly improve the performance of GAs. The results of using same parameter settings of GA Version 3 to even larger problems of choosing a set of 121 piezoelectric actuator locations from 193 possible candidate locations with more than $1.38*10^{54}$ possible solutions are also reported.

4.3 Genetic Algorithms

Genetic algorithms (GAs) are robust stochastic global search techniques based on the mechanics of natural selection and genetics. GAs were invented by Holland in the 1960s, ⁵ and later developed by De Jong, Goldberg, and many others. These algorithms evolve a population of chromosomes using selection and genetic operations such as crossover, mutation, and so on from one generation to another, hopefully to get a better solution. The selection operation is based on the Darwinian principle of the survival of the fittest. Genetic algorithms use only one very general assumption, namely: better individuals can reproduce better offspring more probably than the worst individuals. The concept's simplicity, flexibility and robust performance makes GAs one of the most exciting fields in evolutionary computation.

In the field of shape control, noise and vibration control, buckling control, and aeroelastic control of smart structures, the effectiveness of the control system strongly depends on the actuator locations. In this paper, the improved GA (termed Version 3), developed by the present authors from an earlier GA, the GA Version 2, is employed to solve an important problem in the design of smart structures, namely, the selection of actuator locations. Two key differences between GA Version 3 and GA Version 2 are as follows:

- (i) to apply random-mutation hill climbing to elitists,
- (ii) to apply mutation to micro-genetic algorithms.

---First, random-mutation hill climbing was applied to elitists – the best individuals. This came from our initial intuition that if we don't know which individual can reproduce a better offspring (we need to recall from Ref.6, p.201, "in their purest form, genetic algorithms are blind search procedures", but in practical form GAs are directed search techniques not completely blind.), choosing the best one generally gets the highest probability. Random-mutation hill climbing outperformed the steepest-ascent hill climbing and next-ascent hill climbing (Ref. 5, p.129).

---Second, mutation was applied to micro-genetic algorithms. The micro-genetic algorithms here are the same as the genetic algorithms in the commonly used sense except that they include restart function in the outer loop and usually use a small population size in order to get the effect of faster convergence rate than the large population size usually used in genetic algorithms. In the literature, mutation rate was set 0.0 in most cases whenever the microgenetic algorithms were used. This is probably due to traditional view of the negative effect of mutation – namely that the use of mutation slows down convergence. Recent advances in genetic algorithms show that the mutation operation also has some positive effects –

speeding up convergence as well as providing the diversity of population, thereby avoiding a premature convergence to a local optimum.

4.4 Problem Definition

In the design of next generation of astronomical telescopes, one of the most stringent requirements will be the maintenance of high surface accuracy of the primary mirror during their operation. A promising method is to use a certain number of piezoelectric actuators bonded onto the rear surface of the primary mirror to correct its distortions without imposing a significant weight penalty. The problem is how to find the optimal location of piezoelectric actuators to maximize their effectiveness. Our problem is as follows:

With, *n*, the number of piezoelectric actuators available, determine from a total of 193 candidate locations an optimal placement, and corresponding optimal voltage for each actuator, to obtain the best correction to the surface thermal distortions of a thin hexagonal spherical primary mirror subjected to four different types of thermal loads (Figure 4.1 a)). There are two kinds of optimization problems: one is to find a set of locations and corresponding voltages that get the best correction to the surface thermal distortions under each of the four types of thermal loads; the other is to find one set of locations and corresponding voltages that provide the best possible correction to the surface thermal distortions caused by all the four types of thermal loads. Note that, for the second problem, while the actuator locations are the same for all the four thermal distortions, the corresponding voltages may not be. The second problem is a multi-criterion problem and obviously is a more challenging problem.

These are very large, difficult and computationally intensive combinatorial optimization problems. Total number of different candidate sets are:

$$^{193}C_n = \binom{193}{n} = \frac{193!}{n!(193-n)!}$$

So, for the case of 30 and 121 piezoelectric actuators, the number of different candidate sets are respectively,

$$^{193}C_{30} = 1.28866 * 10^{35}$$

$$^{193}C_{121} = 1.38231*10^{54}$$

The geometry and material properties of the mirror and piezoelectric actuators are given in Table 4.1. The temperature distribution at the lower surface of the mirror is assumed to be in the form of linear combination of the first few terms of the Zernike series expressed in terms of Cartesian coordinates x and y with the origin at the center of the mirror. The Cartesian coordinates used to express the temperature distributions are normalized such that they are in the range [-1,1]. The temperature distributions that are considered in this study are given in Table 2, where the constant C is used to scale the temperature distributions such that the upper, light-reflecting surface is at a lower temperature than the lower surface, with a constant temperature difference ΔT_z °C, taken as 0.2°C. The maximum temperature difference between any two grid points across the lower surface of the mirror, ΔT_{xy} °C, is taken as 0.5°C.

4.5 Finite Element Modeling

A laminated triangular shell element developed in Chapter 2 is used to model the mirror. The element is a combination of the DKT plate bending element and a membrane element derived from the linear strain triangular element with a total of 18 degrees-of-freedom (3 translations and 3 rotations per node). The piezoelectric strips are assumed to be perfectly

bonded on the lower surface of the mirror and are modeled as a separate layer. The finite element model consists of 864 flat shell elements, 469 grid points (Figure 4.1 b)). The mirror segment is assumed to be simple-supported at the six vertices 1, 13, 223, 247, 457, and 469.

4.6 Control Algorithms

The surface thermal distortions or the transverse displacements w of the mirror segment are corrected by applying the voltage across the thickness of the strip, which induces a distributed strain in the strip and hence in the mirror. In this study, the thermal deformation w due to any one type of thermal loads is computed by the finite element analysis. The finite element formulation suggested in Chapter 2 is capable of analyzing panels under thermal loads.

The deformations considered are so small (of order of a few micrometers) that the correction u_i at any point can be assumed to be:

$$u_i = \sum_{j=1}^n \alpha_{ij} V_j$$

where the control input V_j is the voltage applied across the *j*th piezoelectric strip and the influence coefficient α_{ij} is defined as the deformation caused at node *i* due to a unit voltage applied across the *j*th piezoelectric strip alone.

A matrix of influence coefficients of size $m \times n$ (where m represents the total number of grid points in the FEM model, n represents the given number of piezoelectric actuators) is obtained from the finite element model by applying a unit voltage across each of the piezoelectric strips, one at a time. A measure of the overall deviation or the RMS error is given by

$$E = \sqrt{\frac{1}{m} \sum_{i=1}^{m} (w_i + u_i)^2} = \sqrt{\frac{1}{m} \sum_{i=1}^{m} (w_i + \sum_{j=1}^{n} \alpha_{ij} V_j)^2}$$

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To obtain the best correction, setting $\partial E/\partial V_k = 0$ gives

$$\sum_{i=1}^{m} (w_i + \sum_{j=1}^{n} \alpha_{ij} V_j) \alpha_{ik} = 0$$

i.e.
$$[A]{V} = {b}$$
, where $A_{kj} = \sum_{i=1}^{m} \alpha_{ij} \alpha_{ik}$, $b_k = -\sum_{i=1}^{m} w_i \alpha_{ik}$

For each set of locations we can get the optimal voltages to minimize RMS error. Different settings of actuator location have different optimal voltages and corresponding minimum RMS error. Thus, **the first optimization problem** is to find a set of locations and corresponding voltages that minimizes the minimum RMS error for one type of distortions i.e. of the form,

$$E = \mathop{Min}_{L} \mathop{Min}_{V} E(T, L, V)$$

the second optimization problem is to find a set of locations and corresponding voltages that minimizes the maximum of the minimum RMS error for all the four different distortions i.e. of the form,

$$E = \underset{L}{Min} \underset{T}{Max} \underset{V}{Min} E(T, L, V)$$

Obviously, the second problem is a more realistic, but computationally very challenging problem.

4.7 Results and Discussion

In this section, the results obtained by using the GA Version 3 developed by the present authors from the GA Version 2 ¹ to solve the above two kinds of optimization problems are presented. Following parameters were used:

Version 1: Population size 5, crossover rate =0.5, mutation rate=0.0, restart control parameter diffrac=0.06.

Version 2: run1 -- Initial population size 10, population size 5, scale=0.5, random=0, crossover rate =0.5, mutation rate=0.0, restart control parameter diffrac=0.06.

Version 2: run2 -- Initial population size 10, population size 5, scale=0.5, random=0, crossover rate =0.5, mutation rate=0.0, restart control parameter diffrac=0.0.

Version 3: run1 -- Initial population size 10, population size 5, scale=0.5, random=0, crossover rate =0.5, mutation rate=0.01, No_of_max_generation_inner_loop=15, No_of_best_mutation_bits=2.

Version 3: run2 -- Initial population size 10, population size 5, scale=0.5, random=0, crossover rate =0.5, mutation rate=0.01, No_of_max_generation_inner_loop=10, No_of_best_mutation_bits=2.

Note: The new parameter No_of_best_mutation_bits which represents mutation bits for elitists in each generation was introduced in Version 3. The parameter scale is used to adjust the selective pressure. The parameter random is used to control whether the initial population size and population size are randomly generated or not. When the parameter random equals 0, the initial population size and population size equal the preset values respectively; otherwise, they equal the numbers generated randomly. The parameter diffrac is used to check the convergence of population. When this value is less than the preset value, the new population are randomly generated.

In this study, the number of evaluations using genetic algorithms is limited to 15000. For 30 piezoelectric actuators, the performance of different versions of the GAs for the two kinds of optimization problems is shown in Figure 4.2 and Figure 4.3, respectively. Figure 4.2 shows the performance of the GAs versus the number of evaluations for the first optimization problems for each of the four types of thermal loads; Figure 4.3 shows the performance of

the GAs with respect to the number of evaluations for the second optimization problem. The optimal actuator locations obtained by the two runs of the GA version 3 for the two kinds of optimization problems are presented in Figure 4.4 and 4.5, respectively. Figure 4.4 presents the results for the first kind of optimization problems for each of the four types of thermal loads; Figure 4.5 presents the results for the second kind of optimization problem for all the four types of thermal loads. The number in each corner in Figure 4.4 and Figure 4.5 represents the RMS error corresponding to that type of thermal distortion. The set of actuator locations, shown in the top-left corner of Figure 4.4, was obtained to minimize the error due to thermal load T1. The RMS error under thermal load T1 becomes 0.191; the error under T2, T3 and T4 becomes 0.245, 0.332 and 0.326, respectively. Hence it is seen that this set of actuator locations, while best for T1, may perform poorly when used for other types of thermal loads.

The optimal voltages corresponding to the optimization location (Figure 4.5 b)) for each type of thermal loads are listed in Tables 4.3 to 4.6. The negative sign indicates that the voltage is applied in the direction opposite to the direction of polarization of the piezoelectric material.

Regarding the performance of GAs (Figures 4.2 and 4.3), GA Version 3 not only approached near optimal solution very fast, but also found the best solution. For example, when used for the second kind of optimization problem – the multi-criterion optimization problem (Figure 4.3), GA Version 3 run1 approached a near-optimal solution, with distortion RMS error value of 0.251, in less than 2000 evaluations and GA Version 3 run2 obtained the same RMS error in less than 5000 evaluations. Moreover, the latter obtained the best solution, with a distortion RMS error value of 0.230 at the end of 15000 evaluations, but GA

Version 2 run2 did not reach a near-optimal solution, with a distortion RMS error of 0.251 until more than 13000 evaluations. The GA Version 2 run1 and GA Version 1 did not even reach any near-optimal solution with the level of distortion RMS error of 0.251, at the end of stipulated 15000 evaluations. The research shows that the GA Version 3 is more effective than previous versions in solving optimization problem of determining actuator locations for thermal distortion control.

Results for 121 actuators: Figure 4.6 to Figure 4.10 and Table 4.7 to Table 4.10 present the results for the two kinds of optimization problems for the case of 121 piezoelectric actuators. The performance of GA version 3 for the two kinds of optimization problems is shown in Figure 4.6 and Figure 4.7, respectively. We can see from these figures that the GA version 3 still reaches near-optimal solutions very fast even though the search space of 1.38*10⁵⁴ different sets of actuator locations for this case is much larger than that for the case of 30 piezoelectric actuators. Figure 4.8 presents the actuator location for the first kind of optimization problems and corresponding RMS errors obtained using the DeLorenzo's algorithm. Figure 4.9 and Figure 4.10 present the actuator location for the two kinds of optimization problems and corresponding RMS errors obtained by the two runs of the GA version 3. For the case of 121 actuators, we can see in Figure 4.8 and Figure 4.9 that the RMS errors for the first kind of optimization problems for each of the four different types of thermal loads obtained by the two runs of GA version 3 are basically the same as those obtained by the DeLorenzo's algorithm but different sets of actuator locations are found. It is of interest to note that the performance, of a set of actuator locations determined to be best for a given type of thermal loads, does not deteriorate as much for other loads as it did in the case of 30 actuators. This implies that if one can employ a larger number of actuators, the

performance will be better for a large variation in the thermal loads. A large number of actuators will thus provide a more robust set of locations. Moreover, the magnitude of the RMS error reduces as one employs a larger number of actuators. We can see that the RMS error in the worst case T4 for the set of actuator locations in Figure 4.10 a) for the second kind of optimization problem is as good as that in the best case T4 in Figure 4.8 or Figure 4.9 for the first kind of optimization problem. This demonstrates that the search space is highly multi-modal and that the GA version 3 is very powerful to search the solution as good as possible for the multi-criterion optimization problem. The optimal voltages corresponding to the optimization location (Figure 4.10 a)) for each type of thermal loads are listed in Table 4.7 to Table 4.10. We can see that even though the optimal voltages for this case are lower than those for the case of 30 actuators they may be still too high to generate in space. Some promising methods can be used to lower the control voltages. For example, one can select the piezoelectric materials with higher strain constant as actuators or optimize the actuator location and corresponding voltages by applying constraints to electric voltages such as given the maximum voltages that can be provided. The latter is another different large and computationally intensive optimization problem.

4.8 Conclusions

In this study, an improved version of GAs (Version 3) developed from the GA Version 2 by applying random-mutation hill climbing to elitists and applying mutation to micro-genetic algorithms, was used to solve two kinds of combinatorial and continuous large-scale optimization problems for two cases – selecting 30 and 121 actuator locations from 193 candidate locations in the design of a thin hexagonal spherical primary mirror to be used in the next generation of astronomical telescopes (Figure 4.1). One type of optimization

problem is to find a set of locations and the corresponding voltages that gives us the best correction to the surface thermal distortions of the primary mirror under a given type of thermal loads; the other is to find one set of locations and corresponding voltages which provide the best correction to the surface thermal distortions caused by all the four different kinds of thermal loads. The two types of problems are difficult and computationally intensive. The second type is a more challenging, multi-criterion optimization problem. The search space for these problems is highly multi-modal and conventional point-by-point optimization techniques usually get stuck at the local optimum, but population-based GAs are very good at searching such space and more likely get better results than the traditional techniques. The search space of 1.38*10⁵⁴ different sets of actuator locations for the case of 121 piezoelectric actuators is much larger than that for the case of 30 piezoelectric actuators, but the GA version 3 still converged very fast and finally found a very good set of actuator locations which can be used to reduce all the four kinds of thermal distortions. The results show that the two modifications employed in this study significantly improve the performance of GAs. The current version of GAs are more effective than the previous versions in solving optimization problems of determining actuator locations for thermal distortion control.

The problems for this study are computationally intensive. Getting one solution using the GAs with the limit of 15000 evaluations for the second optimization problem took more than a week. GAs are very general and robust optimization methods which can be applied to virtually any optimization problem. Parallel GAs can significantly reduce the time and be more natural to imitate the evolution of the nature, so our current research is to develop parallel GAs.

Acknowledgments

The work was performed under a subcontract from University of Texas at Arlington; themselves working under a grant from the National Science Foundation. We would like to thank the College of engineering for providing computational resources for this research.

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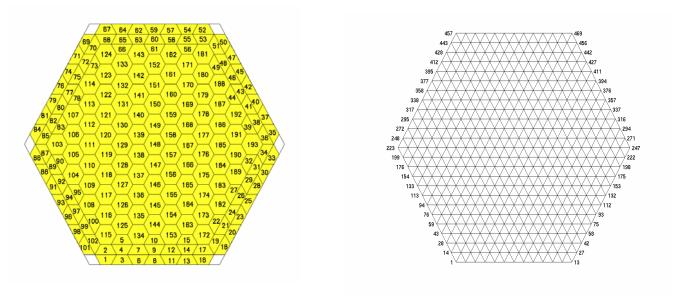


Figure 4.1 a) Piezoelectric actuator candidate locations

b) Finite element mesh

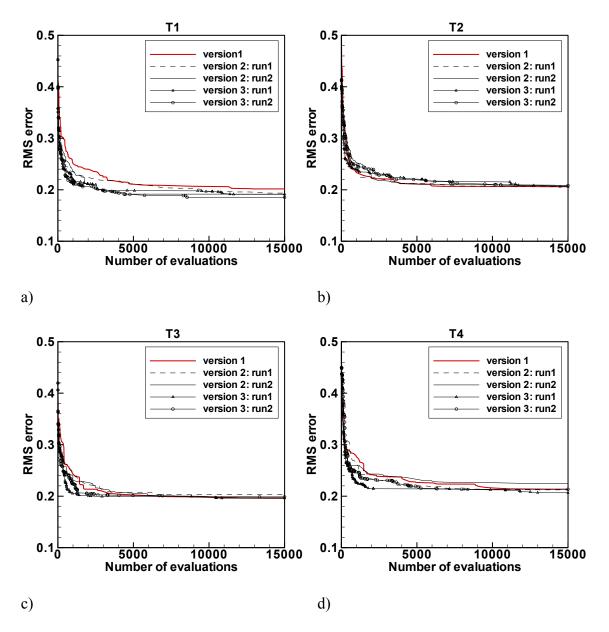


Figure 4.2 Performance of the GAs: The RMS error vs the number of evaluations for the first optimization problem for the case of, a) thermal load T1, b) thermal load T2, c) thermal load T3, and d) thermal load T4 (30 actuators).

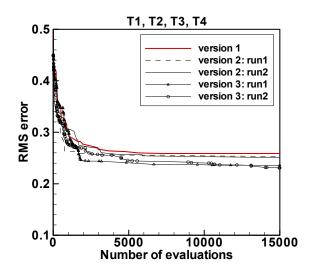


Figure 4.3 Performance of the GAs: Maximum of the four minimum RMS errors vs. the number of evaluations for the second optimization problem (30 actuators).

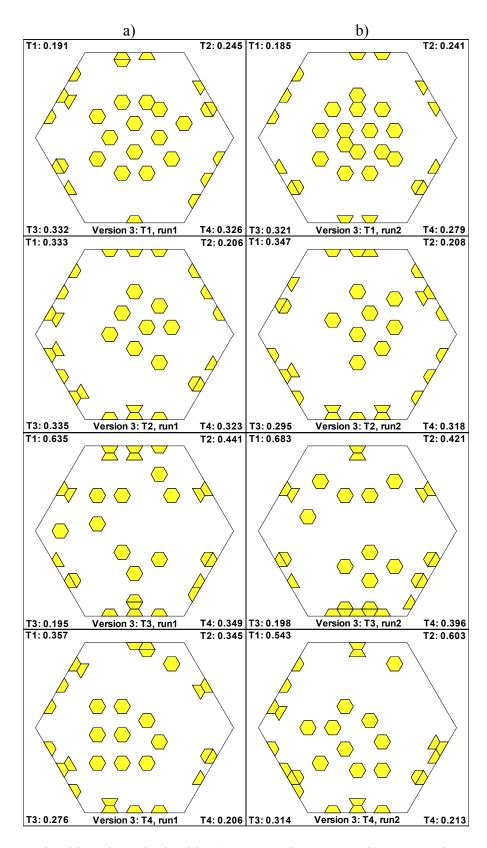


Figure 4.4 Optimal location, obtained by a) GA Version 3, run1; b) GA Version 3, run2 (30 actuators), for the first optimization problem for each of the four thermal loads.

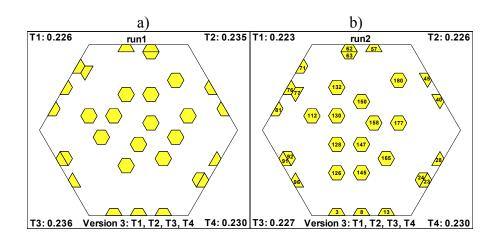


Figure 4.5 Optimal location for the second optimization problem obtained by a) GA Version 3, run1; b) GA Version 3, run2 (30 actuators).

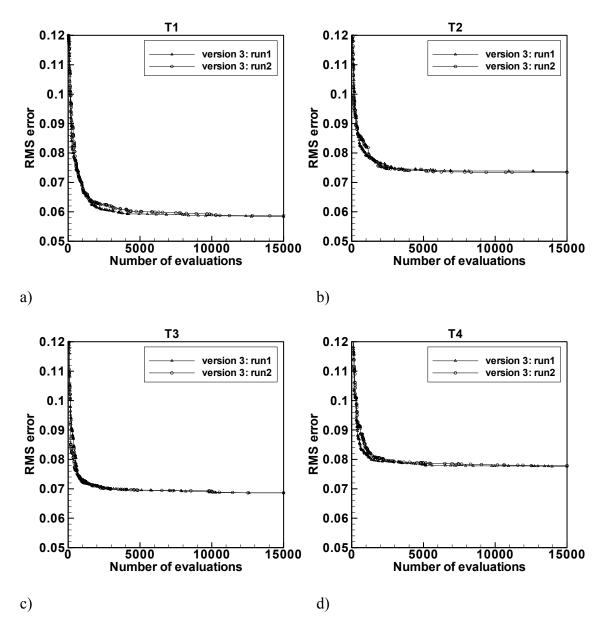


Figure 4.6 Performance of the GAs: The RMS error vs the number of evaluations for the first optimization problem for the case of, a) thermal load T1, b) thermal load T2, c) thermal load T3, and d) thermal load T4 (121 actuators).

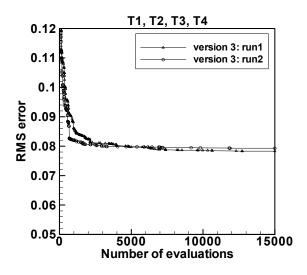


Figure 4.7 Performance of the GAs: Maximum of the four minimum RMS errors vs the number of evaluations for the second optimization problem (121 actuators).

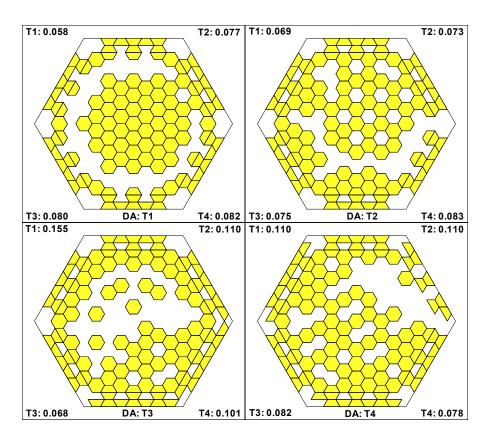


Figure 4.8 Optimal location, obtained by the DeLorenzo's algorithm (121 actuators), for the first optimization problem for each of the four thermal loads.

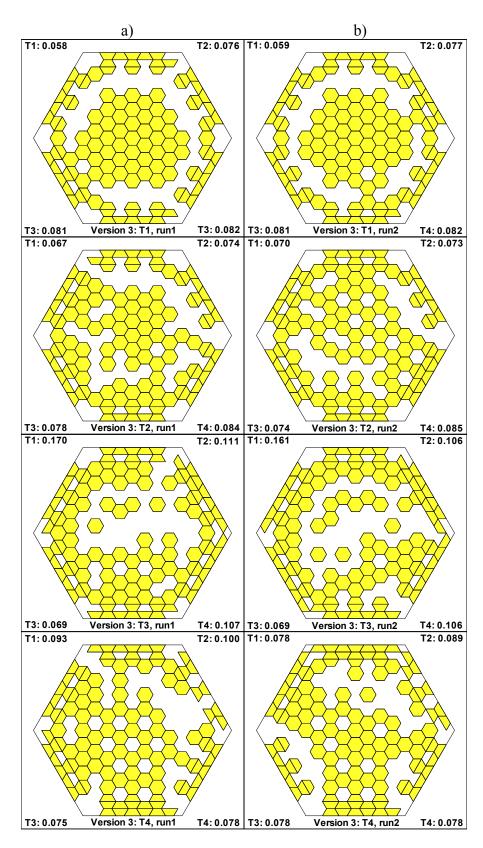


Figure 4.9 Optimal location, obtained by a) GA Version 3, run1; b) GA Version 3, run2 (121 actuators), for the first optimization problems for each of four thermal loads.

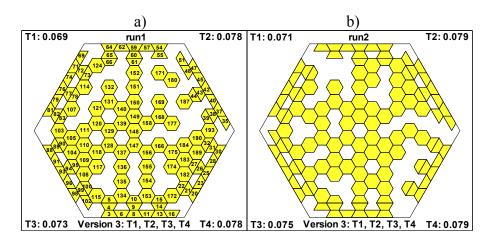


Figure 4.10 Optimal location for the second optimization problem obtained by a) GA Version 3, run1; b) GA Version 3, run2 (121 actuators).

Table 4.1: Properties and geometry of the mirror and piezoelectric actuators

	Mirror	Piezoelectric strips
	(beryllium)	
Young's modulus, GPa	293	63
Poisson's ratio	0.1	0.3
Coefficient of thermal	11.5E-6	0.9E-6
expansion, /°C		
$d_{31}, d_{32}, m/V$		254E-12
Radius, m	10	
Side of the hexagon, m	0.5	0.04166
Thickness, m	0.012	0.25E-3

Table 4.2: Temperature distributions at the lower surface of the mirror

	Temperature distribution
T1	$C[2(x^2+y^2)-1]$
T2	$C[(x+y)(3x^2+3y^2-2)]$
T3	$C(\sum_{i=1}^{i=9}K_iZ_i)^*$
T4	C[x+y+2xy]

^{*} K_i , from Table 2 (back surface), Z_i , terms of Zernike series (p. 216) of Ref. 7.

Table 4.3: Optimal voltages corresponding to optimal location (see Figure 4.5 b)) for thermal load type T1.

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-111	63	474	128	-485
8	-422	71	-246	130	-394
13	-157	76	-903	132	-151
23	-933	77	471	145	-189
24	470	81	-267	147	-430
28	-573	91	-926	150	-491
40	-541	92	367	158	-453
45	-460	96	-466	165	-406
57	-398	112	-209	177	-443
62	-813	126	-137	180	-107

Table 4.4: Optimal voltages corresponding to optimal location (see Figure 4.5 b)) for thermal load type T2.

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-372	63	564	128	-118
8	-769	71	-368	130	-164
13	-400	76	-1112	132	-97
23	-1357	77	544	145	-15
24	721	81	-345	147	-191
28	-784	91	-1614	150	-330
40	-734	92	1026	158	-240
45	-645	96	-607	165	-177
57	-619	112	-20	177	-318
62	-1125	126	57	180	-130

Table 4.5: Optimal voltages corresponding to optimal location (see Figure 4.5 b)) for thermal load type T3.

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-579	63	1337	128	-15
8	-1040	71	-123	130	35
13	-716	76	-902	132	240
23	-1517	77	678	145	-207
24	556	81	-79	147	-98
28	-788	91	-1433	150	193
40	-125	92	1110	158	-25
45	-369	96	-480	165	-261
57	-312	112	195	177	64
62	-1606	126	-48	180	317

Table 4.6: Optimal voltages corresponding to optimal location (see Figure 4.5 b)) for thermal load type T4.

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-563	63	837	128	-301
8	-1021	71	-576	130	-216
13	-682	76	-1394	132	-44
23	-1534	77	615	145	-168
24	765	81	-559	147	-213
28	-667	91	-1569	150	-129
40	50	92	527	158	-212
45	-136	96	-958	165	-230
57	-268	112	-188	177	-23
62	-1097	126	-142	180	207

Table 4.7: Optimal voltages corresponding to optimal location (see Figure 4.10 a)) for thermal load type T1.

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-380	62	-185	121	-127
4	498	64	-385	124	59
5	-259	65	482	128	-154
6	-201	66	-186	129	-155
8	-551	69	17	131	-113
9	647	71	-490	132	-103
10	-348	72	557	134	67
11	-188	73	-250	135	-87
13	-402	74	-225	136	-162
14	493	76	-682	137	-128
15	-256	77	756	139	-181
16	37	78	-356	140	-115
20	-461	79	-230	147	-219
21	551	81	-490	148	-121
22	-286	82	577	149	-56
23	-245	83	-273	150	-181
25	-659	88	-485	151	-152
26	784	89	583	152	-8
27	-495	90	-314	153	66
28	-254	91	-242	154	-86
30	-465	93	-674	155	-165
31	527	94	788	156	-101
32	-249	95	-474	158	-267
35	14	96	-238	166	-241
37	-506	98	-478	168	-53
38	586	99	548	169	-237
39	-279	100	-287	171	-64
40	-219	102	21	172	68
42	-687	103	56	174	-134
43	706	104	53	175	-48
44	-238	105	25	177	-237
45	-215	107	13	180	-41
47	-467	108	74	182	77
48	360	109	55	183	58
51	74	110	-94	184	-146
54	-361	111	-97	187	-61
55	370	114	21	189	83
57	-202	115	68	190	-23
59	-558	117	-127	193	55
60	631	118	-89	170	
61	-236	120	-109		

Table 4.8: Optimal voltages corresponding to optimal location (see Figure 4.10 a)) for thermal load type T2.

thermal load	* *	T ~	T	T ~	T
Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-693	62	-255	121	-51
4	708	64	-558	124	51
5	-351	65	586	128	-42
6	-356	66	-224	129	-58
8	-932	69	-10	131	-67
9	975	71	-594	132	-63
10	-501	72	639	134	97
11	-366	73	-296	135	-15
13	-661	74	-305	136	-45
14	703	76	-818	137	-42
15	-349	77	869	139	-84
16	-35	78	-418	140	-57
20	-638	79	-299	147	-94
21	660	81	-584	148	-50
22	-320	82	632	149	-37
23	-323	83	-283	150	-112
25	-888	88	-631	151	-127
26	980	89	702	152	9
27	-586	90	-371	153	92
28	-332	91	-314	154	-30
30	-644	93	-864	155	-66
31	667	94	955	156	-42
32	-328	95	-526	158	-155
35	-8	96	-333	166	-128
37	-614	98	-606	168	-29
38	666	99	672	169	-189
39	-305	100	-305	171	-70
40	-293	102	-56	172	75
42	-850	103	63	174	-74
43	859	104	77	175	-33
44	-328	105	52	177	-174
45	-298	107	64	180	-54
47	-570	108	85	182	83
48	411	109	60	183	61
51	24	110	-45	184	-113
54	-497	111	-25	187	-51
55	406	114	55	189	84
57	-277	115	98	190	-7
59	-773	117	-27	193	41
60	824	117	-25	173	71
			-39		
61	-352	120	-39		

Table 4.9: Optimal voltages corresponding to optimal location (see Figure 4.10 a)) for thermal load type T3.

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-866	62	-188	121	73
4	839	64	-502	124	79
5	-447	65	466	128	-24
6	-433	66	-29	129	12
8	-1138	69	38	131	7
9	1161	71	-405	132	108
10	-652	72	457	134	81
11	-462	73	-170	135	-81
13	-818	74	-202	136	-88
14	855	76	-522	137	-39
15	-483	77	600	139	3
16	-70	78	-262	140	34
20	-753	79	-202	147	-52
21	759	81	-332	148	-12
22	-414	82	456	149	8
23	-375	83	-198	150	43
25	-1017	88	-505	151	46
26	1098	89	617	152	122
27	-663	90	-333	153	78
28	-363	91	-247	154	-106
30	-746	93	-695	155	-110
31	789	94	807	156	-49
32	-418	95	-452	158	-7
35	77	96	-259	166	-71
37	-373	98	-465	168	-11
38	391	99	550	169	67
39	-53	100	-265	171	65
40	-115	102	-12	172	37
42	-571	103	111	174	-132
43	631	104	84	175	-35
44	-203	105	53	177	-20
45	-122	107	124	180	56
47	-381	108	74	182	76
48	331	109	63	183	81
51	96	110	-37	184	-90
54	-447	111	37	187	80
55	405	114	126	189	69
57	-200	115	53	190	61
59	-719	117	-69	193	33
60	782	118	-8		
61	-275	120	8		

Table 4.10: Optimal voltages corresponding to optimal location (see Figure 4.10 a)) for thermal load type T4.

thermal load t	ř 1	T	1	ı	T
Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-828	62	-173	121	-83
4	841	64	-390	124	68
5	-474	65	457	128	-94
6	-439	66	-137	129	-89
8	-1136	69	-73	131	-62
9	1164	71	-784	132	-44
10	-647	72	814	134	91
11	-474	73	-399	135	-89
13	-833	74	-421	136	-117
14	865	76	-1060	137	-81
15	-473	77	1088	139	-88
16	-86	78	-540	140	-57
20	-665	79	-384	147	-114
21	690	81	-763	148	-62
22	-359	82	766	149	-23
23	-336	83	-356	150	-62
25	-920	88	-827	151	-53
26	1009	89	839	152	53
27	-594	90	-475	153	88
28	-331	91	-405	154	-82
30	-683	93	-1097	155	-110
31	736	94	1176	156	-60
32	-383	95	-707	158	-104
35	101	96	-431	166	-110
37	-260	98	-769	168	-31
38	300	99	833	169	-43
39	-15	100	-437	171	19
40	-42	102	-82	172	60
42	-388	103	17	174	-102
43	456	104	57	175	-28
44	-106	105	52	177	-60
45	-41	107	34	180	42
47	-244	108	81	182	81
48	258	109	79	183	73
51	141	110	-95	184	-87
54	-374	111	-83	187	49
55	383	114	53	189	75
57	-171	115	74	190	57
59	-579	117	-123	193	37
60	654	118	-68	-22	
61	-229	120	-66		
O I		120	00		1

Chapter 5. Extensive Experiments on Genetic Algorithms for the Optimization of Piezoelectric Actuator Locations through Parallel Computation

5.1 Abstract

For investigating the shape control of smart structures, we have developed a series of genetic algorithms (GAs) for an optimal placement of the piezoelectric actuators --- an important issue in the design of smart structures. In this paper, we report results from extensive numerical experiments on the GAs. These numerical experiments were conducted using a course-grain parallel computing not only to answer some concerns expressed by a reviewer of one of our previous publications in the AIAA Journal about some of our findings based on limited runs, but also to determine the extent to which the solution quality is enhanced by incorporating our several ideas into a modified micro-genetic algorithm. We will address the following topics through a thorough comparison of solution quality as obtained by our latest micro-GAs version 3, running on a high performance Sun machine with 17 processors.

- 1. What is the effect of a different random number seed, specifically a comparison between the results obtained using a small seed and a large seed.
- 2. What is the effect of restart criteria in micro-GAs, specifically the two parameters --number of generations used for the inner loop and the level of diversity in the
 population on the solution quality.
- 3. What is the effect on the solution quality as the number of actuators changes from 30 and 121 to 15, 60, and 90.

Using these extended studies, we not only got some better layouts of piezoelectric actuators

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This chapter contains the materials submitted to AIAA/ASME/ASCE/AHS/AHC 46th SDM Conference (in 1st AIAA MDO Specialist Conference). (already accepted, Paper Number: AIAA-2005-1899)

than the ones reported in our previous publications, but also achieved a better understanding of the performance of the genetic algorithms, especially, the micro-genetic algorithms that were already shown to be more efficient than the commonly used genetic algorithms.

Specifically, we look at how they work, what their nature is -- random in nature, what robustness means, how to choose appropriate genetic algorithms, the way to get better results if it becomes necessary, advantages of micro-GAs, how to tune micro-GAs, and the effect of different restart criteria in micro-GAs on the solution quality. Therefore, we can provide an efficient, reliable and robust optimization tool for the challenging problem of choosing optimal location for the large number of actuators or sensors in the design of next generation of smart structures. Our research will be useful for those involved with the application of genetic algorithms to other areas, such as damage identification in structural health monitoring, medical imaging and wing design.

The high performance Sun machine with 17 processors is chosen to run multiple jobs at the same time in order to shorten the investigation time.

5.2 Introduction

During the last two decades, research and development of smart structures has received great attention from both university and industry researchers. ¹⁻⁷ An example is the design of next generation of astronomical telescopes. One of the most stringent requirements for these telescopes is the maintenance of high surface accuracy of the primary mirror during their operation. A promising method is to use a certain number of piezoelectric actuators bonded onto the rear surface of the primary mirror to correct its distortions without imposing a significant weight penalty. The critical issue is how to arrange these actuators to maximize their benefit. If the number of actuators is small, we could easily solve the problem by using

the exhaustively enumerative search method, that is, by checking every possible solution in the search space. But this is not the case. Usually, a large number of actuators are needed for the structure to operate properly, so there are an extremely large number of possible schedules for placing these actuators. In our previous studies, 9-10 we developed a methodology of combining the finite element method and genetic algorithms to solve this challenging problem. We successfully developed a series of GAs, including regular GA and micro-GAs, termed Version 1, 2, and 3, from Carroll's FORTRAN Genetic Algorithm Driver. 11 With limited runs, we showed some very promising results produced by the latest version of our micro-GA codes – Version 3. As pointed out to us by a reviewer, because genetic algorithms are random in nature, the results may be a matter of chance, and thus must be verified by performing a large number of runs. In this paper, we will study these GAs by performing such an extensive numerical verification. With this study using a large number of runs, we aim to get a better understanding of genetic algorithms, which have emerged as a leading global search method for actuator placement. Our research shows that the most distinct natures of genetic algorithms are random and robust. With a few of runs with the number of 15000 evaluations GA version 3 can found a nearly optimal solution from the order of $10^{21} \sim 10^{56}$ possible solutions in all of our test cases. However, to get the highest quality of solution a large number of runs using different seeds are necessary. Overall, the software GA version 3 is reliable and effective in solving the optimal placement of a large number of piezoelectric actuators in the design of next generation of smart structures.

5.3 Genetic Algorithms

Genetic algorithms (GAs), inspired by biology and population genetics, have been definitely accepted as an efficient optimization method to solve a wide variety of problems as

shown by the ever-growing number of scientific papers and conferences dedicated to this subject. Even though there are various GAs proposed, they can be simply classified into two categories: regular GAs and micro-GAs. In their structure the micro-GAs are the same as the regular GAs except that they include an automatically restart function when the population converges to a solution according to some convergence criterion. The significant difference between regular GAs and micro-GAs is that the micro-GAs usually use a much smaller population size in order to achieve a faster convergence rate than the regular GAs that use a large population size. In the GA literature, regular GAs have received more thorough studies in respect to general choice of various parameters than micro-GAs. However, our experiences show that micro-GAs are much more efficient than regular GAs. Moreover, micro-GAs also benefit from the mutation operation. In this paper, we will study these GAs by extensive numerical experiments and further demonstrate that the GA version 3 is an efficient, reliable and robust optimization tool for the challenging problem of choosing optimal location for a large number of actuators or sensors in the design of next generation of smart structures. The results by running our latest micro-GA version 3 on the high performance Sun machine with 17 processors are reported in this paper.

The flow chart for the micro-GAs is shown in Figure 5.1.

The main differences among the regular GAs, micro-GAs version 1, micro-GAs version 2 and micro-GAs version 3 are shown in Table 5.1.

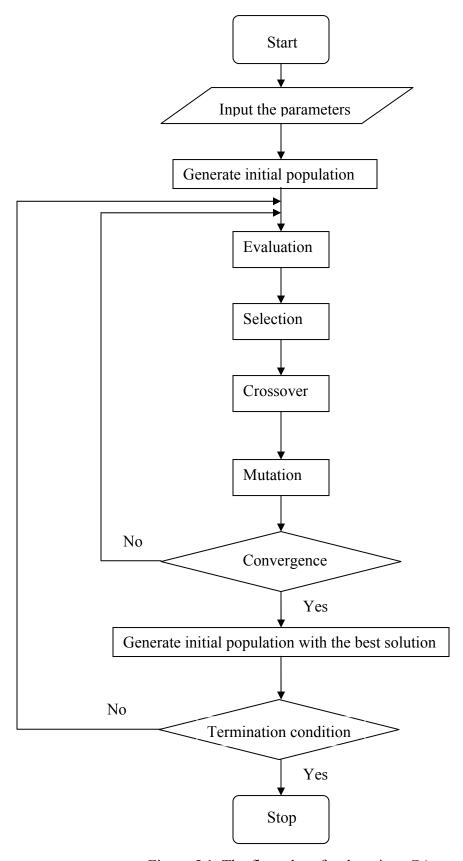


Figure 5.1 The flow chart for the micro-GAs

Table 5.1 The main differences among the regular GAs, micro-GAs version 1, micro-GAs version 2 and micro-GAs version 3

	Regular GAs	Micro_GAs	Micro-GAs	Micro-GAs
		version 1	version 2	version 3
Restart function	No	Yes	Yes	Yes
crossover	Yes	Yes	Yes	Yes
mutation	Yes	No	No	Yes
Individual	N/A	Randomly-	The worst	The worst
replaced by the		selected	individual	individual
best solution		individual		
Size of	Large, fixed	Small, fixed	Small, variable	Small, variable
population				
Hill-climbing	N/A	No	No	Yes

5.4 Problem Definition

The problem in this study is as follows:

With N piezoelectric actuators, the problem is to find an optimal placement from a possible 193 candidate locations to get the best correction to the surface thermal distortions under combining four different types of thermal loads in a thin hexagonal spherical mirror segment to be used in the next generation of astronomical telescopes (Figure 5.2 a)). This is, the problem is to find only one set of locations and corresponding voltages that provide us the best correction for all the four different thermal distortions. Note that only the locations are same for all the four types of loads, not their voltages. This is a challenging multi-criterion optimization problem. Total number of different candidate sets can be computed by the following formula:

$$^{193}C_n = \binom{193}{n} = \frac{193!}{n!(193-n)!}$$

In order to test the stability, reliability and to better understand the performance of our GA version 3, a set of the number of actuators will be tested. We will test the following cases, respectively,

$$^{193}C_{30} = 1.28866 * 10^{35}$$

Case 2: N=121,

$$^{193}C_{121} = 1.38231*10^{54}$$

Case 3: N=15,

$$^{193}C_{15} = 8.4 * 10^{21}$$

Case 4: N=60,

$$^{193}C_{60} = 5.53626 * 10^{50}$$

Case 5: N=90,

$$^{193}C_{90} = 4.65676 * 10^{56}$$

It is clearly seen that it is impossible to use the enumeration method to find the solution from so huge a number of candidate sets even with the most advanced computer in the world.

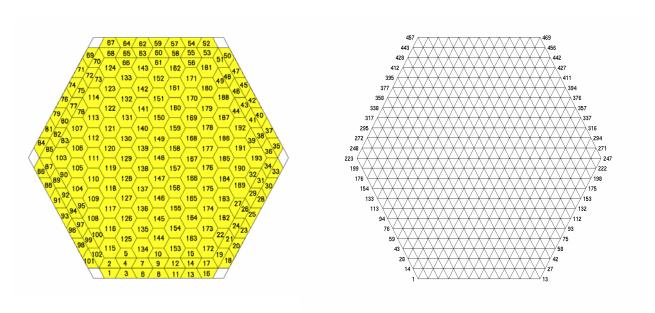


Figure 5.2 a) Piezoelectric actuator candidate locations

b) Finite element mesh

The geometry and material properties of the mirror and piezoelectric actuators are given in Table 5.2. The temperature distribution at the lower surface of the mirror is assumed to be in the form of linear combination of the first few terms of the Zernike series expressed in terms of Cartesian coordinates x and y with the origin at the center of the mirror. The Cartesian coordinates used to express the temperature distributions are normalized such that they are in the range [-1,1]. The temperature distributions that are considered in this study are given in Table 5.3, where the constant C is used to scale the temperature distributions such that the upper, light-reflecting surface is at a lower temperature than the lower surface, with a constant temperature difference $\Delta T_z^{\,o}C$, and the maximum temperature difference between any two grid points across the lower surface of the mirror is $\Delta T_{xy}^{\,o}C$. In this study ΔT_z =0.2°C and ΔT_{xy} =0.5°C. The graphics of these thermal loads are shown in Figure 5.3.

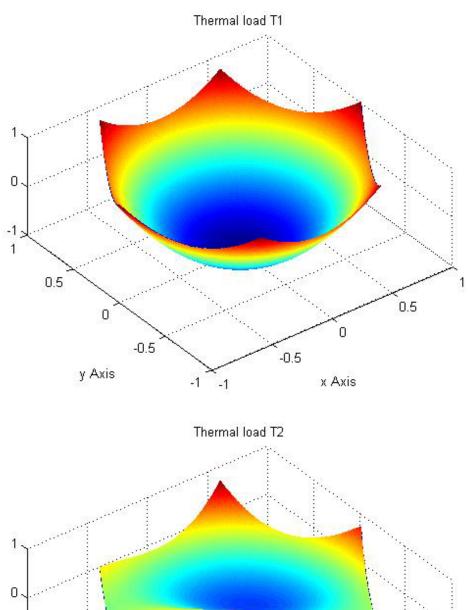
Table 5.2 Properties and geometry of the mirror and piezoelectric actuators

	Mirror	Piezoelectric strips
	(beryllium)	_
Young's modulus, GPa	293	63
Poisson's ratio	0.1	0.3
Coefficient of thermal	11.5E-6	0.9E-6
expansion, /°C		
$d_{31}, d_{32}, m/V$		254E-12
Radius, m	10	
Side of the hexagon, m	0.5	0.04166
Thickness, m	0.012	0.25E-3

Table 5.3 Temperature distributions at the lower surface of the mirror

•	Temperature distribution
T1	$C[2(x^2+y^2)-1]$
T2	$C[(x+y)(3x^2+3y^2-2)]$
T3	$C(\sum_{i=1}^{i=9}K_iZ_i)^*$
T4	C[x+y+2xy]

^{*} K_i , from Table 2 (back surface), Z_i , terms of Zernike series (p. 216) of the reference [Pearson, et al. 1987].



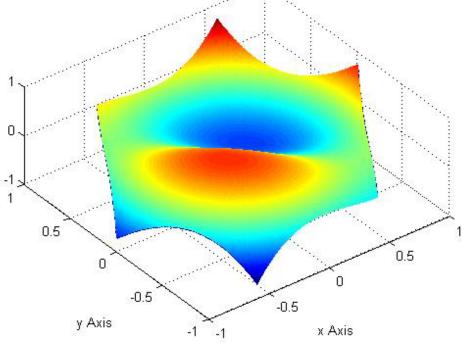
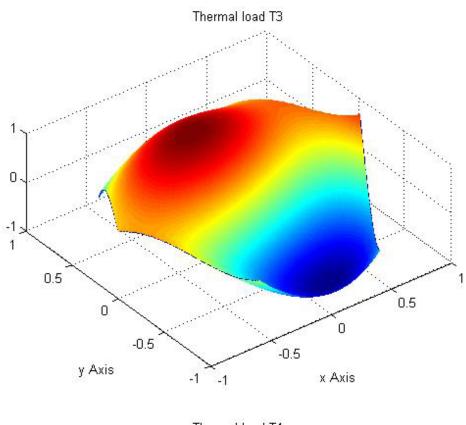


Figure 5.3 a) The thermal loads T1 and T2



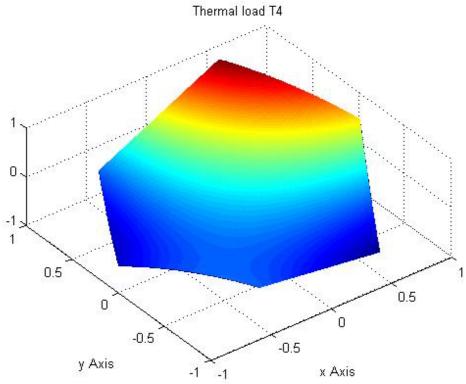


Figure 5.3 b) The thermal loads T3 and T4

5.5 Finite Element Modeling

A laminated triangular shell element developed in Chapter 2 is used to model the mirror. The element is a combination of the DKT plate bending element and a membrane element derived from the linear strain triangular element with a total of 18 degrees of freedom (3 translations and 3 rotations per node). The piezoelectric strips are assumed to be perfectly bonded to the lower surface of the mirror and are modeled as a separate layer. The finite element model consists of 864 flat shell elements, 469 grid points (Figure 5.2 b)). The mirror segment is assumed to be simple-supported at the six vertices 1, 13, 223, 247, 457, and 469.

5.6 Control Algorithms

The surface thermal distortions or the transverse displacements w of the mirror segment are corrected by applying the voltage across the thickness of the strip, which induces a distributed strain in the strip and hence in the mirror. In this study, the thermal deformation w due to a given type of thermal loads is computed by the finite element analysis. The finite element formulation suggested in Ref. 12 is capable of analyzing panels under thermal loads. The deformations considered are so small (of order of a few micrometers) that the correction u_i at any point can be assumed

$$u_i = \sum_{i=1}^n \alpha_{ij} V_j$$

where the control input V_j is the voltage applied across the *j*th piezoelectric strip and the influence coefficient α_{ij} is defined as the deformation caused at node *i* due to a unit voltage applied across the *j*th piezoelectric strip alone.

A matrix of influence coefficients of size $m \times n$ (where m represents the total number of grid points in the FEM model, n represents the given number of piezoelectric actuators) is obtained from the finite element model by applying a unit voltage across each of the

piezoelectric strips, one at a time. A measure of the overall deviation or the RMS error is given by

$$E = E(T, L, V) = \sqrt{\frac{1}{m} \sum_{i=1}^{m} (w_i + u_i)^2} = \sqrt{\frac{1}{m} \sum_{i=1}^{m} (w_i + \sum_{j=1}^{n} \alpha_{ij} V_j)^2}$$

To obtain the best correction, setting $\partial E/\partial V_k = 0$ gives

$$\sum_{i=1}^{m} (w_i + \sum_{j=1}^{n} \alpha_{ij} V_j) \alpha_{ik} = 0$$

i.e.
$$[A]{V} = {b}$$
, where $A_{kj} = \sum_{i=1}^{m} \alpha_{ij} \alpha_{ik}$ $b_k = -\sum_{i=1}^{m} w_i \alpha_{ik}$

For each set of locations we can get the optimal voltages to minimize RMS error. Different settings have different optimal voltages and corresponding minimum RMS error. For one type of thermal distortion we can find a set of locations and corresponding voltages that minimizes the minimum RMS error i.e. of the form,

$$E = \mathop{Min}_{L} \mathop{Min}_{V} E(T, L, V)$$

The optimization problem is to find a set of locations and corresponding voltages that minimizes the maximum of the minimum RMS error for all the four different distortions i.e. of the form,

$$E = \underset{L}{Min} \underset{T}{Max} \underset{V}{Min} E(T, L, V)$$

5.7 Results and Discussions

In this section, the performance of our latest micro-GA version 3 was thoroughly studied through extensive numerical experiments. A high performance Sun machine with 17 processors was used to do this research through parallel computation in order to shorten the time of investigation. The results corresponding to the cases of 30, 121, 15, 60 and 90 actuators are presented in the following.

The number of evaluations is limited to 15000 for all the runs.

Case 1: N=30 actuators

Micro-GA Version 3: Initial population size 10, population size 5, scale=0.5, random=0, crossover rate =0.5, mutation rate=0.01, restart control parameter different level=0.0 No_of_best_mutation_bits=2.

Table 5.4 RMS errors corresponding to optimal placement (30 actuators)

Generations	RMS for		RMS for 5000		RMS for 10000		RMS for 15000	
of inner	evaluatio	n	evaluatio	ns	evaluatio	ns	evaluatio	ns
loop	Seed #1	Seed #2	Seed #1	Seed #2	Seed #1	Seed #2	Seed #1	Seed #2
8			0.25069	0.24362	0.24022	0.23622	0.24022	0.23455
9			0.23541	0.25342	0.23130	0.25026	0.22943	0.24451
10			0.24551	0.24459	0.24102	0.23923	0.23093	0.23399
11			0.23865	0.25410	0.23581	0.25116	0.23186	0.25083
12			0.23831	0.25208	0.23709	0.25020	0.23709	0.24594
13			0.24040	0.24958	0.23924	0.24892	0.23538	0.24764
14	0.87205	0.61079	0.25305	0.24685	0.23455	0.24328	0.23304	0.24243
15			0.24089	0.25215	0.23744	0.24655	0.23604	0.24417
16			0.26163	0.25954	0.25789	0.24619	0.24962	0.24241
17			0.24749	0.25167	0.24473	0.24938	0.24399	0.24938
18			0.25494	0.25040	0.24657	0.24971	0.24657	0.24971
19			0.24287	0.24162	0.24175	0.24162	0.24175	0.23277
20			0.25131	0.25439	0.23615	0.25130	0.22882	0.24636

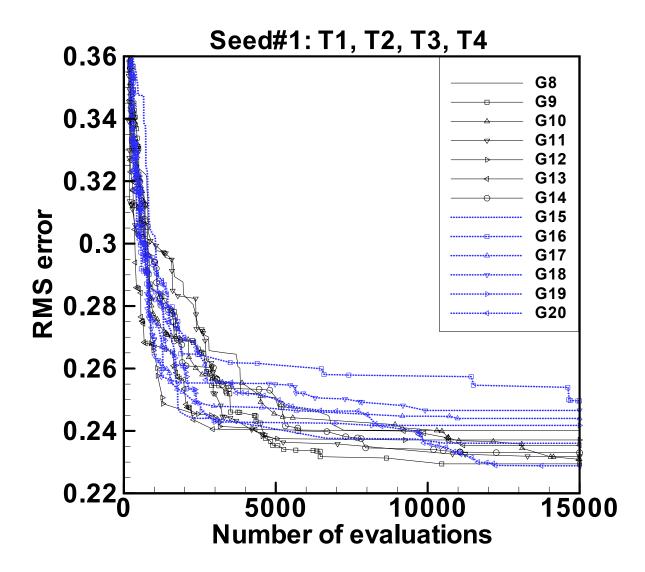


Figure 5.4 a) Performance of the GAs: Maximum of the four minimum RMS errors vs. the number of evaluations (30 actuators, seed#1).

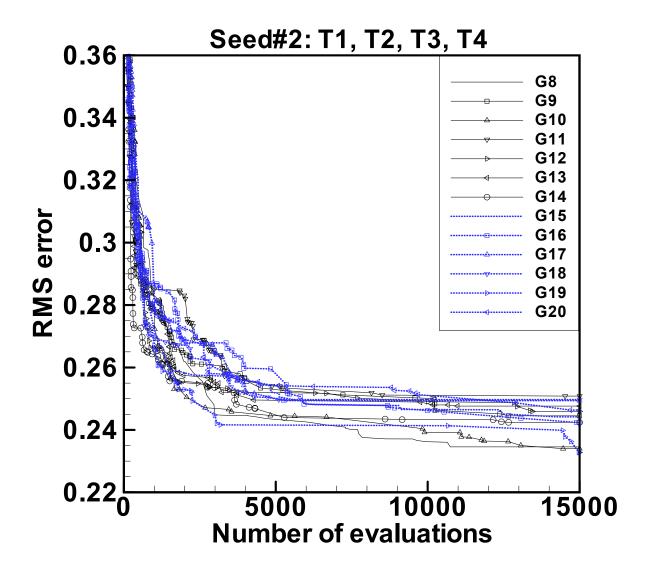


Figure 5.4 b) Performance of the GAs: Maximum of the four minimum RMS errors vs. the number of evaluations (30 actuators, seed#2).

For this case (nstrip=30)

Observation 1: Seed #1 is better than Seed #2 for all the three numbers of evaluations used for termination.

Observation 2: The best number of generations of inner loop is dependent on the number of evaluations used for termination. For example, if 5000 evaluations is chosen as termination condition, the best number of generations of inner loop is 9, but if 15000 evaluations is chosen as termination condition, the best number of generations of inner loop is 20.

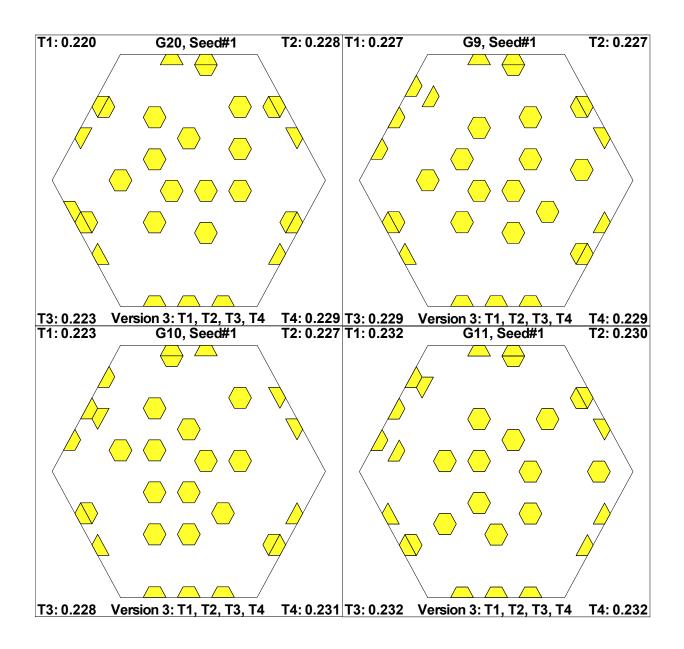


Figure 5.5 RMS error and Optimal location of piezoelectric actuators obtained by GA Version 3 (30 actuators)

Note: The number in each corner in the Figure represents the RMS error corresponding to the specified type of the thermal load.

Table 5.5 Optimal voltages corresponding to optimal location (see Figure 5.5 upper left) for thermal load type T1 (30 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-157	58	427	127	-394
8	-452	62	-385	130	-386
13	-133	74	-1039	132	-152
23	-503	75	541	138	-445
28	-900	79	-535	150	-570
29	-359	88	-53	155	-402
40	-532	91	-994	157	-423
45	-1229	92	540	176	-358
46	-834	96	-497	178	-356
57	-843	111	-247	180	-121

Table 5.6 Optimal voltages corresponding to optimal location (see Figure 5.5 upper left) for thermal load type T2 (30 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-328	58	459	127	-9
8	-758	62	-624	130	-142
13	-358	74	-1375	132	-98
23	-646	75	733	138	-194
28	-1293	79	-637	150	-362
29	523	88	-117	155	-86
40	-750	91	-1483	157	-267
45	-1403	92	967	176	-230
46	823	96	-590	178	-261
57	-1089	111	-14	180	-160

Table 5.7 Optimal voltages corresponding to optimal location (see Figure 5.5 upper left) for thermal load type T3 (30 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-610	58	1305	127	-136
8	-1071	62	-375	130	72
13	-689	74	-1366	132	275
23	-993	75	1168	138	-41
28	-1542	79	-190	150	178
29	747	88	-4	155	-353
40	-217	91	-1353	157	-58
45	-1004	92	963	176	-106
46	676	96	-485	178	130
57	-1592	111	194	180	168

Table 5.8 Optimal voltages corresponding to optimal location (see Figure 5.5 upper left) for thermal load type T4 (30 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-615	58	882	127	-309
8	-1051	62	-338	130	-211
13	-635	74	-1700	132	-64
23	-789	75	741	138	-220
28	-1602	79	-958	150	-164
29	926	88	-311	155	-298
40	-20	91	-1295	157	-240
45	-796	92	584	176	-98
46	713	96	-1041	178	-23
57	-1132	111	-195	180	108

Case 2: N=121 actuators

Micro-GA Version 3: Initial population size 10, population size 5, scale=0.5, random=0, crossover rate =0.5, mutation rate=0.01, restart control parameter different level=0.0 No_of_best_mutation_bits=2.

Table 5.9 RMS errors corresponding to optimal placement (121 actuators)

Generations	RMS for	initial	RMS for 5000		RMS for 10000		RMS for 15000	
of inner	evaluatio	n	evaluatio	ns	evaluations		evaluations	
loop	Seed #1	Seed #2	Seed #1	Seed #2	Seed #1	Seed #2	Seed #1	Seed #2
8			0.07880	0.08017	0.07825	0.07891	0.07802	0.07864
9			0.07930	0.07933	0.07845	0.07874	0.07817	0.07831
10			0.07982	0.07894	0.07942	0.07822	0.07931	0.07810
11			0.07927	0.07951	0.07862	0.07843	0.07824	0.07813
12			0.07866	0.07872	0.07828	0.07812	0.07805	0.07797
13			0.07881	0.07850	0.07802	0.07817	0.07788	0.07809
14	0.20413	0.18827	0.07897	0.07918	0.07854	0.07868	0.07851	0.07853
15			0.07970	0.07902	0.07860	0.07826	0.07831	0.07815
16			0.07903	0.07879	0.07871	0.07855	0.07857	0.07822
17			0.07893	0.07860	0.07840	0.07822	0.07810	0.07809
18			0.07898	0.07892	0.07858	0.07870	0.07832	0.07857
19			0.07861	0.07839	0.07852	0.07831	0.07790	0.07813
20			0.07908	0.07833	0.07888	0.07799	0.07877	0.07795

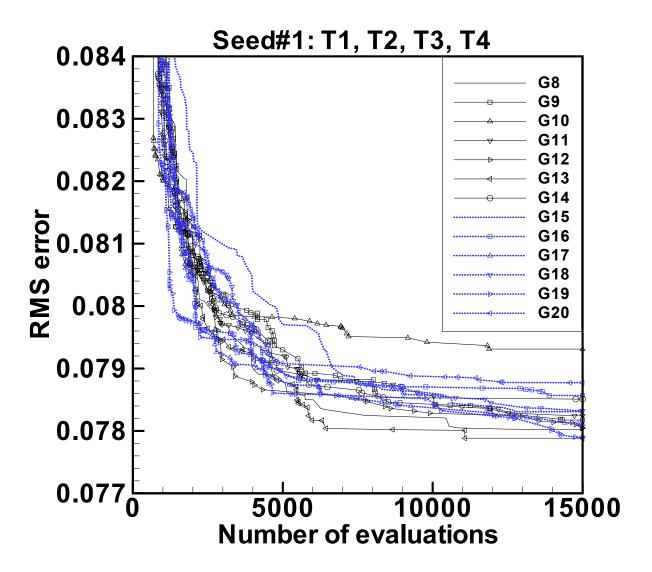


Figure 5.6 a) Performance of the GAs: Maximum of the four minimum RMS errors vs. the number of evaluations (121 actuators, seed#1).

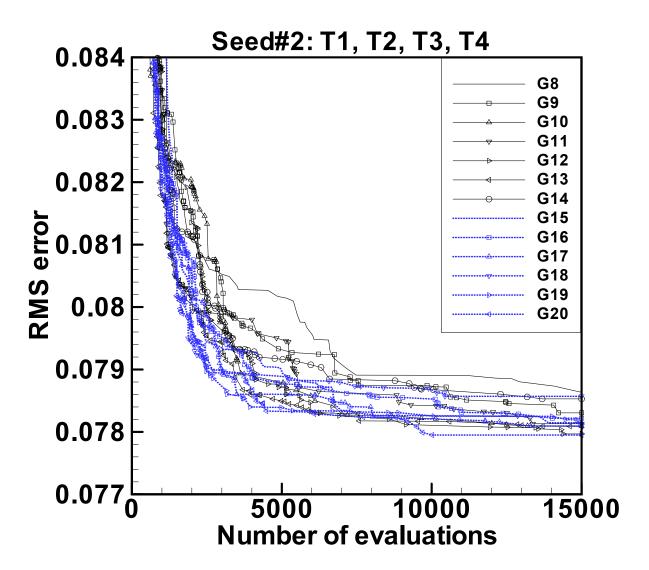


Figure 5.6 b) Performance of the GAs: Maximum of the four minimum RMS errors vs. the number of evaluations (121 actuators, seed#2).

For this case (nstrip=121)

Observation 1: Which seed is better depends on the number of evaluations used for termination. For example, if 5000 or 10000 evaluations are chosen as termination condition,

Seed #2 is better than Seed #1, but if 15000 evaluations are chosen as the termination condition, the conclusion is reverse. However, the differences between the results obtained from two seeds are not substantially different.

Observation 2: The best number of generations of inner loop is dependent on the number of evaluations used for termination. For example, If 5000 or 10000 evaluations are performed for termination, the best number of generations of inner loop is 20, but If 15000 evaluations are performed as the termination condition, the best number of generations of inner loop is 13.

The computation time in the case of 121 actuators is about one week per job running on one processor of the Sun machine. The advantage of using the Sun machine is that we can submit multiple jobs running on different processors simultaneously, so we can study the performance of GAs with different parameter settings at the same time and significantly reduce the time of investigation.

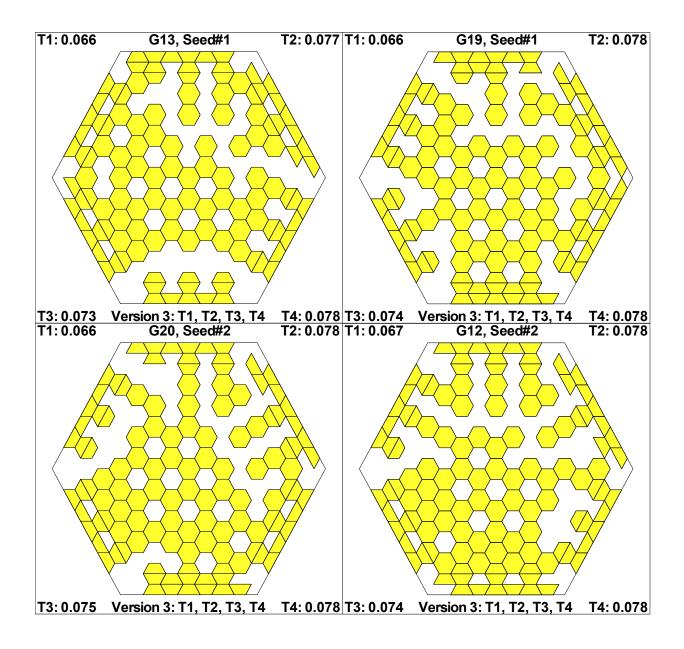


Figure 5.7 RMS error and Optimal location of piezoelectric actuators obtained by GA Version 3 (121 actuators)

Note: The number in each corner in the Figure represents the RMS error corresponding to the specified type of the thermal load.

Table 5.10 Optimal voltages corresponding to optimal location (see Figure 5.7 upper left) for thermal load type T1 (121 actuators)

	type 11 (121 ac		T		1
Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-373	62	-165	120	-104
4	485	64	-419	121	-102
5	-126	65	483	122	-31
6	-218	66	-186	124	60
8	-537	67	48	126	-96
9	638	68	-13	128	-135
10	-261	71	-460	129	-172
11	-217	72	541	131	-74
13	-373	73	-253	132	-105
14	484	74	-239	136	-99
15	-127	76	-662	137	-163
20	-479	77	742	139	-133
21	581	78	-379	140	-238
22	-212	79	-240	145	-119
23	-247	81	-480	147	-155
25	-675	82	562	148	-218
26	766	83	-282	151	-219
27	-369	87	7	152	37
28	-239	88	-482	155	-96
30	-485	89	548	156	-170
31	595	90	-269	158	-102
32	-271	91	-246	159	-279
35	41	93	-666	164	-100
37	-472	94	775	166	-108
38	364	95	-428	167	-233
40	-203	96	-255	170	-155
42	-699	98	-475	171	33
43	777	99	583	173	47
44	-353	100	-228	174	-71
45	-219	103	60	175	-129
47	-486	104	52	178	-218
48	533	107	45	181	48
49	-220	108	23	183	56
51	28	109	49	184	-57
54	-394	110	-89	185	-160
55	510	111	-70	187	39
56	-233	112	-42	188	8
57	-180	114	44	190	53
59	-581	116	45	192	-36
60	667	117	-82	1/2	20
61	-298	118	-106		
01	270	110	100		L

Table 5.11 Optimal voltages corresponding to optimal location (see Figure 5.7 upper left) for thermal load type T2 (121 actuators)

	ype 12 (121 ac		Τ		r
Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-699	62	-252	120	-45
4	661	64	-576	121	-38
5	-152	65	623	122	-19
6	-334	66	-243	124	71
8	-951	67	63	126	-10
9	900	68	-79	128	-38
10	-276	71	-583	129	-64
11	-332	72	638	131	-41
13	-690	73	-320	132	-68
14	668	74	-308	136	-35
15	-184	76	-807	137	-53
20	-649	77	863	139	-56
21	705	78	-437	140	-133
22	-291	79	-304	145	-16
23	-306	81	-579	147	-63
25	-912	82	632	148	-107
26	962	83	-313	151	-165
27	-456	87	-43	152	39
28	-306	88	-612	155	-45
30	-667	89	671	156	-77
31	724	90	-309	158	-52
32	-341	91	-332	159	-186
35	7	93	-863	164	-42
37	-559	94	961	166	-49
38	393	95	-525	167	-146
40	-276	96	-327	170	-143
42	-863	98	-630	171	29
43	925	99	707	173	65
44	-430	100	-315	174	-52
45	-299	103	97	175	-86
47	-609	104	85	178	-176
48	653	107	85	181	44
49	-288	108	58	183	76
51	14	109	72	184	-42
54	-547	110	-43	185	-126
55	624	111	4	187	38
56	-299	112	-21	188	6
57	-258	114	72	190	51
59	-781	116	70	192	-39
60	836	117	-52	1/2	3,
61	-380	118	-23		
U1	200	110	43		<u> </u>

Table 5.12 Optimal voltages corresponding to optimal location (see Figure 5.7 upper left) for thermal load type T3 (121 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-877	62	-185	120	-3
4	826	64	-525	121	62
5	-331	65	494	122	13
6	-410	66	-46	124	94
8	-1172	67	32	124	-68
9	1157	68	-31	128	-21
10	-546	71	-378	129	15
11	-407	72	471	131	0
13	-875	73	-184	132	106
14	841	74	-184	136	-48
15	-405	76	-521	130	-48 -48
		77			-46 -5
20 21	-751 757	78	605 -253	139 140	64
22	-374	79	-198	145	-109
23	-344	81	-333	147	-43
25	-1032	82	445	148	-6
26	-1064	83	-184	151	60
27	-525	87	-2	152	110
28	-340	88	-495	155	-61
30	-750	89	551	156	-68
31	776	90	-236	158	-18
32	-336	91	-252	159	52
35	100	93	-692	164	-109
37	-354	94	796	166	-42
38	325	95	-421	167	-26
40	-138	96	-259	170	56
42	-536	98	-472	171	80
43	642	99	572	173	39
44	-303	100	-251	174	-65
45	-166	103	118	175	-72
47	-372	104	87	178	12
48	419	107	110	181	55
49	-128	108	45	183	76
51	61	109	50	184	-48
54	-494	110	-15	185	-32
55	572	111	50	187	52
56	-192	112	17	188	84
57	-189	114	115	190	74
59	-726	116	40	192	39
60	761	117	-44		
61	-238	118	-19		

Table 5.13 Optimal voltages corresponding to optimal location (see Figure 5.7 upper left) for thermal load type T4 (121 actuators)

	ype 14 (121 ac		Τ		r
Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-841	62	-150	120	-70
4	806	64	-447	121	-53
5	-328	65	518	122	-33
6	-404	66	-151	124	77
8	-1176	67	195	126	-88
9	1142	68	-180	128	-84
10	-504	71	-779	129	-98
11	-409	72	785	131	-48
13	-900	73	-411	132	-39
14	838	74	-401	136	-71
15	-353	76	-1041	137	-99
20	-659	77	1068	139	-72
21	698	78	-571	140	-103
22	-315	79	-392	145	-99
23	-309	81	-752	147	-86
25	-933	82	778	148	-99
26	981	83	-428	151	-81
27	-463	87	-95	152	66
28	-308	88	-798	155	-64
30	-685	89	853	156	-93
31	727	90	-438	158	-51
32	-302	91	-438	159	-88
35	128	93	-1097	164	-84
37	-259	94	1170	166	-53
38	285	95	-684	167	-84
40	-66	96	-407	170	-16
42	-360	98	-799	171	64
43	477	99	824	173	52
44	-212	100	-427	174	-57
45	-78	103	79	175	-70
47	-223	104	70	178	-45
48	283	107	76	181	48
49	-65	107	49	183	82
51	102	109	69	184	-46
54	-405	110	-88	185	-46
55	505	110	-48	187	43
56	-177	112	-46 -45	188	65
57	-177	112	79	190	82
59	-132 -588	114	54	190	39
				192	39
60	652	117	-90		
61	-237	118	-81		

Case 3: N=15 actuators

Micro-GA Version 3: Initial population size 10, population size 5, scale=0.5, random=0, crossover rate =0.5, mutation rate=0.01, restart control parameter different level=0.0 No_of_best_mutation_bits=2.

Table 5.14 RMS errors corresponding to optimal placement (15 actuators)

Generations	RMS for	initial	RMS for	5000	RMS for 10000		RMS for 15000		
of inner	evaluatio	n	evaluatio	evaluations		evaluations		evaluations	
loop	Seed #1	Seed #2	Seed #1	Seed #2	Seed #1	Seed #2	Seed #1	Seed #2	
8			0.42783	0.41276	0.41357	0.40926	0.40345	0.40905	
9			0.40495	0.39477	0.40010	0.38528	0.39804	0.38188	
10			0.39957	0.47201	0.39087	0.44066	0.38581	0.43244	
11			0.40130	0.43442	0.38671	0.41774	0.38314	0.38577	
12			0.41044	0.42707	0.39972	0.42327	0.38018	0.41599	
13			0.41505	0.44012	0.40725	0.41871	0.40618	0.41636	
14	1.66981	1.13175	0.40621	0.44519	0.39955	0.38707	0.39951	0.38342	
15			0.41498	0.43830	0.41037	0.43830	0.40985	0.42657	
16			0.40793	0.44148	0.40190	0.42044	0.40159	0.41474	
17	1	0.41550	0.41523	0.40134	0.41455	0.39573	0.40850		
18			0.41861	0.43983	0.40661	0.41402	0.40340	0.41295	
19			0.41356	0.41859	0.41084	0.40415	0.40651	0.40415	
20			0.41257	0.41963	0.40811	0.40114	0.39851	0.39657	

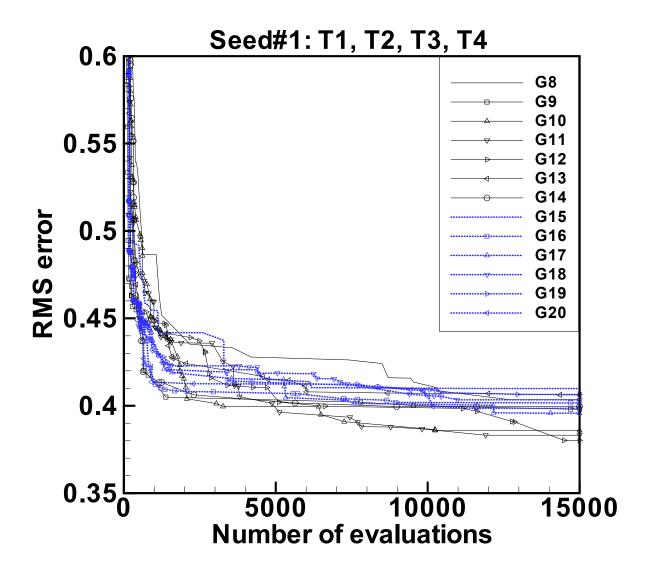


Figure 5.8 a) Performance of the GAs: Maximum of the four minimum RMS errors vs. the number of evaluations (15 actuators, seed#1).

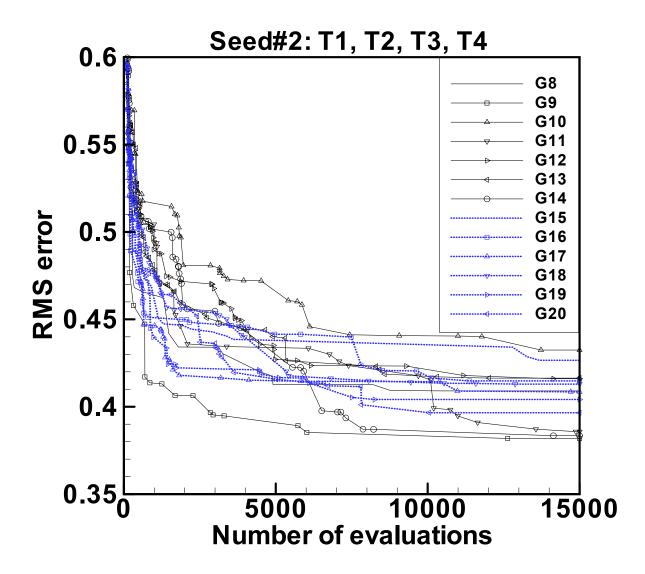


Figure 5.8 b) Performance of the GAs: Maximum of the four minimum RMS errors vs. the number of evaluations (15 actuators, seed#2).

For this case (nstrip=15), following observations are made:

- 1) Which seed is better depends on the number of evaluations used for termination. For example, if 5000 or 10000 evaluations are chosen as termination condition, Seed #2 is better than Seed #1, but if 15000 evaluations are chosen as the termination condition, the conclusion is reverse.
- 2) The best number of generations in the inner loop for each seed is different and maybe dependent on the number of evaluations used for termination. For example, for Seed#1, the best number of generations in the inner loop is 10, 11 and 12 corresponding to the 5000, 10000, and 15000 evaluations as the termination condition, respectively; for Seed#2, the best number of generations in the inner loop is consistently 9 for all the three termination conditions.
- 3) The best number of generations in the inner loop is less than 14 for all the three termination conditions.
- 4) The range of the RMS errors for Seed#2 is obviously larger than that for Seed#1.

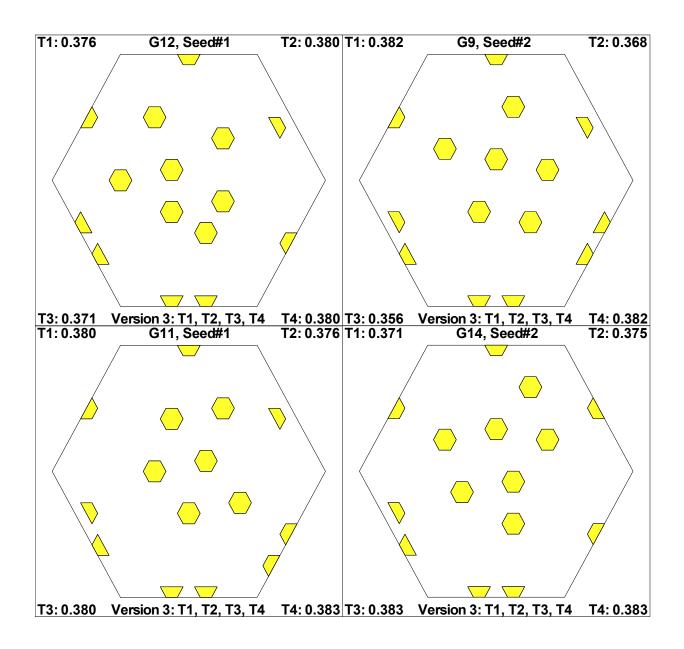


Figure 5.9 RMS error and Optimal location of piezoelectric actuators obtained by GA Version 3 (15 actuators)

Note: The number in each corner in the Figure represents the RMS error corresponding to the specified type of the thermal load.

Table 5.15 Optimal voltages corresponding to optimal location (see Figure 5.9 upper left) for thermal load type T1 (15 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
6	-451	76	-710	137	-589
11	-332	91	-427	139	-825
25	-845	96	-604	155	-256
43	-815	111	-349	166	-735
59	-507	132	-364	169	-672

Table 5.16 Optimal voltages corresponding to optimal location (see Figure 5.9 upper left) for

thermal load type T2 (15 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
6	-644	76	-888	137	-85
11	-697	91	-563	139	-402
25	-1116	96	-536	155	-21
43	-1127	111	6	166	-447
59	-862	132	-237	169	-495

Table 5.17 Optimal voltages corresponding to optimal location (see Figure 5.9 upper left) for

thermal load type T3 (15 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
6	-1049	76	-229	137	-153
11	-1117	91	-362	139	74
25	-1413	96	-540	155	-339
43	-107	111	268	166	-169
59	-323	132	415	169	356

Table 5.18 Optimal voltages corresponding to optimal location (see Figure 5.9 upper left) for

thermal load type T4 (15 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
6	-1116	76	-1396	137	-386
11	-1050	91	-881	139	-376
25	-1151	96	-1099	155	-265
43	208	111	-311	166	-255
59	-255	132	-154	169	-3

Case 4: N=60 actuators

Micro-GA Version 3: Initial population size 10, population size 5, scale=0.5, random=0, crossover rate =0.5, mutation rate=0.01, restart control parameter different level=0.0 No_of_best_mutation_bits=2.

Table 5.19 RMS errors corresponding to optimal placement (60 actuators)

Generations	RMS for	initial	RMS for 5000		RMS for 10000		RMS for 15000		
of inner	evaluatio	n	evaluatio	evaluations		evaluations		evaluations	
loop	Seed #1	Seed #2	Seed #1	Seed #2	Seed #1	Seed #2	Seed #1	Seed #2	
8			0.14527	0.14164	0.13996	0.13690	0.13850	0.13513	
9			0.14180	0.14356	0.13861	0.14016	0.13565	0.13602	
10			0.14266	0.14198	0.13815	0.13931	0.13558	0.13880	
11			0.14130	0.13831	0.13858	0.13606	0.13564	0.13465	
12	0.33310 0.34697	0.14425	0.13835	0.13986	0.13792	0.13769	0.13649		
13		0.14059	0.15004	0.13802	0.13998	0.13697	0.13560		
14		0.14780	0.13319	0.13848	0.13319	0.13797	0.13291		
15			0.14183	0.14884	0.13707	0.14535	0.13540	0.14303	
16			0.14568	0.14158	0.14355	0.14049	0.14311	0.14020	
17			0.14036	0.14150	0.13439	0.13463	0.13358	0.13164	
18		0.14834	0.13944	0.14438	0.13696	0.14244	0.13581		
19			0.14763	0.13736	0.14030	0.13681	0.13810	0.13681	
20			0.14293	0.14318	0.13889	0.13739	0.13441	0.13559	

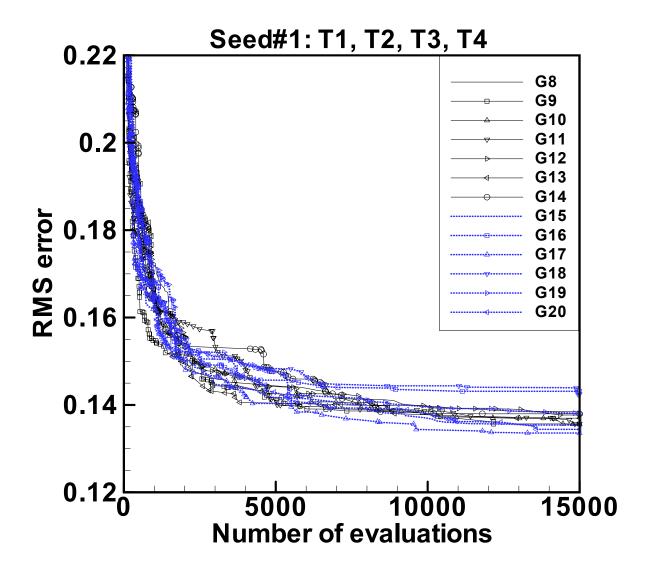


Figure 5.10 a) Performance of the GAs: Maximum of the four minimum RMS errors vs. the number of evaluations (60 actuators, seed#1).

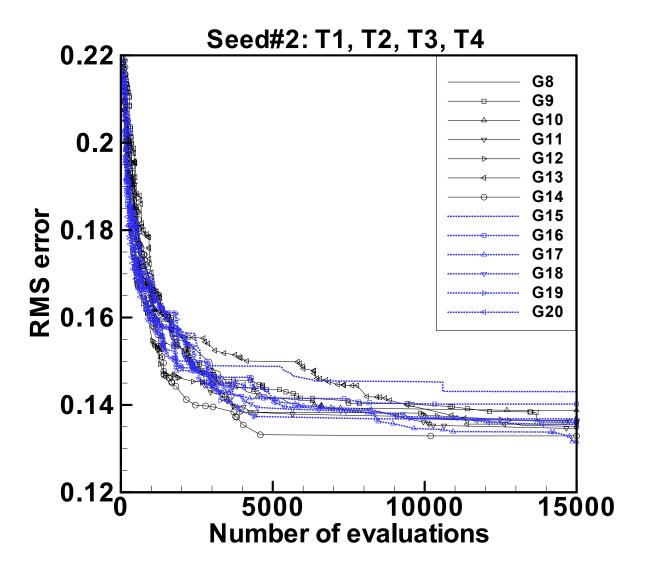


Figure 5.10 b) Performance of the GAs: Maximum of the four minimum RMS errors vs. the number of evaluations (60 actuators, seed#2).

For this case (nstrip=60), following observations are made:

- 1) The best number of generations in the inner loop for Seed #1 is consistently 17 for all the three numbers of evaluations used for termination.
- 2) The best number of generations in the inner loop for seed #2 is 14 for the termination condition of 5000 and 10000 evaluations, and 17 for the termination condition of 15000 evaluations. For Seed#2 and the termination condition of 15000 evaluations, the optimal number 17 for the case of 60 actuators is different from those in the case of 30 and 121 actuators (19 for the case of 30 actuators and 20 for the case of 121 actuators).
- 3) Which seed is better for a specific run is dependant on the number of evaluations. For example, the better seed is Seed #1 for 5000 and 10000 evaluations but Seed #2 for 15000 evaluations for the specific run of generations of inner loop equal to 17.
- 4) All the best results for the three termination conditions of 5000, 10000 and 15000 evaluations are obtained by Seed #2. This is quite different from the case of 30 actuators.

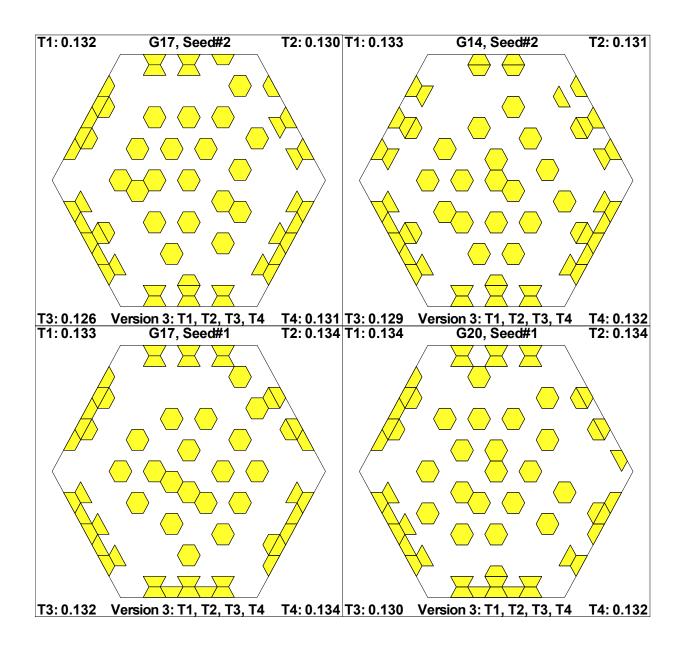


Figure 5.11 RMS error and Optimal location of piezoelectric actuators obtained by GA Version 3 (60 actuators)

Note: The number in each corner in the Figure represents the RMS error corresponding to the specified type of the thermal load.

Table 5.20 Optimal voltages corresponding to optimal location (see Figure 5.11 upper left) for thermal load type T1 (60 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-430	54	-238	99	331
4	340	59	-674	111	-122
8	-760	60	301	119	-142
9	661	64	-446	121	-204
10	-322	65	339	127	-304
13	-426	71	-134	129	-282
14	326	74	-679	132	-153
20	-400	75	469	135	-106
21	366	76	-298	140	-269
23	-282	79	-613	146	-382
25	-467	80	405	148	-479
26	273	81	-175	151	-191
28	-297	88	-426	159	-347
30	-321	89	334	164	-182
31	207	91	-271	166	-319
37	-483	93	-678	170	-167
38	238	94	883	175	-153
42	-868	95	-499	177	-343
43	499	96	-265	181	137
47	-320	98	-398	187	-144

Table 5.21 Optimal voltages corresponding to optimal location (see Figure 5.11 upper left) for thermal load type T2 (60 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-798	54	-383	99	475
4	470	59	-919	111	-8
8	-1302	60	347	119	-30
9	949	64	-617	121	-84
10	-390	65	354	127	-62
13	-774	71	-235	129	-112
14	435	74	-809	132	-101
20	-547	75	576	135	-5
21	405	76	-395	140	-139
23	-342	79	-815	146	-144
25	-694	80	605	148	-223
26	406	81	-230	151	-156
28	-360	88	-608	159	-224
30	-479	89	484	164	-71
31	256	91	-330	166	-152
37	-615	93	-875	170	-157
38	249	94	855	175	-114
42	-1083	95	-251	177	-249
43	551	96	-320	181	120
47	-447	98	-588	187	-148

Table 5.22 Optimal voltages corresponding to optimal location (see Figure 5.11 upper left) for thermal load type T3 (60 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-958	54	-286	99	301
4	440	59	-947	111	109
8	-1607	60	651	119	-30
9	1245	64	-666	121	114
10	-697	65	582	127	-80
13	-906	71	-34	129	7
14	306	74	-873	132	149
20	-588	75	788	135	-129
21	293	76	-181	140	50
23	-391	79	-820	146	-175
25	-798	80	742	148	-40
26	427	81	-10	151	105
28	-396	88	-535	159	41
30	-644	89	560	164	-213
31	443	91	-282	166	-78
37	-371	93	-692	170	117
38	342	94	728	175	-114
42	-671	95	-210	177	-18
43	450	96	-271	181	217
47	-199	98	-391	187	59

Table 5.23 Optimal voltages corresponding to optimal location (see Figure 5.11 upper left) for thermal load type T4 (60 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-901	54	-219	99	334
4	386	59	-734	111	-119
8	-1608	60	466	119	-94
9	1222	64	-447	121	-133
10	-643	65	345	127	-229
13	-957	71	-397	129	-151
14	383	74	-944	132	-72
20	-534	75	634	135	-123
21	340	76	-565	140	-122
23	-354	79	-819	146	-229
25	-739	80	513	148	-230
26	443	81	-397	151	-42
28	-363	88	-735	159	-121
30	-615	89	378	164	-165
31	469	91	-400	166	-152
37	-248	93	-1158	170	18
38	340	94	1297	175	-72
42	-444	95	-709	177	-87
43	383	96	-398	181	200
47	-55	98	-663	187	36

Case 4: N=90 actuators

Micro-GA Version 3: Initial population size 10, population size 5, scale=0.5, random=0, crossover rate =0.5, mutation rate=0.01, restart control parameter different level=0.0 No_of_best_mutation_bits=2.

Table 5.24 RMS errors corresponding to optimal placement (90 actuators)

Generations	RMS for	initial	RMS for 5000		RMS for 10000		RMS for 15000		
of inner	evaluatio	ns	evaluatio	evaluations		evaluations		evaluations	
loop	Seed #1	Seed #2	Seed #1	Seed #2	Seed #1	Seed #2	Seed #1	Seed #2	
8			0.10069	0.09227	0.09348	0.08907	0.09058	0.08658	
9			0.09104	0.09300	0.08775	0.08818	0.08625	0.08724	
10			0.09118	0.09243	0.08657	0.08772	0.08564	0.08632	
11			0.09263	0.08981	0.08831	0.08839	0.08733	0.08705	
12	0.23206 0.24537	0.09141	0.08856	0.09018	0.08713	0.08970	0.08654		
13		0.09096	0.09069	0.08737	0.08661	0.08731	0.08658		
14		0.09143	0.09052	0.08766	0.08896	0.08667	0.08748		
15			0.09060	0.08969	0.08766	0.08793	0.08714	0.08687	
16			0.09003	0.08824	0.08701	0.08652	0.08570	0.08612	
17		0.08990	0.09110	0.08689	0.08858	0.08644	0.08701		
18		0.08924	0.09280	0.08763	0.08403	0.08666	0.08390		
19			0.08950	0.09213	0.08679	0.08757	0.08581	0.08702	
20			0.08941	0.09083	0.08674	0.08666	0.08659	0.08539	

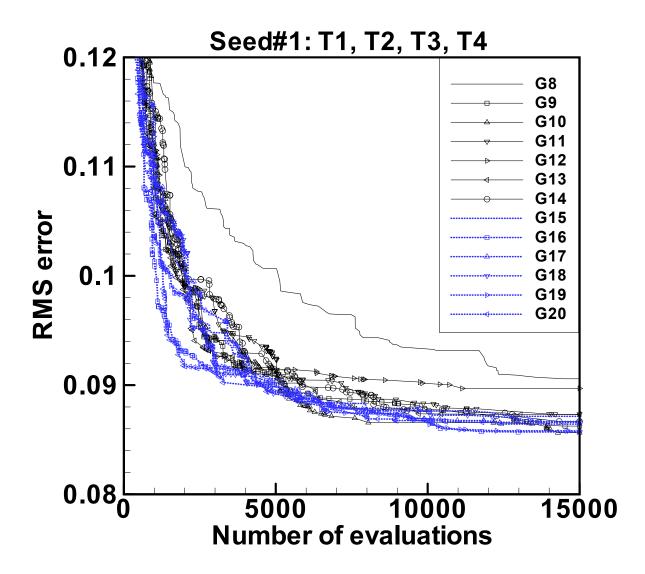


Figure 5.12 a) Performance of the GAs: Maximum of the four minimum RMS errors vs. the number of evaluations (90 actuators, seed#1).

Note: It is the optimization problem in which one set of actuators is chosen for all four thermal distortions.

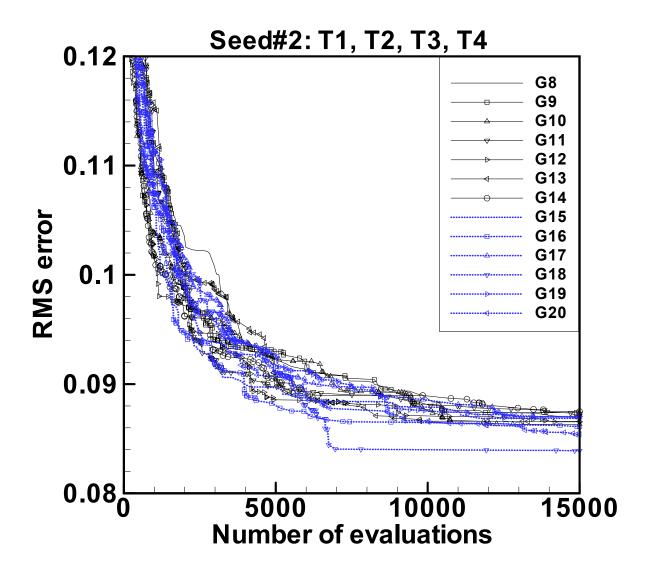


Figure 5.12 b) Performance of the GAs: Maximum of the four minimum RMS errors vs. the number of evaluations (90 actuators, seed#2).

Note: It is the optimization problem in which one set of actuators is chosen for all four thermal distortions.

For this case (nstrip=90), following observations are made:

- 1) When the number of generations in the inner loop is 18, Seed #2 in case of 10000 and 15000 evaluations shows much more superior performance than Seed#1.
- 2) The best number of generations in the inner loop is dependent on not only the number of evaluations used for termination, but also which seed generator was used. For example, in case of Seed #1, the best number of generations in the inner loop is 18, 10 and 10 corresponding to the 5000, 10000, and 15000 evaluations as the termination condition, respectively; In case of Seed #2, the best number of generations in the inner loop is 16, 18 and 18 corresponding to the 5000, 10000, and 15000 evaluations as the termination condition, respectively.
- 3) For the number of generations in the inner loop, the optimal number 18 in case of seed #2 corresponding to the 10000 and 15000 evaluations is also different from those in the case of 15, 30, 60 and 121 actuators.
- 4) The best solution for the case of 90 actuators is very close to that for the case of 121 actuators in terms of RMS errors.

Overall, the RMS value of the error for the minimum distortion is around 0.08 for 90 actuators compared to 0.07 for 121 actuators, 0.23 for 30 actuators; 0.38 for 15 actuators, 0.14 for 60 actuators; and the RMS value of the error reduces with the number of actuators used. The best number of actuators appears to be 90.

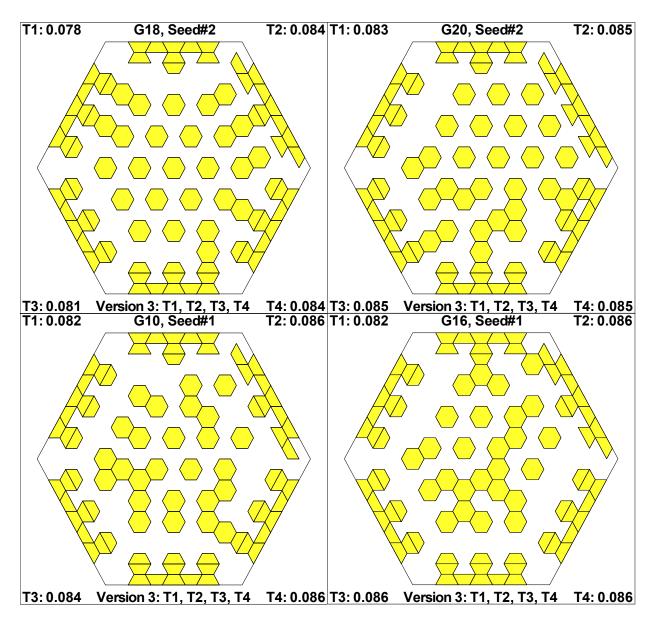


Figure 5.13 RMS error and Optimal location of piezoelectric actuators obtained by GA Version 3 (90 actuators)

Note: The number in each corner in the Figure represents the RMS error corresponding to the specified type of the thermal load.

Table 5.25 Optimal voltages corresponding to optimal location (see Figure 5.13 upper left) for thermal load type T1 (90 actuators)

Strip Loc.	Voltage V	Strip Loc. Voltage		Strip Loc.	Voltage V
3	-370	47	-453	96	-259
4	451	48	333	98	-465
5	-83	51	63	99	521
6	-221	54	-338	100	-122
8	-532	55	312	111	-148
9	616	57	-203	113	25
10	-234	59	-546	118	-198
11	-222	60	644	121	-209
13	-376	61	-260	123	34
14	509	62	-217	126	-134
15	-175	64	-338	129	-313
20	-467	65	344	132	-156
21	534	71	-477	137	-312
22	-137	72	588	140	-306
23	-254	73	-223	145	-188
25	-662	74	-257	148	-363
26	755	76	-663	151	-186
27	-354	77	758	156	-307
28	-249	78	-354	159	-310
30	-470	79	-258	163	39
31	530	81	-477	164	-154
32	-168	82	532	167	-312
35	45	83	-149	170	-138
37	-464	88	-472	175	-208
38	355	89	525	178	-209
40	-229	90	-143	180	-15
42	-693	91	-260	183	28
43	788	93	-651	185	-148
44	-382	94	714	187	23
45	-216	95	-290	191	-15

Table 5.26 Optimal voltages corresponding to optimal location (see Figure 5.13 upper left) for thermal load type T2 (90 actuators)

Strip Loc.	Voltage V	Strip Loc. Voltage V		Strip Loc.	Voltage V
3	-702	47	-561	96	-306
4	642	48	393	98	-623
5	-130	51	13	99	586
6	-335	54	-467	100	-99
8	-954	55	335	111	-17
9	900	57	-278	113	58
10	-273	59	-760	118	-30
11	-337	60	833	121	-106
13	-701	61	-362	123	40
14	741	62	-279	126	-2
15	-304	64	-485	129	-119
20	-635	65	372	132	-107
21	636	71	-618	137	-110
22	-180	72	682	140	-163
23	-310	73	-282	145	-50
25	-900	74	-296	148	-169
26	947	76	-843	151	-141
27	-436	77	899	156	-134
28	-315	78	-394	159	-205
30	-650	79	-295	163	66
31	649	81	-590	164	-78
32	-225	82	582	167	-188
35	11	83	-110	170	-121
37	-558	88	-622	175	-131
38	395	89	585	178	-171
40	-303	90	-92	180	-40
42	-860	91	-309	183	51
43	943	93	-865	185	-108
44	-468	94	810	187	27
45	-299	95	-204	191	-26

Table 5.27 Optimal voltages corresponding to optimal location (see Figure 5.13 upper left) for thermal load type T3 (90 actuators)

Strip Loc.	Voltage V	Strip Loc.	Voltage V	Strip Loc.	Voltage V
3	-879	47	-388	96	-253
4	811	48	336	98	-465
5	-312	51	104	99	500
6	-411	54	-484	100	-121
8	-1173	55	491	111	73
9	1145	57	-207	113	94
10	-528	59	-673	118	-20
11	-411	60	573	121	51
13	-878	61	38	123	85
14	854	62	-230	126	-81
15	-432	64	-510	129	8
20	-740	65	528	132	84
21	725	71	-409	137	-100
22	-321	72	527	140	63
23	-352	73	-116	145	-151
25	-1023	74	-199	148	-21
26	1047	76	-556	151	120
27	-498	77	647	156	-128
28	-342	78	-189	159	36
30	-737	79	-194	163	29
31	697	81	-339	164	-154
32	-209	82	377	167	-44
35	100	83	94	170	78
37	-365	88	-492	175	-122
38	329	89	489	178	6
40	-122	90	13	180	56
42	-571	91	-256	183	43
43	639	93	-684	185	-28
44	-211	94	675	187	71
45	-123	95	-157	191	33

Table 5.28 Optimal voltages corresponding to optimal location (see Figure 5.13 upper left) for thermal load type T4 (90 actuators)

Strip Loc.	Voltage V	Strip Loc. Voltage V		Strip Loc.	Voltage V
3	-841	47	-244	96	-388
4	778	48	250	98	797
5	-293	51	141	99	747
6	-406	54	-387	100	-280
8	-1177	55	417	111	-112
9	1132	57	-170	113	47
10	-488	59	-546	118	-150
11	-414	60	554	121	-145
13	-908	61	-91	123	52
14	882	62	-197	126	-118
15	-428	64	-335	129	-181
20	-648	65	350	132	-85
21	648	71	-849	137	-196
22	-234	72	849	140	-143
23	-315	73	-378	145	-150
25	-926	74	-383	148	-172
26	967	76	-1095	151	-36
27	-442	77	1114	156	-170
28	-310	78	-530	159	-108
30	-672	79	-374	163	49
31	641	81	-770	164	-128
32	-163	82	728	167	-120
35	127	83	-267	170	-4
37	-265	88	-825	175	-117
38	281	89	761	178	-49
40	-49	90	-288	180	55
42	-390	91	-392	183	57
43	474	93	-1112	185	-41
44	-133	94	1073	187	57
45	-44	95	-455	191	41

In summary, the best result for each case of actuators is shown in Table 5.29.

Table 5.29 RMS error from all cases of actuators (best results only)

Number of	Generations	Seed No.	RMS error for various thermal loads			
actuators	of inner loop		T1	T2	Т3	T4
15	12	1	0.37619	0.37962	0.37087	0.38018
30	20	1	0.22012	0.22819	0.22276	0.22882
60	17	2	0.13164	0.13028	0.12643	0.13118
90	18	2	0.07753	0.08390	0.08055	0.08382
121	13	1	0.06621	0.07732	0.07257	0.07788

Comments about the power

The power of the system can be computed by the following formulation,

$$P = \sum_{i=1}^{n} \frac{U^2}{R}$$

where n is the number of the piezoelectric actuators, U is the electric voltage applied across the piezoelectric actuators and R is the resistant of the piezoelectric actuators. The resistance of the piezoelectric actuators is about 30 M Ω , which was measured on a sheet of PZT under static conditions at room temperature by Mr. Eric J. Ruggiero (PhD Candidate) and Mr. Eddie Simmers (Master student) at Center for Intelligent Material Systems and Structures, Virginia Tech. So the maximum value of the power for all of the cases is less than 4 W.

5.8 Conclusions

Through extensive numerical experiments we found that the most distinct natures of genetic algorithms are random and robust. Here we are not considering theoretical genetic algorithms with unlimited or unrealistic big number of evaluations. Right now we are considering practical genetic algorithms with the limited number of evaluations. By random, we mean that we can not precisely predict the performance of genetic algorithms on future evaluations based on their performance on previous evaluations. By robust, we mean that

although genetic algorithms have their random nature the range of their findings are usually small from their different runs after a certain number of evaluations and they are still capable of finding a very good solution by simply adjusting the parameter setting or using another random seed generator. Because of their nonlinear convergence, the genetic algorithms can be used to get an appropriate solution with a few hundreds of evaluations. The genetic algorithms are also easy to use to generate an alternative solution. To get high quality solutions in the design of complex adaptive structures using finite element analysis and genetic algorithm optimization, the multiple runs including different random seed generators are necessary. The time of the investigation can be significantly reduced using a very coarse grain parallel computing, simply running multiple jobs at the same time. Depending on the load of computer and the number of jobs we can run simultaneously, the time of the investigation can be reduced by a factor of n. For example, if we can simultaneously submit 4 jobs, we maybe reduce the time of investigation by a factor of 4 at best. From Table 5.29, although the RMS error reduces with the number of actuators, the reduction of RMS error may not be worth the effort of increasing the number of actuators. From the results of the optimal voltages we can see that some voltages are extremely high. This makes us further consider the optimal problems with voltage constraints. Overall, the methodology of using finite element analysis and genetic algorithm optimization is a very good approach and GA version 3 is efficient, reliable and robust optimization tool for the challenging problem – optimal placements of a large number of actuators in the design of next generation adaptive structures

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Chapter 6. Summary and Conclusions

6.1 Major contributions

The dissertation first reviewed some basic concepts related to finite element method, genetic algorithms optimization and smart structures and proposed a solution methodology for addressing one of major concerns in the design of next generation large-scale adaptive structures--optimal placement of a large number of actuators by a combination of general finite element analysis techniques and genetic algorithms optimization techniques. With regard to the finite element analysis, the dissertation first reviewed some fundamental issues of general reliable finite element technology such as convergence, patch test, locking problem and singularity of finite element results, and then reviewed the state-of-the-art plate and shell finite elements. Also, the dissertation presented the three most frequently used approaches to derive the finite element formulation, including Galerkin weighted residual method, principle of virtual work and general variational principles. The dissertation chose the second approach, that is, the principle of virtual work, to derive the finite element formulation. There are two major methodologies to develop the finite element for plates and shells, using flat shell element that is composed of membrane element and bending element or using curved shell element. The dissertation facilitates the first approach by systematically presenting the relevant key techniques. The Allman triangular element was chosen as the membrane element, but derived from linear strain triangular element by using the Cook's transformation. We clearly demonstrated the transformed element was indeed the Allman triangular element (See Section 2.5). The discrete Kirchhoff triangular (DKT) element was chosen as the bending element. Because the high efficiency and accuracy of DKT element but lack of document of its derivation, we derived DKT elements in detail (See Section 2.6).

Many flat shell elements neglect the effects of anisotropic membrane-bending coupling. In the dissertation, the coupling effect is included in the element formulation. The nonlinear finite element formulation, including the linear formulation, was derived for the analysis of smart structures under thermal and electric loads (See Section 2.7).

With regards to optimization techniques, the dissertation chose the most general global optimization technique – Genetic Algorithms as an optimizer. The series of genetic algorithms were developed, which include the high efficient GA Version 3 with hill-climbing technique (See Chapter 3 to 5, and Appendix A).

The global optimum solution is the solution that outperforms all others within the design space. Searching for the global optimum is largely an academic problem. The engineering designer, faced with all of the complexities including qualitative criteria and uncertainty is searching generally for a solution that best satisfies immediate needs whilst outperforming previous designs. The rapid convergence of genetic algorithms demonstrated they are well suited for these situations

With regards to smart structures, the dissertation first reviewed the smart materials and state-of-the-art smart structural technology, and then studied one of the most critical and challenging problem for developing next generation of large scale smart structures -- optimization of a relatively large number of piezoelectric actuator locations. With regards to solution methodology of large complex design problem of smart structures, the dissertation developed the methodology of combining two major tools—finite elements and genetic algorithms. This is the first dissertation to have successfully solved the large-scale optimization problem of optimal placement of a large number of piezoelectric actuators on the adaptive structures by using the method of combined FEA and GAs, and the efficient

optimizer -- genetic algorithms were developed (See Chapter 3 to 5, and Appendix A, and the best designs for 30, 121, 15, 60 and 90 actuators are shown upper left in Figure 5.4, Figure 5.6, Figure 5.8, Figure 5.10, and Figure 5.12, respectively).

6.2 Perspective and Future directions

Based on our experience of developing the efficient finite elements and genetic algorithms for the design of large scale smart structures, the following future directions are suggested.

- In addition to the thermal loads, the real system is usually subjected to vibration
 environments and may, compared to thickness, undergo large deformations.

 Therefore, the current linear static finite element analysis should be extended to
 include both dynamic analysis and nonlinear analysis, including large deformation
 and nonlinear piezoelectric constitutive relations.
- 2. Neither the membrane element nor the bending element used in dissertation is perfect. Each of them still has some problems. For example, the membrane element has the rotation degree problem i.e. there is no clear definition of ω, not a real rotation, u_t is not quadratic, thus inferior to the LST, and the element has an unusual zero-energy mode. The bending element doesn't account for shear deformation and is not appropriate to analyze the thick plate because the shear deformation is important for thick plates, especially thick laminated plates. Developing more efficient, reliable triangular flat/curved shell element to solve the co-plane singularity and shear locking still needs some effort. Perhaps the best way to do this is through the use of combination of updated membrane and bending elements. Higher order shear deformation theory can also be included. Also, since no experimental data was available for us, it will be a good practice to check our best results using commercial

- software such as MSC/PATRAN/NASTRAN, ABAQUS.
- 3. Among the optimal solutions obtained here, voltages on certain piezoelectric actuators are very high. It would be important to consider the optimization problem under voltage constraints. This would however require large computational resources.
- 4. The future advanced mirror systems should maintain highest accuracy of surface during their operation. The systems must make a rapid response to the changing in the environment. Therefore, more complex optimization problem should include real-time control subsystem, which consists of sensors, actuators and micro-controllers.
- 5. Our research shows that when genetic algorithms are used to solve large complex optimization problems, it is usually a good advice to perform multiple runs from different parameter settings and different seed generators in order to get high quality solutions. It is also shown that the time of total investigation can be significantly reduced by simultaneously running the multiple jobs. From these points it seems that developing parallel GAs is not necessary. However, because the parallel GAs with timely communication best imitate the evolution of nature, they have a greater potential to get better results and are thus worth developing.

Appendix A. Optimization of Piezoelectric Actuator Locations by Finite Element Method and Genetic Algorithms

In this appendix, the performance of the initial combination of FEM and GAs on the first type of optimization problem (GA version I) is documented. The feasibility of using GAs to optimize a large scale of location problem is demonstrated.

A.1 Abstract

Placement of sensors/actuators in an optimal fashion has drawn significant attention recently due to its importance in many applications such as sensing and control of smart structures. One example is to control the surface accuracy of the mirrors to be used in the next generation of astronomical telescopes. A promising method is to use a given number of piezoelectric actuators bonded onto the rear surface of the primary mirror to correct its distortions without imposing a significant weight penalty. The problem is how to find the optimal locations of piezoelectric actuators to maximize their effectiveness. In general, this is a difficult and computationally intensive combinatorial optimization problem. Kapania and Mohan⁽¹⁾ used Dlorenzo algorithm---a heuristic integer programming approach, to study two cases: (i) using 30 strips, and (ii) using 121 strips under 4 different temperature distributions. A drawback of Dlorenzo algorithm is its inability to find optimal distribution for more than one type of disturbances. Also the solution obtained by such an approach may be a local minimum. The present work is focused on the general case using a modified genetic algorithm⁽²⁾---a robust global stochastic search technique based on the mechanism of natural selection and natural genetics, as an optimizer to find the optimal placements for each temperature distribution. A recently developed laminated triangular shell element (3) is used to

This appendix contains parts of the materials presented at ASME Mechanics and Materials Conference, Virginia Tech., June 27-30, 1999.

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model the mirror. The research shows that the modified genetic algorithm gets slightly better results than those given by DeLorenzo's algorithm. The effect of changing the number of actuators is being investigated so as to get the minimum number of actuators needed to meet the required surface accuracy.

A.2 DeLorenzo's algorithm

The DeLorenzo's algorithm, a heuristic integer programming approach, starts with an initial configuration of 193 strips, as shown in Fig.1. The steps in this algorithm are as follows:

- 1) Remove each of the strips from the initial configuration of m strips (m=193) one at time, and determine the RMS error due to m configurations of m-1 strips.
- 2) Rank the strips in ascending order of the RMS error.
- 3) Remove the strip that results in the smallest RMS error, to obtain a new configuration of m-1 strips.
- 4) Repeat steps 1-3 until the desired number n (30 or 121) of strips is reached. The computational cost of DeLorenzo's algorithm is larger than (193+n+1)*((193-n)/2 evaluations of the configuration of desired number. For example, for n=30 strips, the total number of evaluations = 18256; and for n=121 strips, Total number of evaluations = 11340. The disadvantage of DeLorenzo's algorithm is that the obtained shape control could be local minimum.

A.3 Genetic Algorithms

The Genetic Algorithms, GA Version 1 was developed by modifying Carroll's FORTRAN Genetic Algorithm Driver ⁽²⁾, such as applying the constraint to bit strings (chromosome), changing selection scheme, and modifying some important parameters.

Coding: directly mapping, 193 bits

Selection: elite selection, tournament selection

Crossover: uniform crossover

A.4 Case Studies

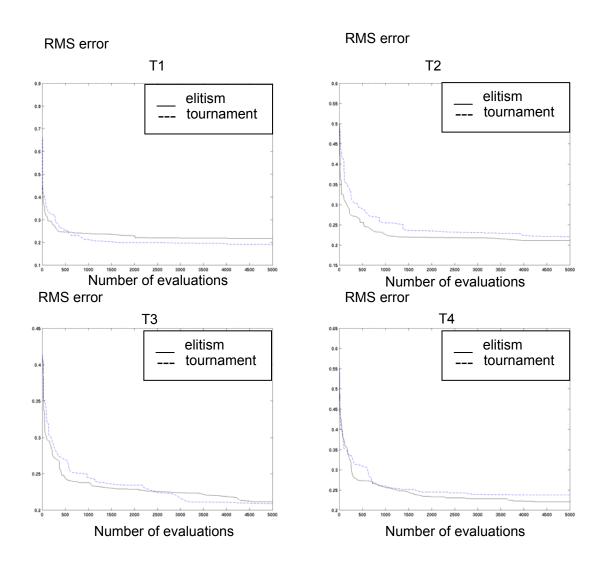


Figure A.1 Performance of the GAs versus the number of evaluations on thermal load T1, T2, T3, and T4, respectively

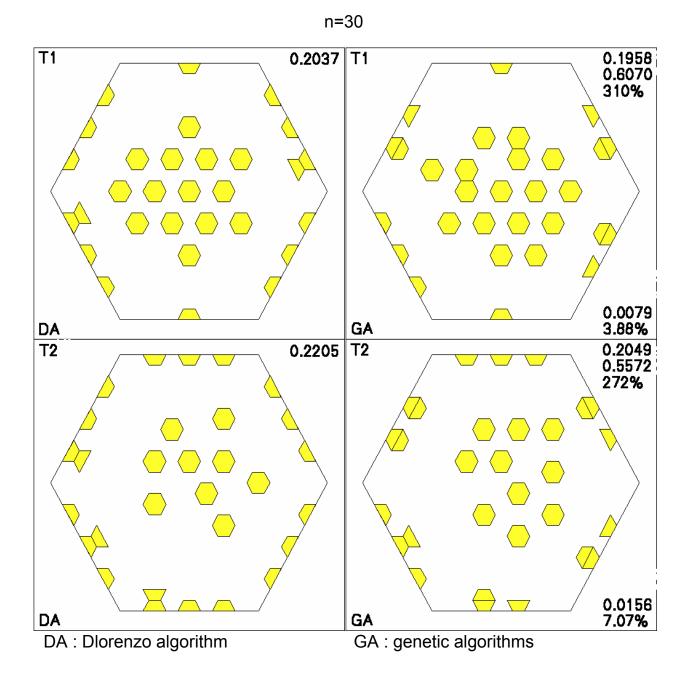


Figure A.2 a) Optimal locations for the thermal load T1 and T2 obtained by DeLorenzo algorithm and Genetic Algorithms (30 actuators)

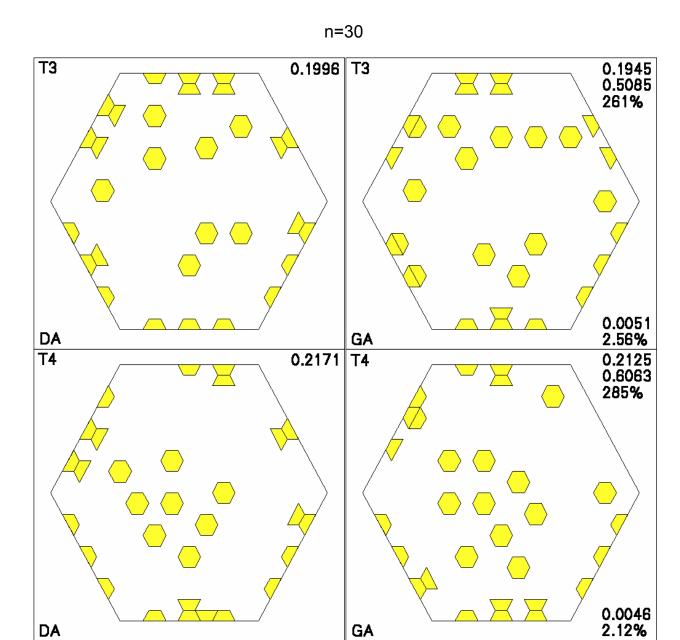


Figure A.2 b) Optimal locations for the thermal load T3 and T4 obtained by DeLorenzo algorithm and Genetic Algorithms (30 actuators)

GA

GA: genetic algorithms

DA

DA: Dlorenzo algorithm

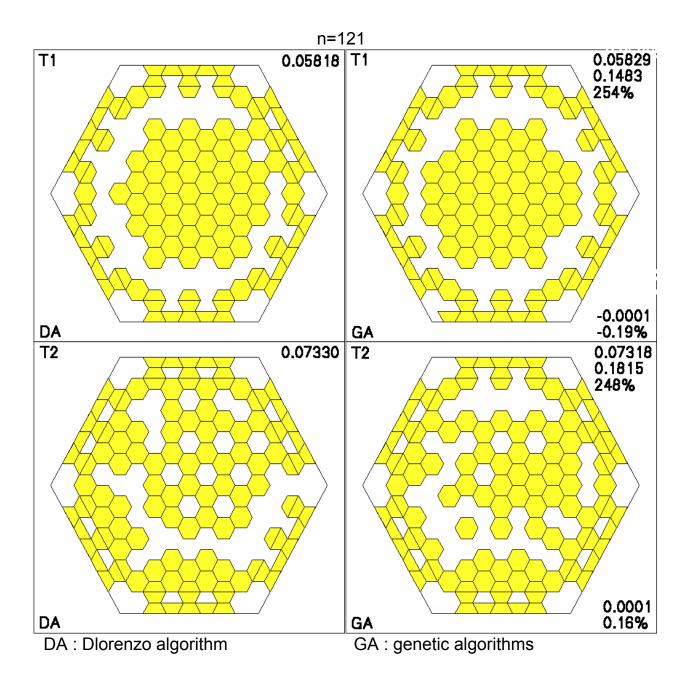


Figure A.3 a) Optimal locations for the thermal load T1 and T2 obtained by DeLorenzo algorithm and Genetic Algorithms (121 actuators)

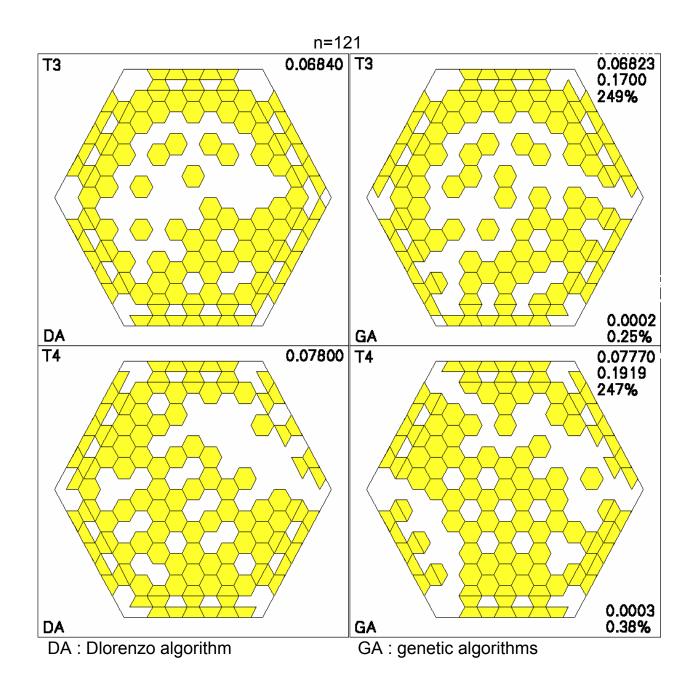


Figure A.3 b) Optimal locations for the thermal load T3 and T4 obtained by DeLorenzo algorithm and Genetic Algorithms (121 actuators)

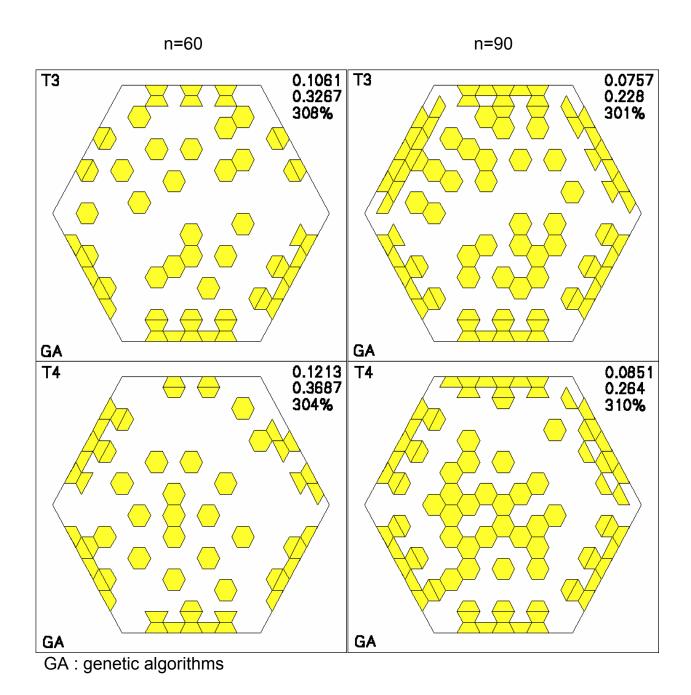


Figure A.4 Optimal locations for the thermal load T3 and T4 obtained by Genetic Algorithms (60 actuators and 90 actuators)

A.5 Major Conclusions

1) GA Version 1 got slightly better results than DeLorenzo's algorithm.

- 2) The cost of the GA Version 1 is less than DeLorenzo's algorithm for the case of 30 actuators, and more than DeLorenzo's algorithm for the case of 121 actuators.
- 3) The feasibility of using GAs to solve the large scale optimization problem of actuator locations was demonstrated.
- 4) The need to further develop the effective GAs for the optimization of actuator locations was proposed.

Appendix B. Integration by Parts

Integration by parts

In one dimension

$$\int_{a}^{b} u dv = -\int_{a}^{b} v du + uv \Big|_{a}^{b}$$

In two dimension

$$\iint_{A} u \frac{\partial v}{\partial x} dx dy = -\iint_{A} v \frac{\partial u}{\partial x} dx dy + \oint_{l} uv n_{x} dl$$

$$\iint_{A} u \frac{\partial v}{\partial y} dx dy = -\iint_{A} v \frac{\partial u}{\partial y} dx dy + \oint_{l} uv n_{y} dl$$

In three dimension

$$\iiint_{V} u \frac{\partial v}{\partial x} dx dy dz = -\iiint_{V} v \frac{\partial u}{\partial x} dx dy dz + \oiint_{S} uvn_{x} dS$$

$$\iiint_{V} u \frac{\partial v}{\partial y} dx dy dz = -\iiint_{V} v \frac{\partial u}{\partial y} dx dy dz + \oiint_{S} uvn_{y} dS$$

$$\iiint_{V} u \frac{\partial v}{\partial z} dx dy dz = - \iiint_{V} v \frac{\partial u}{\partial z} dx dy dz + \oiint_{S} uvn_{z} dS$$

Appendix C. Vector Cross Product, Area of the Triangle, Differential

Area and Volume Element

Vector Cross Product

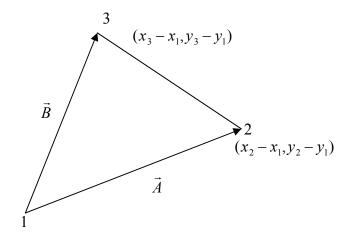


Figure C.1 Vector \vec{A} and \vec{B} in the plane of the triangle

$$\left| \vec{A} \times \vec{B} \right| = \left| \vec{A} \right| \left| \vec{B} \right| \sin \alpha = 2\Delta$$

$$\vec{A} \times \vec{B} = \begin{vmatrix} \vec{i} & \vec{j} & \vec{k} \\ x_2 - x_1 & y_2 - y_1 & 0 \\ x_3 - x_1 & y_3 - y_1 & 0 \end{vmatrix}$$

Area of the triangle

$$\Delta = \frac{1}{2} \begin{vmatrix} 1 & x_1 & y_1 \\ 1 & x_2 & y_2 \\ 1 & x_3 & y_3 \end{vmatrix}$$

Differential area and volume element

$$\vec{A} = \frac{\partial x}{\partial \xi} d\xi \vec{i} + \frac{\partial y}{\partial \xi} d\xi \vec{j}, \qquad \vec{B} = \frac{\partial x}{\partial \eta} d\eta \vec{i} + \frac{\partial y}{\partial \eta} d\eta \vec{j}$$

As the length of the vector resulting from a cross product of $\vec{A} \times \vec{B}$ is equal to the area of the elementary parallelogram, we can write

$$dxdy = \begin{vmatrix} \frac{\partial x}{\partial \xi} & \frac{\partial y}{\partial \xi} \\ \frac{\partial x}{\partial \eta} & \frac{\partial y}{\partial \eta} \end{vmatrix} d\xi d\eta = |J| d\xi d\eta$$

Similarly, a differential volume

$$dxdydz = (\bar{A} \times \bar{B}) \cdot \bar{C} = \begin{vmatrix} \frac{\partial x}{\partial \xi} & \frac{\partial y}{\partial \xi} & \frac{\partial z}{\partial \xi} \\ \frac{\partial x}{\partial \eta} & \frac{\partial y}{\partial \eta} & \frac{\partial z}{\partial \eta} \\ \frac{\partial x}{\partial \zeta} & \frac{\partial y}{\partial \zeta} & \frac{\partial z}{\partial \zeta} \end{vmatrix} d\xi d\eta d\zeta = |J| d\xi d\eta d\zeta$$

Appendix D. Geometrical Verification of the Relations between the Cartesian Coordinates and Area Coordinates of the Triangle

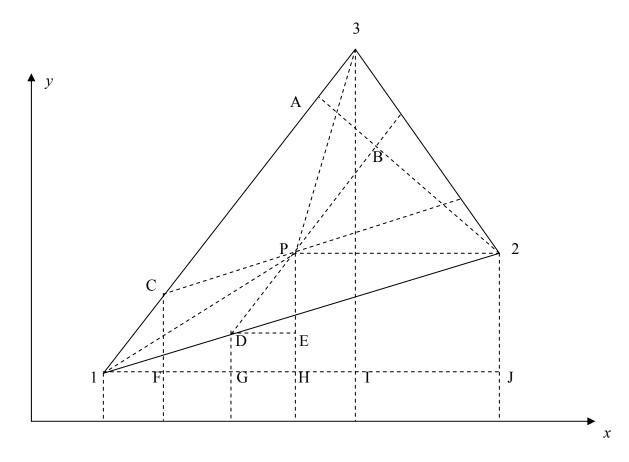


Figure D.1 Cartesian Coordinates and Area Coordinates of the triangle

$$\xi_2 = \frac{S_{\Delta P31}}{S_{\Delta 123}} = \frac{l_{AB}}{l_{A2}} = \frac{l_{1D}}{l_{12}} = \frac{l_{1G}}{l_{1J}}$$

$$\xi_3 = \frac{S_{\Delta P12}}{S_{\Delta 122}} = \frac{l_{1C}}{l_{12}} = \frac{l_{1F}}{l_{1L}}$$

$$x = x_1 + l_{1G} + l_{GH} = x_1 + l_{1G} + l_{DE} = x_1 + l_{1G} + l_{1F} = x_1 + \xi_2(x_2 - x_1) + \xi_3(x_3 - x_1) = \xi_1 x_1 + \xi_2 x_2 + \xi_3 x_3$$

similarly,

$$y = \xi_1 y_1 + \xi_2 y_2 + \xi_3 y_3$$

Appendix E. Exact Integration of a Polynomial Term in terms of Area Coordinates over the Triangular Region

The integral of a polynomial term in terms of area coordinates over the triangular region can be expressed in the following form,

$$\iint_{\Delta} \xi_1^a \xi_2^b \xi_3^c dx dy = \frac{a!b!c!}{(a+b+c+2)!} 2A$$

where A represents the area of the triangular region. The integral along a side of the triangle is given by

$$\int_{l} \xi_{i}^{a} \xi_{j}^{b} dl = \frac{a!b!}{(a+b+1)!} l$$

where a, b, and c are nonnegative integers.

Appendix F. Shape Functions and Their Derivatives of Linear Strain Triangular Element (Quadratic Triangular Element)

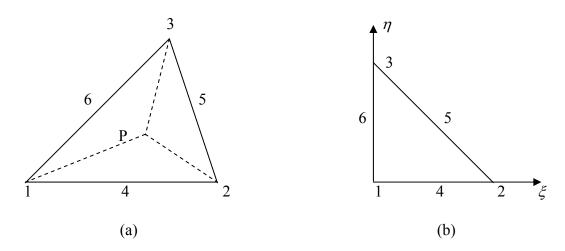


Figure F.1 Linear strain triangular element (Quadratic triangular element) in (a) area coordinate system and (b) natural coordinate system

$$\xi_{1} = 1 - \xi - \eta, \qquad \xi_{2} = \xi, \qquad \xi_{3} = \eta$$

$$N_{1} = \xi_{1}(2\xi_{1} - 1) = 2(1 - \xi - \eta)(0.5 - \xi - \eta)$$

$$N_{2} = \xi_{2}(2\xi_{2} - 1) = \xi(2\xi - 1)$$

$$N_{3} = \xi_{3}(2\xi_{3} - 1) = \eta(2\eta - 1)$$

$$N_{4} = 4\xi_{1}\xi_{2} = 4\xi(1 - \xi - \eta)$$

$$N_{5} = 4\xi_{2}\xi_{3} = 4\eta$$

$$N_{6} = 4\xi_{1}\xi_{3} = 4\eta(1 - \xi - \eta)$$

$$\left\{ N_{,\xi} \right\} = \frac{\partial \left\{ N_{1} \right\}}{\partial \xi} = \begin{cases} \frac{\partial N_{1}}{\partial \xi} \\ \frac{\partial N_{2}}{\partial \xi} \\ \frac{\partial N_{3}}{\partial \xi} \\ \frac{\partial N_{4}}{\partial \xi} \\ \frac{\partial N_{5}}{\partial \xi} \\ \frac{\partial N_{6}}{\partial \xi} \end{cases} = \begin{cases} 4\xi + 4\eta - 3 \\ 4\xi - 1 \\ 0 \\ 4 - 8\xi - 4\eta \\ 4\eta \\ -4\eta \end{cases}$$

$$\left[\frac{\partial N_{1}}{\partial \eta} \right]$$

$$\{N_{,\eta}\} = \frac{\partial \{N\}}{\partial \eta} = \begin{cases} \frac{\partial N_1}{\partial \eta} \\ \frac{\partial N_2}{\partial \eta} \\ \frac{\partial N_3}{\partial \eta} \\ \frac{\partial N_4}{\partial \eta} \\ \frac{\partial N_5}{\partial \eta} \\ \frac{\partial N_6}{\partial \eta} \end{cases} = \begin{cases} 4\xi + 4\eta - 3 \\ 0 \\ 4\eta - 1 \\ -4\xi \\ 4\xi \\ 4 - 4\xi - 8\eta \end{cases}$$

Appendix G. Derivation of the True Rotation in the Theory of Elasticity in the Allman Triangular Element

The displacement field of the Allman triangular element is

$$u = u_{1}\xi_{1} + u_{2}\xi_{2} + u_{3}\xi_{3} + \frac{1}{2}l_{12}\cos\gamma_{12}(\omega_{2} - \omega_{1})\xi_{1}\xi_{2}$$

$$+ \frac{1}{2}l_{23}\cos\gamma_{23}(\omega_{3} - \omega_{2})\xi_{2}\xi_{3} + \frac{1}{2}l_{31}\cos\gamma_{31}(\omega_{1} - \omega_{3})\xi_{3}\xi_{1}$$

$$v = v_{1}\xi_{1} + v_{2}\xi_{2} + v_{3}\xi_{3} + \frac{1}{2}l_{12}\sin\gamma_{12}(\omega_{2} - \omega_{1})\xi_{1}\xi_{2}$$

$$+ \frac{1}{2}l_{23}\sin\gamma_{23}(\omega_{3} - \omega_{2})\xi_{2}\xi_{3} + \frac{1}{2}l_{31}\sin\gamma_{31}(\omega_{1} - \omega_{3})\xi_{3}\xi_{1}$$
(G.1)

where $\sin \gamma_{12} = \frac{x_1 - x_2}{l_{12}}$, $\cos \gamma_{12} = -\frac{y_1 - y_2}{l_{12}}$, etc.

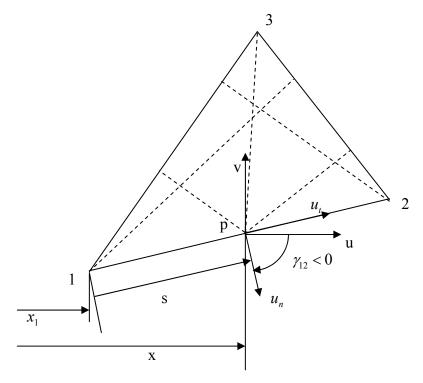


Figure G.1 γ definition along the sides of the triangle

Along the element side 12, for example

$$x = x_1 - s\sin \gamma_{12}$$

$$y = y_1 + s\cos \gamma_{12}$$

Therefore, the partial derivative along an element side is

$$\frac{\partial}{\partial s} = \frac{\partial}{\partial x}\frac{\partial x}{\partial s} + \frac{\partial}{\partial y}\frac{\partial y}{\partial s} = -\sin\gamma\frac{\partial}{\partial x} + \cos\gamma\frac{\partial}{\partial y}$$
(G.2)

In the field of the Allman triangle, eqn. (G.1), the true rotation Ω in the theory of elasticity is

$$\begin{split} &\Omega = \frac{1}{2} (\frac{\partial v}{\partial x} - \frac{\partial u}{\partial y}) \\ &= \frac{1}{2} [(\frac{\partial v}{\partial \xi_{1}} \frac{\partial \xi_{1}}{\partial x} + \frac{\partial v}{\partial \xi_{2}} \frac{\partial \xi_{2}}{\partial x} + \frac{\partial v}{\partial \xi_{3}} \frac{\partial \xi_{3}}{\partial x}) - (\frac{\partial v}{\partial \xi_{1}} \frac{\partial \xi_{1}}{\partial y} + \frac{\partial v}{\partial \xi_{2}} \frac{\partial \xi_{2}}{\partial y} + \frac{\partial v}{\partial \xi_{3}} \frac{\partial \xi_{3}}{\partial y})] \\ &= \frac{1}{2} [(v_{1} + \frac{1}{2} l_{12} \sin \gamma_{12} (\omega_{2} - \omega_{1}) \xi_{2} + \frac{1}{2} l_{31} \sin \gamma_{31} (\omega_{1} - \omega_{3}) \xi_{3}) \frac{b_{1}}{2A} \\ &+ (v_{2} + \frac{1}{2} l_{12} \sin \gamma_{12} (\omega_{2} - \omega_{1}) \xi_{1} + \frac{1}{2} l_{23} \sin \gamma_{23} (\omega_{3} - \omega_{2}) \xi_{3}) \frac{b_{2}}{2A} \\ &+ (v_{3} + \frac{1}{2} l_{23} \sin \gamma_{23} (\omega_{3} - \omega_{2}) \xi_{2} + \frac{1}{2} l_{31} \sin \gamma_{31} (\omega_{1} - \omega_{3}) \xi_{1}) \frac{b_{3}}{2A} \\ &- (u_{1} + \frac{1}{2} l_{12} \cos \gamma_{12} (\omega_{2} - \omega_{1}) \xi_{2} + \frac{1}{2} l_{31} \cos \gamma_{31} (\omega_{1} - \omega_{3}) \xi_{3}) \frac{c_{1}}{2A} \\ &- (u_{2} + \frac{1}{2} l_{12} \cos \gamma_{12} (\omega_{2} - \omega_{1}) \xi_{1} + \frac{1}{2} l_{23} \cos \gamma_{23} (\omega_{3} - \omega_{2}) \xi_{3}) \frac{c_{2}}{2A} \\ &- (u_{3} + \frac{1}{2} l_{23} \cos \gamma_{23} (\omega_{3} - \omega_{2}) \xi_{2} + \frac{1}{2} l_{31} \cos \gamma_{31} (\omega_{1} - \omega_{3}) \xi_{1}) \frac{c_{3}}{2A}] \\ &= \frac{1}{4A} [(x_{2} - x_{3})u_{1} + (x_{3} - x_{1})u_{2} + (x_{1} - x_{2})u_{3} + (y_{2} - y_{3})v_{1} + (y_{3} - y_{1})v_{2} + (y_{1} - y_{2})v_{3}] + R \end{split}$$

where

$$\begin{split} R &= \frac{1}{4} \big[(l_{12} \sin \gamma_{12} (\omega_2 - \omega_1) \xi_2 + l_{31} \sin \gamma_{31} (\omega_1 - \omega_3) \xi_3) \frac{\partial \xi_1}{\partial x} \\ &+ (l_{12} \sin \gamma_{12} (\omega_2 - \omega_1) \xi_1 + l_{23} \sin \gamma_{23} (\omega_3 - \omega_2) \xi_3) \frac{\partial \xi_2}{\partial x} \\ &+ (l_{23} \sin \gamma_{23} (\omega_3 - \omega_2) \xi_2 + l_{31} \sin \gamma_{31} (\omega_1 - \omega_3) \xi_1) \frac{\partial \xi_3}{\partial x} \\ &- (l_{12} \cos \gamma_{12} (\omega_2 - \omega_1) \xi_2 + l_{31} \cos \gamma_{31} (\omega_1 - \omega_3) \xi_3) \frac{\partial \xi_1}{\partial y} \\ &- (l_{12} \cos \gamma_{12} (\omega_2 - \omega_1) \xi_1 + l_{23} \cos \gamma_{23} (\omega_3 - \omega_2) \xi_3) \frac{\partial \xi_2}{\partial y} \\ &- (l_{23} \cos \gamma_{23} (\omega_3 - \omega_2) \xi_2 + l_{31} \cos \gamma_{31} (\omega_1 - \omega_3) \xi_1) \frac{\partial \xi_3}{\partial y} \big] \\ &= \frac{1}{4} \big[l_{12} (\omega_2 - \omega_1) \xi_2 (\sin \gamma_{12} \frac{\partial \xi_1}{\partial x} - \cos \gamma_{12} \frac{\partial \xi_1}{\partial y}) + l_{12} (\omega_2 - \omega_1) \xi_1 (\sin \gamma_{12} \frac{\partial \xi_2}{\partial x} - \cos \gamma_{12} \frac{\partial \xi_2}{\partial y}) \\ &+ l_{31} (\omega_1 - \omega_3) \xi_3 (\sin \gamma_{31} \frac{\partial \xi_1}{\partial x} - \cos \gamma_{31} \frac{\partial \xi_1}{\partial y}) + l_{31} (\omega_1 - \omega_3) \xi_1 (\sin \gamma_{31} \frac{\partial \xi_3}{\partial x} - \cos \gamma_{31} \frac{\partial \xi_3}{\partial y}) \\ &+ l_{23} (\omega_3 - \omega_2) \xi_3 (\sin \gamma_{23} \frac{\partial \xi_2}{\partial x} - \cos \gamma_{23} \frac{\partial \xi_2}{\partial y}) + l_{23} (\omega_3 - \omega_2) \xi_2 (\sin \gamma_{23} \frac{\partial \xi_3}{\partial x} - \cos \gamma_{23} \frac{\partial \xi_3}{\partial y}) \big] \end{split}$$

We note that $\frac{\partial \xi_i}{\partial x}$ and $\frac{\partial \xi_i}{\partial y}$ are constant in the entire field and on the boundaries we have,

refer to eqn. (G.2),

Along the side 12,

$$\xi_1 = \frac{S_{\Delta P23}}{S_{\Delta 123}} = 1 - \frac{s}{l_{12}}, \qquad \xi_2 = \frac{S_{\Delta P31}}{S_{\Delta 123}} = \frac{s}{l_{12}}, \qquad \xi_3 = 0$$

$$\left(\sin\gamma_{12}\frac{\partial\xi_{1}}{\partial x}-\cos\gamma_{12}\frac{\partial\xi_{1}}{\partial y}\right)=-\frac{\partial\xi_{1}}{\partial s}\bigg|_{side12}=\frac{1}{l_{12}},\quad\left(\sin\gamma_{12}\frac{\partial\xi_{2}}{\partial x}-\cos\gamma_{12}\frac{\partial\xi_{2}}{\partial y}\right)=-\frac{\partial\xi_{2}}{\partial s}\bigg|_{side12}=-\frac{1}{l_{12}}$$

Along the side 31,

$$\left(\sin\gamma_{31}\frac{\partial\xi_{1}}{\partial x}-\cos\gamma_{31}\frac{\partial\xi_{1}}{\partial y}\right)=-\frac{\partial\xi_{1}}{\partial s}\bigg|_{side 31}=-\frac{1}{l_{31}},\quad\left(\sin\gamma_{31}\frac{\partial\xi_{3}}{\partial x}-\cos\gamma_{31}\frac{\partial\xi_{3}}{\partial y}\right)=-\frac{\partial\xi_{3}}{\partial s}\bigg|_{side 31}=\frac{1}{l_{31}}$$

Along the side 23,

$$\left(\sin\gamma_{23}\frac{\partial\xi_{2}}{\partial x}-\cos\gamma_{23}\frac{\partial\xi_{2}}{\partial y}\right)=-\frac{\partial\xi_{2}}{\partial x}\bigg|_{side^{23}}=\frac{1}{l_{23}},\quad\left(\sin\gamma_{23}\frac{\partial\xi_{3}}{\partial x}-\cos\gamma_{23}\frac{\partial\xi_{3}}{\partial y}\right)=-\frac{\partial\xi_{3}}{\partial x}\bigg|_{side^{23}}=-\frac{1}{l_{23}}$$

So we have,

$$\begin{split} R &= \frac{1}{4} [(\omega_{2} - \omega_{1})\xi_{2} - (\omega_{2} - \omega_{1})\xi_{1} - (\omega_{1} - \omega_{3})\xi_{3} + (\omega_{1} - \omega_{3})\xi_{1} + (\omega_{3} - \omega_{2})\xi_{3} - (\omega_{3} - \omega_{2})\xi_{2}] \\ &= \frac{1}{4} (2\omega_{1}\xi_{1} - \omega_{1}\xi_{2} - \omega_{1}\xi_{3} + 2\omega_{2}\xi_{2} - \omega_{2}\xi_{1} - \omega_{2}\xi_{3} + 2\omega_{3}\xi_{3} - \omega_{3}\xi_{1} - \omega_{3}\xi_{2}) \\ &= \frac{1}{4} (3\omega_{1}\xi_{1} - \omega_{1} + 3\omega_{2}\xi_{2} - \omega_{2} + 3\omega_{3}\xi_{3} - \omega_{3}) \\ &= \frac{1}{4} \sum_{i=1}^{3} (3\xi_{i} - 1)\omega_{i} \end{split}$$

So the true rotation in the Allman triangular element is

$$\Omega = \Omega_0 + \frac{1}{4} \sum_{i=1}^{3} (3\xi_i - 1)\omega_i$$
 (G.3)

where Ω_0 , the rotation at the centroid $\xi_1 = \xi_2 = \xi_3 = \frac{1}{3}$, is

$$\Omega_0 = \frac{1}{4A} [(x_2 - x_3)u_1 + (x_3 - x_1)u_2 + (x_1 - x_2)u_3 + (y_2 - y_3)v_1 + (y_3 - y_1)v_2 + (y_1 - y_2)v_3]$$

Appendix H. Slope at the Mid-Span of Cubic Beam Element

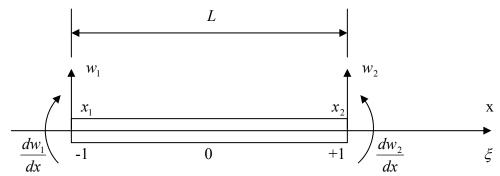


Figure H.1 Cubic beam element

$$\xi = \frac{2x - x_2 - x_1}{x_2 - x_1} = \frac{2x - x_2 - x_1}{L}$$

$$w = w_1 \varphi_1 + \frac{dw_1}{dx} \varphi_2 + w_2 \varphi_3 + \frac{dw_2}{dx} \varphi_4$$

$$\varphi_1 = \frac{1}{4}(2 - 3\xi + \xi^3), \quad \varphi_2 = \frac{L}{8}(1 - \xi - \xi^2 + \xi^3)$$

$$\varphi_3 = \frac{1}{4}(2 + 3\xi - \xi^3), \quad \varphi_4 = \frac{L}{8}(-1 - \xi + \xi^2 + \xi^3)$$

$$\frac{dw}{dx}\bigg|_{\xi=0} = \left(w_1 \frac{d\varphi_1}{d\xi} + \frac{dw_1}{dx} \frac{d\varphi_2}{d\xi} + w_2 \frac{d\varphi_3}{d\xi} + \frac{dw_2}{dx} \frac{d\varphi_4}{d\xi}\right) \frac{d\xi}{dx}\bigg|_{\xi=0} = -\frac{3}{2L} w_1 - \frac{1}{4} \frac{dw_1}{dx} + \frac{3}{2L} w_2 - \frac{1}{4} \frac{dw_2}{dx}$$

Appendix I. Shape Functions and Their Derivatives of Discrete Kirchhoff Triangular (DKT) Element

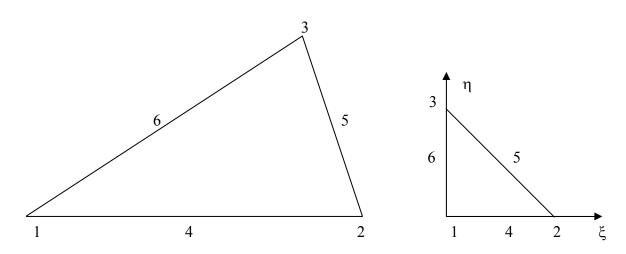


Figure I.1 Node locations of quadratic triangular element

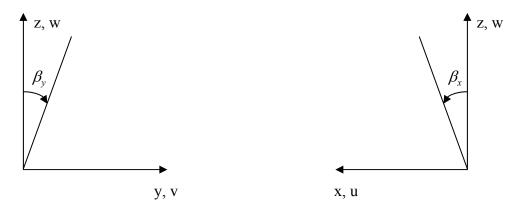


Figure I.2 Positive directions of β_x and β_y in 2D view

$$H_{x_1} = 1.5(a_4N_4 - a_6N_6)$$

$$H_{x_2} = b_4N_4 + b_6N_6$$

$$H_{x_3} = N_1 - c_4N_4 - c_6N_6$$

$$H_{x_4} = 1.5(a_5N_5 - a_4N_4)$$

$$H_{x_5} = b_4N_4 + b_5N_5$$

$$H_{x_6} = N_2 - c_4 N_4 - c_5 N_5$$

$$H_{x_2} = 1.5(a_6N_6 - a_5N_5)$$

$$H_{x_9} = b_5 N_5 + b_6 N_6$$

$$H_{x_9} = N_3 - c_5 N_5 - c_6 N_6$$

$$H_{v_1} = 1.5(d_4N_4 - d_6N_6)$$

$$H_{y_2} = -N_1 + e_4 N_4 + e_6 N_6$$

$$H_{y_3} = -b_4 N_4 - b_6 N_6 = -H_{x_3}$$

$$H_{v_4} = 1.5(d_5N_5 - d_4N_4)$$

$$H_{y_5} = -N_2 + e_4 N_4 + e_5 N_5$$

$$H_{v_6} = -b_4 N_4 - b_5 N_5 = -H_{x_5}$$

$$H_{v_2} = 1.5(d_6N_6 - d_5N_5)$$

$$H_{v_0} = -N_3 + e_5 N_5 + e_6 N_6$$

$$H_{y_9} = -b_5 N_5 - b_6 N_6 = -H_{x_8}$$

$$\beta_{x} = H_{x_{1}}w_{1} + H_{x_{2}}\theta_{x_{1}} + H_{x_{3}}\theta_{y_{1}} + H_{x_{4}}w_{2} + H_{x_{5}}\theta_{x_{2}} + H_{x_{6}}\theta_{y_{2}} + H_{x_{7}}w_{3} + H_{x_{8}}\theta_{x_{3}} + H_{x_{9}}\theta_{y_{3}}$$

$$\beta_{y} = H_{y_{1}} w_{1} + H_{y_{2}} \theta_{x_{1}} + H_{y_{3}} \theta_{y_{1}} + H_{y_{4}} w_{2} + H_{y_{5}} \theta_{x_{2}} + H_{y_{6}} \theta_{y_{2}} + H_{y_{7}} w_{3} + H_{y_{8}} \theta_{x_{3}} + H_{y_{9}} \theta_{y_{3}}$$

$$a_4 = -x_{12}/l_{12}^2$$
, $a_5 = -x_{23}/l_{23}^2$, $a_6 = -x_{31}/l_{31}^2$

$$b_4 = \frac{3}{4} \frac{x_{12}y_{12}}{l_{12}^2}, \quad b_5 = \frac{3}{4} \frac{x_{23}y_{23}}{l_{23}^2}, \quad b_6 = \frac{3}{4} \frac{x_{31}y_{31}}{l_{31}^2}$$

$$c_4 = \left(\frac{1}{4}x_{12}^2 - \frac{1}{2}y_{12}^2\right)/l_{12}^2, \quad c_5 = \left(\frac{1}{4}x_{23}^2 - \frac{1}{2}y_{23}^2\right)/l_{23}^2, \quad c_6 = \left(\frac{1}{4}x_{31}^2 - \frac{1}{2}y_{31}^2\right)/l_{31}^2$$

$$d_4 = -y_{12}/l_{12}^2$$
, $d_5 = -y_{23}/l_{23}^2$, $d_6 = -y_{31}/l_{31}^2$

$$e_4 = (\frac{1}{4}y_{12}^2 - \frac{1}{2}x_{12}^2)/l_{12}^2, \qquad e_5 = (\frac{1}{4}y_{23}^2 - \frac{1}{2}x_{23}^2)/l_{23}^2, \qquad e_6 = (\frac{1}{4}y_{31}^2 - \frac{1}{2}x_{31}^2)/l_{31}^2$$

$$x_{ij} = x_i - x_j$$
, $y_{ij} = y_i - y_j$, $l_{ij}^2 = x_{ij}^2 + y_{ij}^2$

or

$$a_k = -x_{ij} / l_{ij}^2$$

$$b_k = \frac{3}{4} \frac{x_{ij} y_{ij}}{l_{ii}^2}$$

$$c_k = (\frac{1}{4}x_{ij}^2 - \frac{1}{2}y_{ij}^2)/l_{ij}^2$$

$$d_k = -y_{ij}/l_{ij}^2$$

$$e_k = (\frac{1}{4}y_{ij}^2 - \frac{1}{2}x_{ij}^2)/l_{ij}^2$$

where k = 4, 5, 6 for the sides ij = 12, 23, 31, respectively.

$$H_x^T = [H_{x_1}, H_{x_2}, H_{x_3}, H_{x_4}, H_{x_5}, H_{x_6}, H_{x_7}, H_{x_8}, H_{x_9}]$$

$$H_{x,\xi} = \begin{cases} p_4(1-2\xi) + (p_6 - p_4)\eta \\ q_4(1-2\xi) - (q_6 + q_4)\eta \\ -4 + 6(\xi + \eta) + r_4(1-2\xi) - \eta(r_6 + r_4) \\ -p_4(1-2\xi) + \eta(p_5 + p_4) \\ q_4(1-2\xi) - \eta(q_4 - q_5) \\ -2 + 6\xi + r_4(1-2\xi) + \eta(r_5 - r_4) \\ -\eta(p_6 + p_5) \\ \eta(q_5 - q_6) \\ -\eta(r_6 - r_5) \end{cases}$$

$$H_{x,\eta} = \begin{cases} -p_6(1-2\eta) - \xi(p_4 - p_6) \\ q_6(1-2\eta) - \xi(q_6 + q_4) \\ -4 + 6(\xi + \eta) + r_6(1-2\eta) - \xi(r_6 + r_4) \\ \xi(p_5 + p_4) \\ \xi(q_5 - q_4) \\ -\xi(r_4 - r_5) \\ p_6(1-2\eta) - \xi(p_5 + p_6) \\ q_6(1-2\eta) + \xi(q_5 - q_6) \\ -2 + 6\eta + r_6(1-2\eta) + \xi(r_5 - r_6) \end{cases}$$

$$\boldsymbol{H}_{y}^{T} = [\boldsymbol{H}_{y_{1}}, \boldsymbol{H}_{y_{2}}, \boldsymbol{H}_{y_{3}}, \boldsymbol{H}_{y_{4}}, \boldsymbol{H}_{y_{5}}, \boldsymbol{H}_{y_{6}}, \boldsymbol{H}_{y_{7}}, \boldsymbol{H}_{y_{8}}, \boldsymbol{H}_{y_{9}}]$$

$$H_{y,\xi} = \begin{cases} t_4(1-2\xi) + (t_6 - t_4)\eta \\ 1 + r_4(1-2\xi) - (r_6 + r_4)\eta \\ -q_4(1-2\xi) + \eta(q_6 + q_4) \\ -t_4(1-2\xi) + \eta(t_5 + t_4) \\ -1 + r_4(1-2\xi) + \eta(r_5 - r_4) \\ -q_4(1-2\xi) - \eta(q_5 - q_4) \\ -\eta(t_5 + t_6) \\ \eta(r_5 - r_6) \\ -\eta(q_5 - q_6) \end{cases}$$

$$H_{y,\eta} = \begin{cases} -t_6(1-2\eta) - \xi(t_4 - t_6) \\ 1 + r_6(1-2\eta) - \xi(r_6 + r_4) \\ -q_6(1-2\eta) + \xi(q_6 + q_4) \\ \xi(t_5 + t_4) \\ \xi(r_5 - r_4) \\ -\xi(q_5 - q_4) \\ t_6(1-2\eta) - \xi(t_5 + t_6) \\ -1 + r_6(1-2\eta) + \xi(r_5 - r_6) \\ -q_6(1-2\eta) - \xi(q_5 - q_6) \end{cases}$$

where

$$p_4 = -6x_{12}/l_{12}^2 = 6a_4$$
, $p_5 = -6x_{23}/l_{23}^2 = 6a_5$, $p_6 = -6x_{31}/l_{31}^2 = 6a_6$
 $t_4 = -6y_{12}/l_{12}^2 = 6d_4$, $t_5 = -6y_{23}/l_{23}^2 = 6d_5$, $t_6 = -6y_{31}/l_{31}^2 = 6d_6$

$$q_4 = \frac{3x_{12}y_{12}}{l_{12}^2} = 4b_4, \quad q_5 = \frac{3x_{23}y_{23}}{l_{23}^2} = 4b_5, \quad q_6 = \frac{3x_{31}y_{31}}{l_{31}^2} = 4b_6$$

$$r_4 = 3y_{12}^2/l_{12}^2$$
, $r_5 = 3y_{23}^2/l_{23}^2$, $r_6 = 3y_{31}^2/l_{31}^2$

$$x_{ij} = x_i - x_j$$
, $y_{ij} = y_i - y_j$, $l_{ij}^2 = x_{ij}^2 + y_{ij}^2$

or

$$p_k = -6x_{ij} / l_{ij}^2 = 6a_k$$

$$t_k = -6y_{ij} / l_{ij}^2 = 6d_k$$

$$q_k = \frac{3x_{ij}y_{ij}}{l_{ii}^2} = 4b_k$$

$$r_k = 3y_{ij}^2 / l_{ij}^2$$

where k = 4, 5, 6 for the sides ij = 12, 23, 31, respectively.

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